COMPUTATIONAL FINANCE
Numerical Methods for Pricing Financial Instruments

George Levy
To Kathryn
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Preface

It was in late 1995 to early 1996 (shortly after the birth of his first daughter Claire) that the author first began to read the currently available finance books in order to write C/C++ financial software. However, apart from the book Options Futures and Other Derivatives by John Hull, he found very little information of practical help and had to trawl through the original journal articles in the Bodleian library for more information. Even then much information on how to implement and test various models was not included.

The current book aims to provide practical information on basic computational finance. In addition many statistical, financial, and numerical results are derived so that the reader does not need to consult a large number of other books. It should be mentioned that many of the code excerpts assume that the reader has access to NAG Ltd numerical libraries. However, for those who are not so fortunate, equivalent C/C++ software is provided on the accompanying CD ROM.

The book is divided into three parts. Part I considers the type of interfaces to financial functions that can be created using the Microsoft Windows environment. In particular it deals with the use of Dynamic Link Libraries (DLLs) and ActiveX components from languages such as Visual Basic, VBScript, VB.NET, and C#. The author considers that one of the main developments in technical computing over the past ten years has been the emergence of technologies that permit the rapid development of easy to use interfaces to complex functions. At the mouse click of a virtual button complicated computations can be performed.

Part II of the book is concerned with the mathematics of option pricing, and covers computational methods for vanilla options and also simple barrier options. In many cases more exotic options (that for example include complex barriers, lockout periods, rebates, etc.) can be created from these by using them as building blocks. Most of this material can be understood using basic college mathematics and its presentation style is inspired by Numerical Recipies, for instance see Press et al. (1992).

Finally Part III of the book deals with financial econometrics and the modelling of volatility. Although the main emphasis is on GARCH, Levy processes, and stochastic volatility models are also considered.

From an historical point of view the finite-difference methods used in Part II have their origin in the numerical weather forecasting techniques proposed by Lewis Richardson between 1910 and 1930, see Richardson (1910) and Richardson and Gaunt (1927). These were later developed by Phyliss Nicolson (Girton College Cambridge) and John Crank in the 1940s, and their method is known as the
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Crank–Nicolson finite-difference method. GARCH time series methods can trace their roots to earlier work in the 1920s concerned with AR processes. We could continue by discussing the history of Gaussian processes, Levy distributions, etc. However, the reader can read about this elsewhere.

It should be mentioned that this is not a book about how to use and trade in various financial derivatives. In fact the author does not have this experience, and books such as John Hull are a good introduction to this subject.

I would like to take this opportunity to thank my wife Kathryn for putting up with the extra time that a book such as this requires.

I would also like to thank the series editor, Dr Steven Satchell, for his very useful advice concerning the structure of the book, and Mike Cash of Butterworth-Heinemann for his support throughout the project.

In addition I gratefully acknowledge the Risk Waters Group for allowing PDF versions of several journal articles to be placed on the CD ROM.

George Levy
Benson 2003
Part I

Using Numerical Software Components within Microsoft Windows
Chapter 1

Introduction

This part of the book describes a variety of Microsoft technologies that enable software developers to deploy their numerical/financial functions within Microsoft Windows. It would be impossible in such a short space to provide a comprehensive description of Microsoft Windows. One of the reasons is that Microsoft regularly brings to market new and improved products. For instance in 2002 Microsoft launched its release version of .NET; this had been previously available in the form of Beta 1 and Beta 2 releases. This product includes the languages VB.NET, an updated version of Visual Basic, and C#. The main purpose of .NET is to facilitate the easy deployment of Web Service component software over the Internet. Currently (October 2002) the full MSDN documentation and help system (with information on .NET) takes well over 1 Gbyte of computer disc space. Voluminous books have also been written on various aspects of .NET such as: VB.NET, C#, XML, and XSL, and these can be consulted as required. Here we can only aim at providing a short introduction to the use of Microsoft technology for numerical computation. In order to combat information overload we will try here to convey the maximum essential information in the minimum space. To achieve this we will adopt the strategy of supplying well commented code excerpts from real (working) Microsoft projects. It is intended that these code excerpts can be used as templates for the creation of computational finance components. Additional material, including documentation, complete source code and ready to use Microsoft projects can be found on the CD ROM which accompanies this book.

Before embarking on a more detailed description of various Microsoft languages and applications it would be sensible to try and gain an overview of the Microsoft Windows environment and consider the possible benefits to be gained from using it for software development.

To a large extent the Microsoft Windows environment is all about the Visual user-interface. The replacement of command line, DOS based, programming by Microsoft Windows heralded an explosion in the use of computers. Esoteric DOS commands (understood by only a few) gave way to the simple interactive user-interface. Here the user can control a program by (for example) clicking Windows buttons with the mouse and entering values into Windows textboxes. The enormous advantage of this approach (now used by nearly all computational software) is that the user is shielded from complicating factors such as the operating system and the underlying computer languages. All the user needs to do is to enter the correct data and click the appropriate button; the answer then appears on the screen.

Using Windows software can now be made as easy as turning on the television or playing a video player. However, as with the real button on the television or video
player remote control, the virtual button of a Windows application can conceal a great deal of underlying technology. The purpose of this part of the book is to provide information concerning the type of Windows software that may be invoked when a Windows event (such as a mouse click) occurs.

We will consider the ways in which numerical and financial components can be incorporated into various Windows applications. Here we take the term numerical and financial component to mean a self-contained computational object which, given certain inputs, will return various computed results. The inputs and computed results can be single values (scalars), one-dimensional arrays (vectors), two-dimensional arrays (matrices), or higher dimensional arrays. The components described here are designed to be used in mixed language applications. This means that the component is created using a computationally efficient language such as C/C++ or Fortran, and resides in either a Windows Dynamic Link Library (DLL) or COM ActiveX Control. It is then used from another (interface) language such as Visual Basic, which wraps it and provides the Visual interactive interface seen by the user. If the components are to be accessible from the complete range of Microsoft languages it is good programming practice to restrict their data types to the very basic C/C++ types such as real, double, and long (Fortran types REAL, DOUBLE PRECISION, and INTEGER) which have equivalents in all the other Microsoft languages. It should be noted that, in C++, seemingly innocent structures, strings and character parameters can be particularly difficult (if not impossible) to deal with.

The topics covered here include:

- DLL creation using Visual C++.
- Calling C and Fortran routines from Visual Basic, VB.NET, and C#.
- Using ActiveX and COM components from Visual Basic, Internet Web pages, Excel, and Delphi.
- Scripting ActiveX components on Internet Web pages using VBScript and JScript.
- XML and transformation using XSL.

The section on XML data representation and transformation was included because it provides an introduction to viewing data (or computed results) with the Web browser Internet Explorer 6. In Chapter 6 we show how the use of XSL style sheets permits an XML file to be transformed into a HTML file. This transformation can be accomplished automatically when the XML file is loaded into a Web browser (for example by double clicking the XML file with a mouse). By using different XSL files it is thus possible to obtain different views of the numeric values contained within an XML file. For example it may be considered appropriate to generate both a tabular view which gives columns of numeric values, and also a report view which contains fewer numbers and contains graphical plots that summarize the information.

Information is given on how to call components from Visual Basic, Delphi, VB.NET, and C#. In addition we show how numeric components can be used from within Windows applications such as Excel and Internet Explorer.

As previously mentioned we will not consider in any detail the construction of the Visual interface; this information can be readily found in the large selection of Microsoft Windows books that are currently available. We will also concentrate on
the mixed language use of numeric components. This means that although all the examples in this part of the book could have been written in Visual C++, they use a variety of Windows languages such as Visual Basic, VBScript, Delphi, etc.

In practical terms this means that the creation of a computational finance application is a two-step process:

- The creation of the numerical/finance component, using a computationally efficient language such as Visual C++ or Visual Fortran.
- The construction of the application framework and user-interface using Microsoft languages such as Visual Basic, VB.NET, C#, etc.

This separation leads to a natural division of labour. The numerical components are created by an expert mathematician/numerical analyst (with limited knowledge of languages such as C++, Visual Basic, etc.) and the construction of the Visual interface is performed by a computer programmer (with limited numerical knowledge) who is expert in the more complex features of the language chosen for developing the application’s visual interface. For example a numerical analyst may create an option pricing component using Visual C++. A computer programmer may then incorporate this component into a variety of applications such as: Web-based services using VB.NET or C#, spreadsheet applications using Excel, or stand-alone PC applications using Visual Basic, Delphi, etc.

Finally here are just a few remarks concerning the style of the book.

Small example applications have been included in the areas of statistics, linear algebra, financial derivative pricing, portfolio optimisation, and numerical optimisation.

Also some of the examples refer to the NAG C library DLL and also the NAG Fortran DLL. However, the techniques used in these examples can easily be applied to calling functions from other, user-defined, Windows DLLs.

Care has been taken to make all the computer code as simple as possible. We don’t (intentionally) try to be clever; the main consideration is that the code works. Readers can always modify the code to suit their needs and preferences.

Finally some people may find the style rather terse compared to the coverage given in other books. This is intentional, since there is so much the information presented will be limited to the minimum required to obtain working software. The book has been written from the author’s experience that:

* A page of working (and well commented) computer code is worth a hundred pages of explanation.

In spite of all these caveats it is hoped the reader will find the information in the following sections both instructional and useful reference material.
Chapter 2

Dynamic Link Libraries (DLLs)

The Microsoft Windows environment is constructed so that virtually all applications make calls to Dynamic Link Libraries (DLLs). These DLLs may contain system library routines provided by a particular computer vendor or may be customized third party DLLs which provide specialized functions (e.g. graphical, mathematical, text processing, etc.). Section 2.1 shows how DLLs written in C/C++ can be used from Visual Basic and VBA. Sections 2.2 and 2.3 provide information on calling C/C++ DLLs and Fortran DLLs from VB.NET and C# respectively. As previously mentioned the examples given will mainly focus on the use of the NAG C Library DLL and the NAG Fortran Library DLL. However, the information provided is quite general and can be used to interface to a C or Fortran DLL.

2.1 VISUAL BASIC AND EXCEL VBA

The aim of this section is to provide a brief overview of how to use Dynamic Link Library functions from Visual Basic 6 (and earlier), and also from Visual Basic for applications, that is VBA within Microsoft Excel, etc. More detailed information on this subject can be found in Levy (1998), and we will refer to the NAG C library functions mentioned in that report.

2.1.1 Visual Basic types

First we need to consider the Visual Basic data types required to match those that occur in the routine argument lists of a (32-bit) C/C++ DLL. A brief summary of the fundamental types is given in Table 2.1.

In Visual Basic all enumeration variables corresponding to enumeration variables within a C DLL should be declared as type Long. Furthermore a Visual Basic

| Table 2.1 Correspondence between Visual Basic types and those of Fortran 77 and C |
|---------------------------------|-----------------|------------------|
| Visual Basic                   | C               | Fortran 77        | Size in bytes |
| Byte or String*1               | char            | CHARACTER*1       | 1             |
| Long                            | long            | INTEGER           | 4             |
| Long                            | int             | LOGICAL           | 4             |
| Single                          | float           | REAL              | 4             |
| Double                          | double          | DOUBLE PRECISION  | 8             |
variable of type long which has the value 0 is taken as FALSE by the C DLL function; if it has the value 1 it is TRUE.

We now illustrate this in C Code excerpt 2.1 below.

Code excerpt 2.1  C code containing enumeration types

By default, enumerators in a given C enumeration type declaration start at zero and increase by 1 as the declaration is read from left to right. However, if a given enumerator is assigned a value then subsequent enumerators continue the progression from the assigned value. The Visual Basic code corresponding to Code excerpt 2.1 is given below.

Code excerpt 2.2  The Visual Basic corresponding to Code excerpt 2.1

If a C DLL function contains a structure in its parameter list then it is necessary to declare the equivalent structure in Visual Basic. We will now illustrate this with the NAG C library error structure (of type NagError) which is used in nearly all of the NAG C library functions. The definition of this type is given in Code excerpt 2.3.

Code excerpt 2.3  The declaration of the type NagError, used in the NAG C library

The corresponding Visual Basic user-defined type (UDT) is given in Code excerpt 2.4; it can be seen that the pointer to the handler function has been replaced by a structure member of type Long.
8 Using Numerical Software Components within Microsoft Windows

Type NagErrorType
  code As Long
  printm As Long
  Message(511) As String *1
  handler As Long
  errnum As Long
End Type

Code excerpt 2.4 The Visual Basic declaration of a UDT corresponding to NagError in Code excerpt 2.3

2.1.2 Function declarations

The C DLL routines are declared in Visual Basic by using the following syntax:

Declare Function "name" Lib "library name" Alias "decorated name"(arguments)_
As return type

for a C function which returns a value (of type double or long), and

Declare Sub "name" Lib "library name" Alias "decorated name"(arguments)

for a C function which returns void. The ‘decorated name’ is generated from the DLL routine name using the following convention. An underscore (_ ) is prefixed to the routine name. The name is followed by the at-sign (@) character, followed by the number of bytes in the argument list. For instance the NAG C library DLL routines g01aac and f02wec have the function prototypes

```
#define DllExport __declspec(dllexport)
extern DllExport void __stdcall g01aac(long n, double x[], double wt[], long *nvalid, double *xmean, double *xsd, double *xkew, double *xskurt, double *xmin, double *xmax, double *wsum, NagError *iflag);

extern DllExport void __stdcall f02wec(long m, long n, double *a, long tda, long ncolb, double *b, long tdb, Boolean wantq, double *q, long tdq, double *sv, Boolean wantp, double *pt, long tdpt, long *iter, double *e, long *info, NagError *iflag);
```

require Visual Basic declaration statements of the form

```
Declare Sub g01aac Lib "nagcd.dll" Alias "_g01aac@48" (ByVal n As Long, _
  x As Double, wt As double, nvalid As Long, xmean As Double, _
  xsd As Double, xkew As Double, xskurt As Double, xmin As Double, _
  xmax As Double, wsum As Double, iflag As NagErrorType)

Declare Sub f02wec Lib "nagcd.dll" Alias "_f02wec@72" (ByVal m As Long, _
  a As Double, ByVal tda As Long, ByVal ncolb As Long, _
  b As Double, ByVal tdb As Long, ByVal wantq As Long, q As Double, _
  ByVal tdq As Long, sv As Double, ByVal wantp As Long, pt As Double, _
  ByVal tdpt As Long, iter As Long, e As Double, info As Long, iflag As NagErrorType)
```

Code excerpt 2.5 The Visual Basic declaration statements for the NAG C library functions g01aac and f02wec

In C, pointers are used to pass arguments by reference (e.g. double *xsd, long *nvalid, double x[], etc.); here the notation [] is used to denote an array argument. When arguments are passed by value in C the syntax type variable name (e.g. long n, double x, etc.) is used. In Visual Basic, by default, all arguments are passed by reference; the keyword ByVal is required to pass an argument by value. In C
all pointers are 4 bytes long. This means that the function g01aac, in which the first parameter (of type Long) is passed by value and the remaining parameters are passed by reference, has a total byte count of 48 bytes; giving rise to the decorated name ‘_g01aac@48’. In a similar manner the function f02wec, which has 18 parameters of 4 bytes each, has a total byte count of 72 and the decorated name ‘_f02wec@72’.

2.1.3 Null pointers

Many C routines make use of null pointers to indicate that an argument is to be ignored and default action is to be taken. For example the NAG C library routine g01aac has a pointer argument wt which allows the routine to perform statistical computations involving weighted data. If this argument is set to the null pointer then unweighted calculations are performed; all the weights are assumed to be 1. In Visual Basic this can be accomplished by declaring g01aac as shown in Code excerpt 2.6, where the declaration wt As Long (instead of wt As double) has been used to allow this argument to be used as a pointer.

Declare Sub g01aac Lib "nagcd.dll" Alias "_g01aac@48" (ByVal n As Long, _
    x As Double, ByVal wt As Long, nvalid As Long, xmean As Double, _
    xsd As Double, xskew As Double, xkurt As Double, xmin As Double, _
    xmax As Double, wsum As Double, iflag As NagErrorType)

Code excerpt 2.6 A Visual Basic declaration statement which allows a null pointer to be used for the parameter wt in the NAG C library function g01aac

The routine calls

Call g01aac(n, x(0), ByVal 0&, nvalid, xmean, xsd, xskew, xkurt, xmin, xmax, wsum, iflag)

and

Call g01aac(n, x(0), 0, nvalid, xmean, xsd, xskew, xkurt, xmin, xmax, wsum, iflag)

are now both valid and result in unweighted calculations being performed.

2.1.4 Function parameters

In contrast to C, Visual Basic procedures are not allowed to have function arguments. This limitation creates a problem when using and declaring DLL routines that require function parameters such as the objective function for numerical optimization routines. A solution to this problem is the creation of an auxiliary DLL to provide a convenient interface wrapper for both the objective function and optimization routine. Another way around this problem is to use ActiveX COM components as illustrated in Chapters 5 and 6.

2.1.5 Two-dimensional array parameters and storage order

In Visual Basic care must be taken when using one- and two-dimensional arrays. This is because the array indices start at zero (unless Option Base is used to define a
dim a(5) ' declares an array which holds 6 elements: a(0),...a(6)
dim b(3,2) ' declares an array which holds 9 elements:
  b(0,0) b(0,1) b(0,2) b(1,0) b(1,1) b(1,2)
  b(2,0) b(2,1) b(2,2) b(3,0) b(3,1) b(3,2)

The leading dimension of array b is therefore 4, and the trailing dimension of b is 3.
In Visual Basic multidimensional arrays are stored by columns (as in Fortran) rather than stored by rows, which is the C convention.  (Note: In Sections 2.2 and 2.3 we will see that this situation has now been reversed in VB.NET and C# which both store multidimensional arrays using the C convention; that is by rows.) This means that care must be taken when a DLL routine has matrix (two-dimensional array) parameters. For example, assume that a 3 by 2 matrix

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<tbody>
<tr>
<td>11</td>
<td>12</td>
</tr>
<tr>
<td>21</td>
<td>22</td>
</tr>
<tr>
<td>31</td>
<td>32</td>
</tr>
</tbody>
</table>

is stored in a Visual Basic two-dimensional array a in the natural manner, as in the following code fragment.

```
dim a(2, 1) as double
a(0, 0) = 11
a(1, 0) = 21
a(2, 0) = 31
a(0, 1) = 12
a(1, 1) = 22
a(2, 1) = 32
```

The array a consists of 6 elements stored in column order, as follows:

<p>| | | | |</p>
<table>
<thead>
<tr>
<th></th>
<th></th>
<th></th>
<th></th>
</tr>
</thead>
<tbody>
<tr>
<td>11</td>
<td>21</td>
<td>31</td>
<td>12 22 32</td>
</tr>
</tbody>
</table>

However, routines in a C DLL follow the convention that two-dimensional arrays are stored in row order. Suppose the array a were passed to a C routine (for instance the NAG C library DLL routine f02wec, as in the SVD example in Section 2.1.6)

```
call f02wec(3, 2, a(0, 0), ...)
```

where the first two arguments specify the number of rows and columns in the matrix.
The routine would treat the array as representing a 3 by 2 matrix stored in row order

<p>| | |</p>
<table>
<thead>
<tr>
<th></th>
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</thead>
<tbody>
<tr>
<td>11</td>
<td>21</td>
</tr>
<tr>
<td>31</td>
<td>12</td>
</tr>
<tr>
<td>22</td>
<td>32</td>
</tr>
</tbody>
</table>
which is not the intended matrix $A$. One solution (which is used in Sections 2.1.6 and 2.1.7) is to store the matrix in a one-dimensional array $a_1$, with the element $a_1(i,j)$ stored in $a_1((i-1)*\text{tda} + j - 1)$, where tda is the trailing dimension of the matrix (in this case 2).

```
Dim a1(5) As Double
Dim tda As Long
tda = 2
a1(0) = 11
a1(1) = 12
a1(2) = 21
a1(3) = 22
a1(4) = 31
a1(5) = 32
Call f02wec(3, 2, a1(0), tda,...)
```

Another solution is to store the transpose of the matrix $A$ in a two-dimensional array $a_t$, with tda now being the leading dimension of the array $a_t$.

```
Dim at(1, 2) As Double
Dim tda As Long
tda = 3
at(0, 0) = 11
at(0, 1) = 21
at(0, 2) = 31
at(1, 0) = 12
at(1, 1) = 22
at(1, 2) = 32
Call f02wec(3, 2, at(0, 0), tda,...)
```

The Visual Basic array $a_t$ can be larger than is needed to store the 2 by 3 matrix $A^T$; in order that the C routine accesses the correct array elements it is essential that tda is set to the correct value.

```
Dim at(3, 5) As Double
Dim tda As Long
Call f02wec(3, 2, at(0, 0), tda,...)
```

**2.1.6 Singular value decomposition example**

In this example we use the NAG C Library DLL $f02wec$ routine to perform a singular value decomposition (SVD) within Visual Basic.

Briefly the SVD of a matrix consists of the following factorization:

$$A = U\Sigma V^T$$

where $A$ is the original matrix, $\Sigma$ is the diagonal matrix of singular values, $U$ is the matrix containing the left hand singular vectors, and $V$ is the matrix containing the right hand singular vectors. The information obtained from an SVD (see G Golub) can be very valuable and, for example, can be used to perform principal component analysis or least squares regression; both of which have important applications in computational finance.

Here we give the DLL function two Visual Basic function declarations. The function $f02wec\_full$ is used to calculate both the singular values and also the left and right
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Singular vectors of a matrix \( A \); it also requires the declaration of `dummy` arrays for certain array arguments that are not referenced. The function \( f02wec\_ptr \) is used to calculate the singular values only, and contains the declarations \( ByVal q \) As Long, \( ByVal b \) As Long and \( ByVal pt \) As Long within its argument list. This enables assignment of null pointers to these arguments and thus avoids the use of `dummy` array arguments when the routine is called. The example also illustrates how data is assigned to the input array \( a \); note \( tda \) is the second (trailing) dimension of the matrix \( A \).

The following two Visual Basic declaration statements for the C library function \( f02wec \) are used:

```vbnet
Declare Sub f02wec_full Lib "nagcd.dll" Alias "_f02wec@72"(ByVal m As Long, ByVal n As Long, ByVal a As Double, ByVal tda As Long, ByVal ncolb As Long, ByVal b As Double, ByVal tdb As Long, ByVal wantq As Long, ByVal q As Double, ByVal tdq As Long, ByVal sv As Double, ByVal wantp As Long, ByVal pt As Long, ByVal tdpt As Long, ByVal iter As Long, ByVal e As Double, ByVal info As Long, ByVal iflag As NagErrorType)
which requires all the parameters to be supplied, and

Declare Sub f02wec_ptr Lib "nagcd.dll" Alias "_f02wec@72" (ByVal m As Long, ByVal n As Long, ByVal a As Double, ByVal tda As Long, ByVal ncolb As Long, ByVal b As Double, ByVal tdb As Long, ByVal wantq As Long, ByVal q As Double, ByVal tdq As Long, ByVal sv As Double, ByVal wantp As Long, ByVal pt As Long, ByVal tdpt As Long, ByVal iter As Long, ByVal e As Double, ByVal info As Long, ByVal iflag As NagErrorType)
which allows the use of null pointers for the parameters \( b, q, \) and \( pt \).```

Code excerpt 2.7 Illustrating the use of null pointers within Visual Basic when calling the singular value decomposition function \( f02wec \) from the NAG C library
We now give an example of calling a numerical optimization function from Visual Basic.

2.1.7 Numerical optimization example

Since many financial problems involve some form of optimal decision process it is useful to show how to call numerical optimization software from Visual Basic. We will illustrate this by showing how to use the NAG C Library DLL function e04nfc; full documentation at the NAG website, http://www.nag.co.uk. The example we will consider here is taken from Bunch and Kaufman (1980) and consists of estimating the vector $x$ that will minimize the quadratic function:

$$f(x) = c^T x + \frac{1}{2} x^T H x$$

where:

$c = (7.0, 6.0, 5.0, 4.0, 3.0, 2.0, 1.0, 0.0)^T$ and

$$H = \begin{pmatrix}
1.69 & 1.00 & 2.00 & 3.00 & 4.00 & 5.00 & 6.00 & 7.00 \\
1.00 & 1.69 & 1.00 & 2.00 & 3.00 & 4.00 & 5.00 & 6.00 \\
2.00 & 1.00 & 1.69 & 1.00 & 2.00 & 3.00 & 4.00 & 5.00 \\
3.00 & 2.00 & 1.00 & 1.69 & 1.00 & 2.00 & 3.00 & 4.00 \\
4.00 & 3.00 & 2.00 & 1.00 & 1.69 & 1.00 & 2.00 & 3.00 \\
5.00 & 4.00 & 3.00 & 2.00 & 1.00 & 1.69 & 1.00 & 2.00 \\
6.00 & 5.00 & 4.00 & 3.00 & 2.00 & 1.00 & 1.69 & 1.00 \\
7.00 & 6.00 & 5.00 & 4.00 & 3.00 & 2.00 & 1.00 & 1.69 \\
\end{pmatrix}
$$

subject to the bounds:

- $-1.0 \leq x_1 \leq 1.0$
- $-2.1 \leq x_2 \leq 2.0$
- $-3.2 \leq x_3 \leq 3.0$
- $-4.3 \leq x_4 \leq 4.0$
- $-5.4 \leq x_5 \leq 5.0$
- $-6.5 \leq x_6 \leq 6.0$
- $-7.6 \leq x_7 \leq 7.0$
- $-8.7 \leq x_8 \leq 8.0$

and the general constraints:

- $-x_1 + x_2 \geq -1.00$
- $-x_2 + x_3 \geq -1.05$
- $-x_3 + x_4 \geq -1.10$
- $-x_4 + x_5 \geq -1.15$
- $-x_5 + x_6 \geq -1.20$
- $-x_6 + x_7 \geq -1.25$
- $-x_7 + x_8 \geq -1.30$
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The initial point is taken as:

\[ x_0 = (-1.0, -2.0, -3.0, -4.0, -5.0, -6.0, -7.0, -8.0)^T \]

An example of using the function e04nfc from Visual Basic is given in Code excerpt 2.8 below.

```vbnet
Static n, i, j As Long
Static tda As Long
Static tdh As Long
Static x() As Double
Static a() As Double
Static h() As Double
Static cvec() As Double
Static bl() As Double
Static bu() As Double
Static nclin As Long
Static ncnlin As Long
Static options_ptr As Long
Dim iflag As NagErrorType

n = 8 ' Set the number of variables
nclin = 7 ' Set the number of linear constraints
tda = n
tdh = n
ReDim a(nclin * n)
ReDim tda(30)
ReDim x(n)
ReDim h(100)

For i = 0 To nclin - 1 ' Loop on the number of linear constraints
    For j = 0 To n - 1 ' Loop on the number of variables
        a(i * tda + j) = 0# ' Initialise the array a
    Next j
Next i

For i = 0 To nclin - 1 ' Loop on the number of linear constraints
    a(i * tda + i) = -1# ' Set the elements of the constraint matrix a
    b(i * n + i) = -1# - 0.05 * i ' Set the lower bound of the linear constraint
    bu(i * n + i) = 1E+20 ' Set the upper bound of the linear constraint
Next i

For j = 0 To n - 1 ' Loop on the number of variables
    bl(j) = -(j + 1) - 0.1 * j ' Set the lower bounds
    bu(j) = j + 1 ' Set the upper bounds
    cvec(j) = 7 - j ' Set the elements of the vector c
Next j

For i = 0 To nclin - 1 ' Loop on the number of linear constraints
    For j = i + 1 To n - 1 ' Set the upper bounds
        h(i * tda + j) = Abs(i - j)
Next j

MsgBox '' optimum objf = '' & objf ' Output (to the screen) the value of the objective function
MsgBox ' The solution is:'
```
For $i = 0$ to 7 ' Output (to the screen) the individual elements of the solution vector

MsgBox "x(' & i & ' ) = " & x(i)

Next i
End Sub

Code excerpt 2.8 Illustrating how the NAG C library numerical optimization function e04nfc can be used from Visual Basic.

The optimization routine e04nfc is designed to output useful information to the file e04nfce.r. This file contains information such as the optimization settings, and also the Lagrange multipliers and value of the objective function and solution vector $x^*$ at the computed minimum. It can be seen from Exhibit 2.1 that, at the computed minimum, the solution vector is

$$x^* = (-1.0, -2.0, -3.05, -4.15, -5.30, 6.0, 7.0, 8.0)^T$$

and the value of objective function is $-621.4878$.

Parameters to e04nfc

<table>
<thead>
<tr>
<th>Linear constraints</th>
<th>7</th>
</tr>
</thead>
<tbody>
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<tr>
<td>ftol</td>
<td>1.05e-008</td>
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<tr>
<td>rank_tol</td>
<td>1.11e-014</td>
</tr>
<tr>
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<td>Nag_Cold</td>
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<tr>
<td>crash_tol</td>
<td>1.00e+-002</td>
</tr>
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<td>max_df</td>
<td>8</td>
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</tr>
<tr>
<td>ftol</td>
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<tr>
<td>max_df</td>
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<td>max_df</td>
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Results from e04nfc:

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<th>Ninf</th>
<th>Sinf/Obj</th>
<th>Bnd</th>
<th>Lin</th>
<th>Nart</th>
<th>Nrz</th>
<th>Norm Gz</th>
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<td>0.0e+000</td>
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<td>2.3550e+001</td>
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<td>3</td>
<td>0</td>
<td>1.73e+000</td>
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<tr>
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<td>2</td>
<td>U</td>
<td>10 L</td>
<td>4.0e+000</td>
<td>2</td>
<td>1.9600e+001</td>
<td>4</td>
<td>1</td>
<td>3</td>
<td>0</td>
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<tr>
<td>2</td>
<td>4</td>
<td>U</td>
<td>12 L</td>
<td>7.8e+000</td>
<td>1</td>
<td>1.1750e+001</td>
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<td>2</td>
<td>3</td>
<td>0</td>
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<tr>
<td>3</td>
<td>6</td>
<td>U</td>
<td>14 L</td>
<td>1.2e+001</td>
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<td>0.0000e+000</td>
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<td>3</td>
<td>3</td>
<td>0</td>
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Itn 3 -- Feasible point found.

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<th>Jadd</th>
<th>Step</th>
<th>Ninf</th>
<th>Sinf/Obj</th>
<th>Bnd</th>
<th>Lin</th>
<th>Nart</th>
<th>Nrz</th>
<th>Norm Gz</th>
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</thead>
<tbody>
<tr>
<td>3</td>
<td>0</td>
<td>0</td>
<td>0.0e+000</td>
<td>0</td>
<td>8.6653e+002</td>
<td>2</td>
<td>3</td>
<td>2</td>
<td>1</td>
<td>1.52e+002</td>
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<tr>
<td>4</td>
<td>0</td>
<td>9 L</td>
<td>1.0e-001</td>
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<td>4.9824e+001</td>
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<td>5</td>
<td>2</td>
<td>A</td>
<td>11 L</td>
<td>4.5e-001</td>
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<td>-5.6227e+002</td>
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<td>5</td>
<td>1</td>
<td>0</td>
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<tr>
<td>6</td>
<td>1</td>
<td>A</td>
<td>6 U</td>
<td>6.0e-011</td>
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<td>5</td>
<td>0</td>
<td>0</td>
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<tr>
<td>7</td>
<td>14 L</td>
<td>7 U</td>
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<td>4</td>
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</table>

Final solution:

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<tr>
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<th>Value</th>
<th>Lower Bound</th>
<th>Upper Bound</th>
<th>Lagr Mult</th>
<th>Residual</th>
</tr>
</thead>
<tbody>
<tr>
<td>V 1</td>
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<td>-1.00000e+000</td>
<td>-1.00000e+000</td>
<td>1.00000e+000</td>
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<td>0.00e+000</td>
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<tr>
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<tr>
<td>V 3</td>
<td>FR</td>
<td>-3.05000e+000</td>
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<td>0.00000e+000</td>
<td>1.50000e-001</td>
</tr>
<tr>
<td>V 4</td>
<td>F</td>
<td>-4.15000e+000</td>
<td>-4.30000e+000</td>
<td>4.00000e+000</td>
<td>0.00000e+000</td>
<td>1.50000e-001</td>
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<tr>
<td>V 5</td>
<td>F</td>
<td>-5.30000e+000</td>
<td>-5.40000e+000</td>
<td>5.00000e+000</td>
<td>0.00000e+000</td>
<td>1.00000e-001</td>
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<tr>
<td>V 6</td>
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<td>6.00000e+000</td>
<td>-6.50000e+000</td>
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</table>
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<table>
<thead>
<tr>
<th>LCon</th>
<th>State</th>
<th>Value</th>
<th>Lower Bound</th>
<th>Upper Bound</th>
<th>Lagr Mult</th>
<th>Residual</th>
</tr>
</thead>
<tbody>
<tr>
<td>L</td>
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<td>LL</td>
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<td>-1.00000e+00</td>
<td>None</td>
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<td>None</td>
<td>6.443e+001</td>
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<td>None</td>
<td>1.779e+001</td>
</tr>
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<td>None</td>
<td>0.000e+000</td>
</tr>
<tr>
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<td>5</td>
<td>FR</td>
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<td>-1.20000e+000</td>
<td>None</td>
<td>0.000e+000</td>
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<tr>
<td>L</td>
<td>6</td>
<td>FR</td>
<td>1.00000e+000</td>
<td>-1.25000e+000</td>
<td>None</td>
<td>0.000e+000</td>
</tr>
<tr>
<td>L</td>
<td>7</td>
<td>FR</td>
<td>1.00000e+000</td>
<td>-1.30000e+000</td>
<td>None</td>
<td>0.000e+000</td>
</tr>
</tbody>
</table>

Exit after 7 iterations.
Optimal QP solution found.
Final QP objective value = -6.2148783e+002

Exhibit 2.1 The file e04nfce.r which contains information concerning the numerical optimization performed by e04nfc

2.2 VB.NET

In this section we will give details of how to call (and use) Fortran DLLs and C DLLs from VB.NET. Since VB.NET is very similar to Visual Basic, many of the concepts have already been dealt with in Section 2.1. We will therefore concentrate on the important differences (from a numerical view) between Visual Basic Version 6 and VB.NET (also known as Visual Basic Version 7). A brief summary of some important types in VB.NET is displayed in Table 2.2.

We will now list, and briefly comment on, the main differences between VB.NET and Visual Basic. They are as follows:

- All array indices must start at zero. This means that Option Base 0 1, which was available in Visual Basic (Version 6) can no longer be used in VB.NET. Also the number of elements in an array is the same as in Visual Basic. If an array called mya is declared using Dim mya(8) it will have 9 elements; these are mya(0), ..., mya(8). (Note: In VB.NET Beta 1, this was not so. If mya was declared using Dim mya(8) would contain the 8 elements mya(0), ..., mya(7).) This means that the number of elements in VB.NET multi-dimensional arrays (such as matrices) is the same as that described in Section 2.1 for Visual Basic.
- Multidimensional arrays are stored in row order, rather than column order as was the case for Visual Basic. This means that it is now necessary to perform a transpose operation when using passing matrices to Fortran DLL functions; where matrices are stored in column order.
- It is not possible to alter the number of dimensions of an array by using a ReDim statement. For example this means that the two-dimension array mya

| Table 2.2 Correlation between VB.NET types and those of Fortran 77 and C |
|-------------------------|----------------|----------------|-------------|
| VB.NET      | C            | Fortran 77    | Size in bytes |
| Integer      | long         | INTEGER       | 4           |
| Integer      | int          | LOGICAL       | 4           |
| Single       | float        | REAL          | 4           |
| Double       | double       | DOUBLE PRECISION | 8          |
must be declared as `Dim mya(,)` before it can be allocated using a statement such as `ReDim mya(4,3)`. In Visual Basic it was possible to declare the array as `Dim mya( )`, and then allocate it as a two-dimensional array using `ReDim mya(4,3)`.

- In contrast to Visual Basic, VB.NET and C# do not allow fixed length strings to be declared within UDTs (that is structures). This means that it is no longer possible to define a type that corresponds to the NAG C Library type `NagErrorType`, see Section 2.1.1. We will show two ways around this problem. The first is to use a null pointer argument; see Code excerpt 2.9. The second method is to wrap the original C DLL function within another C function which uses an integer parameter to flag errors; see Code excerpts 2.11 and 2.13.

- The type corresponding to a 4 byte integer is `Integer`; in Visual Basic the type was `Long`.

We will now illustrate these language features by describing the computer code contained within a VB.NET project which performs some numerical computations. The example used here is a VB.NET project (Figure 2.1) that computes the singular value decomposition of a given matrix by calling either the NAG Fortran Library DLL function F02WEF, or the NAG C Library DLL function f02wec. The visual user-interface of this project is very simple, and is similar to that shown in Figure 2.2 for the C# example application of Section 2.3. We use radio button controls to choose between the NAG Fortran Library and the NAG C Library.

![Figure 2.1 A view of the VB.NET example project corresponding to Code excerpt 2.9](image-url)
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Figure 2.2  Running the C# example application

```vbnet
Public Class Form1
    Inherits System.Windows.Forms.Form
    Public use_c_dll As Boolean

    #Region "Windows Form Designer generated code"
    #End Region

    Declare Sub F02WEF Lib "NAGF02.DLL" (ByRef m As Integer, ByRef n As Integer,_
    ByRef a As Double, ByRef dim_a As Integer, ByRef ncolb As Integer,_
    ByRef b As Double, ByRef dim_b As Integer, ByRef wantq As Integer,_
    ByRef q As Double, ByRef dim_q As Integer,_
    ByRef sv As Double, ByRef wantp As Integer,_
    ByRef pt As Double, ByRef dim_pt As Integer, ByRef work As Double,_
    ByRef ifail As Integer)

    Declare Sub f02wec Lib "nagc.dll" (ByVal m As Integer, ByVal n As Integer,_
    ByRef a As Double, ByVal dim_a As Integer, ByVal ncolb As Integer,_
    ByVal b As Double, ByVal dim_b As Integer, ByVal wantq As Integer,_
    ByVal q As Double, ByVal dim_q As Integer,_
    ByVal sv As Double, ByVal wantp As Integer,_
    ByVal pt As Double, ByVal dim_pt As Integer, ByVal work As Double,_
    ByVal iflag As Integer)

    Private Sub Button1_Click(ByVal sender As System.Object, ByVal e As System.EventArgs) Handles Button1.Click

        Dim a1(), q1(), sv(), pt1(), b1(), work() As Double
        Dim dim_a, dim_q, dim_b, dim_pt, m, n As Integer
        Dim i, j, ncolb, iflag, wantp, wantq, lwork As Integer
        Dim iter, info As Integer
        Dim evec() As Double
        Dim out_text As String

        FileOpen(2, "c:\BOOK_2002\SVD_DATA.TXT", OpenMode.Input) ' Open the input data file
        If use_c_dll Then ' Open the C DLL results file
            FileOpen(1, "c:\BOOK_2002\VBF02WEC_RESULTS.txt", OpenMode.Output)
            PrintLine(1, "VB.NET Singular Value Decomposition Example: Using C DLL function f02wec")
        Else ' Open the Fortran DLL results file
```
Dynamic Link Libraries (DLLs)

FileOpen(1, "c:\BOOK_2002\VB902WEF_RESULTS.txt", OpenMode.Output)
PrintLine(1, "VB.NET Singular Value Decomposition Example: Using Fortran function F02WEF")
End If
' This example performs a singular value decomposition of an m x n matrix A using either the
' NAG Fortran function F02WEF or the NAG C function f02wec.
' The singular values, left singular vectors and right singular vectors are output to a file.
' When the C function f02wec is used the left hand singular vectors are output in the array a1,
' and the transpose of the right hand singular vectors are output in the array pt.
' When the Fortran function F02WEF is used the transpose of left hand singular vectors are output in the
' array a2, and the right hand singular vectors are output in the array pt.
Input(2, m) ' Input the number of rows of matrix A
Input(2, n) ' Input the number of columns of matrix A
Note: The code assumes that m > n
PrintLine(1)
PrintLine(1)
ReDim sv(n) ' Allocate an array to hold the singular values
If use_c_dll Then ' Using the C DLL so storage is in row order
ReDim a1(m, n) ' Allocate the m x n matrix a1 to represent matrix A
ReDim pt(n, n)
dim_a = UBound(a1, 2) + 1 ' Set the trailing (second) dimension of matrix a1
dim_pt = n + 1 ' Set the dimension of pt
Else ' Using the Fortran DLL so storage is in column order
ReDim a1(n, m) ' Allocate the n x n matrix a1, to represent the transpose of matrix A
ReDim pt(n, n)
dim_a = UBound(a1, 2) + 1 ' This is the trailing (second) dimension of matrix A (the transpose of
' matrix A)
dim_pt = n + 1 ' Set the dimension of pt
End If
dim_b = 1
dim_q = 1
wantq = 1 ' Set WANTQ to TRUE
wantp = 1 ' Set WANTP to TRUE
ncolb = 0 ' Don't want to include a B matrix
For i = 0 To m - 1 ' Loop on the row index
For j = 0 To n - 1 ' Loop on the column index
If use_c_dll Then
Input(2, a1(i, j)) ' Input matrix A
Else
Input(2, a1(j, i)) ' Input the transpose of a1
End If
Next j
Next i
If use_c_dll Then ' Use the NAG C DLL
iflag = 0
ReDim evect(n)
Call f02wec(m, n, a1(0, 0), dim_a, ncolb, b(0), dim_b, wantq, q(0),_dim_q, sv(0), wantp, pt(0, 0), dim_pt, iter, evect(0), info, iflag)
Else ' Use the NAG Fortran DLL
lwork = n * n + 4 * (n - 1)
ReDim work(lwork) ' Allocate the required workspace array
Call F02WEF(m, n, a1(0, 0), dim_a, ncolb, b(0), dim_b, wantq, q(0),_dim_q, sv(0), wantp, pt(0, 0), dim_pt, work(0), iflag)
End If
out_text = "The singular values are:"
i = 0
Do While i < n ' Form a string containing all the singular values
out_text = out_text & " & sv(i)
i = i + 1
Loop
PrintLine(1, out_text) ' Output the text
PrintLine(1) ' Output a blank line
PrintLine(1)
PrintLine(1, "Left-hand singular vectors:")
PrintLine(1)
For i = 0 To m - 1 ' Loop on the row index
out_text = ""
For j = 0 To n - 1 ' Loop on the column index
If use_c_dll Then
out_text = out_text & " & a1(i, j) ' Output a1
Else
out_text = out_text & " & a1(j, i) ' Output the transpose of a1
End If
Next j
Next i
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```vbnet
PrintLine(1)
PrintLine(1, "Right-hand singular vectors: ")
PrintLine(1)
For i = 0 To n - 1 ' Loop on the row index
  out_text = ""
  For j = 0 To n - 1 ' Loop on the column index
    If use_c_dll Then
      out_text = out_text & "" & pt(j, i) ' Output the transpose of pt
    Else
      out_text = out_text & "" & pt(i, j) ' Output pt
    End If
  Next j
  PrintLine(1, out_text) ' Output the text
Next i
FileClose(2)
FileClose(1)
End Sub

Private Sub RadioButton1_CheckedChanged(ByVal sender As System.Object, ByVal e As System.EventArgs)
Handles RadioButton1.CheckedChanged
  use_c_dll = False ' Set the flag which indicates use of the NAG Fortran DLL
End Sub

Private Sub RadioButton2_CheckedChanged(ByVal sender As System.Object, ByVal e As System.EventArgs)
Handles RadioButton2.CheckedChanged
  use_c_dll = True ' Set the flag to indicate use of the NAG C DLL
End Sub
End Class
```

**Code excerpt 2.9** A VB.NET program that computes the singular value decomposition by calling either the NAG Fortran DLL function F02WEF, or the NAG C library DLL function f02wec. The call to f02wec uses a null pointer for the last parameter, which in C is declared as NagErrorType*

In Code excerpt 2.9 we demonstrate the declaration of two-dimensional VB.NET arrays, and their use by Fortran and C DLLs. When a C DLL is used the arrays a1 and pt are declared with the statements ReDim a1(m, n) and ReDim pt(m, n); if a Fortran DLL is called we declare the transposed arrays using ReDim a1(n, m) and ReDim pt(n, m). The C DLL function f02wec is passed the second (or trailing) dimension of the arrays a1 and pt, which is n + 1 for both arrays. The Fortran DLL function F02WEF is passed the first (or leading) dimension of the transposed arrays a1 and pt, which again is n + 1 for both arrays.

Exhibit 2.2 The data file used by the example VB.NET code in Code excerpt 2.9

```
5, 3
2.0, 2.5, 2.5
2.0, 2.5, 2.5
1.6, -0.4, 2.8
2.0, -0.5, 0.5
1.2, -0.3, -2.9
```

**VB.NET Singular Value Decomposition Example: Using Fortran DLL function F02WEF**

The singular values are: 6.56155281280883 2.43844718719117

Left-hand singular vectors:

-0.6011367037189 0.196116135138183 -0.316501382272552
-0.60113670371891 0.196116135138184 -0.316501382272552
0.416640015914854 0.156892908110547 0.694115057121718
0.274211682173158 -0.86291099460801 0.0138769140799509
In this section we provide examples of calling Fortran DLLs and C DLLs from C#.
A summary of some important data types used by C# is given in Table 2.3.
We will begin by mentioning some of the features of C# connected with mixed language programming.

- In C# if the array mya is declared using `double[] mya = new double[8];` it will have 8 elements; these are `mya[0], ..., mya[7].`
- The elements of multidimensional arrays are stored in row order, as is the case for VB.NET and C.
- When a DLL function is called from C# it is necessary to explicitly state which parameters are to be passed by reference; the others are taken as being passed by value. This means that the NAG C DLL function `f02wec` should be called as follows (see Code excerpt 2.12 for more detail):

```csharp
double[] evec = new double[m];
iflag = 0; // Set iflag to zero so use the NAG C library with default error handling.
// This means that, if an error occurs, the C DLL will output a message
f02wec(m, n, ref a[0], dim_a, ncolb, ref b[0], dim_b, wantq, ref q[0], dim_q, ref sv[0], wantp,
ref pt[0], dim_pt, ref iter, ref evec[0], ref info, iflag); // Note iflag is passed by value here
```

This should be compared to the equivalent VB.NET code, where the following will suffice:

```vbnet
iflag = 0
Call f02wec (m, n, a1(0, 0), dim_a, ncolb, b(0), dim_b, wantq, q(0),
            dim_q, sv(0), wantp, pt(0, 0), dim_pt, iter, evec(0), info, iflag)
```

- As previously mentioned in Section 2.2, C# does not allow fixed length strings to be declared within UDTs (that is structures). This means that it is no longer possible to define a type that corresponds to the NAG C Library type `NagErrorType`. Here we illustrate how to overcome this problem by wrapping the original C DLL function within another C function which uses an integer parameter to flag errors; see Code excerpt 2.13.

### Table 2.3 Correspondence between C# types and those of Fortran and C

<table>
<thead>
<tr>
<th>C#</th>
<th>C</th>
<th>Fortran 77</th>
<th>Size in bytes</th>
</tr>
</thead>
<tbody>
<tr>
<td>int</td>
<td>long</td>
<td>INTEGER</td>
<td>4</td>
</tr>
<tr>
<td>int</td>
<td>int</td>
<td>LOGICAL</td>
<td>4</td>
</tr>
<tr>
<td>float</td>
<td>float</td>
<td>REAL</td>
<td>4</td>
</tr>
<tr>
<td>double</td>
<td>double</td>
<td>DOUBLE PRECISION</td>
<td>8</td>
</tr>
</tbody>
</table>
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A C# example application, which calls the Fortran DLL functions G01AAF and F02WEF, and also the C DLL functions g01aac and F02wec, is shown in Figure 2.2.

When the button labelled SVD is clicked the function button3_Click is executed and either the NAG Fortran DLL function F02WEF or the NAG C DLL function f02wec is used to compute the singular value decomposition. When the button labelled ‘Summary stats’ is clicked then the function button1_Click is executed and either the NAG Fortran DLL function G01AAF or the wrapped C DLL function g01aac_wrapped, see Code excerpt 2.13, is used to calculate summary statistics such as mean, standard deviation, etc. The C# code for this example is shown in Code excerpts 2.10 to 2.12.

```csharp
using System;
namespace App2{
    public class Form1 : System.Windows.Forms.Form
    {
        Boolean use_c_dll;
        private System.Windows.Forms.Button button1;
        private System.Windows.Forms.Button button3;
        private System.Windows.Forms.GroupBox groupBox1;
        private System.Windows.Forms.RadioButton radioButton2;
        private System.ComponentModel.Container components = null;
        public Form1()
        {
            InitializeComponent();
            use_c_dll = true;
            this.button1 = new System.Windows.Forms.Button();
            this.button3 = new System.Windows.Forms.Button();
            this.groupBox1.SuspendLayout();
            this.SuspendLayout();
            // button1
            // this.button1.Location = new System.Drawing.Point (48, 24);
            this.button1.Name = "button1";
            this.button1.Size = new System.Drawing.Size(120, 24);
            this.button1.TabIndex = 0;
            this.button1.Text = "Summary stats";
            this.button1.Click += new System.EventHandler(this.button1_Click);
            // button3
            // this.button3.Location = new System.Drawing.Point (294, 24);
            this.button3.Name = "button3";
            this.button3.Size = new System.Drawing.Size(120, 24);
            this.button3.TabIndex = 1;
            this.button3.Text = "Summary stats";
            this.button3.Click += new System.EventHandler(this.button3_Click);
            // groupBox1
            this.groupBox1.SuspendLayout();
            this.SuspendLayout();
            // groupBox1
            this.groupBox1.SuspendLayout();
            // radioButton1
            this.radioButton1.Name = "radioButton1";
            this.radioButton1.Size = new System.Drawing.Size(75, 23);
            this.radioButton1.TabIndex = 0;
            this.radioButton1.Text = "C#";
            this.radioButton1.CheckedChanged += new System.EventHandler(this.radioButton1_CheckedChanged);
            // radioButton2
            this.radioButton2.Name = "radioButton2";
            this.radioButton2.Size = new System.Drawing.Size(75, 23);
            this.radioButton2.TabIndex = 1;
            this.radioButton2.Text = "Fortran";
            this.radioButton2.CheckedChanged += new System.EventHandler(this.radioButton2_CheckedChanged);
            // tabPage1
            this.tabPage1.SuspendLayout();
            // button1_Click
            // [STAThread]
            static void Main()
            {
                Application.Run(new Form1());
            }
            [DllImport("NAGG01.DLL")]
            public static extern void G01AAF(ref int n, ref double x, ref int iwt, ref double wt, ref double xmean, ref double s2, ref double s3, ref double s4, ref double xmin, ref double xmax, ref double wsum, ref int iflag);
            [DllImport("NAGF02.DLL")]
            public static extern void F02WEF(ref int m, ref int n, ref double a, ref int lda, ref int ncolb, ref double b, ref int ldb, ref int wantq, ref double q, ref int tdq, ref double sv, ref int wantp, ref double pt, ref int tdpt, ref int iter, ref double evec, ref int info, ref int iflag);
            [DllImport("nagc.dll")]
            public static extern void f02wec(int m, int n, ref double a, ref double b, ref int lda, ref int wantq, ref double q, ref int dim_q, ref double sv, ref int wantp, ref double pt, ref int Iter, ref int iter, ref int info, ref int iflag);
        }
    }
}
```
Dynamic Link Libraries (DLLs)

Dynamic Link Libraries (DLLs) are shared libraries that can be dynamically loaded into an application at runtime. They are useful for sharing code between applications and for separating the application code from the data it uses. DLLs can be written in any language and can be loaded into an application written in a different language. In this section, we will discuss how to use DLLs in C# programming language.

Code excerpt 2.10 A fragment of C# code which illustrates how the Fortran DLL functions G01AAF and F02WEF, and C DLL functions g01aac, g01aac_wrapped, and f02wec should be declared in C#.

```csharp
private void button1_Click(object sender, System.EventArgs e)
{
    double xmin = 0.0, xsd = 0.0, xskew = 0.0, xkurt = 0.0;
    int iflag = 1, iwt = 0, xmean = 0.0, xmax = 0.0;
    string token = "";
    string filename;
    int len, n1, i, buf_ptr, dummy = 0, nvalid = 0;
    char delim = ' ';
    // Open the input and output files
    FileStream fs_in = new FileStream(@"c:\BOOK_2002\STATS_DATA.TXT", FileMode.Open,
                                    FileAccess.Read, FileShare.None);
    StreamReader sr = new StreamReader(fs_in, Encoding.ASCII);
    StreamWriter sw = new StreamWriter(filename, false, Encoding.ASCII);
    NextLine = sr.ReadLine(); // Read a line from the input file, and store it in the string NextLine
    buf_ptr = 0;
    while (i < n1) // Load the data into the array x
    {
        if (get_token(ref NextLine, delim, ref buf_ptr, ref token))
        {
            x[i] = double.Parse(token); // Assign the data value
            i++;
        }
    }
```
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In these code excerpts we show how both vectors and matrices are declared and passed to Fortran and C DLL functions. Here matrices are declared as one-dimensional arrays and the elements are stored either in row order (if a C DLL function is to be called) or column order (when a Fortran DLL function is used).

Code excerpt 2.11  A fragment of C# code that computes summary statistics by either calling the Fortran DLL function G01AAF, or the C DLL function g01aac_wrapped

In these code excerpts we show how both vectors and matrices are declared and passed to Fortran and C DLL functions. Here matrices are declared as one-dimensional arrays and the elements are stored either in row order (if a C DLL function is to be called) or column order (when a Fortran DLL function is used).
Dynamic Link Libraries (DLLs)

// When the NAG Fortran DLL is used matrix elements are stored in column order, and when the NAG C Library is used matrix elements are stored in row order.

```c
double[] q = new double[1];
double[] b = new double[1];
int dim_a = 0, dim_q = 0, dim_b = 0, dim_pt = 0, m = 0;
int n = 0, iflag = 1, 1, ncolb = 0, j;
int wantp, wantq, buf_ptr;
double[] work = new double[1000]; // Allocate a fixed size workspace
string NextLine, token = " ";
string filename = " ";
char delim;
int info = 0, iter = 0;
if (use_c_dll) // Using the NAG C DLL
{
    filename = @"c:\BOOK_2002\CSF02WEC_RESULTS.TXT"; // Set the results file name for the NAG C DLL
}
else // Using the NAG Fortran DLL
{
    filename = @"c:\BOOK_2002\CSF02WEF_RESULTS.TXT"; // Set the results file name for the NAG Fortran DLL
}
// Open the input and output files
StreamWriter sw = new StreamWriter(@filename, false, Encoding.ASCII);
FileStream fs_in = new FileStream (@"c:\BOOK_2002\SVD_DATA.TXT", FileMode.Open, FileAccess.Read,
FileShare.None);
StreamReader sr = new StreamReader(fs_in, Encoding.ASCII);
NextLine = sr.ReadLine(); // Read a line from the input file
delim = ","; // Set the delimiter to be used when parsing the data
buf_ptr = 0;
get_token(ref NextLine, delim, ref buf_ptr, ref token); // Extract the number of rows m of matrix A
m = int.Parse(token); // Assign the number of rows
get_token(ref NextLine, delim, ref buf_ptr, ref token); // Assign the number of columns
n = int.Parse(token); // Assign the number of columns
double[] sv = new double[n]; // Allocate a vector to hold the singular values
double[] a = new double[n * m]; // Allocate a vector to hold the elements of the matrix A
double[] pt = new double[n * n]; // Allocate a vector to hold the elements of the matrix pt
buf_ptr = 0;
if (use_c_dll) // Use the NAG C DLL
{
    dim_a = n; // Set the trailing (second) dimension of the matrix A
    dim_b = 1;
    dim_q = m;
}
else // Use the NAG Fortran DLL
{
    dim_a = m; // Set the first (leading) dimension of matrix A
    dim_b = 1;
    dim_q = m;
}
dim_pt = n;
wantq = 1; // set WANTQ to TRUE
wantp = 1; // set WANTP to TRUE
ncolb = 0;
for (i = 0; i < n; ++i) // Loop on the row index
{
    buf_ptr = 0;
    NextLine = sr.ReadLine(); // Read another line from the input file
    for (j = 0; j < m; ++j) // Loop on the column index
    {
        get_token(ref NextLine, delim, ref buf_ptr, ref token); // Get the current data element
        if (use_c_dll) // Using the C DLL
        {
            a[i * dim_a + j] = double.Parse(token); // Store elements in row order
        }
        else // Using the Fortran DLL
        {
            a[i + j * dim_a] = double.Parse(token); // Store elements in column order
        }
    }
    buf_ptr = 0;
    sr.ReadLine(); // Read another line from the input file
}
```

iflag = 0; // Set iflag to zero so use the NAG C library with default error handling.
    // This means that, if an error occurs, the DLL will output a message
    f02wec(m, n, ref a[0], dim_a, ncolb, ref b[0], dim_b, wantq, ref q[0], dim_q, ref sv[0], wantp,
    ref pt[0], dim_pt, ref iter, ref wsvec[0], ref info, iflag); // Note iflag is passed by value here
} else // Use the NAG Fortran DLL
    F02WEF(ref m, ref n, ref a[0], ref dim_a, ref ncolb, ref b[0], ref dim_b, ref wantq, ref q[0],
    ref dim_q, ref sv[0], ref wantp, ref pt[0], ref dim_pt, ref work[0], ref iflag);
if (use_c_dll) // Output the file header for the NAG C DLL
    { sw.WriteLine("C# example SVD results: using C library function f02wec");
} else // Output the file header for the NAG Fortran DLL
    { sw.WriteLine("C# example SVD results: using Fortran library DLL function F02WEF");
    sw.WriteLine('''
    sw.WriteLine('''
    sw.WriteLine('The Singular Values are:
    sw.WriteLine();
    for (i = 0; i < n; ++i) // Loop on the number of singular values
        sw.WriteLine("(0.10:F4)", sv[i]);
    sw.WriteLine('''
    sw.WriteLine('''
    sw.WriteLine('Left-hand singular vectors:
    sw.WriteLine('''
    for (i = 0; i < m; ++i) // Loop on the row index
        for (j = 0; j < n; ++j) // Loop on the column index
            if (use_c_dll)
                sw.Write("(0.10:F4)", a[i*dim_a + j]); // Output the elements in row order
            else
                sw.Write("(0.10:F4)", a[i + j*dim_a]); // Output the elements in column order
        sw.WriteLine('''
    sw.WriteLine('''
    sw.WriteLine('Right-hand singular vectors:
    sw.WriteLine('''
    for (i = 0; i < n; ++i) // Loop on the row index
        for (j = 0; j < m; ++j) // Loop on the column index
            if (use_c_dll)
                sw.Write("(0.10:F4)", pt[i*dim_pt + j]); // Output the elements in row order
            else
                sw.Write("(0.10:F4)", pt[i + j * dim_pt]); // Output the elements in column order
        sw.WriteLine('''
    sw.WriteLine('''
    sr.Close();
    sw.Close();
}

Code excerpt 2.12 A fragment of C# code that computes a singular value decomposition by either calling
the Fortran DLL function F02WEF, or the C DLL function f02wec

<p>| | | | | | | | | | |</p>
<table>
<thead>
<tr>
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<th></th>
<th></th>
<th></th>
<th></th>
<th></th>
<th></th>
</tr>
</thead>
<tbody>
<tr>
<td>193.0</td>
<td>215</td>
<td>112.0</td>
<td>161.0</td>
<td>92.0</td>
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<td>473.0</td>
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</tr>
</tbody>
</table>

Exhibit 2.4 The data file used by Code excerpt 2.11
Dynamic Link Libraries (DLLs)

C# example summary statistics results: using the wrapped C library DLL function g01aac_wrapped

no valid cases = 0
mean = 254.25
mean = >> 254.250 <<
standard deviation = >> 433.536 <<
skewness = >> 3.895 <<
kurtosis = >> 14.666 <<
minimum = >> 20.000 <<
maximum = >> 2200.000 <<
sum of weights = >> 24.000 <<

Exhibit 2.5 The results computed by Code excerpt 2.11

C# example SVD results: using Fortran library DLL function F02WEF

The Singular Values are:
6.5616 3.0000 2.4384

Left-hand singular vectors
0.6011 -0.1961 -0.3165
0.6011 -0.1961 -0.3165
0.4166 0.1569 0.6941
0.1688 -0.3922 0.5636
-0.2742 -0.8629 0.0139

Right-hand singular vectors
0.4694 -0.7845 0.6054
0.4324 -0.1961 -0.8801
0.7699 0.5883 0.2471

Exhibit 2.6 The results computed by Code excerpt 2.12

Wrapping the NAG C DLL function:

```c
#include <nag.h>
#include <nag_stdlib.h>
#include <nagg01.h>

#define DLLExport__declspec(dllexport)

void DLLExport__stdcall g01aac_wrapped (long n, double x[], double wt[], long *nvalid, double *mean, double *xsd, double *xskew, double *xkurt, double *xmin, double *xmax, double *wsum, long *iflag)
{
    static NagError stat;
    stat.print = FALSE;
    stat.code = 0;
    g01aac(n, x, wt, nvalid, mean, xsd, xskew, xkurt, xmin, xmax, wsum, &stat);
    *iflag = stat.code;
}
```

Code excerpt 2.13 The C++ DLL wrapper for the function g01aac
Chapter 3

ActiveX and COM

3.1 INTRODUCTION

Here we show how Microsoft ActiveX and COM technology can be used to solve mathematical problems within the Windows environment. It is intended as a general introduction to the subject and shows how to use ActiveX components rather than create them.

To call a DLL routine directly from Visual Basic requires detailed knowledge of both the routine’s arguments and also the manner in which they are passed to the Visual Basic calling program. It is therefore essential that users have access to all the relevant documentation. This approach also has the following disadvantages.

- Currently there are certain restrictions on the use of DLL routines, for instance they cannot be incorporated into an HTML Web page.
- DLLs are not in the spirit of Microsoft’s object-based approach to programming and do not make use of this technology.
- They must be called using low level program statements and cannot be accessed interactively or visually.

By using an Excel Add-In (as shown in Part II, Section 9.3.4) to provide a higher level user-interface to the underlying DLL it is possible to alleviate some of the difficulties previously mentioned. However, it should be mentioned that:

- Not all versions of Excel are compatible.
- There is still the issue of how potential users are to access routines from Visual Basic, Delphi, PowerPoint, etc.
- The underlying framework of the Excel user-interface cannot be changed (since it was created by Microsoft) and can appear rather tedious for routines with large argument lists, etc.

So what is the natural interface to use within Microsoft Windows? Ideally what is needed is an easy-to-use interface that would allow all routines to be called from every Microsoft product. In fact such an interface does already exist: the Component Object Model (COM). It is used by Microsoft, Inprise, Digital Equipment Corporation, and many other companies.

Microsoft has also created the COM-based technologies of ActiveX and OLE to allow Microsoft users the ability to interact with their environment. All the mathematical software described here could have been deployed using custom (user-defined) COM interfaces. However these non-standard COM interfaces would then
require separate documentation and would not automatically integrate into Micro-
soft products such as Visual Basic, Visual C++, etc. To avoid these problems only
the standard Automation interface IDispatch (see Section 3.2) will be considered
here. The IDispatch COM interface allows ActiveX components to be easily used
from languages such as Visual Basic, VBScript, and Inprise Delphi. It also permits
easy incorporation of mathematical software into Excel, Word, PowerPoint, Access,
and HTML Web pages. ActiveX components can also be used from Visual C++,

Some of the advantages of ActiveX components are:

- They can be used by the complete range of Microsoft products and also by other
  Windows software such as Inprise Delphi.
- They support drag and drop technology and so can easily be incorporated into an
  application.
- The properties, methods, and events of a given ActiveX component can be viewed
  using the Microsoft (Inprise) Object Browser.
- Their object-based C++ technology can be used to provide simple user-interfaces
to otherwise complicated routines.

The last point refers to the complete range of C++ class/object-based technology.
This includes optional arguments with default values, data/information hiding within
the object, object initialisation via constructors, and the properties, methods, and
events supported by an object.

This section gives a brief outline of the basic principles of COM and how
ActiveX controls are accessed from Visual Basic and Visual C++ using the
IDispatch interface (also called dispinterface for short). There is not space to
fully explain everything mentioned in this section, but comprehensive information
can be obtained from the available literature on COM and ActiveX, see the
computing references at the end of the book.

ActiveX controls are DLL servers that need to be registered in the Windows
Registry before they can be dynamically linked to by a client. Every registered
ActiveX control has a unique class identifier (CLSID) which allows a client to load
it from the DLL in which it resides and create an instance of the component.

ActiveX controls are COM objects that usually have a visual user-interface and
also support a variety of interfaces including those that allow Automation and
events.

Automation allows an ActiveX control’s properties and methods to be accessed
programmatically from a language such as Visual Basic or C++, and is implemented
using the IDispatch COM interface. Event-handling for events such as Single
(Double) Click is implemented using COM interfaces such as IConnectionPoint
and IConnectionPointContainer.

Since the main purpose of the components described here is to perform numeric
calculations they only need a restricted visual user-interface, and will therefore be
called primitive ActiveX components. In fact a control that maintains an on-screen
window has to manage messages for the window and is therefore slower than a
windowless control. These primitive controls are ideal for use as numeric engine
components since their limited visual user-interface will not interfere with the user-interface of the application into which they are embedded. Mathematical applications with sophisticated user-interfaces can therefore readily be constructed through the incorporation of primitive ActiveX components.

### 3.2 THE COM INTERFACE IDISPATCH

All COM components are derived (in the C++ sense) from an interface called IUnknown. The definition for IUnknown is as follows:

```c
interface IUnknown
{
    virtual HRESULT STDMETHODCALLTYPE QueryInterface(const IID& iid, void** ppv) = 0;
    virtual ULONG STDMETHODCALLTYPE Addref() = 0;
    virtual ULONG STDMETHODCALLTYPE Release() = 0;
};
```

It contains the three virtual functions QueryInterface, Addref, and Release. The function QueryInterface is used to find out whether an object supports a given interface and, if possible, return a pointer to it. For example, an ActiveX control used from within Visual Basic could call QueryInterface to return a pointer to the IDispatch interface. The functions Addref and Release maintain a reference count on the interface of a component and use this to implement memory management. When the reference count reaches zero the component deletes itself from memory.

The definition of the IDispatch interface is:

```c
interface IDispatch : IUnknown
{
    HRESULT TypeInfoCount(UINT* pctinfo);
    HRESULT GetTypeInfo(UNIT iTinfo,
                           LCID lcid,
                           ITypeInfo** ppTInfo);
    HRESULT GetIDsOfNames(const IID& riid,
                           LPCTSTR rgzNames,
                           UINT cNames,
                           LCID lcid,
                           DISPID* rgDispId);
    HRESULT Invoke (DISPID dispIdMember,
                    const IID& riid,
                    LCID lcid,
                    WORD wFlags,
                    DISPPARAMS* pDispParams,
                    VARIANT* pVarResult,
                    EXCEPINFO* pExcepInfo,
                    UINT* puArgErr);
};
```

The functions TypeInfoCount and GetTypeInfo are used to obtain information concerning the methods and properties of the component from its type library (see Section 3.2).

The function GetIDsOfNames converts the Visual Basic name of an Automation object’s properties and methods into a numeric identifier called the DISPID of the property or method.

The function Invoke uses the DISPID to run a given property or method and also passes it the required arguments in a structure of type DISPPARAMS.
3.3 TYPE LIBRARIES

Type libraries are compiled versions of an Object Description Language (ODL) file or an Interface Definition Language (IDL) file. They provide information about the interfaces, methods, properties, and arguments of a COM component and are used by the Visual Basic Object Browser to interactively display this information.

An excerpt (the complete source is given in Appendix A) from the ODL file for the example control NAGDBS.ocx (see Chapter 4) is given below in Code excerpt 3.1.

```cpp
// NAGDBS.odl: type library source for ActiveX Control project.
// This file will be processed by the Make Type Library (mktyplib) tool to
// produce the type library (NAGDBS.tlb) that will become a resource in
// NAGDBS.ocx.

helpstring('Dispatch interface for NAGDBS Control'), hidden

dispinterface_DNAGDBS
{
  properties:
  {//AFX_ODL_PROP(CNAGDBSCtrl)
      [id(1)] METHODTYPE method;
      [id(2)] EXTYPE extype;
      [id(3)] double sigma;
      [id(4)] long numsteps;
      [id(5)] double intrate;
      [id(6)] double dividends;
      [id(7)] double curval;
      [id(8)] double optval;
      [id(9)] double strike;
      [id(10)] PUTCALLTYPE putcall;
      [id(DISPID_CAPTION), bindable, requestedit] BSTR Caption;
      [id(DISPID_BACKCOLOR), bindable, requestedit] OLE_COLOR BackColor;
      [id(DISPID_FORECOLOR), bindable, requestedit] OLE_COLOR ForeColor;
  };//AFX_ODL_PROP

  methods:
  {//AFX_ODL_METHOD(CNAGDBSCtrl)
      [id(12)] void Calculate();
      [id(13)] void greeks(double* greekvals);
  };//AFX_ODL_METHOD
};

Code excerpt 3.1 Fragment of the ODL file for the ActiveX component NAGBS.ocx used in Chapter 4

Figure 3.1 shows the Visual Basic Object Browser using the type library NAGDBS.tlb to interactively display the properties and methods of the component contained in NAGDBS.ocx.

3.4 USING IDISPATCH

The use of the IDispatch interface can be illustrated by considering the following three line Visual Basic program:

```vbnet
Dim NAGDBS1 As Object
Set NAGDBS1 = CreateObject("NAGDBS.NAGDBSCtrl.1")
NAGDBS1.curval = 111.0
```

This program creates an instance of a COM component called NAGDBS1 and assigns a floating-point number to the property `curval`. 

```
The equivalent C++ code for these Visual Basic statements will now be described. The client first needs to retrieve the component’s class identifier (CLSID) from the Windows Registry. This is achieved by using the routine CLSIDFromProgID as follows:

```c++
// Initialise the OLE Library
HRESULT hr = OleInitialize(NULL);

// Get The CLSID for the application
wchar_t progid[] = L"NAGDBS.NAGDBSCtrl1.1";
CLSID clsid;
::CLSIDFromProgID(progid, &clsid);
```

Once the CLSID has been obtained CoCreateInstance can be used to both create an instance of the component and also return a pointer (pIDispatch) to its IDispatch COM interface.

```c++
IDispatch* pDispatch = NULL;
::CoCreateInstance(clsid, NULL, CLSCTX_INPROC_SERVER,
IID_IDispatch, (void**)&pIDispatch);
```

This approach avoids an extra call to QueryInterface for retrieval of the IDispatch interface pointer. The DISPID of the property curval is then obtained
so that curval can be accessed using IDispatch::Invoke. This is achieved by using the IDispatch function GetIDsOfNames as follows:

```c
DISPID dispid;
OLECHAR* name = L"curval";
PIDispatch->GetIDsOfNames(IID_NULL, // Must be IID_NULL
    &name, // Name of the function
    1, // Number of names
    GetUserDefaultLCID(), // Localization info
    &dispid); // Dispatch ID
```

Now that the DISPID for curval has been obtained the assignment statement in the last line of the Visual Basic Program will be discussed. Here the IDispatch function Invoke runs the function curval using its DISPID and passes it the required arguments in a structure of type DISPPARAMS.

The steps in Visual C++ are as follows:

1. Initialize a variable of type VARIANT and assign the value 111.0 to it.
2. Now fill in the DISPPARAMS structure

```c
VARIANTARG varg;
::VariantInit(&varg);  // Initialisethe VARIANT
varg.vt = VT_R8;       // Type of VARIANT data, in this case a double
varg.dblVal = 111.0;   // Set the value of the variant to 111.0
```

Finally the function curval can be run by using Invoke as follows:

```c
hr = pIDispatch->Invoke(dispid,
    IID_NULL,
    GetUserDefaultLCID(),
    DISPATCH_METHOD,
    &param,
    NULL,
    NULL,
    NULL);
```

The use of IDispatch::Invoke to access a component's properties and methods is called run-time binding because the argument types are only checked at run-time. All the example ActiveX controls in this book use the IDispatch interface in this manner even though it is not as efficient as using a dual interface (which can access functions directly through the vtbl).

### 3.5 ACTIVEX CONTROLS AND THE INTERNET

Small scale numerical Internet applications can easily be constructed using the capabilities of Web-based scripting languages such as JavaScript, JScript, or VBScript.
However, the use of the Internet for medium to large numerical applications is still a matter for research. The traditional method of solving these problems is by creating Fortran or C applications which make underlying calls to numerical library subroutines. Although this approach may have the advantages of speed/efficiency it is not based on current Internet technology. A direct consequence of this is that attempts to access such applications from the Internet may suffer from a variety of limitations such as:

- Inability to directly access individual mathematical subroutines.
- Solution is not integrated into the user’s system.
- Limited interactive features.

Here we are concerned with the use of ActiveX components, within the Microsoft Windows environment, to solve medium scale numerical problems on HTML Web pages. These Web pages may either be stored on a computer’s local disk or reside on a remote machine to which there is access via the Intranet or Internet. The Web pages considered here contain HTML and either VBScript or JScript. They are interpreted by means of the Web browser Internet Explorer.

We will now give some of the advantages and disadvantages of ActiveX controls from within HTML Web pages.

**Advantages:**

- Interactive modelling over the Internet.
- Web page VBScript can be pasted into other Microsoft products such as Visual Basic or Excel and used with little or no modification.
- A Web page can be downloaded from the Internet and used to create a working local model which can then be placed back on the Internet.
- Web page models can be easily changed by altering the VBScript code. This would be particularly useful for models where an appropriate CGI script is either not currently available or has restricted access.
- Since ActiveX components can be created using C++ it is not necessary to rewrite complicated numerical algorithms.

**Disadvantages:**

- Microsoft Windows specific.
- A potential lack of efficiency for large-scale numerical problems. This is because all computations are carried out on the user’s local machine, and also VBScript/JScript code is interpreted at run-time by the Web browser.

ActiveX components are therefore expected to be beneficial for small/medium sized mathematical models which require an interactive user-interface.

### 3.6 USING ACTIVEX COMPONENTS ON A WEB PAGE

This section gives brief details on using ActiveX components from within an HTML Web page. ActiveX controls can easily be placed on a Web page by using an interactive tool called ActiveX Control Pad.
Once the control has been placed onto the Web page it is referenced using the information contained in an HTML object tag. The object tag HTML source code for a Microsoft command button is given below:

```html
<OBJECT ID='CommandButtonRed' WIDTH=44 HEIGHT=26
CLASSID='CLSID:D7053240-CE69-11CD-A777-00DD01143C57'>
<PARAM NAME='Caption' VALUE='Red'>
<PARAM NAME='Size' VALUE='2540;846'>
<PARAM NAME='FontCharSet' VALUE='0'>
<PARAM NAME='FontPitchAndFamily' VALUE='2'>
<PARAM NAME='ParagraphAlign' VALUE='3'>
</OBJECT>
```

It can be seen that the unique class identifier for all Microsoft command buttons is D7053240-CE69-11CD-A777-00DD01143C57, and that this particular one, which is referred to in VBScript/JScript as CommandButtonRed, has the caption Red written on it. The size of the component when viewed using a Web browser is controlled by the values of WIDTH and HEIGHT. If an ActiveX control has no interactive user-interface then it is only accessed via its language user-interface and can be made invisible by setting WIDTH and HEIGHT to appropriately small values.

Calling the properties and methods of ActiveX controls on an HTML Web page is similar to using them from other Microsoft products such as Excel, Visual Basic, etc. However, there are slight differences depending on whether VBScript or JScript is used within the Web page. We will now illustrate this using an ActiveX graphical component called Plot1. Code excerpt 3.2 illustrates this using VBScript, and Code excerpt 3.3 gives the equivalent JScript code.

```html
<HTML>
<HEAD>
<TITLE>VBScript GARCH modeller demonstration</TITLE>
</HEAD>
<BODY>
<SCRIPT LANGUAGE='VBScript'>
Sub Calculate_GARCH_Click()
'plot the modelled volatility
Plot1.BrushColor 11,224,230
Plot1.PenColor 0, 0, 255
Plot1.PenWidth = 1
Plot1.text "The modelled GARCH variance",40, bot_pos1 + 5
End Sub
Sub Clean_Click()
Plot1.Clear
End Sub
-->
</SCRIPT>
</HTML>
```

**Code excerpt 3.2**  Scripting an ActiveX component’s properties and methods using VBScript

The corresponding JScript code is now given:

```html
<HTML>
<HEAD>
<TITLE>JScript GARCH modeller demonstration</TITLE>
</HEAD>
<BODY>
</BODY>
</HTML>
```
Using Numerical Software Components within Microsoft Windows

Since ActiveX component technology is based on C++, calls to complicated numerical routines can be simplified through the use of properties, methods, events, object initialization via constructors, data/information hiding within the object, and also optional arguments that take default values.

ActiveX components can be used by the entire range of Microsoft products, from PowerPoint to Internet Web browsers, and also by other Windows products such as Inprise Delphi. It has been shown that it is easy to script ActiveX components on a Web page and that models developed in this way can be easily modified on a local machine and placed on the Internet when appropriate.

Some of the advantages of using ActiveX models on Web pages are:

- VBScript on a Web page can be converted into working Visual Basic code with only minor modifications. Therefore a Web page ActiveX mathematical model can easily be incorporated into Microsoft products such as Microsoft Visual Basic, Microsoft Excel, Microsoft Access, etc.
- Since Microsoft Web browsers are supplied free of charge VBScript or JScript models can be developed at no extra cost (as long as the required ActiveX components are freely available).
- ActiveX components can be created using Visual C++, which means complex mathematical models can be developed with existing numerical software.

Possible disadvantages include:

- ActiveX components are Microsoft Windows specific, and so cannot be used within UNIX.
- There could be a lack of computational efficiency which may become important for certain types of large or complex problems.

From an historical perspective mathematical modelling using ActiveX components and Web script is very similar to the more traditional method of writing Fortran or C programs. Here Fortran or C source code written by the developer is used instead of Web script to call numerical routines (the equivalent of the ActiveX components).
from the appropriate Fortran or C numerical Library. The major differences in the approach outlined here are that:

- The model is easy to construct because it is made up of numeric ActiveX components with a simple language user-interface.
- Although an ActiveX component may have been created using Visual C++, its native language is not relevant when it is called via its IDispatch COM interface. This means that its properties and methods can be accessed directly from VBScript and JScript, and it is not necessary to purchase a Visual C++ compiler.
- General-purpose libraries can be replaced by self-contained ActiveX components.

Through the creation of the necessary ActiveX components and HTML Web script, the majority of numerical models implemented in traditional languages such as Fortran, C, or C++ could be placed on Web pages.
Chapter 4

A financial derivative pricing example

The financial derivative pricing control was chosen to illustrate a control that gives similar importance to both its language and interactive user-interfaces. Its properties can be set interactively at design-time and have associated events, properties and methods. This control was created using Visual C++ and calculates the value of a financial derivative (option) by solving the Black–Scholes partial differential equation, see Part II for more detail. The interface for the control is described by its ODL file which is given in the Appendix A.1. The control is contained in the file NAGDBS.ocx, and its instance in this Visual Basic example is called NAGDBS1.

It is acknowledged that a commercial version of this software would require:

- Comprehensive documentation, both printed and as Help file information.
- Sophisticated interactive design-time and language user-interfaces.

This example is therefore merely provided as a guide to show what is possible using ActiveX and should not be regarded as a definitive statement on what constitutes a good user-interface.

4.1 INTERACTIVE USER-INTERFACE

The interactive user-interface includes Property values that can be set using the Microsoft Properties Window and also Events. Figure 4.1 shows how the background colour of the control can be set interactively at design-time.

Here the ActiveX control uses an event to initiate computation at run-time. No calculations are performed until the control has been clicked by the mouse, as shown in Figure 4.2.

Once the control has been clicked the subroutine NAGDBS1_Click() is invoked and computations are performed; see Figure 4.3. The source code within NAGDBS1_Click() is given below in Section 4.2.

4.2 LANGUAGE USER-INTERFACE

When the control NAGDBS1 is placed on the user’s form, Visual Basic will automatically provide the following template code:

```vbnet
Private Sub NAGDBS1_Click()

End Sub
```
A financial derivative pricing example

Figure 4.1 Selecting the background colour of the control at design-time

Figure 4.2 The user form and control before computations are performed
This subroutine is run whenever the control NAGDBS1 is clicked by the user’s mouse. Here the subroutine contains the following code:

```vbnet
Private Sub NAGDBS1_Click()
    Dim greeks(3) As Double
    Dim S0 As Double
    Dim r As Double
    Dim q As Double
    Dim sigma As Double
    Dim T As Double
    Dim x As Double
    Dim maturity As Double
    Dim i As Long

    x = 8#  
    S0 = 10#  
    r = 0.1  
    sigma = 0.3  
    q = 0.06  
    Font.Bold = True  
    Font.Size = 14  
    Print ""  
    Print "AMERICAN PUT OPTIONS"  
    Print "Time Option Value Delta Gamma Theta"  
    Print "(Years)"  

    NAGDBS1.putcall = 1  
    ' A put option  
    NAGDBS1.curval = S0  
    ' The current asset value  
    NAGDBS1.strike = x  
    ' The strike price
```
A financial derivative pricing example

```vbnet
NAGDBS1.dividends = q ' The continuous dividend yield
NAGDBS1.method = 0 ' Use the standard lattice
NAGDBS1.numsteps = 10 ' The number of time steps
NAGDBS1.intrate = r ' The risk free interest rate
NAGDBS1.exype = 1 ' An american option
NAGDBS1.sigma = sigma ' The volatility

' Construct a table of option values and greeks for different maturities
For i = 1 To 3
    T = i * 0.25
    NAGDBS1.maturity = T ' The maturity, in years
    NAGDBS1.Calculate ' Do the calculations
    opt_val = NAGDBS1.optval ' Get the value of the option
    NAGDBS1.greeks greeks(0) ' Get the calculated hedge statistics (greeks)
Next i

' Now output the results in tabular format
Print ''''; Format(T, ''#0.00''), Format(opt_val, ''#0.0000''),
    Format(greeks(0), ''#0.0000''), Format(greeks(1), ''#0.0000''),
    Format(greeks(2), ''#0.0000'')
End Sub
```

The Visual Basic code used to compute American option values

The code illustrates that the properties NAGDBS1.putcall, NAGDBS1.curval, NAGDBS1.sigma, etc. are used to set up the values for the problem. The method NAGDBS1.Calculate is then used to perform the required calculations, and option values and greeks are returned via the property NAGDBS1.optval and method NAGDBS1.greeks respectively. It can be seen from the output in Figure 4.3 that using NAGDBS1.exype = 1 and NAGDBS1.exype = 2 results in slightly different option values and hedge statistics (greeks). This is because here the partial differential equation is approximated using a lattice with only ten time steps. The most accurate values are expected to be those calculated using the Control Variate method. This method uses the analytic value of the corresponding European option to adjust the answers returned by the lattice. However, as the number of time steps is increased the results from both methods should converge.

4.3 USE WITHIN DELPHI

Views of the Delphi project at design-time and run-time are shown in Figures 4.4 and 4.5 respectively.
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Figure 4.4 The derivative control NAGDBS2 on TForm1

Figure 4.5 The Delphi application with the derivative control’s results
A financial derivative pricing example

Excerpts from the Delphi source code are given below.

```delphi
procedure TForm1.FormClick(Sender: TObject);
var
  greeks: Array[1..5] of double;
  T: double;
  i: integer;
  opt_val: double;
  num_precision: integer;
  num_digits: integer;
  pos: integer;
  val1: String;
begin
  NAGDBS2.putcall := 1;  {A put option}
  NAGDBS2.curval := 10.0; {The current value of the asset}
  NAGDBS2.strike := 8.0; {The strike price for the option}
  NAGDBS2.dividends := 0.06; {The continuous dividend yield}
  Canvas.TextOut(10,80, 'AMERICAN PUT OPTIONS (USING CONTROL VARIATE)');
  Canvas.TextOut(10,140, 'Time');
  for i := 1 To 3 Do
  Begin
    T := i*0.25;
    NAGDBS2.maturity := T;  {Set the maturity of the option, in years}
    NAGDBS2.Calculate;  {Do the calculation}
    NAGDBS2.greeks(greeks[1]);  {Get the hedge statistics, the greeks}
    opt_val := NAGDBS2.optval; {Get the option value}
    val1 := FloatToStrF(T,ffFixed,num_precision,num_digits);
    Canvas.TextOut(10,pos,val1);
    val1 := FloatToStrF (opt_val,ffFixed,num_precision,num_digits);
    Canvas.TextOut(100,pos,val1);
    val1 := FloatToStrF (greeks[1],ffFixed,num_precision,num_digits);
  End;
end;
end;
```

**Code excerpt 4.2**  The Delphi code used to compute American option values, this can be compared with the equivalent Visual Basic code given in Code excerpt 4.1

It can be seen that the option values and hedge statistics computed within Delphi are identical to those obtained (see Figure 4.3) using Microsoft Visual Basic.
Chapter 5

ActiveX components and numerical optimization

In this section we provide two illustrative examples of using numerical optimization components on a Web page. The ‘Ray tracing example’ performs a complicated numerical optimization involving the minimization of an integral, while the ‘Portfolio allocation example’ solves the classic Markowitz portfolio selection problem.

5.1 RAY TRACING EXAMPLE

It is hoped that reader will not think it too indulgent of the author to place this example in a book on mathematical finance and Windows. However, this demonstration (although nothing to do with financial modelling) does illustrate the computational power and flexibility that a numerical optimization ActiveX control, here referred to as OPTIM, allows within an HTML Web page. The optimization component could equally well have been used to demonstrate portfolio optimization, GARCH modelling, etc.

This example plots the path that a ray of light takes in a non-uniform refractive medium. Here there is a choice of three different colours and a non-uniformity (decay) parameter can also be selected. The Web page is constructed using three customized ActiveX components and standard Microsoft controls such as command buttons, labels, and textboxes. The three customized ActiveX components are:

1. A nonlinear numerical optimization control, OPTIM.
2. A graphical plotting control, GRAPH.
3. An integration control, INTG.

The ray tracing problem is modelled by finding the light path which minimizes the integral \( \int n(r, \omega)dr \) where \( n(r, \omega) \) is the spatial/frequency dependent refractive index. A fairly basic model is used here: the light path is assumed to follow a general cubic, which means that there are only four unknown coefficients to be determined. The width of the refractive medium is taken to be 2.5 units and the non-uniformity is assumed to be caused by radial density variation from the centre of the white ball in Figures 5.2 and 5.3. The amount of non-uniformity is controlled by changing the user-specified density decay parameter. Here we consider four rays of light (which can be either red, blue, or green) that start from different heights on the left hand side of the medium, and all pass through the point (0,2.5) on the right hand side of the diagram.

The problem is solved here by using numerical optimization with constraints in conjunction with numerical integration. This means that the call to the integration
control, INTG, is nested within the objective function of the optimization control, OPTIM. Two user-defined functions are therefore required: a user-defined objective function and a user-defined integrand.

Excerpts from the VBScript used for this demonstration are now given, the complete source code is supplied on the CD ROM which accompanies the book. The properties and methods of each control will not be discussed in detail, since these can be worked out from the context in which they occur in the VBScript code.

The VBScript controlling the selection of the light colour is as follows:

```vbnet
Sub CommandButtonBlue_Click()
    red_color = 0
    green_color = 0
    blue_color = 255
    frequency = 7.0
End Sub
```

Here the frequency and RGB plot colour is set to that corresponding to blue light when the command button labelled ‘Blue’ is clicked. Computations are performed when the ‘Calculate’ button is clicked and the subroutine CommandButton1_Click() is run. VBScript excerpts from CommandButton1_Click() are given in Code excerpt 5.1 below.

```vbnet
Sub CommandButton1_Click()
    Dim bl(100) ' holds the upper constraints
    Dim bu(100) ' holds the lower constraints
    Dim loc_x(100)
    Dim g(100)
    Dim a(100)
    Dim n, nclin, ncnlin, tda, num_vars
    Dim y_old, y_new, x_old, x_new
    Dim i, j, k, dx, xtemp, canvas_height, canvas_width, x_start, y_start
    tda = 3
    nclin = 1
    n = 3 ' number of variables
    ncnlin = 0
    num_vars = n
    num_pts = 51 ' number of data points to plot

    ' on the first call initialise the plotting area
    if (first_call = 1) then
        GRAPH.BrushColor 230, 240, 255
        GRAPH.PenColor 0, 0, 0
        GRAPH.PenWidth = 3
        GRAPH.text "Rays that minimise the optical path integral", x_start, y_start + 20
        GRAPH.Rectangle x_start, 20, x_start + 337, y_start
        GRAPH.circle 468, y_start, 10
        first_call = 0
    end if
    i = 0
    For i = 0 To 3 ' loop over the vertical start position of the ray
        y_shift = i ' set the vertical position
        ' set the initial estimates of the coefficients of the cubic
        loc_x(0) = 0.000001
        loc_x(1) = 0.000001
        loc_x(2) = 0.000001
        ' set the bounds and constraints
        For k = 0 To num_vars - 1
            bl(k) = -10.0
            bu(k) = 10.0
        Next
        bl(num_vars) = -y_shift
        bu(num_vars) = -y_shift
        tda = 3
```
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\[ a(0) = 2.5 \times 2.5 \times 2.5 \]
\[ a(1) = 2.5 \times 2.5 \]
\[ a(2) = 2.5 \]

\text{atmospheric\_factor} = \text{TextBox1.Value} \quad \text{set the atmospheric decay factor}

\text{if} (\text{atmospheric\_factor} \neq '') \quad \text{check that the decay factor has been set}

\text{OPTIM.optimize} n, nclin, ncnlin, a(0), tda, g(0), loc_x(0), bl(0), bu(0)

\text{OPTIM.getvarsloc_x(0), n} \quad \text{load the optimal cubic coefficients into loc}_x

\text{work out the optimal path of the ray}

\text{do\_plot()} \quad \text{now plot the optimal path}

\text{end if}

\text{Next}

\text{End Sub}

Code excerpt 5.1 The VBScript code for the subroutine \text{CommandButton1\_Click}

It can be seen that numerical optimization is performed by the optimize method of the ActiveX component OPTIM. The user-defined objective function to be minimized is contained in the routine \text{OPTIM\_Objfunction()}. The complete source code for this routine is given below

Sub \text{OPTIM\_Objfunction()}
\quad \text{The function to optimize.}
\quad \text{Note: The numerical quadrature ActiveX component INTG is called to evaluate the path integral}
\quad \text{Dim x(100)}
\quad \text{Dim obj\_val, num\_vars, a, b}
\quad \text{Dim result, num\_intervals, val, i}
\quad \text{num\_vars = 3}
\quad \text{obj\_val = OPTIM.\_Objval}
\quad \text{OPTIM.getvarsx(0), num\_vars}
\quad \text{num\_intervals = num\_pts}
\quad \text{a = 0.0}
\quad \text{b = 2.5}
\quad \text{For i = 0 To num\_vars - 1}
\quad \quad \text{params(i) = x(i)}
\quad \text{Next}
\quad \text{INTG.\_Integrate a, b, num\_intervals} \quad \text{evaluate the path integral}
\quad \text{val = INTG.\_answer} \quad \text{assign the path integral to val}
\quad \text{obj\_val = val}
\quad \text{OPTIM.\_Objval = obj\_val} \quad \text{make OPTIM\_Objfunction return the value of the path integral}
\quad \text{OPTIM.setvarsx(0), num\_vars}
End Sub

Code excerpt 5.2 Illustrating the use of VBScript to define the objective function that the ActiveX numerical optimization component, OPTIM, will minimize

The optical path length (to be minimized) is calculated by using the integrate method of the numerical quadrature component INTG. The path integral between \(a = 0\) and \(b = 2.5\) is calculated using 51 intervals. The user-defined integrand is specified in the following routine:

Sub \text{INTG\_Integrand()}
\quad \text{This routine is used by INTG to evaluate the path integral}
\quad \text{It specifies the spatial and frequency dependence of the refractive index}
\quad \text{Dim y, x, grad}
\quad \text{Dim index, opt\_path, temp, factor, n_o}
\quad \text{x = INTG.\_getx} \quad \text{set the frequency dependence of the refractive index}
\quad \text{n_o = 10.0 * (1.0 - 0.12 \times \text{frequency})}
\quad \text{y = params(0) * x * x + params(1) * x + y\_shift}
\quad \text{grad = params(0) * x * x + params(1) * x + y\_shift}
\quad \text{temp = y * y + (1.25 - x) * (1.25 - x)}
\quad \text{factor = Sqr(temp)} \quad \text{the spatial dependence of the refractive index}
End Sub
**ActiveX components and numerical optimization**

\[
\text{rindex} = n_o \cdot \exp(-\text{factor} \cdot \text{atmospheric\_factor})
\]
\[
\text{rindex} = \text{rindex} + 1.0
\]
\[
\text{temp} = 1.0 + \text{grad} \cdot \text{grad}
\]
\[
\text{opt\_path} = \text{rindex} \cdot \sqrt{\text{temp}}
\]
\[
\text{value} = \text{opt\_path}
\]
\[
\text{INTG\_getfunval} = \text{value}
\]
End Sub

**Code excerpt 5.3** Illustrating the use of VBScript to define the integrand that the ActiveX numerical quadrature component, INTG will integrate

The optimal rays are plotted using the subroutine `do_plot`.

```vbnet
Sub do_plot()
    ' use the ActiveX control GRAPH to plot the optimal ray
    GRAPH.PenColor red_color, green_color, blue_color
    GRAPH.PenWidth = 2
    GRAPH.Getdata num_pts, x_pos(0), y_pos(0)
End Sub
```

**Code excerpt 5.4** The VBScript that plots the results by calling the ActiveX component GRAPH

Figure 5.1 shows the Web page before any computations are performed. Illustrative results of the ray tracing are presented in Figures 5.2 and 5.3, and indicates that the model behaves as expected. The results of the ray tracing (not presented here in full) show that the model behaves as expected. For a given decay factor, red light is deviated more than blue. Also at sufficiently high decay factors and

![Figure 5.1](image-url)
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Figure 5.2 Plot of optimal red rays, the decay factor is 1

Figure 5.3 Plot of optimal red rays, the decay factor is 15
radial distances the refractive index is effectively unity. This leads to more uniformity in the medium and less curvature in the light ray. It should be mentioned that it would be easy, within a Web browser, to edit the user-defined integrand (within the source of the Web page) so as to model different spatial distributions of the refractive index, or even change the optimization and plot settings to model completely new problems.

5.2 PORTFOLIO ALLOCATION EXAMPLE

This demonstration considers an optimal portfolio selection problem, see Markowitz (1989) and Markowitz (1994), of the type:

\[
\text{minimize } V = X^T CX
\]

subject to the following constraints:

\[
E = \mu X, \quad \sum_{i=1}^{n} X_i = 1, \quad L_i < X_i < U_i, \quad i = 1, \ldots, n
\]

where \( E \) is the expected portfolio return, \( V \) is the portfolio risk, \( \mu \) is the vector of expected asset returns, \( C \) is the covariance matrix, \( X \) is the vector of assets, and \( L_i, U_i \) are the respective lower and upper bounds on the \( i \)th asset.

Here we provide a further example of using the numerical optimization component to solve the Markowitz portfolio optimization problem.

Figure 5.4 shows the Web page before any computations are performed, and Figure 5.5 displays typical results.

The complete code is given in the CD ROM which accompanies the book.

```html
<!DOCTYPE html>
<head>
<title>VBScript demonstration of Markowitz optimization on a web page</title>
</head>
<body>
<script language="VBScript">
<!--
Sub CommandButton1_Click()
    nclin = 2  'number of linear constraints
    n = 4  'number of variables
    ncnlin = 0  'number of nonlinear constraints
    num_pts = 15
    For i = 1 To num_pts - 1  ' Calculate the points on the efficient frontier
        portfolio_return(i) = 0.01 + 0.006*i  ' Set the required portfolio return
        bl(n+1) = portfolio_return(i)
        bu(n+1) = portfolio_return(i)
    Next
    a(6) = asset_returns(2)
    a(7) = asset_returns(3)
    ' perform the numerical optimization
    OPTIM.optimize n, nclin, ncnlin, a(0), tda, g(0), loc_x(0), bl(0), bu(0)
    OPTIM.getvars loc_x(0), n  ' load the optimal portfolio into loc_x
    x_pos(i) = x_start + portfolio_risk(i)*20000
    y_pos(i) = y_start - portfolio_return(i)*2000
Next
GRAPH.text "Return", 50, 15
For i = 1 To num_pts - 1  ' print the values
    GRAPH.text CStr(portfolio_return(i)), 50, 40+i*20
    GRAPH.text CStr(portfolio_risk(i)), 120, 40+i*20
</script>
</body>
</html>
```
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Code excerpt 5.5  VBScript code fragments for the portfolio optimization problem

Figure 5.4  Web page before numerical optimization has been performed
5.3 NUMERICAL OPTIMIZATION WITHIN MICROSOFT EXCEL

Microsoft Excel is widely used in the finance community, and there are many situations where numerical optimization needs to be performed on the data contained within an Excel spreadsheet. Numerical optimization involves the minimization (or maximization) of a specified objective function. It is often helpful to monitor the progress of the optimization using a print function which outputs intermediate values such as:

- The major iteration count.
- The number of minor iterations required by the feasibility and optimality phases of the QP subproblem.
- The step taken along the computed search direction. On reasonably well behaved problems the unit step will be taken as the solution is approached.
- The intermediate solution vector.
- The value of the augmented Lagrangian merit function at the current iterate. This will usually decrease at each iteration. As the solution is approached it will converge to the value of the objective function at the solution.
- The Euclidean norm of the projected gradient. This will be approximately zero in the neighbourhood of a solution.

In this example we illustrate the use of an ActiveX optimization component which permits users to specify both their objective function and print function using the version of Visual Basic (VBA) within Excel (see Figure 5.6). Although this approach is slower than
coding everything in C++ it is far more convenient, and it is still possible to solve optimization problems involving several hundred variables in a few seconds.

Here the objective function is called \texttt{my\_objfun}, and the print function \texttt{printit}. The variable bounds and the upper and lower constraints can be easily changed by altering the contents of the appropriate Excel spreadsheet cells. When the button \texttt{Solve1} is clicked the Visual Basic subroutine \texttt{Solve1\_click()} is run and input data such as the initial values and the upper and lower constraints, are read from the spreadsheet. The numerical optimization is then performed and the intermediate output, Figure 5.7, and computed results are written to the appropriate Excel worksheet.

Illustrative fragments from the Visual Basic code are given in Code excerpt 5.6 below.

```vbnet
Private Sub Solve1_Click()
    Dim x() As Double
    Dim bl() As Double
    Dim bu() As Double
    Dim g() As Double

    ' Input the initial values and bounds from the spreadsheet
    For i = 0 To n - 1
        x(i) = Cells(8 + i, 2).Value ' Initial values for X variables
        bl(i) = Cells(2 + i, 2).Value ' Lower bounds for X variables
        bu(i) = Cells(3 + i, 2).Value ' Upper bounds for X variables
    Next i

    objname = "my\_objfun"
    full_objname = ActiveWorkbook.Name & "!" & objname

    ' Set the name of the print function
    OPTIM1.printfun\_funname "printit"

    ' Set the objective function name
    OPTIM1.objfun\_funname full_objname
```

Figure 5.6 Excel worksheet with component OPTIM1 before numerical optimization is performed.
**ActiveX components and numerical optimization**

Call the optimizer

```vba
OPTIM1.optimize n, nclin, ncnlin, a(0), tda, g(0), x(0), bl(0), bu(0)
objfun_value = OPTIM1.objf ' get the value of the objective function
```

Output the X variable values for the optimal solution

```vba
For i = 0 To n - 1
    Cells(i + 10, 2 + i).Value = x(i)
Next i
Cells(11, 2).Value = objfun_value ' Output the optimal value of the objective function
End Sub
```

Sub my_objfun(num_variables As Long)
    ' The objective function — any valid Visual Basic code is allowed
    objective_value = x(0) * x(3) * (x(0) + x(1) + x(2)) + x(2)
End Sub

Sub printit(n As Long, it_maj_prt As Long, sol_prt As Long, maj As Long, mnr As Long, step As Double, nfun As Long, merit As Double, violtn As Double, norm_gz As Double, cond_hz As Double, x_ptr As Long)
    ' The user-defined print function. The user can decide the format in which any of
    ' the twelve arguments to printit are to be output.
    Dim xp() As Double
    ReDim xp(n)
    If (it_maj_prt) Then ' A major iteration
        Cells(Row, 1).Value = maj ' The major iteration count
        Cells(Row, 2).Value = mnr ' The number of minor iterations of the QP subproblem
        Cells(Row, 3).Value = Format(step, "0.00E+00") ' The step length along the
        ' search direction
        For i = 0 To n - 1 ' Output the current X variable values
            Cells(Row, 4 + i).Value = Format(xp(i), "##.00")
        Next i
        ' Output the value of the augmented Lagrangian merit function at the current point.
    End Sub
```

**Code excerpt 5.6** Fragments of Visual Basic code which illustrate how to call the ActiveX numerical optimization component OPTIM1 from Excel

**Figure 5.7** Excel; intermediate (monitoring) information from component OPTIM1 as numerical optimization is performed
Chapter 6

XML and transformation using XSL

6.1 INTRODUCTION

When numeric computation is performed it is necessary to decide on how the input data and output results will be presented. One approach is for the input data and output results to be contained in simple text files; for instance as comma separated values (CSV). This may have the advantages of simplicity and compact file size for large data sets. There may also be other benefits; for instance CSV files can be directly read into Microsoft Excel. However, this approach has the following disadvantages:

- The files are unstructured, and their contents cannot be easily checked to ensure that they are valid. Possible errors include: a numeric value is not within the required range, a floating point number is where an integer number should be, a string is where a floating point number should be, etc.
- It is not easy to transform the files. For instance it may be necessary to generate a specified subset of the information contained in a file or to visualize its contents graphically.
- The files cannot be directly viewed as Internet Web pages.

Of course many non-standard solutions to these problems can be found, if there are sufficient resources to create the necessary computer programs.

In order to address these issues the World Wide Web Consortium (W3C) provided the specification for an Extensible Markup Language (XML). In this section we discuss how XML files can be used for the structured storage and retrieval of information.

XML files can be directly viewed as Web pages. They can also have their contents validated by using an appropriate schema, and can be automatically (as they are loaded into the Web browser) transformed into HTML by using an Extensible Stylesheet Language (XSL) file.

Here we provide an example of a financial application in which both stock market data and financial analysis information (based on numerical optimization) are contained in a single XML file. The example application uses two XSL files: one displays the stock market data as a table of share prices, the other shows a summary report based on the financial analysis information.
6.2 XML

An XML file contains tagged values; the XML elements. A simple XML element is represented as:

```xml
<TAG>VALUE</TAG>
```

For example a share priced at 170.0 pence and with an annual return of 0.1 could be tagged as follows:

```xml
</PRICE>170.0</PRICE><RETURN>0.1</RETURN>
```

This format is not very useful because we haven’t provided the name of the share. This can be achieved by using an XML element containing an attribute. An XML element with an attribute is represented as:

```xml
<TAG1 ATTRIBUTE=VALUE1><TAG2>VALUE2</TAG2></TAG1>
```

For example if BT.A shares are priced at 170.0 pence, with annual return of 0.1, and BP shares are priced at 440.0 pence, with an annual return of 0.18, then this can be tagged as:

```xml
<ITEM SHARE=“BT.A”></PRICE>170.0</PRICE><RETURN>0.1</RETURN></ITEM>
<ITEM SHARE=“BP”></PRICE>440.0</PRICE><RETURN>0.18</RETURN></ITEM>
```

This technique can be used to describe all the information contained in an XML file. As an example let us consider an XML file that contains daily information concerning the prices and annual returns of nine shares. The file is also assumed to contain the results of analysis which give the average annual returns and optimal holdings (based on portfolio optimization).

In Code excerpt 6.1 we give an outline of the structure of the XML file stockmarket_data.xml, which is used to contain this information. It can be seen that:

- All the information is contained within the XML element STOCK_DATA.
- The XML element STOCK_DATA is composed of the XML elements ALL_DATA and PORTFOLIO_ANALYSIS, which hold the complete share data and portfolio analysis results respectively.
- All the portfolio analysis results are contained in the XML element PORTFOLIO_ANALYSIS which is made up of several PORT_ITEM elements.
- All the stock market data is held in the XML element ALL_DATA. This element is made up of DATA_REC XML elements; one for each day of stock market data. The DATA_REC element is in turn composed of two XML elements: the single element DAY, which gives the day of the month, and an ITEM element for each share, to store the daily price and current annual return.
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Code excerpt 6.1  The overall structure of the file stockmarket_data.xml used to contain both share prices information and portfolio analysis results. String values are denoted by string and floating point numbers are denoted by real

In Code excerpt 6.2 we give a more complete code fragment of the XML file stockmarket_data.xml to show in more detail the information that is actually stored. It can be seen that the XML file makes reference to the schema file stockmarket_data.xdr. This file specifies the allowed XML elements, the order the elements occur in the file, and also the permitted data types contained within the XML elements.
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Code excerpt 6.2  Fragment of the XML file stockmarket_data.xml containing both share prices information and portfolio analysis results. The file uses the schema contained in stockmarket_data.xdr and the XSL in report_style.xsl

More detail concerning schema are given in Section 6.3 below.

6.3 XML SCHEMA

As previously mentioned the structure and contents of an XML file can be checked by using an appropriate schema. There are many different schemas available, here we will consider the XML Data Reduced (XDR) schema that is supported by Microsoft Internet Explorer 6. Some of the commonly used data types supported by this schema are:

- r4: a four byte real number.
- i1: a single byte signed integer.
- i4: a four byte signed integer.
- u1: a single byte unsigned integer.
- string: character data.

The schema for the XML file stockmarket_data.xml is given below. Here both the contents and attributes of XML the elements are defined. XML attributes are defined by using the AttributeType tag. For example the following line:

```
ATTRIBUTE_TYPE name="SHARE" dt:type="string" required="yes" />
```

defines a character string attribute SHARE.

The contents of an XML element are defined by using the ElementType tag. This can take the form of a single line, for example:

```
ELEMENT_TYPE name="DAY" content="textOnly" dt:type="i4" />
```

defines the XML element DAY, which takes a four byte signed integer. It is also possible to define more complex XML elements. For instance:

```
<ElementType name="ITEM" content="eltOnly">
  <attribute type="SHARE"/>
  <element type="PRICE"/>
  <element type="RETURN"/>
</ElementType>
```
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Here the keyword content=“eltOnly” means that the XML element ITEM is only permitted to contain previously defined XML elements. In this example ITEM is defined to have a character string attribute called SHARE, and contain the XML elements PRICE and RETURN; in that order.

The complete XDR schema for the XML file stockmarket_data.xml is given in Code excerpt 6.3 below.

```xml
<?xml version="1.0"?>
<!-- This is the validation file stockmarket_data.xdr -->
<Schema xmlns="urn:schemas-microsoft-com:xml-data" xmlns:dt="urn:schemas-microsoft-com:datatypes">
  <ElementTypename="DAY" content="textOnly" dt:type="i4"/>
  <ElementTypename="PRICE" content="textOnly" dt:type="r4"/>
  <ElementTypename="RETURN" content="textOnly" dt:type="r4"/>
  <AttributeType name="SHARE" dt:type="string" required="yes"/>
  <ElementTypename="AVERAGE_RETURN" content="textOnly" dt:type="r4"/>
  <ElementTypename="OPTIMAL_HOLDING" content="textOnly" dt:type="r4"/>
  <ElementTypename="ITEM" content="eltOnly">
    <attribute type="SHARE"/>
    <element type="PRICE"/>
    <element type="RETURN"/>
  </ElementTypename>
  <ElementTypename="PORT_ITEM" content="eltOnly">
    <attribute type="SHARE"/>
    <attribute type="FULL_NAME"/>
    <element type="AVERAGE_RETURN"/>
    <element type="OPTIMAL_HOLDING"/>
  </ElementTypename>
  <ElementTypename="DATA_REC" content="eltOnly">
    <element type="DAY"/>
    <element type="ITEM"/>
  </ElementTypename>
  <ElementTypename="ALL_DATA" content="eltOnly" order="many">
    <element type="DATA_REC"/>
  </ElementTypename>
  <ElementTypename="PORTFOLIO_ANALYSIS" content="eltOnly">
    <element type="PORT_ITEM"/>
  </ElementTypename>
  <ElementTypename="STOCK_DATA" content="eltOnly" order="seq">
    <element type="ALL_DATA"/>
    <element type="PORTFOLIO_ANALYSIS"/>
  </ElementTypename>
</Schema>
```

Code excerpt 6.3 The XDR schema file stockmarket_data.xdr, used by the XML file stockmarket_data.xml

Once we have defined the schema the XML file can be validated using it. In Figure 6.1 we show the validation error caused when stockmarket_data.xml contains following invalid XML:

```xml
<DAY>1.1</DAY>
<ITEM SHARE="BT.A"><PRICE>170.50</PRICE><RETURN>0.10</RETURN></ITEM>
<ITEM SHARE="OOM"><PRICE>31.73</PRICE><RETURN>0.20</RETURN></ITEM>
```

Here the contents of the XML element DAY, which should be a four byte integer, have instead been replaced by a floating point number.
In this section we will briefly describe the Extensible Stylesheet Language (XSL), and show how it can be used to transform XML files into HTML files. The transformation from XML to HTML occurs dynamically as the XML file is loaded into a Web browser, and is achieved by interpreting the contents of an associated XSL file. This means the manner in which information contained in single XML file is displayed within a Web browser entirely depends on the associated XSL file. We will now describe a few of the features of XSL. It contains the usual features that one might expect, for instance there is:

- Iteration through a list of items using `<xsl:for-each>` and variable assignment using `<xsl:variable>`.

```xml
<xsl:for-each select="stock_xdr:ITEM">
  <xsl:variable name="v1" select="@SHARE"/>
  <xsl:variable name="v2" select="stock_xdr:PRICE"/>
</xsl:for-each>
```

Sets the variable v1 to the value of the attribute SHARE and the variable v2 to the value contained in the child element PRICE.


Figure 6.1 Validation error for the XML file `stockmarket_data.xml`; the value for `DAY` should be an integer but the XML file contains the floating point number 1.1 instead.
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Here if the variable \texttt{return} is less than 0.1 then the background colour of the cell is set to pink to indicate a bad share, but if value of \texttt{return} is greater than 0.2 then the background colour of the cell is set to yellow and red stars are output to indicate that this is a good share. If the value of \texttt{return} is between 0.1 and 0.2 then the code contained in the \texttt{xsl:otherwise} clause is executed and the background colour is just set to yellow.

It is also possible to create procedures in XSL, for instance:

```xml
<xsl:template name="OUTPUT_ELEMENT">
  <xsl:param name="share"/>
  <xsl:param name="price"/>
  <xsl:param name="return"/>
  <xsl:choose>
    <xsl:when test="$return &lt; 0.10">
      <td bgcolor="pink" align="center"><xsl:value-of select="$price"/></td>
    </xsl:when>
    <xsl:otherwise>
      <td bgcolor="yellow" align="center"><xsl:value-of select="$price"/></td>
    </xsl:otherwise>
  </xsl:choose>
</xsl:template>
```

defines an XSL procedure called \texttt{OUTPUT_ELEMENT} with parameters \texttt{share}, \texttt{price}, and \texttt{return}. It can be called using the following syntax:

```xml
<xsl:call-template name="OUTPUT_ELEMENT">
  <xsl:with-param name="share" select="$v1"/>
  <xsl:with-param name="price" select="$v2"/>
  <xsl:with-param name="return" select="$v3"/>
</xsl:call-template>
```

where, for instance, the parameter \texttt{price} is given the value of the XSL variable \texttt{v2}; the variables \texttt{v1}, \texttt{v2}, and \texttt{v3} are assumed to have been set earlier in the code.

6.5 STOCK MARKET DATA EXAMPLE

In this section we give an example of how the contents of an XML file can be displayed in very different ways depending on the XSL stylesheet used. We will only provide short code excerpts; the complete code for this example is provided on the CD ROM.
The XML file used here is called stockmarket_data.xml and has been mentioned earlier in Sections 6.2 to 6.4. This file contains the daily prices and annual returns for nine shares. We assume that the data has been processed by a numerical optimizer which has computed an optimal (Markowitz minimum risk/maximum return) portfolio of these shares, and that the results of these computations have been written to the XML file. As shown in Code excerpt 6.2, the portfolio analysis results are stored between the XML tags <PORTFOLIO_ANALYSIS> and </PORTFOLIO_ANALYSIS>.

Here we will use two different XSL files to visualize the XML file either as data in tabular form (Figure 6.2) or as a report file (Figure 6.3), in which summary information concerning the optimal portfolio is shown.

The XSL stylesheet used to create the report view of the XML data file is shown below in Code excerpt 6.4. All the stock market data is matched using the XSL statement <xsl:template match="stock_xdr:DATA_REC">, and (because we are only interested summary information) produces no output. By contrast the XSL command <xsl:template match="stock_xdr:PORTFOLIO_ANALYSIS"> matches the portfolio analysis results and creates the HTML output seen in Figure 6.3.

```xml
<xsl:stylesheet version="1.0" xmlns:xsl="http://www.w3.org/1999/XSL/Transform"
 xmlns:stock_xdr="x-schema:stockmarket_data.xdr">
  <xsl:template match="/">
    <html>
      <head>
        <title>PORTFOLIO ANALYSIS REPORT</title>
      </head>
      <body>
        <xsl:apply-templates />
      </body>
    </html>
  </xsl:template>
  <xsl:template match="stock_xdr:PORTFOLIO_ANALYSIS">
    <h1 align="center"><i>PORTFOLIO ANALYSIS REPORT</i></h1>
    This report gives details of the optimal (minimum risk, maximum return) portfolio that can be constructed from the share information contained in the file stockmarket_data.xml. The optimal holdings were calculated using numerical optimization. The information is presented in the following format:
    <font color="blue">company name (epic code)</font>,
    <font color="green">annual return</font>,
    <font color="red">optimal portfolio holding</font>.
    High performing companies are starred. <p></p>
    <xsl:for-each select="stock_xdr:PORT_ITEM">
      <xsl:if test="@FULL_NAME &gt; 0.19">
        <b><font color="red">**</font></b>
      </xsl:if>
      <font color="blue"><xsl:value-of select="@FULL_NAME"/></font>,
      <font color="green"><xsl:value-of select="stock_xdr:AVERAGE_RETURN"/></font>,
      <font color="red"><xsl:value-of select="stock_xdr:OPTIMAL_HOLDING"/></font>;
    </xsl:for-each>
  </xsl:template>
  <xsl:template match="stock_xdr:DATA_REC">
    <!-- DO NOT OUTPUT ANY STOCK DATA IN THIS STYLE SHEET-->
  </xsl:template>
</xsl:stylesheet>
```

**Code excerpt 6.4** The XSL file report_style.xsl used to transform the XML file stockmarket_data.xml into the report view shown in Figure 6.3.
Figure 6.2  The tabular view of the XML data file stockmarket_data.xml displayed using the Web browser Internet Explorer 6; the XSL style sheet is available on the CD ROM

Figure 6.3  The report view of XML data file stockmarket_data.xml displayed using the Web browser Internet Explorer 6; the XSL style sheet is given in Code excerpt 6.4
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The report view also includes a Scalable Vector Graphics (SVG) pie chart `report_pie.svg` to display the portfolio composition. Here the SVG graphics were viewed by installing the Adobe SVG Viewer, which can be freely downloaded from [http://www.adobe.com/svg/](http://www.adobe.com/svg/).

The XSL statement for including the SVG graphic is:

```xml
<embed src="report_pie.svg" width="500" height="500" name="SVGEmbed" type="image/svg+xml" pluginspage="http://www.adobe.com/svg/viewer/install/"/>
```

where the image source is specified by using the `src` attribute, and the size of the image is controlled via `width` and `height` attributes.
Chapter 7

Epilogue

7.1 WRAPPING C WITH C++ FOR OO NUMERICS IN .NET

7.1.1 Introduction

A common software requirement is for code written in one computer language to be used by software developed in a different computer language. For instance it may be necessary to access C functions from a .NET or Java application. Here we show how existing C software can be wrapped in C++ and thus made easily accessible from .NET languages such as C# and VB.NET, see Levy (2003). Although our discussions will be concerned with the NAG C library, the method is quite general and can be applied to other C software. We illustrate the technique by considering four NAG C library numeric routines, which have applications in computational finance. These functions are:

- NAG function s15abc; the cumulative normal distribution which is used in analytic option pricing formulae, such as the Black–Scholes equation, see Black and Scholes (1973).
- NAG function f02aac; eigenvalue computation. This has applications in multifactor models, including interest-rate models, and time series, see Rebonato (1998) and Levy (2003).
- NAG function d01ajc; numerical integration, which has applications in risk analysis, see Hull (1997).
- NAG function e04dgc; numerical optimization. This can be used to compute optimal portfolios, see Markowitz (1994).

For more details concerning these functions see NAG Ltd (2003).

7.1.2 COM, .NET assemblies and managed C++

Microsoft COM already enables the creation of numeric components which can be used by the complete range of Windows programming languages, see Levy (2001). In fact COM objects can be used within .NET.

However, from a software developer’s point of view, wrapping C code in COM C++ classes has the disadvantage that there is a lot of visible Microsoft COM baggage that needs to be carried around. This has the effect of obscuring the code and also making it difficult to implement the C++ classes on UNIX platforms. Another limitation is that the classes contained within a COM object cannot be used to create other derived classes.
The recent introduction of .NET assemblies has now substantially improved this situation. Briefly, the classes in an assembly can be coded in any of the .NET languages and then used by any other .NET language. It is thus possible to create assemblies in managed C++ that provide class wrappers for C routines, and then use these from C# and VB.NET software, see Challa and Laksberg (2002).

Code excerpt 7.1 shows the ANSI function prototypes of our four NAG C functions. The managed C++ code used to create an assembly that wraps the C functions is displayed in Code excerpt 7.2. We have called this assembly naglib, and it defines the namespace NAGLIB and the managed class NAG_FUNCTIONS which provides functions to access native C routines contained within the DLL 'nagc'. The Code excerpt 7.2 is only meant for illustrative purposes and is not intended to be a statement of good programming practice. However, we have included some useful features such as flagging errors and setting default parameter values via the constructor NAG_FUNCTIONS().
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[DllImport ("nagc")] extern "C" Double s15abc(Double x);

public _ gc class NAG_FUNCTIONS
{
    public:
        Double QUADRATURE_epsabs;
        Double QUADRATURE_epsrel;
        Int32 QUADRATURE_max_subint;
    
    NAG_FUNCTIONS()
        { // the constructor: set default values
            QUADRATURE_epsabs = 0.0;
            QUADRATURE_epsrel = 0.0001;
            QUADRATURE_max_subint = 200;
        }
    
    void REAL_SYMM_EIGEN (Int32 n, Double *a, Int32 tda, Double *r, Int32 *flag)
        {
            NagError eflag;
            INIT_FAIL(eflag);
            f02aac(n, a, tda, r, &eflag);
            *flag = (Int32)eflag.code;
        }
    
    Double CUM_NORM (Double x) {
        return s15abc(x);
    }
    
    void OPTIMIZE (Int32 n, Double *x, Double *g, Double *objf, Int32 *flag, OBJ_FUN_TYPE *the_fun)
        {
            NagError eflag;
            Nag_E04_Opt options;
            INIT_FAIL (eflag);
            e04xxc (&options);
            options.print_level = Nag_NoPrint;
            options.list = 0;
            options.verify_grad = Nag_NoCheck;
            e04dgc(n, the_fun, x, objf, g, &options, (Int32)0, &eflag);
            *flag = (Int32)eflag.code;
        }
    
    void QUADRATURE (Double a, Double b, Double *result, Double *abserr, Int32 *flag, INTEGRAND_FUN_TYPE *the_fun)
        {
            Nag_QuadProgress qp;
            NagError eflag;
            INIT_FAIL(eflag);
            d01ajc(the_fun, a, b, QUADRATURE_epsabs, QUADRATURE_epsrel, QUADRATURE_max_subint, result, abserr, &qp, &eflag);
            *flag = (Int32)eflag.code;
        }
};

Code excerpt 7.2 The managed C++ code used to create the assembly naglib which contains the namespace NAGLIB, and wraps the NAG C library functions in the class NAG_FUNCTIONS.

It can be seen that the code is almost standard C++ and (in constrast to the equivalent COM approach) could easily be ported to UNIX platforms. We will now consider each non-standard C++ (that is Microsoft specific) feature in turn.

Importing Dynamic Link Library (DLL) functions

The NAG C library routines used are contained in a DLL called ‘nagc’. Here each function is imported into the C++ project by name; for instance:

[DllImport ("nagc")] extern "C" Double s15abc(Double x);
is used to import the function \texttt{s15abc}, which computes the cumulative normal distribution.

**Managed and unmanaged code**

The directive \_\_gc indicates that the code is managed and memory is allocated on the garbage collected (GC) heap; unmanaged code is indicated by \_\_nogc.

**The data types Double and Int32**

In Code excerpt 7.2 the .NET data types \texttt{Double} and \texttt{Int32} have been used so that the assembly can be accessed by both C\# and VB.NET code.

All managed .NET code, written in VB.NET, C\#, and C++, is compiled to the same intermediate language (IL) code. In order to permit interoperability within .NET there is a common type system (CTS) which standardizes the basic data types across all languages.

A summary of the .NET data types corresponding to the C\++ types \texttt{double} and \texttt{long} is given in the Table 7.1.

**Delegates**

In the case of numerical integration and optimization a user-defined function, or callback function, needs to be passed as a parameter to the NAG C library routine. This is achieved in .NET by declaring a delegate with the same signature (that is return type and parameter types) as the callback function. For example

```csharp
public delegate double INTEGRAND_FUN_TYPE(Double x);
```

declares the delegate \texttt{INTEGRAND\_FUN\_TYPE} with a signature corresponding to functions that return a \texttt{Double} and have a single \texttt{Double} parameter passed by value. This delegate is used by the numerical integration routine \texttt{d01ajc} for defining the integrand. It can be seen that the declaration of a delegate is similar to the declaration of a function prototype with the additional words public (or private) and delegate. Also the declaration and use of delegates in Code excerpt 7.2 has similarities with the declaration and use of function pointers in Code excerpt 7.1.

A more complicated delegate example is:

```csharp
public delegate void OBJ_FUN_TYPE(Int32 n, double *x, double *objf, double *g, Int32 comm);
```

**Table 7.1** The correspondence between data types used by C++\+, C\#, VB.NET, and the .NET CTS

<table>
<thead>
<tr>
<th>C++</th>
<th>C#</th>
<th>VB.NET</th>
<th>CTS</th>
<th>Size in bytes</th>
</tr>
</thead>
<tbody>
<tr>
<td>long</td>
<td>int</td>
<td>Integer</td>
<td>Int32</td>
<td>4</td>
</tr>
<tr>
<td>double</td>
<td>double</td>
<td>Double</td>
<td>Double</td>
<td>8</td>
</tr>
</tbody>
</table>
which declares the delegate `OBJ_FUN_TYPE`, with a signature that applies to subroutines (that is a functions which return void) with parameters of type `Int32` and `double*`. As can be seen, here it was found necessary to use `double*` instead of the more general `Double*`. This delegate is used by the numerical optimization routine `e04dgc` for specifying the objective function to be minimized.

### 7.1.3 Accessing the assembly `naglib` from C#

In this section we show how the previously described assembly `naglib` can be accessed from a C# console project created using Visual Studio .NET. The C# code is presented in Code excerpt 7.3 and a screen view of the project is shown in Figure 7.1.

It can be seen that the C# code defines the two classes `DCLASS` and `RUNIT`. `DCLASS` is derived from the class `NAG_FUNCTIONS` and supplies the definitions for the callback functions used by the member functions `OPTIMIZE` and `QUADRATURE`.

The class `RUNIT` only contains the member function `Main`. This function is run by the example console application, and all the computations are performed by a single numeric object (called `tt`) of type `DCLASS`.

![Figure 7.1](image-url) A view of the C# example project. The Object Browser displays the assembly `naglib`, the namespace `NAGLIB`, the delegates `INTEGRAND_FUN_TYPE`, and `OBJ_FUN_TYPE`, and also the member functions of the class `NAG_FUNCTIONS`: `CUM_NORM`, `OPTIMIZE`, `QUADRATURE`, and `REAL_SYMM_EIGEN`
using System;
using System.Runtime.InteropServices;
using NAGLIB;

namespace USE_NAGLIBS
{
    class DCLASS : NAG_FUNCTIONS
    {
        public unsafe void objfun (Int32 n, Double *x, Double *objf, Double *g, Int32 comm)
        {
            Double ex1, x1, x2;
            ex1 = Math.Exp(x[0]);
            x1 = x[0];
            x2 = x[1];
            *objf = ex1*(4.0*x1*x1 + 2.0*x2*x2 + 4.0*x1*x2 + 2.0*x1 + 1.0);
            g[0] = 2.0*ex1*(2.0*x2 + 2.0*x1 + 1.0);
            g[1] = 2.0*ex1*(2.0*x2 + 2.0*x1 + 1.0);
        }
        public Double the_integrand_c(Double x)
        {
            Double pi;
            pi = Math.PI;
            Double val;
            val = (x*Math.Sin(x*30.0)/1.0)/x*x/(pi*pi*4.0);
            return val;
        }
    }
    class RUNIT
    {
        static unsafe void Main(string[] args)
        {
            Int32 tda = 4, n = 4, n2 = 2, j, flag = 0;
            Double [] r = new Double[30];
            Double [] x2 = new Double[2];
            Double [] g = new Double[2];
            Double [] a, b1, objf = 0.0, abserr = 0.0, the_answer, x;
            Double [,] a = new Double[n,n];
            DCLASS tt = new DCLASS();
            x = 1.0;
            the_answer = tt.CUM_NORM(x);
            Console.WriteLine("The value of the cumulative normal = {0,8:F4}", the_answer);
            INTEGRAND_FUN_TYPE myfun_c = new INTEGRAND_FUN_TYPE(tt.the_integrand_c);
            OBJ_FUN_TYPE myobjfun = new OBJ_FUN_TYPE(tt.objfun);
            a1 = 0.0;
            b1 = Math.PI*2.0;
            flag = 0;
            the_answer = 0.0;
            tt.QUADRATURE(a1, b1, ref the_answer, ref abserr, ref flag, ref myfun_c);
            Console.WriteLine("The integral (default maximum number of subintervals) = {0,8:F6}", the_answer);
            flag = 0;
            tt.QUADRATURE_max_subint = 3;
            tt.QUADRATURE(a1, b1, ref the_answer, ref abserr, ref flag, ref myfun_c);
            Console.WriteLine("The integral (maximum number of subintervals set to 3) = {0,8:F6}", the_answer);
            x2[0] = -1.0;
            x2[1] = 1.0;
            n2 = 2;
            flag = 0;
            tt.OPTIMIZE(n2, ref x2[0], ref g[0], ref objf, ref flag, ref myobjfun);
            Console.Write("The optimization solution vector is:");
            for (j = 0; j < 2; ++j) {
                Console.WriteLine("{0,8:F4}", x2[j]);
            }
            Console.WriteLine();
            Console.WriteLine("The value of the objective function is: {0,8:E4}", objf);
            flag = 0;
            //first row
            a[0,0] = 0.5;
            a[0,1] = 0.0;
            a[0,2] = 2.3;
            a[0,3] = -2.6;
            //second row
            a[1,0] = 0.0;
            a[1,1] = 0.5;
        }
    }
}
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```csharp
// third row
a[2,0] = 2.3;
a[2,1] = -1.4;
a[2,2] = 0.5;
a[2,3] = 0.0;
// fourth row
a[3,0] = -2.6;
a[3,1] = -0.7;
a[3,2] = 0.0;
a[3,3] = 0.5;
tt.REAL_SYMM_EIGEN(n, ref a[0,0], td, ref r[0], ref flag);
Console.Write("The Eigenvalues are:");
for (j = 0; j < 3; ++j) {
    Console.Write("{0,8:F4}", r[j]);
}
Console.WriteLine();
```

Code excerpt 7.3 Example C# code which uses the assembly naglib in a C# console application

The value of the cumulative normal = 0.1587
The integral (default maximum number of subintervals) = -2.303835
The integral (maximum number of subintervals set to 3) = -3.168259
The optimization solution vector is: 0.5000 / 1.0000
The value of the objective function is: 2.7457E-014
The Eigenvalues are: 3.0000 / 1.0000 2.0000 4.0000

Code excerpt 7.4 The output from Code excerpt 7.3

We will now briefly discuss some of the important features of the code, for more information on C# see Robinson et al. (2001).

General points

The assembly containing the namespace NAGLIB is accessed with the statement using NAGLIB; which occurs on the third line of the C# code listing.

We use `Math.PI` to return the value of π, `Math.Sin(x)` to compute sin(x), and `Math.Exp(x)` to evaluate exp(x). These functions are members of the class `Math` which is contained in the namespace `System`.

The keyword unsafe

This directive is necessary because C# does not really support pointers. The keyword `unsafe` allows us to use pointers and thus easily pass scalars and arrays by reference to the managed C++ class `NAG_FUNCTIONS` contained in the namespace NAGLIB.

Declaring numeric objects and using simple member functions

The statement `DCLASS tt = new DCLASS()` creates a numeric object `tt` with the type of the derived class `DCLASS`. Since `DCLASS` was derived from `NAG_FUNCTIONS` it allows access not only to the public member functions `objfun` and `the_integrand_c`,
but also the public member functions of NAG_FUNCTIONS: CUM_NORM, OPTIMIZE, QUADRATURE, and REAL_SYM_EIGEN. This means that we can compute the cumulative normal distribution, and perform eigenvalue computations by using statements of the form:

```csharp
the_answer = tt.CUM_NORM(x);
flag = 0;
// first row
a[0,0] = 0.5;
a[0,1] = 0.0;
// fourth row
a[3,0] = 2.6;
a[3,1] = -0.7;
a[3,2] = 0.0;
a[3,3] = 0.5;
// first row
a[0,0] = 0.5;
a[0,1] = 0.0;
// fourth row
a[3,0] = 2.6;
a[3,1] = -0.7;
a[3,2] = 0.0;
a[3,3] = 0.5;
tt.REAL_SYM_EIGEN(n, ref a[0,0], tda, ref r[0], ref flag);
```

We note that the keyword ref is used to pass the address of a[0, 0], r[0], and flag to the member function REAL_SYM_EIGEN.

Using numeric objects with member functions requiring delegates

We will now consider how to call the numerical integration function QUADRATURE. This is achieved using the following C# statement:

```csharp
INTEGRAND_FUN_TYPE myfun_c = new INTEGRAND_FUN_TYPE (tt.the_integrand_c);
```

to declare (and also define) the delegate myfun_c, of type INTEGRAND_FUN_TYPE, which corresponds to the user-defined function the_integrand_c contained in the derived class DCLASS. The next step is to pass the appropriate parameters to the function tt.QUADRATURE; for example:

```csharp
a1 = 0.0;
b1 = Math.PI*2.0;
flag = 0;
the_answer = 0.0;
// first row
a[0,0] = 0.5;
a[0,1] = 0.0;
// fourth row
a[3,0] = 2.6;
a[3,1] = -0.7;
a[3,2] = 0.0;
a[3,3] = 0.5;
tt.REAL_SYM_EIGEN(n, ref a[0,0], tda, ref r[0], ref flag);
```

The method of calling the numerical optimization member function is very similar. For instance in the example code we use:

```csharp
OBJ_FUN_TYPE myobjfun = new OBJ_FUN_TYPE (tt.objfun);
x2[0] = 1.0;
x2[1] = 1.0;
n2 = 2;
flag = 0;
tt.OPTIMIZE(n2, ref x2[0], ref g[0], ref objf, ref flag, myobjfun);
```

The initial parameter estimates and computed optimal values are contained in the array x2. The estimated gradient at the solution point is returned in the array g and the parameter objf contains the value of the minimized objective function.
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7.1.4 Accessing the assembly naglib from VB.NET

Here we illustrate how the assembly naglib can be used from VB.NET; for more details on VB.NET see Barwell et al. (2002).

The assembly naglib can be used from VB.NET in a similar manner to that described for C#. This is illustrated below in Code excerpt 7.5.

Imports System
Imports System.Runtime.InteropServices
Imports NAGLIB

Module Module1
    Public Class DCLASS
        Inherits NAG_FUNCTIONS

        Public Function the_integrand_c(ByVal x As Double) As Double
            Dim pi As Double
            Dim val As Double
            pi = Math.PI
            val = (x * Math.Sin(x) / 1 - x * x / (pi * pi * 4))
            Return val
        End Function
    End Class

    Sub Main()
        Dim x, the_answer As Double
        Dim flag, j, tda, n As Integer
        Dim a(,), r(), a1, b1, abserr As Double
        Dim tt As New DCLASS()
        Dim myfun As New INTEGRAND_FUN_TYPE(AddressOf tt.the_integrand_c)
        a1 = 0
        b1 = Math.PI * 2
        flag = 0
        tt.QUADRATURE(a1, b1, the_answer, abserr, flag, myfun)
        Console.WriteLine("The integral (default number of subintervals) = {0:F4}", the_answer)
        tt.QUADRATURE_max_subint = 3
        flag = 0
        tt.QUADRATURE(a1, b1, the_answer, abserr, flag, myfun)
        Console.WriteLine("The integral (number of subintervals set to 3) = {0:F4}", the_answer)
        x = -1
        the_answer = tt.CUM_NORM(x)
        Console.WriteLine("The value of the cumulative normal = {0:F4}", the_answer)

        flag = 0
        n = 4
        tda = n
        ReDim r(n - 1)
        ReDim a(n - 1, n - 1)

        'first row
        a(0, 0) = 0.5
        a(0, 1) = 0
        a(0, 2) = 2.3
        a(0, 3) = -2.6

        'second row
        a(1, 0) = 0
        a(1, 1) = -0.5
        a(1, 2) = -1.4
        a(1, 3) = -0.7

        'third row
        a(2, 0) = 2.3
        a(2, 1) = -1.4
        a(2, 2) = 0.5
        a(2, 3) = 0

        'fourth row
        a(3, 0) = -2.6
        a(3, 1) = -0.7
        a(3, 2) = 0
        a(3, 3) = 0.5
tt.REAL_SYM_EIGEN(n, a(0, 0), tda, r(0), flag)
Console.WriteLine("The Eigenvalues are:")
For j = 0 To n - 1
    Console.WriteLine("{0,8:F4}". r(j))
Next j
Console.WriteLine()
End Sub
End Module

Code excerpt 7.5  Example of using the numeric objects from VB.NET

The integral (default number of subintervals) = 2.3038
The integral (number of subintervals set to 3) = 3.1683
The value of the cumulative normal = 0.1587
The Eigenvalues are: 3.0000 1.0000 2.0000 4.0000

Code excerpt 7.6  The output from Code excerpt 7.5

7.1.5 Conclusions

We have shown how to wrap C code in a managed C++ assembly, which can then be used from within either a C# or VB.NET project.

A major benefit of this approach over COM is that the managed C++ wrapper code can with little effort, be used on UNIX platforms. In addition, unlike COM, it is possible to create C# or VB.NET derived classes from the managed C++ (base) classes.

As more software supports .NET (for example Excel 2003 will) the future of OO numerics in .NET looks increasingly promising.

7.2 FINAL REMARKS

In this part of the book we have discussed various ways in which the Windows environment can be used to develop financial software. The creation of DLLs and their incorporation into Visual Basic, VB.NET, C#, and Excel has been considered. We have also described how numeric ActiveX components, with primitive visual user-interfaces, can be used from within Visual Basic, Delphi, and HTML Web pages. In addition examples of how XML, XSL, and SVG can be used to represent and display financial information from within a Web browser have been given.

With so many choices now available a software developer needs to carefully consider which is the most appropriate technology to use for a particular task. In making this decision the relative importance of the following will need to be made: the user-interface, computational speed, Internet access, speed of development, software portability, and the computer language(s) to use.

More information on these subjects can be found in the citations provided in the computing bibliography at the end of the book.
Part II

Pricing Assets
Chapter 8

Introduction

8.1 AN INTRODUCTION TO OPTIONS AND DERIVATIVES

In general, an option (also called a derivative, or contingent claim) is a contract whose value depends on the future values that specified underlying quantities take over a given time span. One use of options is as a means of providing insurance against certain events which may happen in the future. For instance an airport, which wants to insure against climatic risk, takes out a weather option. The contract for this option may pay out a given amount of cash when the outside temperature either exceeds or goes below certain prescribed levels.

This part of the book is concerned with financial options; that is options that are based on the future value of various financial quantities that can be determined from the financial markets. A put option is an agreement to sell an asset in the future for a fixed price the strike price, and a call is an agreement to buy an asset in the future for a given price. Furthermore European options can only be exercised at option maturity, whereas American options have greater flexibility and can be exercised at any time up to option maturity.

Here we will discuss options whose value depends on the future prices of various stocks and shares; these are called equity options. There are many different types of equity options, see Hull (1997) for more detail.

If we want to buy or sell an equity option it is very important to determine its fair value today. This will depend on the expected future values of the underlying stock values, based on our current (and historical) information.

To do this it is necessary to model how the stock value changes with time. In Part II of this book we will consider valuation models that are based on the assumption that the asset price can be described by Brownian motion. In Part III we consider more complex time series models for the asset price changes.

We will mainly be concerned with vanilla put and call options, however we do provide some detailed coverage of barrier options. In most cases it should not be too difficult to value more exotic options by modifying the supplied code.

We will consider the following computational methods for pricing options:

• Analytic methods and analytic approximations.
• Finite-difference lattices.
Finite-difference grids.
Simulation: Monte Carlo, using pseudorandom and quasirandom numbers.

We will discuss Brownian Motion and derive the Black–Scholes formula which is used for pricing European options. We will also derive formulae for the value of some commonly used European barrier options.

The value of a standard vanilla option depends on:

- The volatility of the underlying stock.
- The time to maturity.
- The strike price.
- The riskless interest rate.
- The dividends.
- Current value of the stock.

In the Black–Scholes setting the asset prices are assumed to follow a lognormal process. This means that the logarithm of the asset prices has a Gaussian distribution, and the asset returns can be modelled as a Brownian process.

8.2 BROWNIAN MOTION

Brownian motion is named after the botanist Robert Brown who used a microscope to study the fertilization mechanism of flowering plants. He first observed the random motion of pollen particles (obtained from the American species Clarkia pulchella) suspended in water, and wrote:

The fovilla or granules fill the whole orbicular disk but do not extend to the projecting angles. They are not spherical but oblong or nearly cylindrical, and the particles have manifest motion. This motion is only visible to my lens which magnifies 370 times. The motion is obscure yet certain (Robert Brown, 12 June 1827; see Ramsbottom, 1932)

It appears that Brown considered this motion no more than a curiosity (he believed that the particles were alive) and continued undistracted with his botanical research. The full significance of his observations only became apparent about eighty years later when it was shown, Einstein (1905), that the motion is caused by the collisions that occur between the pollen grains and the water molecules. In 1908 Perrin, see Perrin (1909), was finally able to confirm Einstein’s predictions experimentally. His work was made possible by the development of the ultramicroscope by Zsigmondy and Siedentopf in 1903. He was able to work out from his experimental results and Einstein’s formula the size of the water molecule and a precise value for Avogadro’s number. His work established the physical theory of Brownian motion and ended the skepticism about the existence of atoms and molecules as actual physical entities. Many of the fundamental properties of Brownian motion were discovered by Levy (1939, 1948), and the first mathematically rigorous treatment was provided by Wiener (1923, 1924). Karatzas and Shreve (1988) is an excellent text book on the theoretical properties of Brownian motion, while Shreve et al. (1997) provides much useful information concerning the use of Brownian processes within finance.
Brownian motion is also called a random walk, a Wiener process, or sometimes (more poetically) the drunkard’s walk.

In formal terms a process \( Z = (Z_t : t \geq 0) \) is (one-dimensional) Brownian motion if:

(i) \( Z_t \) is continuous, and \( Z_0 = 0 \)

(ii) \( Z_t \sim N(0, t) \)

(iii) The increment \( dZ_{dt} = Z_{t+dt} - Z_t \) is normally distributed as, \( dZ_{dt} \sim N(0, dt) \), so \( E[dZ_{dt}] = 0 \) and \( \text{Var}(dZ_{dt}) = dt \). The increment \( dZ_{dt} \) is also independent of the history of the process up to time \( t \).

From (iii) we can further state that, since the increments \( dZ_{dt} \) are independent of past values \( Z_t \), a Brownian process is also a Markov process. In addition we shall now show that Brownian motion is also a Martingale process.

In a Martingale process \( P_t, t \geq 0 \), the conditional expectation \( E(P_{t+dt} | F_t) = P_t \), where \( F_t \) is called the filtration generated by the process and contains the information learned by observing the process up to time \( t \). Since for Brownian motion we have

\[
E(Z_{t+dt} | F_t) = E((Z_{t+dt} - Z_t) + Z_t | F_t) = E(Z_{t+dt} - Z_t) + Z_t = E(dZ_{t+dt}) + Z_t = Z_t
\]

where we have used the fact that \( E[dZ_{t+dt}] = 0 \). Since \( E(Z_{t+dt} | F_t) = Z_t \), the Brownian motion \( Z \) is a Martingale process.

We will now consider the Brownian increments over the time interval \( dt \) in more detail. Over the time interval \( dt \) we have:

\[
dX_{dt} = dZ_{dt}
\] (8.1)

where \( dZ_{dt} \) is a random variable drawn from a normal distribution with mean zero and variance \( dt \), which we denote as \( dZ_{dt} \sim N(0, dt) \). Equation 8.1 can also be written in the equivalent form:

\[
dX_{dt} = \sqrt{dt} \epsilon
\] (8.2)

where \( \epsilon \) is a random variable drawn from a standard normal distribution (that is a normal distribution with zero mean and unit variance), and we use the notation \( \epsilon \sim N(0, 1) \).

Equations 8.1 and 8.2 give the incremental change in the value of \( X \) over the time interval \( dt \) for standard Brownian motion.

We shall now generalize these equations slightly by introducing the extra (volatility) parameter \( \sigma \) which controls the variance of the process. We now have:

\[
dX_{dt} = \sigma dZ_{dt}
\] (8.3)

where \( dZ_{dt} \sim N(0, dt) \), and \( dX_{dt} \sim N(0, \sigma^2 dt) \). Equation 8.3 can also be written in the equivalent form:

\[
dX_{dt} = \sigma \sqrt{dt} \epsilon_i, \quad \epsilon_i \sim N(0, 1)
\] (8.4)
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or equivalently

\[ dX_{it} = \sqrt{dt} \epsilon_i', \quad \epsilon_i' \sim N(0, \sigma^2) \] (8.5)

We are now in a position to provide a mathematical description of the movement of the pollen grains in water observed by Robert Brown in 1827. We will start by assuming that the container of water is perfectly level. This will ensure that there is no drift of the pollen grains in any particular direction. Let us denote the position of a particular pollen grain at time \( t \) by \( X_t \), and set the position at \( t = 0, X_0 \), to zero. The statistical distribution of the grain’s position, \( X_T \), at some later time \( t = T \), can be found as shown below.

We divide the time \( T \) into \( n \) equal intervals \( dt = T/n \). Since the position of the particle changes by the amount \( dX_i = \sigma \sqrt{dt} \epsilon_i \) over the \( i \)th time interval \( dt \), the final position \( X_T \) is given by:

\[ X_T = \sum_{i=1}^{n} \left( \sigma \sqrt{dt} \epsilon_i \right) = \sigma \sqrt{dt} \sum_{i=1}^{n} \epsilon_i \]

Since \( \epsilon_i \sim N(0, 1) \), by the Law of Large numbers, see Appendix F.1, we have that the expected value of position \( X_T \) is:

\[ E[X_T] = \sigma \sqrt{dt} E \left[ \sum_{i=1}^{n} \epsilon_i \right] = 0 \]

The variance of the position \( X_T \) is:

\[ Var[X_T] = Var \left[ \sigma \sqrt{dt} \sum_{i=1}^{n} \epsilon_i \right] = \sigma^2 dt \sum_{i=1}^{n} Var[\epsilon_i] \] (8.6)

Using the fact that \( Var[\epsilon_i] = 1 \) and that

\[ Var \left[ \sum_{i=1}^{n} X_i \right] = \sum_{i=1}^{n} Var[X_i], \]

see Appendix F.3, we have:

\[ Var[X_T] = \sigma^2 dt \sum_{i=1}^{n} Var[\epsilon_i] = \sigma^2 dt \sum_{i=1}^{n} 1 \] (8.7)

which gives:

\[ Var[X_T] = \sigma^2 n dt = T \sigma^2 \] (8.8)

So, at time \( T \), the position of the pollen grain, \( X_T \) is distributed as \( X_T \sim N(0, T \sigma^2) \).

If the water container is not perfectly level then the pollen grains will exhibit drift in a particular direction. We can modify Equation 8.4 to take this into account as follows:

\[ dX_{it} = \mu dt + \sigma \sqrt{dt} \epsilon, \quad \epsilon_i \sim N(0, 1) \] (8.9)
or equivalently
\[ dX_{dt} = \mu dt + \sigma dZ_t, \quad dZ_t \sim N(0, dt) \] (8.10)

where we have included the constant drift \( \mu \). Proceeding in a similar manner to that for the case of zero drift Brownian motion we have:

\[ X_T = \sum_{i=1}^{n} \left( \mu dt + \sigma \sqrt{dt} \epsilon_i \right) = \mu \sum_{i=1}^{n} dt + \sigma \sqrt{dt} \sum_{i=1}^{n} \epsilon_i = \mu T + \sigma \sqrt{dt} \sum_{i=1}^{n} \epsilon_i \]

which gives

\[ E[X_T] = E\left[ \mu T + \sigma \sqrt{dt} \sum_{i=1}^{n} \epsilon_i \right] = \mu T + \sigma \sqrt{dt} E\left[ \sum_{i=1}^{n} \epsilon_i \right] = \mu T \]

The variance of the position \( X_T \) is:

\[ Var[X_T] = Var\left[ \mu T + \sigma \sqrt{dt} \sum_{i=1}^{n} \epsilon_i \right] = \sigma^2 \sum_{i=1}^{n} \epsilon_i \]

Here we have used the fact (see Appendix F.3) that \( Var[a + bX] = b^2 Var[X] \), where \( a = \mu T \), and \( b = 1 \). From Equations 8.6 to 8.8 we have:

\[ Var[X_T] = \sigma^2 \sum_{i=1}^{n} \epsilon_i = T \sigma^2 \]

So, at time \( T \), the position of the pollen grain, \( X_T \) is distributed as \( X_T \sim N(\mu T, T \sigma^2) \).

### 8.3 A BROWNIAN MODEL OF ASSET PRICE MOVEMENTS

In the previous section we showed how Brownian motion can be used to describe the random motion of small particles suspended in a liquid. The first attempt at using Brownian motion to describe financial asset price movements was provided by Bachelier (1900). This however only had limited success because the significance of a given absolute change in asset price depends on the original asset price. For example a £1 increase in the value of a share originally worth £1.10 is much more significant than a £1 increase in the value of a share originally worth £100. It is for this reason that asset price movements are generally described in terms of relative or percentage changes. For example if the £1.10 share increases in value by 11 pence and the £100 share increases in value by £10, then both of these price changes have the same significance, and correspond to a 10 per cent increase in value. The idea of relative price changes in the value of a share can be formalized by defining a quantity called the return, \( R_t \), of a share at time \( t \). The return \( R_t \) is defined as follows:

\[ R_t = \frac{S_{t+dt} - S_t}{S_t} = \frac{dS_t}{S_t} \] (8.11)
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where $S_{t+dt}$ is the value of the share at time $t+dt$, $S_t$ is the value of the share at time $t$, and $dS_t$ is the change in value of the share over the time interval $dt$. The percentage return $R_t^r$, over the time interval $dt$ is simply defined as $R_t^r = 100 \times R_t$.

We are now in a position to construct a simple Brownian model of asset price movements, further information on Brownian motion within finance can be found in Shreve et al. (1997).

The asset return at time $t$ is now given by:

$$ R_t = \frac{dS_t}{S_t} = \mu dt + \sigma dZ_t, \quad dZ_t \sim N(0, dt) \quad (8.12) $$

or equivalently:

$$ dS_t = S_t \mu dt + S_t \sigma dZ_t \quad (8.13) $$

The process given in Equations 8.11 and 8.12 is termed Geometric Brownian Motion; which we will abbreviate as GBM. This is because the relative (rather than absolute) price changes follow Brownian motion.

We will now use Ito’s lemma (see Section 8.4) which allows us to write down the process followed by the function $\phi(S, t)$, if the asset price $S$ follows GBM. Ito’s formula states, see Equation 8.21, that:

$$ d\phi = \left( \mu S \frac{\partial \phi}{\partial S} + \frac{\sigma^2 S^2}{2} \frac{\partial^2 \phi}{\partial S^2} + \frac{\partial \phi}{\partial t} \right) dt + \sigma S \frac{\partial \phi}{\partial S} dZ $$

where $d\phi$ denotes the increment in the function $\phi(S, t)$ over the time interval $dt$. This means that if we choose $\phi(S, t) = \log(S)$, then we have:

$$ \frac{\partial \phi}{\partial S} = \frac{\partial \log(S)}{\partial S} = \frac{1}{S}, \quad \frac{\partial^2 \phi}{\partial S^2} = \frac{\partial}{\partial S} \left( \frac{\partial \log(S)}{\partial S} \right) = \frac{\partial}{\partial S} \left( \frac{1}{S} \right) = -\frac{1}{S^2} $$

$$ \frac{\partial \phi}{\partial t} = \frac{\partial \log(S)}{\partial t} = 0 $$

Therefore if we let $Y = \log(S)$ we have:

$$ dY = \log \left( \frac{S_{t+dt}}{S_t} \right) = \log(S_{t+dt}) - \log(S_t) = \left( \mu - \frac{\sigma^2}{2} \right) dt + \sigma dZ, \quad dZ \sim N(0, dt) $$

or equivalently

$$ dY \sim N \left( \left\{ \mu - \frac{1}{2} \sigma^2 \right\} dt, \sigma^2 dt \right) $$

If we now substitute the riskless interest rate, $r$, for the drift in the asset price, $\mu$, we obtain the following two equations:

$$ \log \left( \frac{S_{t+dt}}{S_t} \right) = \left( r - \frac{\sigma^2}{2} \right) dt + \sigma dZ, \quad dZ \sim N(0, dt) \quad (8.14) $$

and

$$ dY \sim N \left( \left\{ r - \frac{1}{2} \sigma^2 \right\} dt, \sigma^2 dt \right) \quad (8.15) $$
We have therefore shown that if the asset price follows GBM, then the logarithm of the asset price $Y$ follows standard Brownian motion. Another way of stating this is that, over the time interval $dt$, the change in the logarithm of the asset price is a Gaussian distribution with mean $(r - \sigma^2/2)dt$ and variance $\sigma^2 dt$.

This is a very important result and will be referred to in later sections of the book.

8.4 ITO’S LEMMA IN ONE DIMENSION

In this section we will derive Ito’s formula, a more rigorous treatment can be found in Shreve (1988).

Let us consider the stochastic process $X$: \[
dX = adt + b\sqrt{dt} \epsilon, \quad \epsilon \sim N(0, 1), \quad dZ \sim N(0, dt) \quad (8.16)\]

where $a$ and $b$ are constants. We want to find the process followed by a function of the stochastic variable $X$, that is $\phi(X, t)$. This can be done by applying a Taylor expansion, up to second order, in the two variables $X$ and $t$ as follows:

\[
\phi^* = \phi + \frac{\partial \phi}{\partial X} dX + \frac{\partial \phi}{\partial t} dt + \frac{1}{2} \frac{\partial^2 \phi}{\partial X^2} dX^2 + \frac{1}{2} \frac{\partial^2 \phi}{\partial t^2} dt^2 + \frac{\partial \phi}{\partial X} \frac{dX}{dt} dt \quad (8.17)\]

where $\phi^*$ is used to denote the value $\phi(X + dX, t + dt)$, and $\phi$ denotes the value $\phi(X, t)$. We will now consider the magnitude of the terms $dX^2$, $dX dt$, and $dt^2$ as $dt \to 0$. First

\[
dX^2 = (adt + b\sqrt{dt} \epsilon)(adt + b\sqrt{dt} \epsilon) = a^2 dt^2 + 2ab \sqrt{dt} \epsilon + b^2 dt \epsilon^2
\]

then

\[
dX dt = adt^2 + b dt^{3/2} \epsilon
\]

So as $dt \to 0$, and ignoring all terms in $dt$ of order greater than 1, we have:

\[
dX^2 \sim b^2 \epsilon^2, \quad dt^2 \sim 0, \quad \text{and} \quad dX dt \sim 0
\]

If we now replace $dX^2$ by its expected value $E[dX^2]$ we then have:

\[
dX^2 \sim E[dX^2] = E[b^2 \epsilon^2] = b^2 dt E[\epsilon^2] = b^2 dt
\]

where we have used the fact that, since $\epsilon \sim N(0, 1)$, the variance of $\epsilon$, $E[\epsilon^2]$, is by definition equal to 1. Using these values in Equation 8.17 and substituting for $dX$ from Equation 8.16, we obtain:

\[
d\phi = \frac{\partial \phi}{\partial X} (adt + bdZ) + \frac{\partial \phi}{\partial t} dt + \frac{b^2}{2} \frac{\partial^2 \phi}{\partial X^2} dt \quad (8.18)
\]
where \( d\phi = \phi^* - \phi \). This gives Ito’s formula

\[
d\phi = \left( a \frac{\partial \phi}{\partial X} + \frac{\partial \phi}{\partial t} + \frac{b^2 \partial^2 \phi}{2 \partial X^2} \right) dt + \frac{\partial \phi}{\partial X} b dZ
\]

(8.19)

In particular if we consider the Geometric Brownian process:

\[
dS = \mu S dt + \sigma S dZ
\]

(8.20)

where \( \mu \) and \( \sigma \) are constants then substituting \( X = S, a = \mu S, \) and \( b = \sigma S \) into Equation 8.19 yields:

\[
d\phi = \left( \mu S \frac{\partial \phi}{\partial S} + \frac{\partial \phi}{\partial t} + \frac{\sigma^2 S^2 \partial^2 \phi}{2 \partial S^2} \right) dt + \frac{\partial \phi}{\partial S} \sigma S dZ
\]

(8.21)

Equation 8.21 describes the change in value of a function \( \phi(S, t) \) over the time interval \( dt \), when the stochastic variable \( S \) follows GBM. This result has very important applications in the pricing of financial derivatives. Here the function \( \phi(S, t) \) is taken as the price of a financial derivative, \( f(S, t) \), that depends on the value of an underlying asset \( S \), which is assumed to follow GBM. In Section 9.3 we will use Equation 8.21 to derive the (Black–Scholes) partial differential equation that is satisfied by the price of a financial derivative.

### 8.5 ITO’S LEMMA IN MANY DIMENSIONS

We will now consider the \( n \)-dimensional stochastic process:

\[
dX_i = a_i dt + b_i \sqrt{dt} \epsilon_i = a_i dt + b_i dZ_i, \quad i = 1, \ldots, n
\]

(8.22)

or in vector form:

\[
dX = A dt + B \sqrt{dt} \mathcal{E} = A dt + B dZ
\]

(8.23)

where \( A \) and \( B \) are \( n \) element vectors respectively containing the constants, \( a_i, i = 1, \ldots, n \) and \( b_i, i = 1, \ldots, n \). The stochastic vector \( X \) contains the \( n \) stochastic variables \( X_i, i = 1, \ldots, n \), the vector \( \mathcal{E} \) contains the \( n \) shocks \( \epsilon_i, i = 1, \ldots, n \), and the vector \( dZ \) contains the \( n \) shocks \( \sqrt{dt} \epsilon_i, i = 1, \ldots, n \).

We will assume that the random vector \( \mathcal{E} \) is drawn from a multivariate normal distribution with zero mean and covariance matrix \( C \). That is we can write:

\[
\mathcal{E} \sim N(0, C) \quad \text{and} \quad dZ \sim N(0, dt C)
\]

Since the diagonal elements of \( C \) are all unity

\[
C_{ii} = E[\epsilon_i^2] = 1, \quad i = 1, \ldots, n
\]

the matrix \( C \) is in fact a correlation matrix with off-diagonal elements given by:

\[
C_{ij} = E[\epsilon_i \epsilon_j] = \rho_{ij}, \quad i = 1, \ldots, n, \quad j = 1, \ldots, n, \quad i \neq j
\]

where \( \rho_{ij} \) is the correlation coefficient between the \( i \)th and \( j \)th variates.
As in Section 8.4 we want to find the process followed by a function of the stochastic vector \( X \), that is the process followed by \( \phi(X, t) \). This can be done by applying any \( n \)-dimensional Taylor expansion, up to second order, in the variables \( X \) and \( t \) as follows:

\[
\phi^* = \phi + \sum_{i=1}^{n} \frac{\partial \phi}{\partial X_i} dX_i + \frac{\partial \phi}{\partial t} dt + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} \frac{\partial^2 \phi}{\partial X_i \partial X_j} dX_i dX_j + \frac{1}{2} \frac{\partial^2 \phi}{\partial t^2} dt^2
\]

\[\text{(8.24)}\]

where \( \phi^* \) is used to denote the value \( \phi(X + dX, t + dt) \), and \( \phi \) denotes the value \( \phi(X, t) \).

We will now consider the magnitude of the terms \( dX_i dX_j \), \( dX_i dt \), and \( dt^2 \) as \( dt \to 0 \).

Expanding the terms \( dX_i dX_j \) and \( dX_i dt \) we have:

\[
dX_i dX_j = (a_i dt + b_i \sqrt{dt} \epsilon_i)(a_j dt + b_j \sqrt{dt} \epsilon_j)
\]

\[
\therefore dX_i dX_j = a_i a_j dt^2 + a_i b_j dt^{3/2} \epsilon_j + a_j b_i dt^{3/2} \epsilon_i + b_i b_j dt \epsilon_i \epsilon_j
\]

\[\text{(8.25)}\]

So as \( dt \to 0 \), and ignoring all terms in \( dt \) of order greater than 1, we have:

\[
dX_i dt \sim 0 \quad \text{and} \quad dX_i dX_j \sim b_i b_j dt \epsilon_i \epsilon_j
\]

If we now replace \( dX_i dX_j \) by its expected value \( E[dX_i dX_j] \) we then have:

\[
E[dX_i dX_j] = E[b_i b_j dt \epsilon_i \epsilon_j] = b_i b_j dt = E[\epsilon_i \epsilon_j] = b_i b_j \rho_{ij} dt
\]

where \( \rho_{ij} \) is the correlation coefficient between the \( i \)th and \( j \)th assets.

Using these values in Equation 8.24, and substituting for \( dX_i \) from Equation 8.22, we obtain:

\[
d\phi = \sum_{i=1}^{n} \frac{\partial \phi}{\partial X_i} (a_i dt + b_i \epsilon_i) + \frac{\partial \phi}{\partial t} dt + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} b_i b_j \rho_{ij} dt \frac{\partial^2 \phi}{\partial X_i \partial X_j}
\]

\[\text{(8.26)}\]

where we have used \( d\phi = \phi^* - \phi \). This gives Ito’s \( n \)-dimensional formula:

\[
d\phi = \left\{ \frac{\partial \phi}{\partial t} + \sum_{i=1}^{n} a_i \frac{\partial \phi}{\partial X_i} + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} b_i b_j \rho_{ij} \frac{\partial^2 \phi}{\partial X_i \partial X_j} \right\} dt + \sum_{i=1}^{n} \frac{\partial \phi}{\partial X_i} b_i dZ_i
\]

\[\text{(8.27)}\]

In particular if we consider the GBM:

\[
dS_i = \mu_i S_i dt + \sigma_i S_i dZ_i, \quad i = 1, \ldots, n
\]
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where \( \mu_i \) is the constant drift of the \( i \)th asset and \( \sigma_i \) is the constant volatility of the \( i \)th asset, then substituting \( X_i = S_i \), \( a_i = \mu_i S_i \), and \( b_i = \sigma_i S_i \) into Equation 8.27 then yields:

\[
d\phi = \left\{ \frac{\partial \phi}{\partial t} + \sum_{i=1}^{n} \mu_i S_i \frac{\partial \phi}{\partial S_i} + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} \sigma_i \sigma_j S_i S_j \rho_{ij} \frac{\partial^2 \phi}{\partial S_i \partial S_j} \right\} dt \\
+ \sum_{i=1}^{n} \frac{\partial \phi}{\partial S_i} \sigma_i S_i dZ_i
\]

\text{(8.28)}
Chapter 9

Analytic methods and single asset European options

9.1 INTRODUCTION

A European option taken out at current time \( t \) gives the owner the right (but no obligation) to do something when the option matures at time \( T \). This could for example be the right to buy or sell stocks at a particular strike price. The option would of course only be exercised if it was in the owner’s interest to do so. For example a single asset European vanilla put option, with strike price \( E \) and expiry time \( T \), gives the owner the right at time \( T \) to sell a particular asset for \( E \). If the asset is worth \( S_T \) at maturity then the value of the put option at maturity, known as the payoff, is thus \( \max(E - S_T, 0) \). By contrast a single asset European vanilla call option, with strike price \( E \) and expiry time \( T \), gives the owner the right at time \( T \) to buy an asset for \( E \); the payoff at maturity for a call option is \( \max(S_T - E, 0) \).

The owner of an American option has the right (but no obligation) to exercise the option at any time from current time \( t \) to option maturity. These options are more difficult to value than European options because of this extra flexibility. Even the simple single asset American vanilla put has no analytic solution and requires finite-difference or lattice methods to estimate its value. Many European options on the other hand take the form of a relatively easy definite integral from which it is possible to compute a closed form solution. The valuation of multiasset European options, dependent on a large number of underlying assets, is more complicated but can conveniently be achieved by using Monte Carlo simulation to compute the required multidimensional definite integral.

The expected current value of a single asset European vanilla option will depend on the current asset price at time \( t \), \( S \), the duration of the option, \( \tau = T - t \), the strike price, \( E \), the riskless interest rate, \( r \), and the probability density function of the underlying asset price at maturity, \( p(S_T) \). The fair price (expected current value) of a vanilla call is thus:

\[
c(S, E, \tau, r, p(S_T)) = \exp(-r\tau)E[\max(S_T - E, 0)]
\]

\[
= \exp(-r\tau) \int_{-\infty}^{\infty} p(S_T) \max(E - S_T, 0) dS_T
\]

(9.1)
and that of the put is:

\[ p(S, E, r, p(S_T)) = \exp(-r\tau)E[\max(E - S_T, 0)] \quad \text{(9.3)} \]

\[ = \exp(-r\tau) \int_{-\infty}^{\infty} p(S_T) \max(E - S_T, 0) dS_T \quad \text{(9.4)} \]

It can be seen from Equations 9.1 to 9.4 that the fair price of a European option is its payoff, at time \( T \), discounted by the riskless interest rate, \( r \), to current time \( t \).

Since we assume that \( r \) is constant throughout the duration of the option and also that the underlying asset has a given distribution (usually lognormal), we will denote the value of a European vanilla call option by \( c(S, E, \tau) \), and that of a European put option by \( p(S, E, \tau) \).

In this section we will consider:

- The put–call parity relationship for European options.
- The differential equation obeyed by single asset and multiasset European options.
- The Black–Scholes option pricing formula for a single asset European option.
- The pricing formulae for some European barrier options.

The notation used will be that which we have previously outlined.

### 9.2 PUT–CALL PARITY

#### 9.2.1 Discrete dividends

Here we consider single asset European put and call options, and derive the following relationship between their values in the presence of cash dividends:

\[ c(S, E, \tau) + E \exp(-r\tau) + D = p(S, E, \tau) + S \quad \text{(9.5)} \]

where \( D \) is the present value of the dividends that are paid during the life of the option. That is:

\[ D = \sum_{k=1}^{n} D_k \exp(-r(t_k - t)) \]

with \( D_k \) the \( k \)th cash dividend paid at time \( t_k \); the other symbols have already been defined in the section introduction.

This result can be proved by considering the following two investments:

**Portfolio A**

One European call, \( c(S, E, \tau) \), and cash of value \( E \exp(-r\tau) + D \).

**Portfolio B**

One European put, \( p(S, E, \tau) \), and one share of value \( S \).

At option maturity, time \( T \), the value of the call and put are \( c(S_T, E, 0) \) and \( p(S_T, E, 0) \) respectively; also at time \( T \) the value of the dividends paid during the life of the option is \( D \exp(r\tau) \).
We now consider the value of both portfolios at option maturity, time $T$, under all possible conditions.

**If $S_T \geq E$**

Portfolio A is worth:

$$\max(S_T - E, 0) + \exp(r\tau)\{E \exp(-r\tau) + D\} = S_T - E + E + D \exp(r\tau)$$

$$= S_T + D \exp(r\tau)$$

Portfolio B is worth:

$$\max(E - S_T, 0) + S_T + D \exp(r\tau) = 0 + S_T + D \exp(r\tau) = S_T + D \exp(r\tau)$$

**If $S_T < E$**

Portfolio A is worth:

$$\max(S_T - E, 0) + \exp(r\tau)\{E \exp(-r\tau) + D\} = 0 + E + D \exp(r\tau) = E + D \exp(r\tau)$$

Portfolio B is worth:

$$\max(E - S_T, 0) + S_T + D \exp(r\tau) = E - S_T + S_T + D \exp(r\tau) = E + D \exp(r\tau)$$

We have therefore shown that under all conditions the value of portfolio A is the same as that of portfolio B.

### 9.2.2 Continuous dividends

Here we consider single asset European put and call options, and derive the following relationship:

$$c(S, E, \tau) + E \exp(-q\tau) = p(S, E, \tau) + S \exp(-q\tau)$$  \hspace{1cm} (9.6)

where $q$ is the asset’s continuous dividend yield that is paid during the life of the option. The result can be proved by considering the following two investments:

**Portfolio A**

One European call, $c(S, E, \tau)$, and cash of value $E \exp(-r\tau)$.

**Portfolio B**

One European put, $p(S, E, \tau)$, and one share of value $S \exp(-q\tau)$.

At option expiry, time $t$, the value of the call and put are $c(S_T, E, 0)$ and $p(S_T, E, 0)$ respectively. Also, if the value of the share at time $t$ is denoted by $S$, the combined value of shares and dividends at time $T$ is $S \exp(q\tau)$. Note that $q$ is treated in a similar manner to the continuously compounded riskless interest rate $r$.

As in Section 9.2.1 we will now consider the value of portfolios A and B at time $T$ under all possible conditions:
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If \( S_T \geq E \)
Portfolio A is worth:
\[
\max(S_T - E, 0) + \exp(rT)E \exp(-rT) = S_T - E + E = S_T
\]
Portfolio B is worth:
\[
\max(E - S_T, 0) + S_T \exp(-qT) \exp(qT) = 0 + S_T = S_T
\]
where \( S_T \exp(-qT) \exp(qT) \) is the combined value of the shares and dividends at option maturity.

If \( S_T < E \)
Portfolio A is worth:
\[
\max(S_T - E, 0) + \exp(rT)E \exp(-rT) = 0 + E = E
\]
Portfolio B is worth:
\[
\max(E - S_T, 0) + S_T \exp(-qT) \exp(qT) = E - S_T + S_T = E
\]
We have therefore shown that under all conditions the value of portfolio A is the same as that of portfolio B.

9.3 VANILLA OPTIONS AND THE BLACK–SCHOLES MODEL

9.3.1 The option pricing partial differential equation

In this section we will derive the (Black–Scholes) partial differential equation that is obeyed by options written on a single asset.

Previously, in Sections 8.4 and 8.5, we derived Ito’s lemma, which provides an expression for the change in value of the function \( \phi(X, t) \), where \( X \) is a stochastic variable. When the stochastic variable, \( X \), follows GBM, the change in the value of \( \phi \) was shown to be given by Equation 8.21. Here we will assume that the function \( \phi(S, t) \) is the value of a financial option and that the price of the underlying asset, \( S \), follows GBM.

If we denote the value of the financial derivative by \( f \), then its change, \( df \), over the time interval \( dt \) is given by:
\[
df = \left( \mu_S \frac{\partial f}{\partial S} + \frac{\partial f}{\partial t} + \frac{\sigma^2 S^2}{2} \frac{\partial^2 f}{\partial S^2} \right) dt + \frac{\partial f}{\partial S} \sigma S dZ, \quad dZ \sim N(0, dt)
\]
The discretized version of this equation is:
\[
\Delta f = \Delta t \left( \mu_S \frac{\partial f}{\partial S} + \frac{\partial f}{\partial t} + \frac{\sigma^2 S^2}{2} \frac{\partial^2 f}{\partial S^2} \right) + \frac{\partial f}{\partial S} \sigma S \Delta Z, \quad \Delta Z \sim N(0, \Delta t) \quad (9.7)
\]
where the time interval is now \( \Delta t \) and the change in derivative value is \( \Delta f \).

If we assume that the asset price, \( S \), follows GBM we also have:
\[
\Delta S = \mu S \Delta t + \sigma S \Delta Z, \quad \Delta Z \sim N(0, \Delta t) \quad (9.8)
\]
where \( \mu \) is the constant drift and the definition of the other symbols is as before. Let us now consider a portfolio consisting of \(-1\) derivative and \(\partial f / \partial S\) units of the underlying
stock. In other words we have gone short (that is sold) a derivative on an asset and have $\partial f / \partial S$ stocks of the (same) underlying asset. The value of the portfolio, $\Pi$, is therefore:

$$\Pi = -f + \frac{\partial f}{\partial S} S$$

(9.9)

and the change, $\Delta \Pi$, in the value of the portfolio over time $\Delta t$ is:

$$\Delta \Pi = -\Delta f + \frac{\partial f}{\partial S} \Delta S$$

(9.10)

Substituting Equations 9.7 and 9.8 into Equation 9.10 we obtain:

$$\Delta \Pi = -\left(\left(\mu S \frac{\partial f}{\partial S} + \frac{\partial f}{\partial t} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 f}{\partial S^2}\right) \Delta t - \sigma S \Delta Z \frac{\partial f}{\partial S} + \frac{\partial f}{\partial S} \left(\mu S \Delta t + \sigma S \Delta Z\right)\right)$$

$$\therefore \Delta \Pi = -\mu S \Delta t \frac{\partial f}{\partial S} - \Delta t \frac{\partial f}{\partial t} - \frac{1}{2} \Delta t \sigma^2 S^2 \frac{\partial^2 f}{\partial S^2} - \sigma S \Delta Z \frac{\partial f}{\partial S}$$

$$+ \mu S \Delta t \frac{\partial f}{\partial S} + \sigma S \Delta Z \frac{\partial f}{\partial S}$$

(9.11)

Cancelling terms we obtain:

$$\Delta \Pi = -\Delta t \left\{\frac{\partial f}{\partial t} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 f}{\partial S^2}\right\}$$

(9.12)

If this portfolio is risk neutral then it grows at the riskless interest rate, $r$ and we have:

$$r \Pi \Delta t = \Delta \Pi$$

So we have that:

$$r \Pi \Delta t = -\Delta t \left\{\frac{\partial f}{\partial t} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 f}{\partial S^2}\right\}$$

(9.13)

Substituting for $\Pi$ and we obtain:

$$r \Delta t \left( f - S \frac{\partial f}{\partial S} \right) = -\Delta t \left\{\frac{\partial f}{\partial t} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 f}{\partial S^2}\right\}$$

(9.14)

On rearranging we have:

**The Black–Scholes partial differential equation**

$$\frac{\partial f}{\partial t} + \frac{\partial f}{\partial S} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 f}{\partial S^2} = r f$$

(9.15)

Let us now consider put and call options on the same underlying asset. If we let $c$ be the value of a European call option and $p$ that of a European put option then we have the following equations:

$$\frac{\partial p}{\partial t} + S \frac{\partial p}{\partial S} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 p}{\partial S^2} = rp$$

(9.16)
and
\[ \frac{\partial c}{\partial t} + S \frac{\partial c}{\partial S} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 c}{\partial S^2} = rc \]  
(9.17)

If we now form a linear combination of put and call options, \( \Psi = a_1 c + a_2 p \), where both \( a_1 \) and \( a_2 \) are constants, then \( \Psi \) also obeys the Black–Scholes equation:
\[ \frac{\partial \Psi}{\partial t} + S \frac{\partial \Psi}{\partial S} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 \Psi}{\partial S^2} = r \Psi \]  
(9.18)

We will now prove that \( \Psi \) satisfies Equation 9.15.

First we rewrite Equation 9.15 as:
\[ \frac{\partial (a_1 c + a_2 p)}{\partial t} + S \frac{\partial (a_1 c + a_2 p)}{\partial S} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 (a_1 c + a_2 p)}{\partial S^2} = r(a_1 c + a_2 p) \]  
(9.19)

and use the following results from elementary calculus:
\[ \frac{\partial (a_1 c + a_2 p)}{\partial t} = a_1 \frac{\partial c}{\partial t} + a_2 \frac{\partial p}{\partial t} \]
\[ \frac{\partial (a_1 c + a_2 p)}{\partial S} = a_1 \frac{\partial c}{\partial S} + a_2 \frac{\partial p}{\partial S} \]

and
\[ \frac{\partial^2 (a_1 c + a_2 p)}{\partial S^2} = a_1 \frac{\partial^2 c}{\partial S^2} + a_2 \frac{\partial^2 p}{\partial S^2} \]

If we denote the left hand side of Equation 9.15 by LHS, then we have:
\[ \text{LHS} = a_1 \left\{ \frac{\partial c}{\partial t} + S \frac{\partial c}{\partial S} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 c}{\partial S^2} \right\} + a_2 \left\{ \frac{\partial p}{\partial t} + S \frac{\partial p}{\partial S} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 p}{\partial S^2} \right\} \]  
(9.20)

We now use Equations 9.13 and 9.14 to substitute for the values in the curly brackets in Equation 9.19, and we obtain:
\[ \text{LHS} = a_1 rc + a_2 rp \]  
(9.21)

which is just the LHS of Equation 9.21; so we have proved the result. It should be noted that this result is also true for American options, since they also obey the Black–Scholes equation.

The above result can be generalized to include a portfolio consisting of \( n \) single asset options. Here we have:
\[ \Psi = \sum_{j=1}^{n} a_j f_j, \quad j = 1, \ldots, n \]

where \( f_j \) represents the value of the \( j \)th derivative and \( a_j \) is the number of units of the \( j \)th derivative. To prove that \( \Psi \) follows the Black–Scholes equation we simply partition the portfolio into sectors whose options depend on the same underlying asset. We then proceed as before by showing that the value of each individual sector obeys the Black–Scholes equation and thus the value of the complete portfolio (the sum of the values of all the sectors) obeys the Black–Scholes equation. It should be
mentioned that this result applies for both American and European options and it
doesn’t matter whether we have bought or sold the options.

In Section 10.3.2 we will use the fact that the difference between the value of a
European option and the equivalent American option obeys the Black–Scholes
equation. We can see this immediately by considering the following portfolios that
are long in an American option and short (that is have sold) a European option:

\[ \Psi^p = P - p, \quad \Psi^c = C - c \]

where \( P \) and \( C \) are the values of American put and call options. \( \Psi^p \) and \( \Psi^c \) both obey
the Black–Scholes equations, and are the respective differences in value of American/
European put options and American/European call options.

### 9.3.2 The multiasset option pricing partial differential equation

In this section we will derive the multiasset (Black–Scholes) partial differential equation
that is obeyed by options written on \( n \) assets. Proceeding as in Section 9.3.1 we will use
the \( n \)-dimensional version of Ito’s lemma to find the process followed by the the value of
a multiasset financial derivative. We will denote the value of this derivative by
\( f(S, t) \), where \( S \) is a \( n \) element stochastic vector containing the prices of the underlying assets,
\( S_i, i = 1, \ldots, n \). If we assume that \( S \) follows \( n \)-dimensional GBM then the change in the value of the derivative, \( df \), is (see Section 8.5, Equation 8.28) given by:

\[
\frac{df}{dt} = \left\{ \frac{\partial f}{\partial t} + \sum_{i=1}^{n} \mu_i S_i \frac{\partial f}{\partial S_i} + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} \sigma_i \sigma_j S_i S_j \rho_{ij} \frac{\partial^2 f}{\partial S_i \partial S_j} \right\} dt + \sum_{i=1}^{n} \frac{\partial f}{\partial S_i} \sigma_i S_i dZ_i \quad (9.22)
\]

The discretized version of this equation is:

\[
\Delta f = \left\{ \frac{\partial f}{\partial t} + \sum_{i=1}^{n} \mu_i S_i \frac{\partial f}{\partial S_i} + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} \sigma_i \sigma_j S_i S_j \rho_{ij} \frac{\partial^2 f}{\partial S_i \partial S_j} \right\} \Delta t + \sum_{i=1}^{n} \frac{\partial f}{\partial S_i} \sigma_i S_i \Delta Z_i \quad (9.23)
\]

where the time interval is now \( \Delta t \) and the change in derivative value is \( \Delta f \).

Let us now consider a portfolio consisting of \(-1\) derivative and \( \partial f / \partial S_i \) units of the
\( i \)th underlying stock. In other words we have gone short (that is sold) a derivative that
depends on the price, \( S_i, i = 1, \ldots, n \), of \( n \) underlying assets, and have \( \partial f / \partial S_i \) units of the
\( i \)th asset. The value of the portfolio, \( \Pi \), is therefore:

\[
\Pi = -f + \sum_{i=1}^{n} \frac{\partial f}{\partial S_i} S_i \quad (9.24)
\]

and the change, \( \Delta \Pi \), in the value of the portfolio over the time interval \( \Delta t \) is:

\[
\Delta \Pi = -\Delta f + \sum_{i=1}^{n} \frac{\partial f}{\partial S_i} \Delta S_i \quad (9.25)
\]

Since the stochastic variables \( S_i, i = 1, \ldots, n \) follow \( n \)-dimensional GBM the change in the \( i \)th asset price, \( \Delta S_i \) over the time interval \( \Delta t \) is given by:

\[
\Delta S_i = \mu_i S_i \Delta t + \sigma_i S_i \Delta Z_i, \quad i = 1, \ldots, n \quad (9.26)
\]
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where \( \Delta Z_i = \epsilon_i \sqrt{\Delta t} \) and, as in Section 8.5, we write:

\[
E[\epsilon_i^2] = 1, \quad i = 1, \ldots, n
\]

and

\[
E[\epsilon_i \epsilon_j] = \rho_{ij}, \quad i = 1, \ldots, n, \quad j = 1, \ldots, n, \quad i \neq j
\]

Substituting Equations 9.23 and 9.26 into Equation 9.25 we obtain:

\[
\Delta \Pi = - \left\{ \frac{\partial f}{\partial t} + \sum_{i=1}^{n} \mu_i S_i \frac{\partial f}{\partial S_i} + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} \sigma_i \sigma_j \rho_{ij} S_i S_j \frac{\partial^2 f}{\partial S_i \partial S_j} \right\} \Delta t
\]

\[
- \sum_{i=1}^{n} \sigma_i S_i \Delta Z_i \frac{\partial f}{\partial S_i} + \sum_{i=1}^{n} \frac{\partial f}{\partial S_i} \{ \mu_i S_i \Delta t + \sigma_i \Delta S_i \}
\]

\[
: \Delta \Pi = - \sum_{i=1}^{n} \mu_i S_i \Delta t \frac{\partial f}{\partial S_i} - \Delta t \frac{\partial f}{\partial t} - \frac{1}{2} \Delta t \sum_{i=1}^{n} \sum_{j=1}^{n} \sigma_i \sigma_j \rho_{ij} S_i S_j \frac{\partial^2 f}{\partial S_i \partial S_j}
\]

\[
- \sum_{i=1}^{n} \sigma_i S_i \Delta Z_i \frac{\partial f}{\partial S_i} + \sum_{i=1}^{n} \frac{\partial f}{\partial S_i} \{ \mu_i S_i \Delta t + \sigma_i \Delta S_i \}
\]

(9.27)

Cancelling terms we obtain:

\[
\Delta \Pi = - \Delta t \left\{ \frac{\partial f}{\partial t} + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} \sigma_i \sigma_j \rho_{ij} S_i S_j \frac{\partial^2 f}{\partial S_i \partial S_j} \right\}
\]

(9.28)

If this portfolio is to grow at the riskless interest rate, \( r \) we have:

\[
r \Pi \Delta t = \Delta \Pi
\]

So from Equation 9.28 we have that:

\[
r \Pi \Delta t = - \Delta t \left\{ \frac{\partial f}{\partial t} + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} \sigma_i \sigma_j \rho_{ij} S_i S_j \frac{\partial^2 f}{\partial S_i \partial S_j} \right\}
\]

(9.29)

Substituting for \( \Pi \) and we obtain:

\[
r \Delta t \left\{ f - \sum_{i=1}^{n} S_i \frac{\partial f}{\partial S_i} \right\} = - \Delta t \left\{ \frac{\partial f}{\partial t} + \frac{1}{2} \sum_{i=1}^{n} \sum_{j=1}^{n} \sigma_i \sigma_j \rho_{ij} S_i S_j \frac{\partial^2 f}{\partial S_i \partial S_j} \right\}
\]

(9.30)
Rearranging Equation 9.30 gives:

### The $n$-dimensional Black–Scholes partial differential equation

\[
\frac{\partial f}{\partial t} + \sum_{i=1}^{n} S_i \frac{\partial f}{\partial S_i} + \frac{1}{2} \sum_{i,j=1}^{n} \sigma_i \sigma_j \rho_{ij} S_i S_j \frac{\partial^2 f}{\partial S_i \partial S_j} = rf \tag{9.31}
\]

#### 9.3.3 The Black–Scholes formula

In this section we will derive the Black–Scholes formula for pricing European put and call options on a single asset which follows GBM. The approach we will adopt here is to first derive an expression for the value of a European call option, and then use the put/call parity relationships of Section 9.2 to obtain the value of the corresponding European put option. If we denote the current time by $t$ and the expiry time of the option by $T$, then the duration of the option is $\tau = T - t$. Since the asset is assumed to follow GBM we can use a discretized version of Equations 8.14 and 8.15 in Section 8.3 to write:

\[
\log \left( \frac{S_{t+\Delta t}}{S_t} \right) \sim N \left( \left( r - \frac{1}{2} \sigma^2 \right) \Delta t, \sigma^2 \Delta t \right) \tag{9.32}
\]

Here we use the following notation:

$\Delta t = \tau$, $S_t = S$, and $S_{t+\Delta t} = S_T$

where $S$ is the asset value at the current time $t$, and $S_T$ is the asset value at option maturity. We will now introduce the variable $X$ which we define as follows:

\[
X = \log \left( \frac{S_T}{S} \right) \quad \text{or equivalently} \quad S_T = S \exp(X)
\]

From Equation 9.32 we have that

\[
X \sim N \left( (r - \sigma^2/2) \tau, \sigma^2 \tau \right)
\]

The probability density function of $X, f(X)$, is thus the Gaussian:

\[
f(X) = \frac{1}{\sigma \sqrt{\tau \sqrt{2\pi}}} \exp \left( - \frac{(X - (r - \sigma^2/2) \tau)^2}{2\sigma^2 \tau} \right)
\]

The value of a European call option, $c(S, E, \tau)$, with strike price $E$, is the expected value of the option’s payoff at maturity discounted to the current time by the riskless interest rate $r$. That is:

\[
c(S, E, \tau) = \exp(-r\tau)E[S_T - E]
\]
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This can be rewritten in terms of the probability density function of $S_T$ as follows:

$$c(S, E, \tau) = \exp(-r\tau) \int_{S_T = E}^{\infty} f(S_T)(S_T - E) dS_T$$

(9.33)

Instead of integrating over values of $S_T$, as above, we will use $S_T = S \exp(X)$ and then integrate over $X$. Equation 9.33 then becomes:

$$c(S, E, \tau) = \frac{\exp(-r\tau)}{\sigma \sqrt{\tau} 2\pi} \int_{X = \log(E/S)}^{\infty} (S \exp(X) - E) \times \exp\left(-\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau}\right) dX$$

(9.34)

where we have used $S \exp(X) = E$, giving $X = \log(E/S)$, to obtain the lower limit of the integral. This integral is evaluated by splitting it into the two parts:

$$c(S, E, \tau) = I_A - I_B$$

(9.35)

where

$$I_A = \frac{S \exp(-r\tau)}{\sigma \sqrt{\tau} 2\pi} \int_{X = \log(E/S)}^{\infty} \exp(X) \exp\left(-\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau}\right) dX$$

(9.36)

and

$$I_B = \frac{E \exp(-r\tau)}{\sigma \sqrt{\tau} 2\pi} \int_{X = \log(E/S)}^{\infty} \exp\left(-\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau}\right) E dX$$

(9.37)

To evaluate these integrals we will make use of the fact that the univariate cumulative normal function $N_1(x)$ is:

$$N_1(x) = \frac{1}{\sqrt{2\pi}} \int_{u = -\infty}^{x} \exp\left(-\frac{u^2}{2}\right) du$$

by symmetry we have $N_1(-x) = 1 - N_1(x)$ and

$$\frac{1}{\sqrt{2\pi}} \int_{x}^{\infty} \exp\left(-\frac{u^2}{2}\right) du = \frac{1}{\sqrt{2\pi}} \int_{-\infty}^{x} \exp\left(-\frac{u^2}{2}\right) du = N_1(-x)$$

We will first consider $I_B$, which is the easier of the two integrals.

$$I_B = \frac{E \exp(-r\tau)}{\sigma \sqrt{\tau} 2\pi} \int_{X = \log(E/S)}^{\infty} \exp\left(-\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau}\right) dX$$

If we let $u = (X - (r - \sigma^2/2)\tau)/\sigma \sqrt{\tau}$ then $dX = \sigma \sqrt{\tau} du$. So

$$I_B = \frac{E \exp(-r\tau) \sigma \sqrt{\tau}}{\sigma \sqrt{2\pi} \sigma \sqrt{\tau}} \int_{u = k_2}^{\infty} \exp\left(-\frac{u^2}{2}\right) du$$

where the lower integration limit is $k_2 = (\log(E/S) - (r - \sigma^2/2)\tau)/\sigma \sqrt{\tau}$.
Therefore the value of a European call is:

\[ c(S, E, \tau) = SN_1(-k_1) - E \exp(-r\tau)N_1(-k_2) \]  

(9.41)
which gives the usual form of the Black–Scholes formula for a European call as:

The Black–Scholes formula for a European call

\[ c(S, E, \tau) = SN_1(d_1) - E \exp(-r\tau)N_1(d_2) \] (9.42)

where

\[ d_1 = \frac{\log(S/E) + (r + \sigma^2/2)\tau}{\sigma\sqrt{\tau}} \]
\[ d_2 = \frac{\log(S/E) + (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}} = d_1 - \sigma\sqrt{\tau} \] (9.43)

To gain some insight into the meaning we will rewrite the above equation in the following form:

\[ c(S, E, \tau) = \exp(-r\tau)\{SN_1(d_1) \exp(r\tau) - EN_1(d_2)\} \] (9.44)

The term \( N_1(d_2) \) is the probability that the option will be exercised in a risk-neutral world, so that \( EN_1(d_2) \) is the strike price multiplied by the probability that the strike price will be paid. The term \( SN_1(d_1) \exp(r\tau) \) is the expected value of a variable, in a risk neutral world, that equals \( S_T \) if \( S_T > E \) and is otherwise zero.

The corresponding formula for a put can be shown using put-call parity, see Section 9.2, to be:

The Black–Scholes formula for a European put

\[ p(S, E, \tau) = E \exp(-r\tau)N_1(-d_2) - SN_1(-d_1) \] (9.45)

where

\[ d_1 = \frac{\log(S/E) + (r + \sigma^2/2)\tau}{\sigma\sqrt{\tau}} \]
\[ d_2 = \frac{\log(S/E) + (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}} = d_1 - \sigma\sqrt{\tau} \] (9.46)

or equivalently, using \( N_1(-x) = 1 - N_1(x) \) we have

\[ p(S, E, \tau) = E \exp(-r\tau)\{1 - N_1(d_2)\} - S\{1 - N_1(d_1)\} \] (9.47)

The inclusion of continuous dividends

The effect of dividends on the value of a European option can be dealt with by assuming that the asset price is the sum of a riskless component involving known dividends that will be paid during the life of the option, and a risky (stochastic) component; see Hull (1997).
As dividends are paid the stock price is reduced by the same amount, and by the
time the European option matures, all the dividends will have been paid leaving only
the risky component of the asset price.

This means that, in the case of a continuous dividend yield $q$, European put/call
options can be priced using Equations 9.42 and 9.45 but with $S$ replaced by $S \exp (-q \tau)$.

This results in:

The Black–Scholes formula with continuous dividends

$$c(S, E, \tau) = S \exp (-q \tau) N_1(d_1) - E \exp (-r \tau) N_1(d_2)$$

and the corresponding formula for a put can be shown (using put–call parity) to be:

$$p(S, E, \tau) = E \exp (-r \tau) N_1(-d_2) - S \exp (-q \tau) N_1(-d_1)$$

or equivalently, using $N_1(-x) = 1 - N_1(x)$, we have

$$p(S, E, \tau) = E \exp (-r \tau) \{1 - N_1(d_2)\} - S \exp (-q \tau) \{1 - N_1(d_1)\}$$

where

$$d_1 = \frac{\log(S/E) + (r - q + \sigma^2/2) \tau}{\sigma \sqrt{\tau}}$$

and

$$d_2 = \frac{\log(S/E) + (r - q - \sigma^2/2) \tau}{\sigma \sqrt{\tau}} = d_1 - \sigma \sqrt{\tau}$$

The above values of $d_1$ and $d_2$ are obtained by simply substituting $S = S \exp (-q \tau)$
into Equation 9.43 as follows:

$$d_1 = \frac{\log(S \exp(-q \tau)/E) + (r + \sigma^2/2) \tau}{\sigma \sqrt{T-t}} = \frac{\log(S/E) - q \tau + (r + \sigma^2/2) \tau}{\sigma \sqrt{\tau}}$$

$$d_2 = \frac{\log(S \exp(-q \tau)/E) + (r - \sigma^2/2) \tau}{\sigma \sqrt{\tau}} = \frac{\log(S/E) - q \tau + (r - \sigma^2/2) \tau}{\sigma \sqrt{\tau}}$$

The inclusion of discrete dividends

Here we consider $n$ discrete cash dividends $D_i, i = 1, \ldots, n$, paid at times $t_i,
i = 1, \ldots, n$ during the life of the option. In these circumstances the Black–Scholes
formula can be used to price European options, but with the current asset value $S$
reduced by the present value of the cash dividends.

This means that instead of $S$ we use the quantity $S_D$ which is computed as

$$S_D = S - \sum_{i=1}^{n} D_i \exp(-r t_i)$$
where $r$ is the (in this case constant) riskless interest rate. The formulae for European puts and calls are then

$$c(S, E, \tau) = S_D N_1(d_1) - E \exp(-r\tau) N_1(d_2)$$  \hspace{1cm} (9.51)

$$p(S, E, \tau) = E \exp(-r\tau) \{1 - N_1(d_2)\} - S_D \{1 - N_1(d_1)\}$$  \hspace{1cm} (9.52)

where

$$d_1 = \frac{\log(S_D/E) + (r + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}$$

and

$$d_2 = \frac{\log(S_D/E) + (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}} = d_1 - \sigma\sqrt{\tau}$$  \hspace{1cm} (9.53)

In Section 10.2.3 we give results for perpetual European options.

**The greeks**

Now that we have derived formulae to price European vanilla puts and calls it is possible to work out their partial derivatives (hedge statistics). We will now merely quote expressions for the Greeks (hedge statistics) for European options. Here the subscript $c$ refers to a European call, and the subscript $p$ refers to a European put. Complete derivations of these results can be found in Appendix C.

**Gamma**

$$\Gamma_c = \frac{\partial^2 c}{\partial S^2} = \Gamma_p = \frac{\partial^2 p}{\partial S^2} = \exp(-q\tau) \frac{n(d_1)}{S\sigma\sqrt{\tau}}$$  \hspace{1cm} (9.54)

**Delta**

$$\Delta_c = \frac{\partial c}{\partial S} = \exp(-q\tau) N_1(d_1), \quad \Delta_p = \frac{\partial p}{\partial S} = \exp(-q\tau) \{N_1(d_1) - 1\}$$  \hspace{1cm} (9.55)

**Theta**

$$\Theta_c = \frac{\partial c}{\partial t} = q \exp(-q\tau) SN_1(d_1) - rE \exp(-r\tau) N_1(d_2) - \frac{Sn(d_1)\sigma \exp(-q\tau)}{2\sqrt{\tau}}$$

$$\Theta_p = \frac{\partial p}{\partial t} = -q \exp(-q\tau) SN_1(-d_1) + rE \exp(-r\tau) N_1(-d_2) - \frac{Sn(d_1)\sigma \exp(-q\tau)}{2\sqrt{\tau}}$$  \hspace{1cm} (9.56)

**Rho**

$$\rho_c = \frac{\partial c}{\partial r} = E\tau N_1(d_2), \quad \rho_p = \frac{\partial p}{\partial r} = -E\tau N_1(-d_2)$$  \hspace{1cm} (9.57)

**Vega**

$$\nu_c = \frac{\partial c}{\partial \sigma} = \nu_p = \frac{\partial p}{\partial \sigma} = S \exp(-q\tau) n(d_1)\sqrt{\tau}$$  \hspace{1cm} (9.58)

where $n(x) = (1/\sqrt{2\pi}) \exp(-x^2/2)$. 

100  **Pricing Assets**
We now present, in Code excerpt 9.1, a computer program to calculate the Black–Scholes option value and Greeks given in Equations 9.54 to 9.57. The routine uses the NAG C library macro X02AJC to identify whether the arguments are too small, and also the NAG C library function s15abc to compute the cumulative normal distribution function.

```c
void black_scholes(double *value, double greeks[], double s0, double x,
        double sigma, double t, double r, double q, Integer put, Integer *iflag)
{
    /* Input parameters:
       s0 — the current price of the underlying asset
       x — the strike price
       sigma — the volatility
       t — the time to maturity
       r — the interest rate
       q — the continuous dividend yield
       put — an integer indicating whether it is a call (put=0) or a put (put=1)
    
    Output parameters:
       value — the value of the option
       greeks[] — the hedge statistics output as follows: greeks[0] is gamma, greeks[1] is delta
       iflag — an error indicator
    */
    double one=1.0, two=2.0, zero=0.0;
    double eps,d1,d2,temp,temp1,temp2,pi,np;
    eps = X02AJC;
    if( (x < eps) || (sigma < eps) || (t < eps) ) { /* Check if any of the the input
        arguments are too small */
        *iflag = 2;
        return;
    }
    temp = log(s0/x);
    d1 = temp + (r-q+(sigma*sigma/two))*t;
    d1 = d1/(sigma*sqrt(t));
    d2 = d1-sigma*sqrt(t);
    /* evaluate the option price */
    if (put==0) {
        *value = (s0*exp(-q*t)*s15abc(d1) - x*exp (-r * t) *s15abc (d2));
    } else
        *value = (-s0*exp(-q*t)*s15abc(-d1) + x*exp(-r*t)*s15abc (-d2));
    if (greeks){/* then calculate the greeks */
        temp1 = -d1/d2/2;
        d2 = d1-sigma*sqrt(t);
        pi = X01AAC;
        np = (one/sqrt(two*pi)) * exp(temp1);
        if (put==0) { /* a call option */
            greeks[0] = np*exp(-q*t); /* delta */
            greeks[1] = (s15abc(d1))*exp(-q*t); /* theta */
            greeks[2] = -s0*exp(-q*t)*np*sigma/(two*sqrt(t))
                       + s0*s15abc(d1)*np*(-q*t) - (r+x)*s15abc (-d2);/* rho */
            greeks[3] = x*t*exp(-r*t)*s15abc(d2); /* rho */
        } else { /* a put option */
            greeks[0] = s15abc(d1) - one)*exp(-q*t); /* delta */
            greeks[1] = (s15abc(d1)) - one)*exp(-q*t); /* delta */
            greeks[2] = -s0*exp(-q*t)*np*sigma/(two*sqrt(t))
                       - q*s0*s15abc(d1)*exp(-q*t) + (r+x)*s15abc (-d2);/* rho */
            greeks[3] = -x*t*exp(-r*t)*s15abc (-d2); /* rho */
        }
        greeks[4] = s15abc(d2)*np*exp(-q*t); /* vega */
    }
    return;
}
```

**Code excerpt 9.1** Function to compute the Black–Scholes value for European options
It can be seen in Tables 9.1 and 9.2 that the values for gamma and vega are the same for both puts and calls. We can also demonstrate that the option values are consistent by using put–call parity.

\[ c(S, E, \tau) + E \exp(-r\tau) = p(S, E, \tau) + S \exp(-q\tau) \]

For example when \( \tau = 1.0 \), we have \( c(S, E, \tau) = 12.952 \) and \( p(S, E, T) = 9.260 \). So: \( c(S, E, \tau) + E \exp(-r\tau) = 12.952 + 100 \times \exp(-0.1) = 103.436 \) and \( p(S, E, \tau) + S \exp(-q\tau) = 9.260 + 100 \times \exp(-0.06) = 103.436 \).

### 9.3.4 Historical and implied volatility

Obtaining the best estimate of the volatility parameter, \( \sigma \), in the Black–Scholes formula is of crucial importance. There are many different approaches to volatility estimation. These include:

- Historical estimation
- Implied volatility
- Time series methods.

<table>
<thead>
<tr>
<th>( \tau )</th>
<th>Value</th>
<th>Delta</th>
<th>Gamma</th>
<th>Theta</th>
<th>Vega</th>
<th>Rho</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.100</td>
<td>3.558</td>
<td>-0.462</td>
<td>0.042</td>
<td>-16.533</td>
<td>12.490</td>
<td>-4.971</td>
</tr>
<tr>
<td>0.200</td>
<td>4.879</td>
<td>-0.444</td>
<td>0.029</td>
<td>-10.851</td>
<td>17.487</td>
<td>-9.860</td>
</tr>
<tr>
<td>0.300</td>
<td>5.824</td>
<td>-0.431</td>
<td>0.024</td>
<td>-8.298</td>
<td>21.204</td>
<td>-14.663</td>
</tr>
<tr>
<td>0.400</td>
<td>6.571</td>
<td>-0.419</td>
<td>0.020</td>
<td>-6.758</td>
<td>24.241</td>
<td>-19.377</td>
</tr>
<tr>
<td>0.500</td>
<td>7.191</td>
<td>-0.408</td>
<td>0.018</td>
<td>-5.698</td>
<td>26.832</td>
<td>-24.004</td>
</tr>
<tr>
<td>0.600</td>
<td>7.720</td>
<td>-0.399</td>
<td>0.016</td>
<td>-4.909</td>
<td>29.100</td>
<td>-28.544</td>
</tr>
<tr>
<td>0.700</td>
<td>8.179</td>
<td>-0.390</td>
<td>0.015</td>
<td>-4.292</td>
<td>31.118</td>
<td>-32.997</td>
</tr>
<tr>
<td>0.800</td>
<td>8.582</td>
<td>-0.381</td>
<td>0.014</td>
<td>-3.792</td>
<td>32.935</td>
<td>-37.364</td>
</tr>
<tr>
<td>0.900</td>
<td>8.940</td>
<td>-0.373</td>
<td>0.013</td>
<td>-3.377</td>
<td>34.585</td>
<td>-41.646</td>
</tr>
<tr>
<td>1.000</td>
<td>9.260</td>
<td>-0.366</td>
<td>0.012</td>
<td>-3.025</td>
<td>36.093</td>
<td>-45.843</td>
</tr>
</tbody>
</table>

<table>
<thead>
<tr>
<th>( \tau )</th>
<th>Value</th>
<th>Delta</th>
<th>Gamma</th>
<th>Theta</th>
<th>Vega</th>
<th>Rho</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.100</td>
<td>3.955</td>
<td>0.532</td>
<td>0.042</td>
<td>-20.469</td>
<td>12.490</td>
<td>4.929</td>
</tr>
<tr>
<td>0.200</td>
<td>5.667</td>
<td>0.544</td>
<td>0.029</td>
<td>-14.724</td>
<td>17.487</td>
<td>9.744</td>
</tr>
<tr>
<td>0.300</td>
<td>6.996</td>
<td>0.552</td>
<td>0.024</td>
<td>-12.109</td>
<td>21.204</td>
<td>14.451</td>
</tr>
<tr>
<td>0.400</td>
<td>8.121</td>
<td>0.558</td>
<td>0.020</td>
<td>-10.508</td>
<td>24.241</td>
<td>19.054</td>
</tr>
<tr>
<td>0.500</td>
<td>9.113</td>
<td>0.562</td>
<td>0.018</td>
<td>-9.387</td>
<td>26.832</td>
<td>23.557</td>
</tr>
<tr>
<td>0.600</td>
<td>10.007</td>
<td>0.566</td>
<td>0.016</td>
<td>-8.539</td>
<td>29.100</td>
<td>27.962</td>
</tr>
<tr>
<td>0.700</td>
<td>10.826</td>
<td>0.569</td>
<td>0.015</td>
<td>-7.863</td>
<td>31.118</td>
<td>32.271</td>
</tr>
<tr>
<td>0.800</td>
<td>11.584</td>
<td>0.572</td>
<td>0.014</td>
<td>-7.305</td>
<td>32.935</td>
<td>36.485</td>
</tr>
<tr>
<td>0.900</td>
<td>12.290</td>
<td>0.574</td>
<td>0.013</td>
<td>-6.832</td>
<td>34.585</td>
<td>40.608</td>
</tr>
<tr>
<td>1.000</td>
<td>12.952</td>
<td>0.576</td>
<td>0.012</td>
<td>-6.422</td>
<td>36.093</td>
<td>44.640</td>
</tr>
</tbody>
</table>
Here we will consider both historical and implied volatility estimation. Part III of this book deals with the more complex issues connected with time series volatility estimation.

**Historical volatility**

In this method we calculate the volatility using \( n + 1 \) historical asset prices, \( S_i, i = 0, \ldots, n \), and we assume that the asset prices are observed at the regular time interval, \( d\tau \). Since the asset prices are assumed to follow GBM, the volatility is computed as the *annualized* standard deviation of the \( n \) continuously compounded returns, \( u_i, i = 1, \ldots, n \), where

\[
S_i = S_{i-1} \exp(u_i) \quad \text{or} \quad u_i = \log\left( \frac{S_i}{S_{i-1}} \right)
\]

We already know, see Section 8.3, Equation 8.15, that the expected standard deviation of the asset returns over the time interval is \( \sigma \sqrt{d\tau} \). This means that we obtain the following expression for \( \hat{\sigma} \), the estimated volatility

\[
\hat{\sigma} \sqrt{d\tau} = \sqrt{\frac{1}{n-1} \sum_{i=1}^{n} (u_i - \bar{u})^2} \quad (9.59)
\]

or

\[
\hat{\sigma} = \sqrt{\frac{1}{(n-1)d\tau} \sum_{i=1}^{n} (u_i - \bar{u})^2} \quad (9.60)
\]

The estimated standard error in \( \hat{\sigma} \) is, see for example Hull (1997), given by

\[
\hat{\sigma}_{std} = \hat{\sigma} \sqrt{\frac{1}{2(n-1)}} \quad (9.61)
\]

A computer program to perform these calculations is given below in Code excerpt 9.2.

```c
void hist_vol(double *sigma, double *err, double data[], Integer n, double dt, Integer *ifail)
{
    /*Input parameters:
        data[] — the data, which consists of n asset prices
        n — the number of data points
        dt — the (constant) time spacing between the data points (in years)
    
    Output parameters:
        sigma — the computed historical volatility
        err — the standard error in the volatility estimate sigma
        iflag — an error indicator
    */

define DATA(1) data[(n)-1]

double mean=0.0, sum=0.0;
```

---

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double temp, tn;
Integer i;
for(i = 2; i <= n; i++)
    mean = mean + log(DATA(i)) - log(DATA(i-1));
mean = mean/(double)(n-1);
for(i = 2; i <= n; i++)
    temp = log(DATA(i)) - log(DATA(i-1));
    sum = sum + (temp - mean) * (temp - mean);
} 
sum = sum/(double)(n-2);
*sigma = sqrt(sum/dt); 
*err = *sigma/sqrt(tn);
return;
}

Code excerpt 9.2 Function to compute the historical volatility from asset data

**Implied volatility**

The implied volatility of a European option is the volatility which, when substituted into the Black–Scholes formula, yields the market value quoted for the same option.

The routine provided in Code excerpt 9.2 uses Newton’s method to calculate the implied volatility for a European option from its market price. We will now illustrate this technique for a European call option with market value \(\text{opt\_value}\). The implied volatility, \(\sigma\), is then that value which satisfies:

\[ K(\sigma) = c(S, E, \tau, \sigma) - \text{opt\_value} = 0 \]

where \(c(S, E, \tau, \sigma)\) represents the value of the European call and the other symbols have their usual meaning.

From Newton’s method we have:

\[ \sigma_{i+1} = \sigma_i - \frac{F(\sigma_i)}{F'(\sigma_i)} \]

where

\[ F'(\sigma_i) = \frac{\partial F}{\partial \sigma} = \frac{\partial c(S, E, \tau, \sigma)}{\partial \sigma} = \mathcal{V}_c \]

Therefore the iterative procedure is

\[ \sigma_{i+1} = \sigma_i - \frac{c(S, E, \tau, \sigma) - \text{opt\_value}}{\mathcal{V}_c} \]

where \(\sigma_0\) is the initial estimate, and \(\sigma_{i+1}\) is the improved estimate of the implied volatility based on the \(i\)th estimate \(\sigma_i\). Termination of this iteration occurs when \(\text{ABS}(\sigma_{i+1} - \sigma_i) < \text{tol}\), for a specified tolerance, \(\text{tol}\).

It can be seen that as \(\sigma \to 0, d_1 \to \infty, d_2 \to \infty\) and, from Equation 9.58 we have \(\mathcal{V}_c \to 0\). Under these circumstances Newton’s method fails.

The same procedure can be used to compute the implied volatility for a European put, in this case we just replace \(c(S, E, \tau, \sigma)\) by \(p(S, E, \tau, \sigma)\), the value of a European put; from Equation 9.58 \(\mathcal{V}_c = \mathcal{V}_p\).
void implied_volatility(double value, double s0, double x, double sigma[],
  double t, double r, double q, Integer put, Integer *iflag)
{
  /* Input parameters: */
  value — the current value of the option
  s0 — the current price of the underlying asset
  x — the strike price
  sigma[] — the input bounds on the volatility: sigma[0], the lower bound and, sigma[1], the upper bound
  t — the time to maturity
  r — the interest rate
  q — the continuous dividend yield
  put — if put is 0 then a call option, otherwise a put option

  Output parameters:
  sigma[] — the element sigma[0] contains the estimated implied volatility
  iflag — an error indicator */
  double zero = 0.0;
  double fx, sig1, sig2;
  double val, tolx;
  double temp, eps, epsqrt, temp1, v1;
  Integer max_iters, i, ind, ir;
  double greeks[5], c[20], sig, vega;
  Boolean done;
  eps = X02AJC;
  tolx = eps;
  epsqrt = sqrt(eps);
  if (put == 0) /* a call option */
    temp1 = MAX(s0*exp(-q*t)-x*exp(-r*t),zero);
  else /* a put option */
    temp1 = MAX(x*exp(-r*t)-s0*exp(-q*t),zero);
  v1 = FABS(value-temp1);
  if (v1 <= epsqrt)
    /* the volatility is too small */
    *iflag = 3;
    return;
  *iflag = 0;
  i = 0;
  max_iters = 50;
  done = FALSE;
  sig = sigma[0]; /* initial estimate */
  val = value;
  while ((i < max_iters) && (!done))
    /* Newton iteration */
    black_scholes(&val, greeks, s0, x, sig, r, q, put, iflag); /* compute the Black–Scholes option value, val */
    vega = greeks[4]; /* and vega. */
    sigl = sig = ((val - value)/vega); /* compute the new estimate of sigma */
    done = (tolx > FABS ((sigl - sig)/sigl)); /* check whether the specified accuracy has been reached */
    sig = sigl;
    ++i;
  } /* done = (tolx > FABS ((sigl - sig)/sigl)); */
  sigma[0] = sigl; /* return the estimate for sigma */
  return;
}

Code expert 9.3  Function to compute the implied volatility of European options

If the implied volatility of American options is required, the procedure is exactly the same. However, instead of using the Black–Scholes formula to compute both the option value and Vega, we use a binomial lattice to do this. The use of binomial lattices to obtain option prices and the Greeks is described in Section 10.4.

Below, in Code excerpt 9.4, is provided a simple test program which illustrates the use of the function implied_volatility; the results are presented in Table 9.3.
Pricing Assets

\[ S = 10.0; \]
\[ X = 10.5; \]
\[ r = 0.1; \]
\[ \text{sigmat} = 0.1; \]
\[ q = 0.04; \]
\[ \text{put} = 0; \]

\[
\text{printf}('' \text{Time option value implied volatility (Error)}
'');
\]

\[
\text{for}(i = 1; i < 6; ++i){}
\]

\[
\begin{align*}
T &= (\text{double})(0.5); \\
\text{black_scholes}(&\text{value}, \text{NULL}, S, X, \text{sigmat}, T, r, q, \text{put}, &\text{flag}); \\
\text{sigma}[0] &= 0.05; \\
\text{sigma}[1] &= 1.0; \\
\text{implied_volatility}(\text{value}, S, X, \text{sigmat}, T, r, q, \text{put}, &\text{flag}); \\
\text{printf}('' \%8.4f \%15.4f \%15.4f (\%8.4e)\n'', T, \text{value}, \text{sigma}[0], \text{FABS}((\text{sigmat} - \text{sigma}[0])); \\
\text{sigmat} &= \text{sigmat} + 0.1; \\
\end{align*}
\]

Code excerpt 9.4 Simple test program for function \text{implied_volatility}

9.3.5 Pricing options with Microsoft Excel

In this section we show how the Visual Basic within Excel can be used to create powerful derivative pricing applications based on the Black–Scholes formula. We will explain how Excel’s Visual Basic can be used to create an application that prices a selection of simple European put and call options at the press of a button.

In Section 9.3.3 we derived the Black–Scholes formula:

\[
c(S, E, \tau) = S N_1(d_1) - e^{-\tau} E N_1(d_2) \quad \text{and} \quad p(S, E, \tau) = -SN_1(-d_1) + e^{-\tau} EN_1(-d_2)
\]

where \( d_1 = \frac{\log(S/E)(r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}} = d_1 - \sigma\sqrt{\tau} \)

where \( S \) is the current value of the asset and \( \sigma \) is the volatility of the asset, and

\[
N_1(x) = \frac{1}{\sqrt{2\pi}} \int_{-\infty}^{x} e^{-x^2/2}dx
\]

The univariate cumulative standard normal distribution, \( N_1(x) \), can be evaluated in Excel by using its built in function NORMDIST. The definition of this function is as follows:

\[
\text{NORMDIST}(x, \text{mean, standard}\_\text{dev, cumulative})
\]

This function returns the normal cumulative distribution for the specified mean and standard deviation.

**Table 9.3** Calculated option values and implied volatilities from Code excerpt 9.4

<table>
<thead>
<tr>
<th>Time (in years)</th>
<th>Option value</th>
<th>True ( \sigma )</th>
<th>Error in estimated ( \sigma )</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.5</td>
<td>0.1959</td>
<td>0.1</td>
<td>( 2.7756 \times 10^{-16} )</td>
</tr>
<tr>
<td>1.0</td>
<td>0.8158</td>
<td>0.2</td>
<td>( 2.2204 \times 10^{-16} )</td>
</tr>
<tr>
<td>1.5</td>
<td>1.5435</td>
<td>0.3</td>
<td>( 3.8888 \times 10^{-16} )</td>
</tr>
<tr>
<td>2.0</td>
<td>2.3177</td>
<td>0.4</td>
<td>( 5.5511 \times 10^{-17} )</td>
</tr>
<tr>
<td>2.5</td>
<td>3.1033</td>
<td>0.5</td>
<td>( 1.1102 \times 10^{-16} )</td>
</tr>
</tbody>
</table>
Function parameters

- \( x \), is the value for which you want the distribution;
- \( \text{mean} \), is the arithmetic mean of the distribution;
- \( \text{standard_dev} \), is the standard deviation of the distribution;
- \( \text{cumulative} \), is a logical value that determines the form of the function. If cumulative is \( \text{TRUE} \), \( \text{NORMDIST} \) returns the cumulative distribution function; if \( \text{FALSE} \), it returns the probability density function.

If \( \text{mean} = 0 \) and \( \text{standard_dev} = 1 \), \( \text{NORMDIST} \) returns the standard normal distribution.

This function can be used to create the following Visual Basic function to calculate European option values within Excel.

```vbnet
Function bs_opt(S0 As Double, ByVal X As Double, sigma As Double, T As Double, r As Double, q As Double, ByVal putcall As Long) As Double
    ' Visual Basic Routine to calculate the value of either a European Put or European Call option.
    Dim temp As Double
    Dim d1 As Double
    Dim d2 As Double
    Dim SQT As Double
    Dim value As Double
    temp = Log(S0 / X)
    d1 = temp + (r - q + (sigma * sigma / 2#)) * T
    SQT = Sqr(T)
    d1 = d1 / (sigma * SQT)
    d2 = d1 - sigma * SQT
    If (putcall = 0) Then ' a call option
        value = S0 * Exp(q * T) * WorksheetFunction.NormDist(d1, 0#, 1#, True) / WorksheetFunction.NormDist(d2, 0#, 1#, True) * X * Exp(r * T)
    Else ' a put option
        value = S0 * Exp(-q * T) * WorksheetFunction.NormDist(-d1, 0#, 1#, True) + _
        X * WorksheetFunction.NormDist(-d2, 0#, 1#, True) * Exp(-r * T)
    End If
    bs_opt = value
End Function
```

**Code excerpt 9.5** Visual Basic code to price European options using the Black–Scholes formula

Once the function has been defined it can be accessed interactively using the Paste Function facility within Excel as shown in Figure 9.1.

The function `bs_opt` can also be incorporated into other Visual Basic code within Excel. To illustrate, if the following Visual Basic subroutine is defined:

```vbnet
Private Sub MANY_EUROPEANS_Click()
    Dim i As Long
    Dim putcall As Long
    Dim S0 As Double
    Dim q As Double
    Dim sigma As Double
    Dim T As Double
    Dim r As Double
    q = 0#
    T = 1.5
```
Pricing Assets

\[ r = 0.1 \]
\[ \sigma = 0.2 \]

For \( i = 1 \) To 22
\[ S_0 = \text{Sheet1.Cells}(i + 1, 1).value \]
\[ X = \text{Sheet1.Cells}(i + 1, 2).value \]
\[ \text{putcall} = \text{Sheet1.Cells}(i + 1, 3).value \]
\[ \text{Sheet1.Cells}(i + 1, 4).value = \text{bs_opt}(S_0, X, \sigma, T, r, q, \text{putcall}) \]
Next i
End Sub

Code excerpt 9.6  Visual Basic code that uses the function \text{bs_opt}

When the button labelled ‘CALCULATE OPTIONS’ is clicked, the values of 22 European options will be calculated using the data in columns 1–3 on worksheet 1. This is shown in Figures 9.2 and 9.3.

The cumulative standard normal distribution can also be used to provide analytic solutions for a range of other exotic options such as: Barrier options, Exchange options, Lookback options, Binary options, etc. A quick reference guide of the formulae for various options is included in Appendix B.

Figure 9.1  Using the function \text{bs_opt} interactively within Excel. Here a call option is priced with the following parameters: \( S = 10.0 \), \( X = 9.0 \), \( q = 0.0 \), \( T = 1.5 \), \( r = 0.1 \), and \( \sigma = 0.2 \)
Figure 9.2  Excel worksheet before calculation of the European option values

Figure 9.3  Excel worksheet after calculation of the European option values
9.4 BARRIER OPTIONS

9.4.1 Introduction

Barrier options are derivatives where the payoff depends on whether the asset price reaches a given barrier level, \( B \). Knockout options become worthless (cease to exist) if the asset price reaches the barrier, whereas knockin options come into existence when the asset price hits the barrier. We will consider the following single asset barrier options:

- **Down and out call**: A knockout vanilla call option, value \( c_{do}(S, B, E, \tau) \), which ceases to exist when the asset price reaches or goes below the barrier level.
- **Up and out call**: A knockout vanilla call option, value \( c_{uo}(S, B, E, \tau) \), which ceases to exist when the asset price reaches or goes above the barrier level.
- **Down and in call**: A knockin vanilla call option, value \( c_{di}(S, B, E, \tau) \), which comes into existence when the asset price reaches or goes below the barrier level.
- **Up and in call**: A knockin vanilla call option, value \( c_{ui}(S, B, E, \tau) \), which comes into existence when the asset price reaches or goes above the barrier level.

Since the following expressions must be true:

\[
  c(S, E, \tau) = c_{uo}(S, B, E, \tau) + c_{ui}(S, B, E, \tau) \quad (9.62)
\]

\[
  c(S, E, \tau) = c_{do}(S, B, E, \tau) + c_{di}(S, B, E, \tau) \quad (9.63)
\]

we need to only derive expressions for both the knockout options, and then use the above equations to calculate the value of the corresponding knockin options.

The notation that we will use is as follows:

- \( E \) is the strike price,
- \( S \) is the current value of the asset,
- \( B \) the barrier level,
- the symbol \( t \) represents the current time,
- \( T \) represents the time at which the option matures and \( \tau = T - t \), the duration of the option.
- The symbol \( s \), with constraint \( t \leq s \leq T \), is any intermediate time during which the option is alive.

9.4.2 Down and out call

If we consider Brownian motion (with zero drift) \( X_t \sim N(0, (s - t)\sigma^2) \), \( t \leq s \leq T \) which starts at \( X_t = 0 \) and, after time \( \tau = T - t \), ends at the point \( X_T = \) then (e.g. Freedman, 1983) the probability density function for this motion not to exceed the value \( X = b \) (where \( b > 0 \)) during time \( \tau \) is given by:

\[
  f(b \geq X_{s_{\text{max}}}, X) = \sqrt{\frac{2}{\pi}} \exp\left(\frac{2b(X - b)}{\sigma^2\tau}\right) \exp\left(-\frac{X^2}{2\sigma^2\tau}\right) \quad (9.64)
\]

where for convenience we have used \( \Omega = (2b - X)/\sigma^3\tau^{3/2} \), and \( X_{s_{\text{max}}} = \max(X_s, t \leq s \leq T) \).

Since \( X_s \) is Brownian motion without drift, and volatility \( \sigma \) then \(-X_s\) is identical Brownian motion. Therefore by substituting \( X \to -X\), and \( b \to -b \) in the above equation we obtain:

\[
  f(b \leq X_{s_{\text{min}}}, X) = -\sqrt{\frac{2}{\pi}} \exp\left(\frac{2b(X - b)}{\sigma^2\tau}\right) \exp\left(-\frac{X^2}{2\sigma^2\tau}\right) \quad (9.65)
\]
Analytic methods and single asset European options

where we have used \( X_s^{\text{min}} = \min(X_s, t \leq s \leq T) \). Equation 9.65 is the probability density function of \(-X_s\) staying above the value \( X = b \), where \( b < 0 \). These results can be generalized to include drift (e.g. Musiela and Rutkowski, 1998, p. 212), so that \( X_s \sim N((r - \sigma^2/2)(s - t), \sigma(s - t)) \), for \( t \leq s \leq T \). We now have the following results:

\[
f(b \geq X_s^{\text{max}}, X) = \Omega \sqrt{\frac{2}{\pi}} \exp\left(\frac{2b(X - b)}{\sigma^2} \right) \exp\left(-\frac{(X - (r - \sigma^2/2))^2}{2\sigma^2} \right) \tag{9.66}
\]

\[
f(b \leq X_s^{\text{min}}, X) = -\Omega \sqrt{\frac{2}{\pi}} \exp\left(\frac{2b(X - b)}{\sigma^2} \right) \exp\left(-\frac{(X - (r - \sigma^2/2))^2}{2\sigma^2} \right) \tag{9.67}
\]

A European down and out barrier option with maturity \( \tau \) and a barrier at \( X = B \) will cease to exist (become worthless) if at any time \( X_s \leq B \), for \( t \leq s \leq T \). The probability density function that the barrier option will continue to exist at time \( T \) if the end point is \( X \) is therefore:

\[
f(X > B) = -\sqrt{\frac{2}{\pi}} \int_{b = \log(B/S)}^{b = \infty} \Omega \exp\left(\frac{2b(X - b)}{\sigma^2} \right) \times \exp\left(-\frac{(X - (r - \sigma^2/2))^2}{2\sigma^2} \right) db \tag{9.68}
\]

or

\[
f(X > B) = -\sqrt{\frac{2}{\pi}} \exp\left(-\frac{(X - (r - \sigma^2/2))^2}{2\sigma^2} \right) \\
\times \int_{b = \log(B/S)}^{b = X} \Omega \exp\left(\frac{2b(X - b)}{\sigma^2} \right) db \tag{9.69}
\]

where we have integrated over all possible values of \( b \) (i.e. \( B < b < X \)) that keep the option alive. Recalling that:

\[
-\int_{b = \log(B/S)}^{b = X} \Omega \exp\left(\frac{2b(X - b)}{\sigma^2} \right) db = \int_{b = \log(B/S)}^{b = X} \frac{(X - 2b)}{\sigma^2} \exp\left(\frac{2b(X - b)}{\sigma^2} \right) db
\]

and noting that:

\[
\frac{\partial}{\partial b} \exp\left(\frac{2b(X - b)}{\sigma^2} \right) = \frac{2(X - 2b)}{\sigma^2} \exp\left(\frac{2b(X - b)}{\sigma^2} \right)
\]

we have:

\[
\int_{b = \log(B/S)}^{b = X} \frac{2(X - 2b)}{\sigma^2} \exp\left(\frac{2b(X - b)}{\sigma^2} \right) db = \left[ \exp\left(\frac{2b(X - b)}{\sigma^2} \right) \right]_{b = \log(B/S)}^{b = X} = \left\{ 1 - \exp\left(\frac{2\log(B/S)(X - \log(B/S))}{\sigma^2} \right) \right\}
\]
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So the value of the option is given by:

\[
f(X > B) = \frac{1}{\sigma \sqrt{2\pi}} \exp \left( -\frac{\left( X - (r - \sigma^2/2)\tau \right)^2}{2\sigma^2\tau} \right) \times \left\{ 1 - \exp \left( \frac{2 \log(B/S)(X - \log(B/S))}{\sigma^2\tau} \right) \right\}
\]

This integral is evaluated in Appendix G.1; here we merely state the result.

**Down and out call option**

\[
c_{do} = S \left( N_1(d_1) - N_1(d_4) \left( \frac{B}{S} \right)^{2r/\sigma^2+1} \right) - E \exp(-r\tau) \left( N_1(d_2) - N_1(d_3) \left( \frac{B}{S} \right)^{2r/\sigma^2-1} \right)
\]

(9.70)

where \( S \) is the current asset value, \( E \) the strike price, \( B \) the barrier level, \( \sigma \) the volatility, \( r \) the riskless interest rate, \( \tau \) the duration of the option, and:

\[
d_1 = \frac{\log(S/E) + (r + \sigma^2/2)\tau}{\sigma \sqrt{\tau}}, \quad d_2 = \frac{\log(S/E) + (r - \sigma^2/2)\tau}{\sigma \sqrt{\tau}},
\]

\[
d_3 = \frac{\log(B^2/SE) + (r - \sigma^2/2)\tau}{\sigma \sqrt{\tau}}, \quad \text{and} \quad d_4 = \frac{\log(B^2/ES) + (r + \sigma^2/2)\tau}{\sigma \sqrt{\tau}}
\]

In Code excerpt 9.7 below we provide the function `bs_opt_barrier_downout_call` which uses Equation 9.70 to price a down and out European call option. This routine will be used in Sections 10.6.3 and 10.6.6 to measure the accuracy achieved by using various finite-difference grid techniques to solve the Black–Scholes equation.

```c
void bs_opt_barrier_downout_call(double *value, double barrier_level, double s0, double x, double sigma, double t, double r, Integer *iflag) {
    /* Input parameters:
       barrier_level — the level of the barrier
       s0 — the current price of the underlying asset
       x — the strike price
       sigma — the volatility
       t — the time to maturity
       r — the interest rate
    Output parameters:
       == == == == ==
       value — the value of the option
       iflag — an error indicator */
    double one=1.0,two=2.0,zero=0.0;
    double eps,temp,temp1,temp2,a,b,d1,d2,d3,d4,d5,d6,d7,d8;
    double fac;
    /* Input parameters:
       == == == == ==
       barrier_level — the level of the barrier
       s0 — the current price of the underlying asset
       x — the strike price
       sigma — the volatility
       t — the time to maturity
       r — the interest rate
    Output parameters:
       == == == == ==
       value — the value of the option
       iflag — an error indicator */
    double one=1.0,two=2.0,zero=0.0;
    double eps,temp,temp1,temp2,a,b,d1,d2,d3,d4,d5,d6,d7,d8;
    double fac;
    /* Input parameters:
       == == == == ==
       barrier_level — the level of the barrier
       s0 — the current price of the underlying asset
       x — the strike price
       sigma — the volatility
       t — the time to maturity
       r — the interest rate
    Output parameters:
       == == == == ==
       value — the value of the option
       iflag — an error indicator */
```
eps = 0.02AJC;
if (x < eps) /* then strike price (X) is too small */
    printf("ERROR X is too small \n");
    return;
}
if (sigma < eps) /* then volatility (sigma) is too small */
    printf("ERROR sigma is too small \n");
    return;
if (t < eps) /* then time to expiry (t) is too small */
{ifail = 3;
    printf("ERROR option maturity is too small \n");
    return;
}
if (barrier_level == 0)
{printf("ERROR barrier must be > zero \n");
    return;
}
fac = sigma*sqrt(t);
temp1 = one + (two*r/(sigma*sigma));
temp2 = barrier_level/s0;
a = pow(temp2,temp1);
temp1 = one + (two*r/(sigma*sigma));
b = pow(temp2,temp1);
if (x > barrier_level){
d1 = (log(s0/x)+(r+0.5*sigma*sigma)*t)/fac;
d2 = (log(s0/x)+(r+0.5*sigma*sigma)*t)/fac;
temp = (s0*x)/(barrier_level*barrier_level);
d7 = (log(temp)-(r+0.5*sigma*sigma)*t)/fac;
d8 = (log(temp)-(r+0.5*sigma*sigma)*t)/fac;
temp1 = s0*(s15abc(d1)-b*(one-s15abc(d8)));
temp2 = x*exp(-r*t)*(s15abc(d2)-a*(one-s15abc(d7)));
    value = temp1/temp2;
}
else /* x < = barrier_level */
{d3 = (log(s0/barrier_level)+(r-0.5*sigma*sigma)*t)/fac;
d6 = (log(s0/barrier_level)-(r-0.5*sigma*sigma)*t)/fac;
d4 = (log(s0/barrier_level)+(r+0.5*sigma*sigma)*t)/fac;
d5 = (log(s0/barrier_level)-(r+0.5*sigma*sigma)*t)/fac;
temp1 = s0*(s15abc(d3)-b*(one-s15abc(d6)));
temp2 = x*exp(-r*t)*(s15abc(d4)-a*(one-s15abc(d5)));
    value = temp1/temp2;
}
return;
}

Code excerpt 9.7  Function to compute the value for European down and out call options

9.4.3 Up and out call

Here we will obtain an expression for an up and out European call option in a similar manner to that used in Section 9.3.5 for the down and out European call option. A European up and out barrier option with maturity \( T \) and a barrier at \( X = B \) will cease to exist (become worthless) if at any time \( X_s < B \), for \( t \leq s \leq T \). The probability density function that the barrier option will continue to exist at time \( T \) if the end point is \( X \) is therefore:

\[
 f(X < B) = \sqrt{\frac{2}{\pi}} \int_{b = X}^{B = \exp(b)} \Omega \exp\left(\frac{2b(X - b)}{\sigma^2 t}\right)
 \times \exp\left(-\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau}\right) db
\]

(9.71)
or

\[ f(X < B) = \sqrt{\frac{2}{\pi}} \exp \left( -\frac{(X - (r - \sigma^2/2)t)^2}{2\sigma^2t} \right) \]

\( \times \int_{b=x}^{b=\log(B/S)} \Omega \exp \left( \frac{2b(X - b)}{\sigma^2t} \right) db \) \hspace{1cm} (9.72)

where as in Section 9.3.5 we have used \( \Omega = (2b - X)/\sigma^3\sqrt{\tau} \) and have integrated overall possible values of \( b \) (i.e. \( B > b > X \)) that keep the option alive. Recalling that:

\[ \int_{b=x}^{b=\log(B/S)} \Omega \exp \left( \frac{2b(X - b)}{\sigma^2t} \right) db = \int_{b=x}^{b=\log(B/S)} \frac{(2b - X)}{\sigma^3\sqrt{\tau}} \exp \left( \frac{2b(X - b)}{\sigma^2t} \right) db \]

and noting:

\[ -\frac{\partial}{\partial b} \exp \left( \frac{2b(X - b)}{\sigma^2t} \right) = \frac{2(X - 2b)}{\sigma^2t} \exp \left( \frac{2b(X - b)}{\sigma^2t} \right) \]

we have:

\[ \int_{b=x}^{b=\log(B/S)} \frac{2(2b - X)}{\sigma^3\sqrt{\tau}} \exp \left( \frac{2b(X - b)}{\sigma^2t} \right) db = \left[ -\exp \left( \frac{2b(X - b)}{\sigma^2t} \right) \right]_{b=x}^{b=\log(B/S)} \]

\[ = \left\{ 1 - \exp \left( \frac{2\log(B/S)(X - \log(B/S))}{\sigma^2t} \right) \right\} \]

Therefore:

\[ f(X < B) = \frac{1}{\sigma\sqrt{\tau}\sqrt{\pi}} \exp \left( -\frac{(X - (r - \sigma^2/2)t)^2}{2\sigma^2t} \right) \]

\[ \times \left\{ 1 - \exp \left( \frac{2\log(B/S)(X - \log(B/S))}{\sigma^2t} \right) \right\} \] \hspace{1cm} (9.74)

We will now derive the formula for an up and out call option when \( E < B \). In fact if \( E > B \) then the option is worthless, since at the current time \( t \) the call option’s payout, \( \max(S_t - E, 0) = 0 \), and if \( S_t > E \) then the option will be knocked out.

\[ c_{uo} = \frac{\exp(-rt)}{\sigma\sqrt{\tau}\sqrt{\pi}} \int_{X=\log(E/S)}^{\infty} \{S \exp(X) - E\} f(X < B) dX \] \hspace{1cm} (9.75)

Taking into account the fact the option becomes worthless when \( S \exp(X) > B \), (i.e. \( X > \log(B/S) \)) we have:

\[ c_{uo} = \frac{\exp(-rt)}{\sigma\sqrt{\tau}\sqrt{\pi}} \int_{X=\log(E/S)}^{\log(B/S)} \{S \exp(X) - E\} f(X < B) dX \] \hspace{1cm} (9.76)
This integral is evaluated in Appendix G.2, and the value of the up and out call option $cuo$ is:

Up and out call option

$$cuo = \{N_1(k_7) - N_1(k_8)\}S \left(\frac{B}{S}\right)^{2r/\sigma^2+1}$$

$$- \{\exp(-r\tau)N_1(k_5) - N_1(k_6)\}E \left(\frac{B}{S}\right)^{2r/\sigma^2-1}$$

$$+ S\{N_1(k_2) - N_1(k_1)\} - E \exp(-r\tau)\{N_1(k_4) - N_1(k_3)\}$$

where $S$ is the current asset value, $E$ the strike price, $B$ the barrier level, $\sigma$ the volatility, $r$ the riskless interest rate, $\tau$ the duration of the option, and:

$$k_1 = \frac{\log(E/S) - (r + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}$$

$$k_2 = \frac{\log(B/S) - (r + \sigma^2/2)\tau}{\sqrt{\tau}}$$

$$k_3 = \frac{\log(E/S) - (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}}$$

$$k_4 = \frac{\log(B/S) - (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}}$$

$$k_5 = \frac{\log(ES/B^2) - (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}}$$

$$k_6 = \frac{\log(S/B) - (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}}$$

$$k_7 = \frac{\log(ES/B^2) - (r + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}$$

and

$$k_8 = \frac{\log(S/B) - (r + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}$$
Chapter 10

Numeric methods and single asset American options

10.1 INTRODUCTION

In Chapter 9 we discussed single asset European options and the analytic formulae which can be used to price them. Here we will consider the valuation of single asset American style options using both numeric methods and analytic formulae; in addition we will discuss the use of numerical techniques to value certain European options. The coverage in this section is as follows:

- Analytic methods applied to perpetual European and American options.
- Analytic approximation techniques for the valuation of American options.
- Binomial lattice techniques used for the valuation of American and European options.
- The valuation of American and European vanilla and barrier options using finite-difference grids.
- The valuation of American options via Monte Carlo simulation.

It should be mentioned that although much of the discussion here concerns the valuation of vanilla European and American puts and calls, the techniques used can be modified without much difficulty to include more exotic options with customized payoffs and early exercise features.

10.2 PERPETUAL OPTIONS

10.2.1 The perpetual American put

Here we derive the value, $P(S, E)$, for a perpetual American put with strike price $E$ on an asset of current value $S$. This option can be exercised at any time, and so there is no expiry date. Since the option is perpetual its payoff is time independent (see Merton (1973)) and the Black–Scholes equation reduces to the following second order ordinary differential equation:

$$\frac{\sigma^2 S^2}{2} \frac{d^2 V}{dS^2} + (r - q) \frac{dV}{dS} - rV = 0$$

(10.1)

where as usual $S$ is the asset price, $V$ is the option value, $\sigma$ is the volatility of the asset, $r$ is the riskless interest rate and $q$ is the continuous dividend yield.
If we substitute $S = \exp(X)$ we then have:

\[
\frac{dV}{dS} = \frac{dV}{dX} \frac{dX}{dS} = \exp(-X) \frac{dV}{dX}
\]

\[
\frac{d^2V}{dS^2} = \frac{d}{dS} \frac{d}{dX} \left\{ \frac{dV}{dX} \exp(-X) \right\} = \exp(-2X) \frac{d^2V}{dX^2} - \frac{dV}{dX} \exp(-2X)
\]

Substituting the above results into Equation 10.1 we obtain:

\[
\frac{\sigma^2 \exp(2X) \exp(-2X) \left\{ \frac{d^2V}{dX^2} - \frac{dV}{dX} \right\}}{2} + (r - q) \exp(X) \exp(-X) \frac{dV}{dX} - rV = 0
\]

\[
\frac{\sigma^2}{2} \frac{d^2V}{dX^2} + \left\{ \frac{2r - q}{\sigma^2} - 1 \right\} \frac{dV}{dX} - \frac{2r}{\sigma^2} V = 0
\]

So

\[
\frac{d^2V}{dX^2} + \left\{ \frac{2r - q}{\sigma^2} - 1 \right\} \frac{dV}{dX} - \frac{2r}{\sigma^2} V = 0 \quad (10.2)
\]

Equation 10.2 is a homogeneous equation with constant coefficients, so we can look for solutions of the form $V = \exp(mX)$. This gives:

\[
m^2 + \left\{ \frac{2(r - q)}{\sigma^2} - 1 \right\} m - \frac{2r}{\sigma^2} = 0 \quad (10.3)
\]

which can be solved to yield:

\[
m_1 = \frac{1}{2} \left\{ \frac{-2(r - q)}{\sigma^2} + 1 \right\} + \frac{1}{2} \sqrt{\left\{ \frac{2(r - q)}{\sigma^2} - 1 \right\}^2 + \frac{8r}{\sigma^2}} \quad (10.4)
\]

and

\[
m_2 = \frac{1}{2} \left\{ \frac{-2(r - q)}{\sigma^2} + 1 \right\} - \frac{1}{2} \sqrt{\left\{ \frac{2(r - q)}{\sigma^2} - 1 \right\}^2 + \frac{8r}{\sigma^2}} \quad (10.5)
\]

The general solution to Equation 10.2 is therefore:

\[
V(X) = A_1 \exp(m_1X) + A_2 \exp(m_2X) \quad (10.6)
\]

However, since we are solving Equation 10.1 we would like the solution in terms of the asset price $S$. So re-substituting $S = \exp(X)$, and using the fact that $\exp(aX) = \exp(X)^a$, we obtain:

\[
A_1 \exp(m_1X) = A_1 (\exp(X))^{m_1} = A_1 S^{m_1}
\]

and

\[
A_2 \exp(m_2X) = A_2 (\exp(X))^{m_2} = A_2 S^{m_2}
\]
The general solution of Equation 10.2 as a function of $S$ is therefore:

$$V(S) = A_1 S^{m_1} + A_2 S^{m_2}$$ (10.7)

If we assume that $(2(r - D)/\sigma^2) > 1$ then $m_1 > 0$ and $m_2 < 0$. (Note: When $(2(r - D)/\sigma^2) < 1$, $m_1 < 0$ and $m_2 > 0$.)

For the perpetual American put as $S \to \infty$ we have $P(S, E) \to 0$. This means that the coefficient $A_1$ in Equation 10.7 must be zero, and $P(S, E) = A_2 S^{m_2}$. Suppose we decide that we will exercise the option when $S \leq S^*$, where $S^*$ is termed the critical value of $S$, then the payoff (which is positive) at $S = S^*$ will be

$$P(S^*, E) = E - S^*$$ (10.8)

This gives

$$P(S^*, E) = A_2 (S^*)^{m_2} = E - S^*$$ (10.9)

Solving for $A_2$ gives:

$$A_2 = \frac{E - S^*}{(S^*)^{m_2}}$$ (10.10)

So we have:

$$P(S, E) = (E - S^*) \left( \frac{S}{S^*} \right)^{m_2}$$ (10.11)

We are now going to find the value of $S^*$ which maximizes the option value at any time before exercise. Differentiating Equation 10.11 and setting the value to zero we have:

$$\frac{\partial}{\partial S^*} \left\{ (E - S^*) \left( \frac{S}{S^*} \right)^{m_2} \right\} = \frac{1}{S^*} \left( \frac{S}{S^*} \right)^{m_2} \left\{ -S^* - m_2(E - S^*) \right\} = 0$$

and

$$-S^* - m_2(E - S^*) = 0, \quad \text{so} \quad S^* = \frac{E}{1 - 1/m_2}$$

So substituting into Equation 10.10 results in:

$$A_2 = -\frac{1}{m_2} \left( \frac{E}{1 - 1/m_2} \right)^{1-m_2}$$

When there are no dividends, $q = 0$, we have from Equation 10.5 that

$$m_2 = \frac{1}{2} \left\{ -\frac{2r}{\sigma^2} + 1 \right\} - \frac{1}{2} \sqrt{\left\{ \frac{2r}{\sigma^2} - 1 \right\}^2 + \frac{8r}{\sigma^2}}$$ (10.12)

but

$$\left\{ \frac{2r}{\sigma^2} - 1 \right\}^2 + \frac{8r}{\sigma^2} = \left( 1 + \frac{2r}{\sigma^2} \right)^2$$
Therefore

\[ m_2 = \frac{1}{2} \left\{ -\frac{2r}{\sigma^2} + 1 - \frac{2r}{\sigma^2} - 1 \right\} \quad \text{and} \quad m_2 = -\frac{2r}{\sigma^2} \]  \hspace{1cm} (10.13)

Substituting for \( m_2 \) and \( A_2 \) in Equation 10.9 we thus obtain the value for a perpetual American put without dividends as:

\[ P(S, E) = \frac{\sigma^2 S - 2r/\sigma^2}{2r} \left\{ \frac{E}{1 + (\sigma^2/2r)} \right\}^{1+(2r/\sigma^2)} \]  \hspace{1cm} (10.14)

see Merton (1973), Equation 52, p. 174.

10.2.2 The perpetual American call

Here we derive the value, \( C(S, E) \), for a perpetual American call with strike price \( E \) on an asset of current value \( S \). For the perpetual American call as \( S \to 0 \) we have \( C(S, E) \to 0 \). In the previous section we mentioned that \( m_2 < 0 \) which means that the \( A_2 S^{m_2} \to \infty \) as \( S \to 0 \). Thus if Equation 10.7 is to yield a finite solution for the perpetual American call we must set \( A_2 = 0 \) and look for solutions of the form:

\[ C(S, E) = A_1 S^{m_1} \]

The payoff for the call option is max \((S - E, 0)\), so when \( S^* = S \) we have:

\[ C(S^*, E) = S^* - E = A_1 (S^*)^{m_1} \]  \hspace{1cm} (10.15)

and

\[ A_1 = \left( \frac{S^* - E}{S^*} \right)^{m_1} \]  \hspace{1cm} (10.16)

This gives

\[ C(S, E) = (S^* - E) \left( \frac{S}{S^*} \right)^{m_1} \]  \hspace{1cm} (10.17)

As in Section 10.2.1 we find the value \( S^* \) which maximizes the option value by differentiating Equation 10.17 w.r.t. \( S^* \) and setting the value to zero. This yields:

\[ \frac{\partial}{\partial S^*} \left\{ (E - S^*) \left( \frac{S}{S^*} \right)^{m_1} \right\} = \frac{1}{S^*} \left( \frac{S}{S^*} \right)^{m_1} \{ S^* - m_1 (S^* - E) \} = 0 \]

and

\[ S^* - m_1 (S^* - E) = 0, \quad \text{so} \quad S^* = \frac{E}{1 - 1/m_1} \]  \hspace{1cm} (10.18)
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Now using \( A_1 = (S^* - E)/(S^*)^{m_1} \) we obtain

\[
A_1 = \frac{E^{1/(1 - 1/m_1)} - 1}{E^{m_1 - 1}(1 - 1/m_1)(1 - 1/m_1)^{m_1 - 1}} = \frac{1}{E^{m_1 - 1}} \frac{1}{(1 - 1/m_1)} (1 - 1/m_1)^{m_1 - 1}
\]

\[
A_1 = \frac{1}{m_1} \left( \frac{1 - 1/m_1}{E} \right)^{m_1 - 1} = \frac{1}{m_1} \left( \frac{E}{1 - 1/m_1} \right)^{1-m_1}
\]

Therefore the value of the perpetual American call option is:

\[
C(S, E) = \frac{1}{m_1} \left( \frac{E}{1 - 1/m_1} \right)^{1-m_1} S^{m_1} \tag{10.19}
\]

When there are no dividends, \( q = 0 \), we have from Equation 10.4 that

\[
m_1 = \frac{1}{2} \left\{ \frac{-2r}{\sigma^2} + 1 \right\} + \frac{1}{2} \sqrt{\left( \frac{2r}{\sigma^2} - 1 \right)^2 + \frac{8r}{\sigma^2}} \tag{10.21}
\]

but

\[
\left\{ \frac{2r}{\sigma^2} - 1 \right\}^2 + \frac{8r}{\sigma^2} = \left( 1 + \frac{2r}{\sigma^2} \right)^2
\]

so substituting into Equation 10.21 we obtain

\[
m_1 = \frac{1}{2} \left\{ \frac{-2r}{\sigma^2} + 1 + \frac{2r}{\sigma^2} + 1 \right\} = 1 \tag{10.23}
\]

Setting \( m_1 = 1 \) in Equation 10.18 we thus find that \( S^* = \infty \). Therefore from Equation 10.16:

\[
A_1 = \frac{(S^* - E)}{(S^*)^{m_1}} = \frac{(S^* - E)}{(S^*)} = 1 \tag{10.24}
\]

This means that the value of a perpetual American call with zero dividends is:

\[
C(S, E) = A_1 S^{m_1} = 1 \times S = S \tag{10.25}
\]

10.2.3 Perpetual European options

We can easily derive expressions for perpetual European options by using the Black–Scholes formulae given in Section 9.3.3. It can be seen that as the option maturity, \( \tau \), tends to infinity \( d_1 \to \infty \) and \( d_2 \to -\infty \). This means that for perpetual options we should use \( N_1(d_1) \sim 1 \) and \( N_1(d_2) \sim 0 \) in the Black–Scholes formulae. Therefore when \( q > 0 \), we have \( c(S, E) \sim 0 \) and \( p(S, E) \sim 0 \). Also when \( q = 0 \) we have \( c(S, E) \sim S \) and \( p(S, E) \sim 0 \).

The value of a European call (when \( q = 0 \)) is therefore:

\[
c(S, E) = C(S, E) = S \tag{10.26}
\]

which means that, when there are no dividends, the perpetual American call and the perpetual European call options have the same value; the current asset price \( S \).
10.2.4 Perpetual European down and out call

Here we find the value of a perpetual down and out European call barrier option, see Merton (1973).

Let the exercise price be $E$ and the barrier be at $B$ where $B < E$.

Since the Black–Scholes partial differential equation governs the price of the option we can, as before, look for solutions of the form:

$$c(S, E)_{do} = A_1 S^{m_1} + A_2 S^{m_2}$$  \hspace{1cm} (10.27)

subject to the boundary conditions: (i) $c_{do}(B, E) = 0$ and (ii) $c(\infty, E)_{do} = S$, see the previous section.

From (i) we have:

$$c_{do}(B, E) = A_1 B^{m_1} + A_2 B^{m_2} = 0, \quad \text{so} \quad A_1 = -A_2 B^{m_2 - m_1}$$

Therefore

$$c_{do}(S, E) = -A_2 B^{m_2 - m_1} S^{m_1} + A_2 S^{m_2}$$

From (ii), as $S \to \infty$:

$$c_{do}(S, E) = -A_2 B^{m_2 - m_1} S^{m_1} + A_2 S^{m_2} = S$$

However, since $m_2 < 0$, we have $A_2 S^{m_2} \to 0$, as $S \to \infty$, giving

$$c_{do}(S, E) = -A_2 B^{m_2 - m_1} S^{m_1} = S$$

So

$$A_2 = \frac{S^{1-m_1}}{B^{m_2-m_1}} \quad \text{and} \quad c_{do}(S, E) = \frac{S^{1-m_1} S^{m_1} B^{m_2-m_1}}{B^{m_2-m_1}} - \frac{S^{1-m_1} S^{m_2}}{B^{m_2-m_1}}$$

which results in:

$$c_{do}(S, E) = S - \frac{S^{1+m_2-m_1}}{B^{m_2-m_1}}$$  \hspace{1cm} (10.28)

When there are no dividends ($q = 0$) we have already shown in Sections 10.2.1 and 10.2.2 that $m_1 = 1$ and $m_2 = -2r/\sigma^2$ so the value of a perpetual down and out call is (see Merton (1973)):

$$c_{do}(S, E) = S - \frac{S^{m_2}}{B^{m_2-1}} = S - B \left( \frac{S}{B} \right)^{-2r/\sigma^2}$$  \hspace{1cm} (10.29)

10.3 APPROXIMATIONS FOR VANILLA AMERICAN OPTIONS

10.3.1 American call options with cash dividends

In this section we will consider the valuation of vanilla American call options with cash dividends, and discuss both the Roll, Geske, and Whaley method and also the Black (1975) method. We will first consider the Roll, Geske, and Whaley method.
The Roll, Geske, Whaley approximation

This method uses the work of Roll (1977), Geske (1979), and Whaley (1981). Let $S$ be the current (time $t$) price of an asset which pays a single cash dividend $D_1$ at time $t_1$. At the ex-dividend date, $t_1$, there will be a decrease in the asset’s value from $S_{t_1}$ to $S_{t_1} - D_1$. Also the current asset price net of escrowed dividends is:

$$S_D = S - D_1 \exp(-r(t - t_1))$$ (10.30)

where $r$ is the riskless interest rate.

Now consider an American call option, with strike price $E$ and expiry time $T$, which is taken out on this asset. At $t_1$ there will be a given ex-dividend asset price, $S^*$, above which the option will be exercised early. This value can be found by solving the following equation:

$$c(S^*, E, \tau_1) = S^* + D_1 - E$$ (10.31)

where $c(S^*, E, \tau_1)$ is the Black–Scholes value of a European call option with strike price $E$ and maturity $\tau_1 = T - t_1$, on an asset with current value $S^*$ at time $t_1$. If just prior to the ex-dividend date $S_{t_1} > S^*$, then the American option will be exercised and realize a cash payoff of $S_{t_1} + D_1 - E$. On the other hand if $S_{t_1} \leq S^*$ then the option is worth more unexercised and it will be held until option maturity at time $T$.

We can rewrite Equation 10.31 so that $S^*$ is the root of the following equation:

$$K(S^*) = c(S^*, E, \tau_1) - S^* - D_1 + E = 0$$ (10.32)

where $K(S^*)$ denotes the function in the single variable $S^*$.

A well-known technique for solving Equation 10.32 is Newton’s method, which in this case takes the form:

$$S^*_{i+1} = S^*_i - \frac{K(S^*_i)}{K'(S^*_i)}$$ (10.33)

where $S^*_i$ is the $i$th approximation to $S^*$, and $S^*_{i+1}$ is the improved $(i + 1)$th approximation.

If we now consider the terms in Equation 10.33 we have that

$$K(S^*_i) = c(S^*_i, E, \tau_1) - S^*_i - D_1 + E$$

and

$$K'(S^*_i) = \frac{\partial K(S^*_i)}{\partial S^*_i} = \frac{\partial c(S^*_i, E, \tau_1)}{\partial S^*_i} - 1$$

Also from Equation C.14 in Appendix C.3

$$\frac{\partial c(S^*_i, E, \tau_1)}{\partial S^*_i} = N(d_1(S^*_i))$$

We note that here the continuous dividend yield, $q = 0$.

So

$$K'(S^*_i) = \frac{\partial K(S^*_i)}{\partial S^*_i} = N(d_1(S^*_i)) - 1, \text{ where } d_1 = \frac{\log(S^*_i/E) + (r + \sigma^2/2)(\tau_1)}{\sigma \sqrt{T - t_1}}$$
Numeric methods and single asset American options

Substituting these results into Equation 10.33 gives:

\[ S_{i+1}^* = S_i^* - \left\{ \frac{c(S_i^*, E, \tau_1) - (S_i^* + D_1 - E)}{N(d_1(S_i^*))) - 1} \right\} \]

On rearrangement this yields

\[ S_{i+1}^* = S_i^* \frac{N_1(d_1(S_i^*))) - c(S_i^*, E, \tau_1) + D_1 - E}{N(d_1(S_i^*))) - 1}, \quad \text{for} \quad i = 0, \ldots, \text{max\_iter} \tag{10.34} \]

where a convenient initial approximation is to choose \( S_0^* = E \), and \( \text{max\_iter} \) is the maximum number of iterations that are to be used.

We will now quote the Roll, Geske, and Whaley formula for the current value of an American call which pays a single cash dividend \( D_1 \) at time \( t_1 \), it is:

\[
C(S, E, \tau) = S_D \left\{ N_1(b_1) + N_2(a_1, -b_1, \sqrt{(t_1 - t)/\tau}) \right\} + D_1 \exp(-r(t_1 - t))N_1(b_2) - E \exp(-r\tau) \left\{ N_1(b_2) \exp(r(\tau)) + N_2(a_2, -b_2, -\sqrt{(t_1 - t)/\tau}) \right\} \tag{10.35}
\]

where \( S_D \) is given by Equation 10.30, \( E \) is the exercise price, \( T \) is the option expiry date, \( t \) represents the current time, \( \tau \) is the option maturity, \( N_1(a) \) is the univariate cumulative normal density function with upper integral limit \( a \), and \( N_2(a, b, \rho) \) is the bivariate cumulative normal density function with upper integral limits \( a \) and \( b \) and correlation coefficient \( \rho \). The other symbols used in Equation 10.35 are defined as

\[
a_1 = \frac{\log(S/E) + (r + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}, \quad a_2 = a_1 - \sigma\sqrt{\tau}
\]

\[
b_2 = \frac{\log(S/S^*) + (r + \sigma^2/2)(t_1 - t)}{\sigma\sqrt{(t_1 - t)}}, \quad b_2 = b_1 - \sigma\sqrt{(t_1 - t)}
\]

and \( S \) is the current (time \( t \)) asset price, \( S^* \) is found using Equation 10.34, \( r \) is the riskless interest rate, \( \sigma \) is the asset’s volatility, \( \tau = T - t \) and \( \tau_1 = T - t_1 \).

To compute the value of an American call option which pays \( n \) cash dividends \( D_i, i = 1, \ldots, n \) at times \( t_i, i = 1, \ldots, n \), we can use the fact that optimal exercise normally only ever occurs at the final ex-dividend date \( t_n \), see for example Hull (1997). Under these circumstances Equation 10.35 can still be shown to value the American call but now \( t - 1 \) should be set to \( t_n \), \( D_1 \) should be set to \( D_n \), and \( S_D \) is given by:

\[
S_D = S - \sum_{i=1}^n D_i \exp(-r(t_i - t)) \tag{10.36}
\]

A program to compute the Roll, Geske, and Whaley approximation for an American call option with multiple cash dividends is given in Code excerpt 10.1. Here the NAG C library functions s15abc and g01hac are used to calculate the values of \( N_1(a) \) and \( N_2(a, b, \rho) \) respectively. Code excerpt 10.3 was used to compute the values presented in Table 10.1. These compare the Roll, Geske, and Whaley approximation with the Black approximation, which we will now briefly discuss.
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Table 10.1 A comparison of the computed values for American call options with dividends, using the Roll, Geske, and Whaley approximation, and the Black approximation. The parameters used were: $E = 100.0$, $r = 0.04$, $\sigma = 0.2$, $\tau = 2.0$ and there is one cash dividend of value 5.0 at time $t = 1.0$. The current stock price, $S$, is varied from 80.0 to 120.0. The results are in agreement with those given in Table 1 of Whaley (1981)

<table>
<thead>
<tr>
<th>Stock price</th>
<th>Critical price, $S^*$</th>
<th>RGW approximation</th>
<th>Black approximation</th>
</tr>
</thead>
<tbody>
<tr>
<td>80.0</td>
<td>123.582</td>
<td>3.212</td>
<td>3.208</td>
</tr>
<tr>
<td>85.0</td>
<td>123.582</td>
<td>4.818</td>
<td>4.808</td>
</tr>
<tr>
<td>90.0</td>
<td>123.582</td>
<td>6.839</td>
<td>6.820</td>
</tr>
<tr>
<td>95.0</td>
<td>123.582</td>
<td>9.276</td>
<td>9.239</td>
</tr>
<tr>
<td>100.0</td>
<td>123.582</td>
<td>12.111</td>
<td>12.048</td>
</tr>
<tr>
<td>105.0</td>
<td>123.582</td>
<td>15.316</td>
<td>15.215</td>
</tr>
<tr>
<td>110.0</td>
<td>123.582</td>
<td>18.851</td>
<td>18.703</td>
</tr>
<tr>
<td>115.0</td>
<td>123.582</td>
<td>22.676</td>
<td>22.470</td>
</tr>
<tr>
<td>120.0</td>
<td>123.582</td>
<td>26.748</td>
<td>26.476</td>
</tr>
</tbody>
</table>

void RGW_approx(double *opt_value, double *critical_value, Integer n_divs, double dividends[], double Divs_T[], double S0, double X, double sigma, double T, double r, Integer *iflag)
{
    /* Input parameters:
    n_divs — the number of dividends
    dividends[] — the dividends: dividends[0] contains the first dividend, dividend[1] the second etc.
    Divs_T[] — the times at which the dividends are paid: Divs_T[0] is the time at which the first
    dividend is paid Divs_T[1] is the time at which the second dividend is paid, etc.
    S0 — the current value of the underlying asset
    X — the strike price
    sigma — the volatility
    T — the time to maturity
    r — the interest rate
    *
    Output parameters:
    opt_value — the value of the option
    critical_value — the critical value
    iflag — an error indicator
    */
    double A_1,A_2,S_star,a1,a2,nt1,t1,S;
    double b1,b2,d1,alpha,h,div,beta,temp,temp1,temp2,temp3;
    double pdf,b,eur_val,fac,tol,loc_q,err,zero = 0.0;
    Boolean iterate;
    Integer i,iflagx,putx;
    static NagError nagerr;
    loc_q = 0.0;
    temp = 0.0;
    for (i = 0; i < n_divs; ++i) { /*Check the Divs_T array*/
        if ((Divs_T[i] <= temp) || (Divs_T[i] > T) || (Divs_T[i] <= zero)) {
            *iflag = 2;
            return;
        }
        temp = Divs_T[i];
    }
    /* calculate the present value of the dividends (excluding the final one) */
    temp = 0.0;
    for (i = 0; i < n_divs-1; ++i) {
        temp = fac * dividends[i] * exp(-r*Divs_T[i]);
    }
    t1 = Divs_T[n_divs-1];
    /* decrease the stock price by the present value of all dividends */
    div = dividends[n_divs-1];
    S = S0-temp-div*exp(-r*t1);
    iterate = TRUE;
    tol = 0.000001;
    S_star = X;
    while (iterate) { /* calculate S_star, iteratively */
        /* calculate the Black-Scholes value of a European call */
### Numeric methods and single asset American options

\[
d_1 = \frac{(\log(S_{\star}/X) + (r + (\sigma \cdot \sigma/2.0)) \cdot (T-t_1)) / (\sigma \cdot \sqrt{T}/C_0)}{t_1};
\]
\[
\text{put}_X = 0;
\]
\[
\text{loc}_Q = 0.0;
\]
\[
\text{black_scholes}(&\text{eur_val}, \text{NULL}, S_{\star}, X, \sigma, T/C_0, t_1, r, \text{loc}_Q, \text{put}_X, &\text{iflag});
\]
\[
S_{\star} = \frac{(S_{\star}/s_{15abc}(d_1) - \text{eur_val} + \text{div}/C_0)}{s_{15abc}(d_1)/C_0};
\]
\[
\text{err} = \frac{\text{FABS(eur_val} - (S_{\star} + \text{div}/C_0))}{X};
\]
\[
\text{if (err < tol) iterate = FALSE;}
\]

\[
d_1 = \frac{(\log(S/X) + (r + (\sigma \cdot \sigma/2.0)) \cdot T)}{(\sigma \cdot \sqrt{T})};
\]
\[
a_2 = \frac{a_1}{\sigma \cdot \sqrt{T}};
\]
\[
b_1 = \frac{(\log(S/S_{\star}) + (r + (\sigma \cdot \sigma/2.0)) \cdot t_1)}{(\sigma \cdot \sqrt{t_1})};
\]
\[
b_2 = \frac{b_1}{\sigma \cdot \sqrt{t_1}};
\]
\[
\text{nt1} = \sqrt{t_1/T};
\]
\[
\text{temp1} = S \cdot (s_{15abc}(b_1) + g01hac(a_1, b_1, \text{nt1}, &\text{nagerr}));
\]
\[
\text{temp2} = X \cdot \exp(-r \cdot T) \cdot g01hac(a_2, -b_2, -\text{nt1}, \text{nagerr}) \cdot (X - \text{div}) \cdot \exp(-r \cdot t_1) \cdot s_{15abc}(b_2);
\]
\[
\text{*opt_value} = \text{temp1} + \text{temp2};
\]
\[
\text{*critical_value} = S_{\star};
\]

#### Code excerpt 10.1

Function to compute the Roll, Geske, and Whaley approximation for the value of an American call option with discrete dividends.

We will now consider the Black approximation.

### Black's approximation

The Black (1975) approximation for an American call with cash dividends is simpler than the Roll, Geske, and Whaley method we have just described. For an American call option which expires at time \( T \), with \( n \) discrete cash dividends \( D_i \), \( i = 1, \ldots, n \), at times \( t_i \), \( i = 1, \ldots, n \), it involves calculating the prices of European options that mature at times \( T \) and \( t_n \), and then setting the option price to the greater of these two values, see for example Hull (1997).

The Black approximation, \( C_{BL} \), can be expressed more concisely in terms of our previously defined notation as:

\[
C_{BL}(S, E, \tau) = \max(v_1, v_2)
\]

where \( v_1 \) and \( v_2 \) are the following European calls

\[
v_1 = c(S_D, E, \tau) \quad \text{and} \quad v_2 = c(S_D^+, E, \tau), \quad \tau = T - t \quad \tau_1 = T - t_n
\]

and

\[
S_D = S - \sum_{i=1}^{n} D_i \quad \text{and} \quad S_D^+ = S - \sum_{i=1}^{n-1} D_i
\]

Code excerpt 10.2 below computes the Black approximation.

```c
void black_approx(double *value, Integer n_divs, double dividends[], double Divs_T[],
 double S0, double X, double sigma, double T, double r, Integer put, Integer *ifail) {

  /* Input parameters:
   * n_divs — the number of dividends
   * dividends[] — the dividends, dividends[0] contains the first dividend, dividend[1] the second etc.
   * Divs_T[] — the times at which the dividends are paid, Divs_T[0] is the time at which the first
   * dividend is paid, Divs_T[1] is the time at which the second dividend is paid, etc.
   * S0 — the current value of the underlying asset
   */

  // ...
```
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- **X** — the strike price
- **sigma** — the volatility
- **T** — the time to maturity
- **r** — the interest rate
- **put** — if put is 0 then a call option, otherwise a put option

Output parameters:

- `value` — the value of the option
- `iflag` — an error indicator

```c
double zero = 0.0;
double beta, temp, temp1, temp2, temp3;
double tn, val_T, val_tn, tol, loc_q, err, fac;
Integer i, ifailx;
loc_q = 0.0;
temp = 0.0;
for (i = 0; i < n_divs; ++i) {
    if (Divs_T[i] < temp)
        printf("Error in Divs_T array, elements not increasing \n");
    if (Divs_T[i] > T)
        printf("Error in Divs_T array element has a value greater than T \n");
    if (Divs_T[i] <= zero)
        printf("Error in Divs_T array element <= zero \n");
    temp = Divs_T[i];
}
/* calculate the present value of the dividends */
fac = 0.0;
for (i = 0; i < n_divs; ++i) {
    fac = fac + dividends[i] * exp(-r * Divs_T[i]);
}
temp = S0 - fac;
/* calculate the value of the option on expiry */
black_scholes(&val_T, NULL, temp, X, sigma, T, r, loc_q, put, &ifailx);
/* calculate the value of the option on last dividend date */
if (Divs_T[n_divs - 1] < t1)
    printf("Error in Divs_T array element less than t1 \n");
val_tn = black_value;
/* calculate the value of the option on last dividend date */
val_T = MAX(val_tn, val_T);
```

**Code excerpt 10.2**  Function to compute the value of the Black approximation for the value of an American call option with discrete dividends

**Code excerpt 10.3** below uses the same values as in Whaley (1981) and compares the Roll, Geske, and Whaley approximation with that of Black; the results are presented in Table 10.1.

```c
double q, r, temp, loc_r;
Integer i, m, m2, m_acc;
double S0, E, T, sigma, t1, delta, value, ad_value, put_value;
Integer is_american, ifail, put;
double bin_greeks[5], greeks[5], bin_value, bs_value;
double opt_value, critical_value, E1, E2, crit1, crit2;
double black_value;
double Divs_T[3], dividends[3];
Integer n_divs, put;

E = 100.0;
r = 0.04;
sigma = 0.2;
T = 2.0;
t1 = 1.0;
put = 0;
/* check using the same parameters as in Whaley (1981) */
Divs_T[0] = 1.0;
dividends[0] = 5.0;
n_divs = 1;
printf("\nPrice S RGW Approximation Black Approximation \n\n");
for (i = 0; i < 9; ++i) {
    put = 0;
    S0 = 80.0 + (double)i * 5.0;
    /*...*/
```
Code excerpt 10.3  Simple test program to compare the results of function opt_RGW_approx with function opt_black_approx, the parameters used are the same as in Whaley (1981)

We will now consider a more general technique for pricing both American puts and calls.

10.3.2 The MacMillan, Barone-Adesi, and Whaley method

Here we consider a method of pricing American options which relies on an approximation that reduces a transformed Black–Scholes equation into a second order ordinary differential equation, see Barone-Adesi and Whaley (1987) and MacMillan (1986). It thus provides an alternative way of evaluating American options that can be used instead of computationally intensive techniques such as finite-difference methods. Although the method prices American options it is really based on the value of an American option relative to the corresponding European option value (which can readily be computed using the Black–Scholes pricing formula).

Since an American option gives more choice its value is always at least that of its European counterpart. This early exercise premium \( \nu(S, E, \tau) \geq 0 \) is now defined more precisely for American puts and calls. If at current time \( t \) the asset price is \( S \), then the early exercise premium for an American call which expires at time \( T \), and therefore has maturity \( t = T - \tau, \) is:

\[
\nu_c(S, E, \tau) = C(S, E, \tau) - c(S, E, \tau) \geq 0
\]

where \( C(S, E, \tau) \) denotes the value of the American call and \( c(S, E, \tau) \) denotes the value of the corresponding European call. The early exercise premium of an American put option, \( \nu_p(S, E, \tau) \), is similarly defined as:

\[
\nu_p(S, E, \tau) = P(S, E, \tau) - p(S, E, \tau) \geq 0
\]

where \( P(S, E, \tau) \) is the value of the American put, and \( p(S, E, \tau) \) is the value of the corresponding European put. The key insight provided by the MacMillan, Barone-Adesi, and Whaley method is that since both the American and European option values satisfy the Black–Scholes partial differential equation so does the early exercise premium, \( \nu(S, E, \tau) \); see Section 9.3.1. This means that we can write:

\[
\frac{\partial \nu}{\partial t} + (r - q)S \frac{\partial \nu}{\partial S} + \frac{\sigma^2 S^2}{2} \frac{\partial^2 \nu}{\partial S^2} = r\nu
\]

where as usual \( S \) is the asset price, \( r \) the continuously compounded interest rate, \( q \) the continuously compounded dividend, \( \sigma \) the volatility, and time \( t \) increases from the current time to the expiry time \( T \).
We will now introduce the variable \(h(\tau) = 1 - \exp(-r\tau)\) and use the factorization \(\nu(S, E, \tau) = h(\tau)g(S, E, h)\). From standard calculus we obtain:
\[
\frac{\partial \nu}{\partial t} = g \frac{\partial h}{\partial t} + h \frac{\partial g}{\partial t} = rg(h - 1) + h \frac{\partial g}{\partial h} \frac{\partial h}{\partial t} = rg(h - 1) + hr(h - 1) \frac{\partial g}{\partial h}
\]

and also
\[
\frac{\partial \nu}{\partial S} = h \frac{\partial g}{\partial S} \quad \text{and} \quad \frac{\partial^2 \nu}{\partial S^2} = h \frac{\partial^2 g}{\partial S^2}
\]

Substituting these results into Equation 10.39 yields the following transformed Black–Scholes equation:
\[
\frac{S^2 \sigma^2 h \frac{\partial^2 g}{\partial S^2}}{2} + (r - q)Sh \frac{\partial g}{\partial S} + rg(h - 1) \frac{\partial g}{\partial h} = rgh
\]  
(10.40)

which can be further simplified to give:
\[
S^2 \sigma^2 \frac{\partial^2 g}{\partial S^2} + 2(r - q)S \frac{\partial g}{\partial S} - \frac{2rg}{h \sigma^2} - \frac{2r(1 - h)}{\sigma^2} \frac{\partial g}{\partial h} = rgh
\]  
(10.41)

or
\[
S^2 \frac{\partial^2 g}{\partial S^2} + \beta S \frac{\partial g}{\partial S} - \frac{\alpha}{h} g - (1 - h) \frac{\partial g}{\partial h} = 0
\]  
(10.42)

where \(\alpha = 2r/\sigma^2\) and \(\beta = (2(r - q))/\sigma^2\).

We now consider the last term of Equation 10.42 and note that when \(\tau\) is large, \(1 - h(\tau) \sim 0\). Also when \(\tau \to 0\) the option is close to maturity, and the value of both the European and American options converge; which means that \(\nu(S, E, \tau) \sim 0\) and \(\partial g/\partial h \sim 0\). It can thus be seen that the last term is generally quite small and, the MacMillan, Barone-Adesi and Whaley approximation assumes that it can be ignored. This results in the following equation:
\[
S^2 \frac{\partial^2 g}{\partial S^2} + \beta S \frac{\partial g}{\partial S} - \frac{\alpha}{h} g = 0
\]  
(10.43)

which is a second order differential equation with two linearly independent solutions of the form \(aS^\gamma\). They can be found by substituting \(g(S, E, h) = aS^\gamma\) into Equation 10.43 as follows:
\[
\frac{\partial g}{\partial S} = \gamma S^{\gamma - 1} \quad \frac{\partial^2 g}{\partial S^2} = a\gamma(\gamma - 1)S^{\gamma - 2} = a\gamma^2 S^{\gamma - 2} - a\gamma S^{\gamma - 2}
\]

so
\[
S^2 \frac{\partial^2 g}{\partial S^2} = a\gamma^2 S^{\gamma - 1} - a\gamma S^{\gamma} = \gamma^2 g - \gamma g
\]

and
\[
\beta S \frac{\partial g}{\partial S} = \beta Sa\gamma S^{\gamma - 1} = \beta \gamma S^{\gamma} = \beta \gamma g
\]
When the above results are substituted in Equation 10.43 we obtain the quadratic equation:

$$
\gamma^2 g - \gamma g + \beta \gamma g - \alpha/h = g(\gamma^2 - \gamma + (\beta - 1)\gamma - \alpha/h) = 0
$$

or

$$
\gamma^2 - \gamma + (\beta - 1)\gamma - \alpha/h = 0
$$

(10.44)

which has the two solutions

$$
\gamma_1 = \frac{1}{2}\left\{-(\beta - 1) - \sqrt{(\beta - 1)^2 + 4(\alpha/h)}\right\}
$$

(10.45)

and

$$
\gamma_2 = \frac{1}{2}\left\{-(\beta - 1) + \sqrt{(\beta - 1)^2 + 4(\alpha/h)}\right\}
$$

(10.46)

where we note that since $\alpha/h > 0$, we have $\gamma_1 < 0$ and $\gamma_2 > 0$.

The general solution to Equation 10.43 is thus:

$$
g(S, E, h) = a_1 S^{\gamma_1} + a_2 S^{\gamma_2}
$$

(10.47)

We will now derive the appropriate solutions pertaining to American call options and American put options.

**American call options**

Here we use the fact that both the value and the early exercise premium $(\nu_\tau(S, E, \tau) = h g(S, E, h))$ of an American call tend to zero as the asset price $S \to 0$. This means that as $S \to 0$, $g(S, E, h) \to 0$.

However, since $\gamma_1 < 0$, the only way this can be achieved in Equation 10.47 is if $a_1 = 0$. So $g(S, E, h) = a_2 S^{\gamma_2}$, and the value of an American call is:

$$
C(S, E, \tau) = c(S, E, \tau) + ha_2 S^{\gamma_2}
$$

(10.48)

An expression for $a_2$ can be found by considering the critical asset price (point on the early exercise boundary), $S^*$, above which the American option will be exercised. For $S < S^*$, the value of the American call is governed by Equation 10.48, and when $S > S^*$ we have $C(S, E, \tau) = S - E$.

Now, since the value of the American option is continuous, at the critical asset value $S^*$ the following equation applies:

$$
S^* - E = c(S^*, E, \tau) + ha_2 S^{\gamma_2}
$$

(10.49)

Furthermore, since the gradient of the American option value is also continuous, at $S^*$ we have:

$$
\frac{\partial(S^* - E)}{\partial S} = \frac{\partial}{\partial S^*}\left\{c(S^*, E, \tau) + ha_2 S^{\gamma_2}\right\}
$$

(10.50)
which gives:

\[ 1 = \exp(-q\tau)N_1(d_1(S^*)) + \gamma_2 h_2 S^*^{q(2-1)} \]  

(10.51)

where we have used the value of the hedge parameter \( \Delta_c \), see Section 9.3.3, for a European call

\[ \Delta_c = \frac{\partial c(S^*, E, \tau)}{\partial S^*} = \exp(-q\tau)N_1(d_1(S^*)) \]

Equation 10.51 can therefore be written as:

\[ h_2 S^*^{q2} = \frac{S^*}{\gamma_2} \{ 1 - \exp(-q\tau)N_1(d_1(S^*)) \} \]  

(10.52)

When the left hand side of the above equation is substituted into Equation 10.49 we obtain the following equation for \( S^* \):

\[ S^* - E = c(S^*, E, \tau) + \frac{S^*}{\gamma_2} \{ 1 - \exp(-q\tau)N_1(d_1(S^*)) \} \]  

(10.53)

This equation can be solved for \( S^* \) using standard iterative methods (see the section on the numerical solution of critical asset values). Once \( S^* \) has been found Equation 10.52 gives:

\[ h_2 = A_2 S^*^{q2} \]  

where \( A_2 = \frac{S^*}{\gamma_2} \{ 1 - \exp(-q\tau)N_1(d_1(S^*)) \} \)

From Equation 10.48 the value of an American call is thus of the form:

\[
\begin{align*}
\text{MacMillan, Barone-Adesi, and Whaley method: American call option} \\
C(S, E, \tau) &= c(S, E, \tau) + A_2 \left( \frac{S^*}{S^*} \right)^{q2} \quad \text{when } S < S^* \\
C(S, E, \tau) &= S - E \quad \text{when } S \geq S^*
\end{align*}
\]  

(10.54)

(10.55)

**American put options**

For an American put option we proceed in a similar manner to that for the American call. We now use fact that both the value and early exercise premium, \( \nu_p(S, E, \tau) = h g_p(S, E, h) \), of an American put tend to zero as the asset price \( S \to \infty \). So \( g_p(S, E, h) \to 0 \) as \( S \to \infty \). Since \( \gamma_2 > 0 \) the only way this can be achieved by Equation 10.47 is if \( a_2 = 0 \). This gives \( g_p(S, E, h) = a_1 S^{q1} \) and the value of an American put is:

\[ P(S, E, \tau) = p(S, E, \tau) + h a_1 S^{q1} \]  

(10.56)

An expression for \( a_1 \) can be found by considering the critical asset price, \( S^{**} \), below which the American option will be exercised. For \( S > S^{**} \) the value of the American put is given by Equation 10.56, and for \( S < S^{**} \) we have \( P(S, E, \tau) = E - S \).

Continuity of the American option value at the critical asset price gives:

\[ E - S^{**} = p(S^{**}, E, \tau) + h a_1 S^{**q1} \]  

(10.57)
and continuity of the option value’s gradient at the critical asset price yields:
\[
\frac{\partial (E - S^* \gamma_1)}{\partial S^*} = \frac{\partial}{\partial S^*} \{ p(S^*, E, \tau) + ha_1 S^{*-\gamma_1} \} \tag{10.58}
\]
which can be simplified to:
\[
-1 = -N_1(-d_1(S^*)) \exp(-q\tau) + \gamma_1 a_1 S^{*-\gamma_1} \tag{10.59}
\]
where we have used the value of hedge parameter \( \Delta_p \) for a European put (see section on the Greeks):
\[
\Delta_p = \frac{\partial p(S^*, E, \tau)}{\partial S^*} = \{ N_1(d_1(S^*)) - 1 \} \exp(-q\tau) = -N_1(-d_1(S^*)) \exp(-q\tau)
\]
Equation 10.59 can therefore be written as:
\[
ha_1 S^{*-\gamma_1} = -S^* \left\{ 1 - N_1(-d_1(S^*)) \exp(-q\tau) \right\} \tag{10.60}
\]
When the left hand side of the above equation is substituted into Equation 10.57 we obtain the following equation for \( S^* \):
\[
E - S^* = p(S^*, E, \tau) + \{ 1 - \exp(-q\tau)N[(-d_1(S^*))] \} \frac{S^*}{\gamma_1} \tag{10.61}
\]
which can be solved iteratively to yield \( S^* \) (see the section on the numerical solution of critical asset values). Once \( S^* \) has been found Equation 10.60 gives:
\[
ha_1 = A_1 S^{*-\gamma_1} \text{ where } A_1 = -\left( \frac{S^*}{\gamma_1} \right) \{ 1 - \exp(-q\tau)N_1(-d_1(S^*)) \}
\]
We note here that \( A_1 > 0 \) since, \( \gamma_1 < 0, S^* > 0 \) and \( N_1(-d_1(S^*)) \exp(-q\tau) < 1 \). From Equation 10.56 the value of an American put is thus:

**MacMillan, Barone-Adesi, and Whaley method: American put option**

\[
P(S, E, \tau) = p(S, E, \tau) + A_1 \left( \frac{S}{S^*} \right)^{\gamma_2} \text{ when } S > S^* \tag{10.62}
\]
\[
P(S, E, \tau) = E - S \text{ when } S \leq S^* \tag{10.63}
\]

**Numerical solution of critical asset values**

We now provide details on how to iteratively solve for the critical asset price in Equations 10.53 and 10.61.

**American call options**

For American call options we need to solve Equation 10.53, which is:
\[
S^* - E = c(S^*, E, \tau) + \left( \frac{S^*}{\gamma_2} \right) \{ 1 - \exp(-q\tau)N_1(d_1(S^*)) \}
\]
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We denote the $i$th approximation to the critical asset value $S^*$ by $S^*_i$, and represent the left hand side of the equation by:

$$LHS(S^*_i, E, \tau) = S^*_i - E$$

and the right hand side of the equation by:

$$RHS(S^*_i, E, \tau) = c(S^*_i, E, \tau) + \frac{S^*_i}{\gamma_2} \left\{ 1 - \exp(-q\tau)N_1(d_1(S^*_i)) \right\}$$

If we let $K(S^*_i, E, \tau) = RHS(S^*_i, E, \tau) - LHS(S^*_i, E, \tau)$ then we want to find the value of $S^*_i$ which (to a specified tolerance) gives $K(S^*_i, E, \tau) \approx 0$. This can be achieved with Newton’s root finding method, in which a better approximation, $S^*_{i+1}$, can be found using:

$$S^*_{i+1} = S^*_i - \frac{K(S^*_i, E, \tau)}{K'(S^*_i, E, \tau)}$$

(10.64)

where

$$K'(S^*_i, E, \tau) = \frac{\partial}{\partial S^*_i} \left\{ RHS(S^*_i, E, \tau) - LHS(S^*_i, E, \tau) \right\}$$

$$= \frac{\partial}{\partial S^*_i} \left\{ RHS(S^*_i, E, \tau) \right\} - \frac{\partial}{\partial S^*_i} \left\{ LHS(S^*_i, E, \tau) \right\}$$

$$= b_i - 1$$

Here we have used $b_i = (\partial/\partial S^*_i)\left\{ RHS(S^*_i, E, \tau) \right\}$, and the expression for $b_i$ is given by Equation 10.66, which is derived at the end of this section.

Substituting for $K(S^*_i, E, \tau)$ and $K'(S^*_i, E, \tau)$ into Equation 10.64 we therefore obtain:

$$S^*_{i+1} = S^*_i - \frac{(RHS(S^*_i, E, \tau) - LHS(S^*_i, E, \tau))}{(b_i - 1)}$$

$$= S^*_i - \frac{(RHS(S^*_i, E, \tau) - (S^*_i - E))}{(b_i - 1)}$$

$$= \frac{b_i S^*_i - RHS(S^*_i, E, \tau) - E}{(b_i - 1)}$$

The final iterative algorithm for the American call is therefore:

$$S^*_{i+1} = \frac{E + RHS(S^*_i, E, \tau) - b_i S^*_i}{(1 - b_i)}$$

(10.65)

where we can use $S^*_0 = E$ for the initial estimate of the critical value, see computer Code excerpt 10.4.

The expression for $b_i$ in an American call

Here we derive an expression for the term $b_i$ which is used in Equation 10.65.

$$b_i = \frac{\partial c(S^*_i, E, \tau)}{\partial S^*_i} + \frac{1}{\gamma_2} \left\{ 1 - \exp(-q\tau)N_1(d_1(S^*_i)) \right\} - \frac{S^*_i}{\gamma_2} \frac{\partial N_1(d_1(S^*_i))}{\partial d_1(S^*_i)} - \frac{\partial}{\partial S^*_i}$$
We will now quote the following results which are derived in Appendix C:

Equation C.3:

\[
\frac{\partial N_1(d_1(S_i^*)))}{\partial d_1(S_i^*)} = n(d_1(S_i^*))
\]

Equation C.6:

\[
\frac{\partial d_1(S_i^*)}{\partial S_i^*} = \frac{1}{S_i^* \sigma \sqrt{\tau}}
\]

Equation C.14:

\[
\Delta c = \frac{\partial c(S_i^*, E, \tau)}{\partial S_i^*} = \exp(-q\tau)N_1(d_1(S_i^*))
\]

Substituting these results into the above expression we obtain:

\[
b_i = \exp(-q\tau)N_1(d_1(S_i^*)) + \frac{1}{\gamma_2} - \frac{\exp(-q\tau)n(d_1(S_i^*))}{\gamma_2} - \frac{\exp(-q\tau)n(d_1(S_i^*))}{\gamma_2 \sigma \sqrt{\tau}}
\]

which can be rearranged to yield:

\[
b_i = \exp(-q\tau)N_1(d_1(S_i^*)) \left\{ 1 - \frac{1}{\gamma_2} \right\} + \frac{1}{\gamma_2} \left\{ 1 - \frac{\exp(-q\tau)n(d_1(S_i^*))}{\sigma \sqrt{\tau}} \right\}
\]

(10.66)

**American put options**

For American put options we need to solve Equation 10.61 which is:

\[
E - S_i^* = p(S_i^{**, E, \tau}) - \frac{S_i^{**, E, \tau}}{\gamma_1} \left\{ 1 - N_1(-d_1(S_i^{**})) \exp(-q\tau) \right\}
\]

If we let \( S_i^{**, E, \tau} \) denote the \( i \)th approximation to the critical asset value \( S^{**, E, \tau} \), then we can represent the left hand side of the equation by:

\[
LHS(S_i^{**}, E, \tau) = E - S_i^{**}
\]

and the right hand side of the equation by:

\[
RHS(S_i^{**}, E, \tau) = p(S_i^{**}, E, \tau) - \frac{S_i^{**}}{\gamma_1} \left\{ 1 - N_1(-d_1(S_i^{**})) \exp(-q\tau) \right\}
\]

\[
= p(S_i^{**}, E, \tau) - \frac{S_i^{**}}{\gamma_1} \left\{ 1 - [1 - N_1(d_1(S_i^{**}))] \exp(-q\tau) \right\}
\]

\[
= p(S_i^{**}, E, \tau) - \frac{S_i^{**}}{\gamma_1} \left\{ 1 - \exp(-q\tau) + N_1(d_1(S_i^{**})) \exp(-q\tau) \right\}
\]
We then denote \( K(S_i^{**}, E, \tau) = RHS(S_i^{**}, E, \tau) - LHS(S_i^{**}, E, \tau) \), and using Newton’s method we obtain:

\[
S_{i+1}^{**} = S_i^{**} - \frac{K(S_i^{**}, E, \tau)}{K'(S_i^{**}, E, \tau)}
\]  

(10.67)

where as before:

\[
K'(S_i^{**}, E, \tau) = \frac{\partial}{\partial S_i^{**}} \left\{ RHS(S_i^{**}, E, \tau) - LHS(S_i^{**}, E, \tau) \right\}
\]

So \( K'(S_i^{**}, E, \tau) = 1 + b_i \), where \( b_i = (\partial(RHS(S_i^{**}, E, \tau))/\partial S_i^{**}) \), and the expression for \( b_i \) is given by Equation 10.69, which is derived at the end of this section.

Equation 10.67 can therefore be written as:

\[
S_{i+1}^{**} = S_i^{**} - \frac{(RHS(S_i^{**}, E, \tau) - LHS(S_i^{**}, E, \tau))}{1 + b_i} = S_i^{**} (1 + b_i) - RHS(S_i^{**}, E, \tau) + E - S_i^{**}
\]

The final iterative algorithm for the American put is therefore:

\[
S_i^{**} = \frac{E - RHS(S_i^{**}, E, \tau) + b_i S_i^{**}}{1 + b_i}
\]  

(10.68)

where we can use \( S_0^{**} = E \) for the initial estimate of the critical asset value, see computer Code excerpt 10.4.

**The expression for \( b_i \) in an American put**

Here we derive an expression for the term \( b_i \) which is used in Equation 10.67. Since

\[
b_i = \frac{\partial}{\partial S_i^{**}} \left\{ p(S_i^{**}, E, \tau) - \frac{S_i^{**}}{\gamma_1} (1 - \exp(-q\tau) + N_1(d_1(S_i^{**})) \exp(-q\tau)) \right\}
\]

we have

\[
b_i = \frac{\partial p(S_i^{**}, E, \tau)}{\partial S_i^{**}} - \frac{1}{\gamma_1} \{ 1 - \exp(-q\tau) \} - \frac{1}{\gamma_1} \exp(-q\tau) N_1(d_1(S_i^{**})) - \frac{S_i^{**} \exp(-q\tau) \partial N_1(d_1(S_i^{**})) \partial d_1(S_i^{**})}{\gamma_1} \frac{\partial d_1(S_i^{**})}{\partial S_i^{**}}
\]

We will now quote the following results which are derived in Appendix C: Equation C.3:

\[
\frac{\partial N_1(d_1(S_i^{**}))}{\partial d_1(S_i^{**})} = n(d_1(S_i^{**}))
\]
Equation C.6:
\[
\frac{\partial d_1(S_i^{**})}{\partial S_i^{**}} = \frac{1}{S_i^{**} \sigma \sqrt{\tau}}
\]

Equation C.16:
\[
\Delta_p = \frac{\partial p(S_i^{**}, E, \tau)}{\partial S_i^{**}} = \exp(-q\tau)\left\{N_1(d_1(S_i^{**})) - 1\right\}
\]

Substituting these results into the above expression we therefore obtain:
\[
b_i = \exp(-q\tau)\left\{N_1(d_1(S_i^{**})) - 1\right\} - \frac{1}{\gamma_1} \left\{1 - \exp(-q\tau) + N_1(d_1(S_i^{**}))\exp(-q\tau)\right\}
\]
\[
- \frac{S_i^{**} \exp(-q\tau) \partial N_1(d_1(S_i^{**})) \partial d_1(S_i^{**})}{\gamma_1
\]
\[
= \exp(-q\tau)\left\{N_1(d_1(S_i^{**})) - 1\right\} - \frac{1}{\gamma_1} \left\{1 - \exp(-q\tau) + N_1(d_1(S_i^{**}))\exp(-q\tau)\right\}
\]
\[
- \frac{S_i^{**} \exp(-q\tau)n(d_1(S_i^{**}))}{\gamma_1 \sigma \sqrt{\tau}}
\]

which can be rearranged to yield:
\[
b_i = \exp(-q\tau)N_1(d_1(S_i^{**}))\left\{1 - \frac{1}{\gamma_1}\right\}
\]
\[
+ \frac{1}{\gamma_1} \left\{\exp(-q\tau) - 1 - \frac{\exp(-q\tau)n(d_1(S_i^{**}))}{\sigma \sqrt{\tau}}\right\} - \exp(-q\tau)
\]

(10.69)

The computer code to implement the MacMillan, Barone-Adesi, and Whaley method is provided below.

```c
void MBW_approx(double *opt_value, double *critical_value, double S0, double X, double sigma, double T, double r, double q, Integer put, Integer *iflag)
{
    /* Input parameters:
    S0      — the current value of the underlying asset
    X       — the strike price
    sigma   — the volatility
    T       — the time to maturity
    r       — the interest rate
    q       — the continuous dividend yield
    put     — if put is 0 then a call option, otherwise a put option
    * Output parameters:
    opt_value — the value of the option
    critical_value — the critical value
    iflag — an error indicator*/
    double A_1, A_2, S_star, gamma_2, gamma_1;
    double d1, alpha, h, beta, temp, temp1;
    double pdf, pi, b, rhs, h, err;
    Boolean iterate;
    Integer iflagx, putx;
```
pi = X01AAC;
beta = 2.0 * (r - q) / (sigma * sigma);
alpha = 2.0 * r / (sigma * sigma);
h = 1.0 - exp(-r*T);
temp = beta - 1.0;
iterate = TRUE;
tol = 0.000001;
if (!put){/* An American call */
gamma_2 = (temp + sqrt((temp*temp) + (4.0*alpha/h)));
gamma_2 = gamma_2 / 2.0;
S_star = X;
while (iterate){/* calculate S_star, iteratively */
d1 = log(S_star/X) + (r-q)/(sigma/sqrt(T)));
pdf = (1.0/sqrt(2.0*pi))*exp(-d1*d1/2.0);
temp = exp(-q*T) * s15abc(d1) * (1.0 - 1.0/gamma_2));
temp1 = (1.0 - 1.0/(exp(-q*T)*pdf)/(sigma/sqrt(T)))/gamma_2;
b = temp + temp1;
/* calculate the Black–Scholes value of a European call */
putx = 0;
black_scholes(&eur_val,NULL,S_star,X,sigma,T,r,q,putx,&iflagx);
rhs = eur_val + (1.0 - exp(-q*T) * s15abc(d1)) * S_star/gamma_2;
S_star = (X + rhs - b*S_star)/(1.0 - b);
err = FALSE(S_star - X) - rhs)/X;
if (err < tol) iterate = FALSE;
}
A_2 = (S_star/gamma_2) * (1.0 - exp(-q*T) * s15abc(d1));
if (SO < S_star) {
temp1 = SO/S_star;
black_scholes(&temp,NULL,S0,X,sigma,T,r,q,putx,&iflagx);
*opt_value = temp + A_2 * pow(temp1,gamma_2);
} else {
*opt_value = SO - X;
}
}
else {/* An American put */
gamma_1 = (temp + sqrt((temp*temp) + (4.0*alpha/h)));
gamma_1 = gamma_1 / 2.0;
S_star = X;
while (iterate){/* calculate S_star, iteratively */
d1 = log(S_star/X) + (r-q)/(sigma/sqrt(T)));
pdf = (1.0/sqrt(2.0*pi))*exp(-d1*d1/2.0);
temp = exp(-q*T) * s15abc(d1) * (1.0 - 1.0/gamma_2));
temp1 = (1.0 - 1.0/(exp(-q*T)*pdf)/(sigma/sqrt(T)))/gamma_2;
b = temp + temp1;
/* calculate the Black–Scholes value of a European put */
putx = 1;
black_scholes(&eur_val,NULL,S_star,X,sigma,T,r,q,putx,&iflagx);
rhs = eur_val + (1.0 - exp(-q*T) * exp(-q*T) * s15abc(d1)) * S_star/gamma_1;
S_star = (X + rhs + b*S_star)/(1.0 + b);
err = FALSE(X - S_star) - rhs/X;
if (err < tol) iterate = FALSE;
}
A_1 = (S_star/gamma_1) * (1.0 - exp(-q*T) * s15abc(-d1));
if (SO > S_star) {
temp1 = SO/S_star;
black_scholes(&temp,NULL,S0,X,sigma,T,r,q,putx,&iflagx);
*opt_value = temp + A_1 * pow(temp1,gamma_2);
} else {
*opt_value = X - SO;
}
}
critical_value = S_star;

Code excerpt 10.4 The function MBW_approx which computes the MacMillan, Barone-Adesi, and Whaley approximation for American options

Tables 10.2 and 10.3 present the results of using the function MBW_approx to compute the values of various American options.
10.4 LATTICE METHODS FOR VANILLA OPTIONS

10.4.1 Binomial lattice

In this section we will derive equations for a binomial lattice that describes the GBM movement of asset price changes. The approach that we will adopt is based on the work of Cox, Ross, and Rubinstein (1979), and will be referred to as the CRR lattice.

From Section 8.3 Equation 8.15 we know that if the price of an asset, $S_t$, follows GBM then the change in value of its price over time interval $\Delta t$, has the following distribution:

$$\log \left( \frac{S_{t+\Delta t}}{S_t} \right) \sim N \left( (r - \frac{\sigma^2}{2}) \Delta t, \sigma^2 \Delta t \right)$$

---

Table 10.2 The MacMillan, Barone-Adesi, and Whaley method for American option values computed by the routine MBW_approx. The parameters used were: $\tau = 0.5$, $E = 100.0$, $r = 0.1$, $q = 0.06$, $\sigma = 0.2$. The accurate value was calculated using a standard lattice with 2000 time steps, and the error was the MacMillan, Barone-Adesi, and Whaley estimate minus the accurate value

<table>
<thead>
<tr>
<th>Stock price</th>
<th>Call Accurate value</th>
<th>Call Error</th>
<th>Put Accurate value</th>
<th>Put Error</th>
</tr>
</thead>
<tbody>
<tr>
<td>86.0</td>
<td>1.2064</td>
<td>5.54 $\times 10^{-4}$</td>
<td>14.0987</td>
<td>$-3.69 \times 10^{-2}$</td>
</tr>
<tr>
<td>89.0</td>
<td>1.8838</td>
<td>1.95 $\times 10^{-4}$</td>
<td>11.5120</td>
<td>$-4.85 \times 10^{-2}$</td>
</tr>
<tr>
<td>92.0</td>
<td>2.7890</td>
<td>7.03 $\times 10^{-4}$</td>
<td>9.2478</td>
<td>$-3.58 \times 10^{-2}$</td>
</tr>
<tr>
<td>95.0</td>
<td>3.9427</td>
<td>1.16 $\times 10^{-3}$</td>
<td>7.3031</td>
<td>$-1.66 \times 10^{-2}$</td>
</tr>
<tr>
<td>98.0</td>
<td>5.3522</td>
<td>1.15 $\times 10^{-3}$</td>
<td>5.6674</td>
<td>$7.19 \times 10^{-4}$</td>
</tr>
<tr>
<td>101.0</td>
<td>7.0119</td>
<td>1.10 $\times 10^{-3}$</td>
<td>4.3209</td>
<td>$1.35 \times 10^{-2}$</td>
</tr>
<tr>
<td>104.0</td>
<td>8.9043</td>
<td>2.21 $\times 10^{-3}$</td>
<td>3.2362</td>
<td>$2.22 \times 10^{-2}$</td>
</tr>
<tr>
<td>107.0</td>
<td>11.0072</td>
<td>2.63 $\times 10^{-3}$</td>
<td>2.3823</td>
<td>$2.63 \times 10^{-2}$</td>
</tr>
<tr>
<td>110.0</td>
<td>13.2905</td>
<td>4.20 $\times 10^{-3}$</td>
<td>1.7235</td>
<td>$2.80 \times 10^{-2}$</td>
</tr>
<tr>
<td>113.0</td>
<td>15.7264</td>
<td>4.77 $\times 10^{-3}$</td>
<td>1.2272</td>
<td>$2.66 \times 10^{-2}$</td>
</tr>
</tbody>
</table>

Table 10.3 The MacMillan, Barone-Adesi, and Whaley critical asset values for the early exercise boundary of an American put computed by the routine MBW_approx. The parameters used were: $S = 101.0$, $E = 101.0$, $r = 0.1$, $q = 0.06$, and $\sigma = 0.2$

<table>
<thead>
<tr>
<th>Time to expiry, $\tau$</th>
<th>Critical asset value, $S^*$</th>
<th>Time to expiry, $\tau$</th>
<th>Critical asset value, $S^*$</th>
</tr>
</thead>
<tbody>
<tr>
<td>1.0</td>
<td>82.1510</td>
<td>0.50</td>
<td>85.1701</td>
</tr>
<tr>
<td>0.95</td>
<td>82.3751</td>
<td>0.45</td>
<td>85.6199</td>
</tr>
<tr>
<td>0.90</td>
<td>82.6115</td>
<td>0.40</td>
<td>86.1176</td>
</tr>
<tr>
<td>0.85</td>
<td>82.8618</td>
<td>0.35</td>
<td>86.6740</td>
</tr>
<tr>
<td>0.80</td>
<td>83.1273</td>
<td>0.30</td>
<td>87.3049</td>
</tr>
<tr>
<td>0.75</td>
<td>83.4098</td>
<td>0.25</td>
<td>88.0333</td>
</tr>
<tr>
<td>0.70</td>
<td>83.7115</td>
<td>0.20</td>
<td>88.8959</td>
</tr>
<tr>
<td>0.65</td>
<td>84.0349</td>
<td>0.15</td>
<td>89.9568</td>
</tr>
<tr>
<td>0.60</td>
<td>84.3830</td>
<td>0.10</td>
<td>91.3469</td>
</tr>
<tr>
<td>0.55</td>
<td>84.7598</td>
<td>0.05</td>
<td>93.4260</td>
</tr>
</tbody>
</table>
If we use the notation:

\[ X = \frac{S_{t+\Delta t}}{S_t}, \quad \text{and} \quad \eta = (r - \sigma^2/2)\Delta t \quad \nu^2 = \sigma^2\Delta t \]

the above equation becomes:

\[ \log(X) \sim N(\eta, \nu^2) \quad \text{or equivalently} \quad X \sim \Lambda(\eta, \nu^2) \]

where \( \Lambda(\eta, \nu^2) \) is the lognormal distribution derived from a Gaussian distribution with mean \( \eta \), and variance \( \nu^2 \). It is well known, see for example Evans et al. (2000), that the first two moments of a variable \( X \) drawn from a lognormal distribution are:

**Lognormal mean**

\[ E[X] = \exp(\eta + \nu^2/2) \quad (10.70) \]

substituting for \( \eta \) and \( \nu^2 \) gives

\[ E[X] = \exp\left\{ \left( r - \frac{\sigma^2}{2} \right)\Delta t + \frac{\sigma^2}{2} \Delta t \right\} \quad (10.71) \]

**Lognormal variance**

\[ Var[X] = E[(X - E[X])^2] = E[X^2] - (E[X])^2 = \exp(2\eta + \nu^2)\{\exp(\nu^2) - 1\} \quad (10.72) \]

substituting for \( \eta \) and \( \nu^2 \) gives

\[ Var[X] = \exp\left\{ 2r\left( r - \frac{\sigma^2}{2} \right)\Delta t + \sigma^2\Delta t \right\} \]

which can be simplified to yield

\[ Var[X] = \exp\{2r\Delta t\}\{\exp(\sigma^2\Delta t) - 1\} \quad (10.73) \]

Since we can assume that the expected value of \( X \) grows at the riskless interest rate, \( r \), we can also write:

\[ E[X] = \exp(r\Delta t) \quad (10.74) \]

The above results can be used to find the first two moments of the asset price distribution \( S_{t+\Delta t} \), given that we know the asset price, \( S_t \), at time instant \( t \). To do this we will use, see Appendix F.3 for a proof, the fact that for a random variable \( G \) we have:

\[ E[a + bG] = E[a] + bE[G] \quad \text{and} \quad Var[a + bG] = b^2Var[G] \]

where \( a \) and \( b \) are constants. Applying this to the variable \( X \) gives:

\[ E[X] = E\left[ \frac{S_{t+\Delta t}}{S_t} \right] = \frac{1}{S_t} E[S_{t+\Delta t}] \quad (10.75) \]
and

\[ \text{Var}[X] = \text{Var}\left[ \frac{S_{t+\Delta t}}{S_t} \right] = \frac{1}{S_t^2} \text{Var}[S_{t+\Delta t}] \]  \hspace{1cm} (10.76) \]

where we have used \( a = 0 \) and \( b = 1/S_t \). Note that it is also easy to show that:

\[ \text{Var}[S_{t+\Delta t}] = \text{Var}[\Delta S] \]  \hspace{1cm} (10.77) \]

where the change in asset price over the time interval \( \Delta t \) is denoted by \( \Delta S = S_{t+\Delta t} - S_t \). This elementary result sometimes is used without proof, see for example Hull (1997) p. 344. The proof is simple:

\[ \text{Var}[S_{t+\Delta t}] = \text{Var}[S_t + \Delta S] = \text{Var}[\Delta S] \]

where again we have used \( \text{Var}[a + bG] = b^2 \text{Var}[G] \) this time with \( a = 0 \) and \( b = 1 \).

To find expressions for the mean and variance of \( S_{t+\Delta t} \) we simply substitute Equation 10.74 into Equation 10.75 and obtain:

\[ \mathbb{E}[S_{t+\Delta t}] = S_t \exp(r \Delta t) \]  \hspace{1cm} (10.78) \]

and substitute Equation 10.73 into Equation 10.76 to yield:

\[ \text{Var}[S_{t+\Delta t}] = S_t^2 \exp(2r \Delta t) \{ \exp(\sigma^2 \Delta t) - 1 \} \]  \hspace{1cm} (10.79) \]

Since we are modelling asset price movements with a binomial lattice, the asset price, \( S_t \), at any given node is only permitted to either jump up or jump down in value over the next time step \( \Delta t \). Here we will assume that the new asset price, \( S_{t+\Delta t} \), is \( S_t u \) for an up jump and \( S_t d \) for a down jump; where \( u \) and \( d \) are constants that apply to all lattice nodes. If we further denote the probability of an up jump by \( p \) then the probability of a down jump must (by definition) be \( 1 - p \).

Now that we have specified the lattice parameters we will use these to match the first two moments of the lognormal distribution. This results in the following equation for the mean:

\[ \mathbb{E}[S_{t+\Delta t}] = pS_t u + (1 - p)S_t d = S_t \exp(r \Delta t) \]  \hspace{1cm} (10.80) \]

The corresponding equation for the variance requires a little more work:

\[ \text{Var}[S_{t+\Delta t}] = \mathbb{E}[(S_{t+\Delta t})^2] - \mathbb{E}^2[S_{t+\Delta t}] \]  \hspace{1cm} (10.81) \]

Since

\[ \mathbb{E}[(S_{t+\Delta t})^2] = p(S_t u)^2 + (1 - p)(S_t d)^2 = S_t^2 (pu^2 + (1 - p)d^2) \]  \hspace{1cm} (10.82) \]

and, from Equation 10.80, we have

\[ (\mathbb{E}[S_{t+\Delta t}])^2 = \{S_t \exp(r \Delta t)\}^2 = S_t^2 \exp(2r \Delta t) \]  \hspace{1cm} (10.83) \]
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We can substitute Equations 10.82 and 10.83 into Equation 10.81 to obtain

\[ \text{Var}[S_t + \Delta t] = S_t^2 \exp(2r\Delta t) \{ \exp(\sigma^2 \Delta t) - 1 \} \]

\[ = S_t^2 (pu^2 + (1 - p)d^2) - S_t^2 \exp(2r\Delta t) \]

We therefore have:

\[ \exp(2r\Delta t)(\exp(\sigma^2 \Delta t) - 1) = pu^2 + (1 - p)d^2 - \exp(2r\Delta t) \] (10.84)

So, restating Equation 10.80 and simplifying Equation 10.84, we obtain the following two equations:

\[ pu + (1 - p)d = \exp(r\Delta t) \] (10.85)

\[ \exp(2r\Delta t + \sigma^2 \Delta t) = pu^2 + (1 - p)d^2 \] (10.86)

which we will use to solve for the three parameters \( u \), \( d \), and \( p \). Since there are three unknowns and only two equations, we can impose an additional constraint to obtain a unique solution. The constraint used in the CRR binomial model is:

\[ u = \frac{1}{d} \] (10.87)

We now use the following notation:

\[ a = \exp(r\Delta t) \quad \text{and} \quad b^2 = \exp(2r\Delta t) \{ \exp(\sigma^2 \Delta t) - 1 \} = a^2 \{ \exp(\sigma^2 \Delta t) - 1 \} \] (10.88)

This means that Equation 10.85 can be written as

\[ a = pu + (1 - p)d, \quad \text{which gives} \quad p = \frac{a - d}{u - d} \] (10.89)

From Equation 10.86 we have

\[ \exp(2r\Delta t + \sigma^2 \Delta t) = a^2 \exp(\sigma^2 \Delta t) = a^2 + b^2 \]

and so

\[ a^2 + b^2 = pu^2 + (1 - p)d^2 \]

Rearranging we have

\[ pu^2 + (1 - p)d^2 - a^2 = b^2 \]

\[ pu^3 + (1 - p)d^2 u - a^2 u - b^2 u = 0 \]

but

\[ (1 - p)d^2 u = (1 - p)d = a - pu \]

so

\[ pu^3 = (a - pu) - a^2 u - b^2 u = 0 \]

or

\[ p(u^3 - u) + a - a^2 - b^2 u = 0 \]

Now

\[ p(u^3 - u) = u^2 p(u - d) = u^2 (a - d) = u^2 a - u \]
which gives
\[ au^2 - u + a - a^2 u - b^2 u = 0 \]

So we obtain the following quadratic equation in \( u \):
\[ au^2 - u(1 + a^2 + b^2) + a = 0 \]

A solution is:
\[ u = \frac{(1 + a^2 + b^2) + \sqrt{(1 + a^2 + b^2)^2 - 4a^2}}{2a} \]

If \( \Delta t \) is small we can obtain a reasonable approximation to the solution by neglecting terms of order higher than \( \Delta t \).

In these circumstances we have:
\[ a^2 + b^2 + 1 = \exp(2r\Delta t) + \exp(2r\Delta t)\{\exp(\sigma^2\Delta t) - 1\} + 1 \]
\[ \sim 1 + 2r\Delta t + (1 + 2r\Delta t)\sigma^2\Delta t + 1 \sim 2 + 2r\Delta t + \sigma^2\Delta t \]

Therefore
\[ \sqrt{(a^2 + b^2 + 1)^2 - 4a^2} \sim \sqrt{(2 + 2r\Delta t + \sigma^2\Delta t)^2 - 4(1 + 2r\Delta t)} \]
\[ \sim \sqrt{4 + 8r\Delta t + 4\sigma^2 - 4 - 8r\Delta t} = \sqrt{4\sigma^2\Delta t} = 2\sigma\sqrt{\Delta t} \]

and so
\[ u \sim \frac{2 + 2r\Delta t + \sigma^2\Delta t + 2\sigma\sqrt{\Delta t}}{2 \exp(r\Delta t)} \]
\[ u \sim \left(1 + r\Delta t + \frac{\sigma^2\Delta t}{2} + \sigma\sqrt{\Delta t}\right)(1 - r\Delta t) \]
\[ u \sim 1 + r\Delta t + \frac{\sigma^2\Delta t}{2} + \sigma\sqrt{\Delta t} - r\Delta t = 1 + \sigma\sqrt{\Delta t} + \frac{\sigma^2\Delta t}{2} \]
which to order \( \Delta t \) gives: \[ u = \exp(\sigma\sqrt{\Delta t}) \] (10.90)
since
\[ \exp(\sigma\sqrt{\Delta t}) = 1 + \sigma\sqrt{\Delta t} + \frac{\sigma^2\Delta t}{2} + \frac{\sigma^3(\Delta t)^{3/2}}{6} + \cdots \] (10.91)

which gives:
\[ d = \frac{1}{u} = \exp(-\sigma\sqrt{\Delta t}) \] (10.92)

It is interesting to note that when \( r = 0 \) we have \( p \to 1/2 \).

Now that we know the values of the lattice parameters \( u, d, \) and \( p \) we can use these to build a lattice with a specified number of time steps. Once this has been constructed it can be used to compute the values and Greeks for various types of financial options. These could simply be American/European vanilla options, or more exotic options that may incorporate features such as: lockout periods, barriers, and nonstandard payoff functions.
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We will now discuss how to create a lattice which can be used to value American and European vanilla options.

If the current value of the underlying asset is $S$, and the duration of the option is $\tau$ and we use a lattice with $n$ equally spaced time intervals $\Delta t$, then we have:

$$\Delta t = \frac{\tau}{n}$$

The values of the asset price at various nodes in the lattice can easily be computed. This is illustrated in Figure 10.1, for a lattice with six time steps (that is seven lattice levels).

The asset values at the labelled nodes are:

Lattice level 1: Time $t$

$$S_R = S$$

Lattice level 2: Time $t + \Delta t$

$$S_S = Su \quad S_T = Sd$$

Lattice level 6: Time $t + 5\Delta t$

$$S_A = Su^5 \quad S_B = Su^3 \quad S_C = Su \quad S_D = Sd \quad S_E = S \quad S_F = Sd^5$$

Lattice level 7: Time $t + 6\Delta t$

$$S_G = Su^6 \quad S_H = Su^4 \quad S_I = Su^2 \quad S_J = S \quad S_K = Sd^2 \quad S_L = Sd^4 \quad S_M = Sd^6$$

![Figure 10.1](image-url)
In general, at time $t + i \Delta t$, there are $i + 1$, stock prices; these are:

$$S_{t,i} = Su^id^{i-j}, \quad j = 0, 1, \ldots, i$$

We note, that since $u = 1/d$, an up movement followed by a down movement gives the same stock price as a down movement followed by an up movement; for instance $Su^2d = Su$. This means that the tree recombines, and the number of nodes required to represent all the different asset prices is significantly reduced.

### 10.4.2 Constructing and using the binomial lattice

In this section we are concerned with the practical details of how to construct, and then use, a *standard* one-dimensional binomial lattice to value American and European options. Since this lattice forms the basis for other one-dimensional and multi-dimensional lattice techniques we will discuss its construction in some detail.

A complete computer program for a standard binomial lattice is given in Code excerpt 10.11, and we will use this as a basis for our discussions. For easy reference we will now list the input parameters used by this computer program:

- $S_0$: the current price of the underlying asset, $S$
- $X$: the strike price
- $\sigma$: the volatility of the asset
- $r$: the risk free interest rate
- $q$: the continuous dividend yield
- put: if put equals 1 then the option is a put option, if put equals 0 then it is a call option
- is_american: if is_american equals 1 then it is an American option, if is_american equals 0 then it is a European option
- $M$: the number of time steps in the lattice

We will now discuss in more detail the computational issues involved in each stage of the calculation.

**Compute the values of the constants used by the lattice**

First calculate the values of various constants that will be used.

```c
dt = T/(double)M;
t1 = sigma*sqrt(dt);
u = exp(t1);
d = exp(-t1);
a = exp((r - q)*dt);
p = (a - d)/(u - d);
if ((p < zero) || (p > 1.0)) printf("Error p out of range\n");
discount = exp(-r*dt);
p_u = discount*p;
p_d = discount*(1.0-p);
```

Code excerpt 10.5

For convenience, we have used the variables $p_u$ and $p_d$ to store respectively the up and down jump probabilities discounted by the interest rate $r$ over one time step; these values will be used later on when we work backwards through the lattice to calculate the current option value.
Assign the asset values to the lattice nodes

We will now show that the number of different asset prices, $LS_n$, for an $n$ step recombining lattice is $2n + 1$.

The nodes in a recombining lattice can be considered as being composed of two kinds: those corresponding to an even time step, and those corresponding to an odd time step.

This is because the set of node asset values, $ET$, for an even time step is distinct from the set of node asset values, $OT$, for an odd time step. Although $ET \cap OT = \emptyset$, the elements of $ET$ and $OT$ for any consecutive pair of time steps, are related by the simple constant multiplicative factor $d$. Also for an even time step there is a central node corresponding to the current asset price $SO$, and the remaining nodes are symmetrically arranged about this. These features are illustrated in Figure 10.1, for a standard lattice with six time steps.

The number of distinct asset prices in a lattice is therefore the sum of the number of nodes in the last two time steps. Since the number of nodes in the $ith$ time step, $S_i$, is $i + 1$ (see Figure 10.1), for an $n$ time step lattice we have:

$$S_n = n + 1 \text{ and } S_{n-1} = n$$

This means that the number of different asset values in an $n$ time step lattice is:

$$LS_n = S_n + S_{n-1} = 2n + 1$$

The number of nodes in an $n$ time step lattice, $LN_n$, is:

$$LN_n = \sum_{i=0}^{n} (i + 1) = \frac{(n + 1)(n + 2)}{2}$$

where we have used the fact that $LN_n$ is the sum of an arithmetic progression with first term 1, increment 1 and last term $n + 1$.

One might initially think that, in order to price options, it is necessary to store the asset value of each lattice node; which would entail storing $LN_n$ values. However, this is not the case. We only need to store the number of different asset values in the lattice; that is $LS_n$ values.

Storing $LS_n$ values instead of $LN_n$ can result in dramatic economies of storage. For example an accurate, 1000 step lattice, has $LN_n = 2001 \times 2002 \times \frac{1}{2} = 2003001$, while the corresponding value of $LS_n$ is only $2 \times 1000 + 1 = 2001$.

```java
s[M] = S0;
for (i = 1; i < M; ++i) {
    s[M+i] = u*s[M+i-1];
    s[M-i] = d*s[M-i+1];
}
```

**Code excerpt 10.6** A code fragment which assigns the different binomial lattice asset values to the storage array $s$ by using the up and down jump ratios $u$ and $d$ defined in Section 10.4.1. The current asset value $S$ is assigned to the central array element $s[M]$, where $M$ is the number of time steps in the lattice. The array elements above centre are $S[M+i] = Su^i, i = 1, \ldots, M$, and the array elements below centre are $S[M-i] = Sd^i, i = 1, \ldots, M$. 

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Compute the option payoff at the terminal nodes

The current value of an option is evaluated by starting at option maturity, the end of the tree and working backwards. The option values for the terminal nodes of the tree are just given by the payoff (at maturity) of the option; this is independent of whether the option is an American or European. For a lattice with $n$ time steps there are $n+1$ terminal nodes, with option values, $f_{n,j}, j = 0, \ldots, n$.

To compute the values of vanilla American and European options, with exercise price $E$, then we will start with the following terminal node values:

For put options

$$f_{n,j} = \max(E - S_t d^{m-j}, 0), \quad j = 0, \ldots, n$$

and for call options

$$f_{n,j} = \max(S_t d^{m-j} - E, 0), \quad j = 0, \ldots, n$$

The computer code used to achieve this is:

```c
if (((M+1)/2) == (M/2)) {/* then M is even */
if (put)
  v[M/2] = MAX(X - s[M], zero);
else
  v[M/2] = MAX(s[M] - X, zero);
} P1 = 2*M;
P2 = 0;
for (i = 0; i < (M+1)/2; ++i){
  if (put){
    v[M-i] = MAX(X - s[P1], zero);
    v[i] = MAX(X - s[P2], zero);
  }
  else{
    v[M-i] = MAX(s[P1] - X, zero);
    v[i] = MAX(s[P2] - X, zero);
  }
P1 = P1 - 2;
P2 = P2 + 2;
}
```

**Code excerpt 10.7** A code fragment that computes the payouts for puts and calls at the lattice terminal nodes. The payoffs are assigned to elements of the array $v$ and are computed using the strike price, $X$, and the previously computed asset values stored in array $s$, as before $M$ is the number of time steps in the lattice.

Iterate backwards through the lattice

The probability of moving from node $(i, j)$ at time $i\Delta t$ to node $(i + 1, j + 1)$ at time $(i + 1)\Delta t$ is $p$, and the probability of moving from node $(i, j)$ at time $i\Delta t$ to the node $(i + 1, j)$ at time $(i + 1)\Delta t$ is $1 - p$. If we assume that there is no early exercise then:

$$f_{i,j}^E = \exp(-r\Delta t) \{ pf_{i+1,j+1} + (1 - p)f_{i+1,j} \}, \quad j \leq i \leq n - 1 \quad 0 \leq j \leq i \quad (10.93)$$

When early exercise, for an American option, is taken into account we have:

$$f_{i,j}^A = \max \{ E - S_{i,j}, f_{i,j}^E \} \quad (10.94)$$
or for an American call option:

$$f_{i,j}^A = \max \left\{ S_{i,j} - E, f_{i,j}^E \right\}, \quad j \leq i \leq N - 1 \quad 0 \leq j \leq i$$

(10.95)

where $f_{i,j}^E$ is given by Equation 10.93.

The following code works backward through the lattice and uses the array $v$ to store the option values.

```c
P2 = 0;
for (m1 = M-1; m1 >= 2; --m1) {
    P2 = P2 + 1;
    P1 = P2;
    for (n = 0; n <= m1; ++n){
        if ((v[n] == zero) && (v[n+1] == zero)){
            hold = zero;
        } else
            hold = p_d*v[n] + p_u*v[n+1];
        if (is_american){
            if (put)
                v[n] = MAX(hold, X*s[P1]);
            else
                v[n] = MAX(hold, s[P1]-X);
        } else
            v[n] = hold;
        P1 = P1 + 2;
    }
}
```

**Code excerpt 10.8** Computer code that works iteratively backward through the lattice computing the option values at each time step. The array $v$ contains the option values computed from the previous time step, and these are overwritten with option values computed for the current time step. The iteration stops at second time step, since we do not want to overwrite values in the array $v$ which are required for calculating the Greeks in the neighbourhood of the root node.

At each time step the newly calculated option values overwrite those computed by the previous time step. This process is continued until the second time step ($m1 = 2$) is reached. A different technique is then used, which doesn’t overwrite the option values and thus allows the Greeks to be computed in the vicinity of the root lattice node $R$. In cases where we are not interested in calculating the Greeks (see for example Code excerpt 12.6) we continue working backward through the lattice until the root node $R$ ($m1 = 0$) is reached, and the current value of the option is then given by $v[0]$ (or its multidimensional equivalent).

The option values at all lattice nodes in time steps 0, 1, and 2 are made accessible by the following code:

```c
jj = 2;
for (m1 = 2; m1 >= 1; --m1){
    ind = M-m1+1;
    for (n = 0; n < m1; ++n){
        hold = p_d*v[5-jj-m1-1] + p_u*v[5-jj-m1];
        if (is_american){
            if (put)
                v[5-jj] = MAX(hold, X*s[ind]);
            else
                v[5-jj] = MAX(hold, s[ind]-X);
        } else
            v[5-jj] = hold;
    }
}
```
Code excerpt 10.9  Code fragment illustrating how the option values are stored for the first two time steps so that the Greeks can be computed in the vicinity of the root node R

Figure 10.2 presents the results for the valuation of an American put option.

Computing the greeks: $\Delta$, $\Gamma$, and $\Theta$

We will now describe how to calculate the option’s hedge statistics (Greeks).

Let the option value and asset value at lattice node $k$ be denoted by $f_k$ and $S_k$ respectively. So, for instance, $S_T$ represents the asset price at node $T$, and $f_T$ is the corresponding option value at node $T$. Table 10.4 supplies details of the lattice node values in the vicinity of the root node $R$.

---

Figure 10.2  The error in the estimated value, $est\_val$, of an American put using a standard binomial lattice. The parameters used were: $T = 1.0$, $S = 105.0$, $X = 105.0$, $r = 0.1$, $q = 0.02$, $\sigma = 0.3$. The very accurate value ($acc\_val$) was 9.2508 and was computed using a 6000 step standard binomial lattice. The error in the estimated value was obtained as $est\_val - acc\_val$
The computation of each Greek is now considered.

**Delta**

The definition of $\Delta$ is the rate of change of the option value with asset price; all other parameters remaining fixed. Thus

$$\Delta = \frac{\partial f}{\partial S} = \frac{\Delta f}{\Delta S}$$

where $\Delta f$ is the change option value corresponding to the change in the asset price $\Delta S$. Ideally we would like to evaluate this partial derivative at the root node $R (m_1 = 0)$, however we can’t because we need at least two lattice nodes to compute a value. The best we can do is to evaluate the derivative at the first time step $(m_1 = 1)$ as follows:

$$\Delta = \frac{f_S - f_T}{S_S - S_T} = \frac{v[4] - v[3]}{s[M + 1] - s[M - 1]}$$

**Gamma**

The definition of $\Gamma$ is the rate of change of $\Delta$ with asset price; all other parameters remaining fixed. Thus

$$\Gamma = \frac{\partial^2 f}{\partial S^2} = \frac{\partial \Delta}{\partial S}$$

In order to evaluate $\Gamma$ we require at least two values of $\Delta$. The nearest this can be achieved to the root node $R$ is at time step 2, where we have:

$$\Gamma = \frac{\Delta_{UV} - \Delta_{VW}}{S_{UV}^* - S_{VW}^*}$$

with the midpoints

$$S_{UV}^* = \frac{1}{2} (S_U + S_V)$$

and the values of $\Delta$ at the midpoints $S_{UV}^*$ and $S_{VW}^*$ denoted by $\Delta_{UV}^*$ and $\Delta_{VW}^*$ respectively. Since
\[ \Delta^*_U = \frac{f_U - f_V}{S_U - S_V}, \quad \Delta^*_V = \frac{f_V - f_W}{S_V - S_W} \]

and

\[ S^*_U - S^*_V = \frac{1}{2} (S_U - S_W) \]

we have

\[ \Delta^*_U = \frac{v[2] - v[1]}{s[M + 2] - s[M]}, \quad \Delta^*_V = \frac{v[1] - v[0]}{s[M] - s[M - 2]} \]

The value of \( \Gamma \) can therefore be approximated as:

\[ \Gamma = \frac{2\{\Delta^*_U - \Delta^*_V\}}{s[M + 2] - s[M - 2]} \]

**Theta**

The definition of \( \Theta \) is the rate of change of option value with time; all other parameters remaining fixed. Thus

\[ \Theta = \frac{\partial f}{\partial t} = \frac{\Delta f}{\Delta t} \]

The nearest to the root node \( R \) that can be computed is over the time interval from time step 0 to time step 2. We then obtain the following approximation:

\[ \Theta = \frac{f_V - f_R}{2\Delta t} = \frac{v[1] - v[5]}{2\Delta t} \]

The Code excerpt 10.10 computes the \( \Delta, \Gamma, \) and \( \Theta \) by using the approximations we have just discussed.

**Vega**

The definition of \( \mathcal{V} \) is the rate of change of the option value with volatility.

\[ \mathcal{V} = \frac{\partial f}{\partial \sigma} \]

In a standard binomial lattice \( \mathcal{V} \) cannot be computed directly. A simple approach is to use two binomial lattices as follows

\[ \mathcal{V} = \frac{f_{\sigma + \Delta \sigma} - f_{\sigma}}{\Delta \sigma} \]

where \( f_{\sigma + \Delta \sigma} \) is the option value computed using a binomial lattice with volatility \( \sigma + \Delta \sigma \) and \( f_{\sigma} \) is the option value computed using another binomial lattice with a volatility of \( \sigma \); all other lattice parameters remain constant.

```c
if(greeks){
    /* assign the value of delta (obtained from ml = 1) */
    greeks[1] = (v[4]-v[3])/(s[M+1]-s[M-1]);
    /* assign the value of gamma (use the values at time step ml = 2) */
```
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dv1 = v[2] - v[1];
ds1 = s[M+2] - s[M];
dv2 = v[1] - v[0];
ds2 = s[M] - s[M-2];

\[ h = 0.5 \times (s[M+2] - s[M-2]); \]
\[ \text{greeks[0]} = \frac{(dv1/ds1) - (dv2/ds2))}{h}; \]

\[ \text{greeks[2]} = \frac{(v[1] - \text{value})/(2.0 \times \text{dt})}; \]

/* can also write: greeks[2] = \frac{(v[1] - v[5])/(2.0 \times \text{dt})}; */

### Code excerpt 10.10

A code fragment that computes the values of the Greeks (*Delta*, *Gamma*, and *Theta*) in the vicinity of the root lattice node R

```c
void standard_lattice(double *value, double greeks[], double S0, double X, double sigma, double T, double r, double q, Integer put, Integer is_american, Integer M, Integer *iflag)
{

/* Input parameters:
   = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = =
   S0 — the current price of the underlying asset
   X — the strike price
   sigma — the volatility
   T — the time to maturity
   r — the interest rate
   q — the continuous dividend yield
   put — if put is 0 then a call option, otherwise a put option
   is_american — if is_american is 0 then a European option, otherwise an American option
   M — the number of time steps

   Output parameters:
   = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = = =
   value — the value of the option,
   greeks[] — the hedge statistics output as follows: greeks[0] is gamma, greeks[1] is delta,
              greeks[2] is theta,
   iflag — an error indicator.

   */

dt = T/(double)M;
t1 = sigma*sqrt(dt);
u = exp(t1);
d = exp(-t1);
a = exp((r-q)*dt);
p = (a-d)/(u-d);
if (p < zero) || (p > 1.0) printf ("Error p out of range\n");
discount = exp(-r*dt);
p_u = discount*p;
p_d = discount*(1.0-p);

/* assign the 2*M+1 asset values */
s[M] = S0;
for (i = 1; i <= M; ++i){
    s[M+i] = u*s[M+i-1];
s[M-i] = d*s[M-i+1];
}
/* Find out if the number of time steps, M, is odd or even */
if (((M+1)/2) == (M/2)) {/* then M is even */
    if (put)
        v[M/2] = MAX(X - s[M], zero);
    else
        v[M/2] = MAX(s[M] - X, zero);
}
/* Calculate the option values at maturity */
P1 = 2*M;
P2 = 0;
for (i = 0; i < (M+1)/2; ++i) {
    v[M-i] = MAX(X - s[F1], zero);
v[i] = MAX(X - s[F2], zero);
}
else {
    v[M-i] = MAX(s[F1]-X, zero);
v[i] = MAX(s[F2]-X, zero);
}
```

---

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P1 = P1 - 2;
P2 = P2 + 2;

/* now work backwards through the lattice to calculate the current option value */
P2 = 0;
for (m1 = M-1; m1 > 2; --m1){
P2 = P2 + 1;
P1 = P2;
for (n = 0; n <= m1; ++n){
if ((v[n] == zero) && (v[n+1] == zero)) {
    hold = zero;
  } else
    hold = p_d*v[n] + p_u*v[n+1];
if (is_american) {
    if (put) v[n] = MAX(hold, X-s[P1]);
    else v[n] = MAX(hold, s[P1]-X);
  } else
    v[n] = hold;
P1 = P1 + 2;
}
}

/* The values v[0], v[1] & v[2] correspond to the nodes for m1 = 2, v[3] & v[4] correspond to the nodes for m1 = 1 and the option value (*value) is the node for m1 = 0, v[5]. For a given time step v[0] corresponds to the lowest asset price, v[1] to the next lowest etc. */
jj = 2;
for (m1 = 2; m1 > 1; --m1) {
  ind = M-m1+1;
  for (n = 0; n < m1; ++n) {
    hold = p_d*v[5-jj-m1-1] + p_u*v[5-jj+m1];
    if (is_american) {
      if (put) v[5-jj] = MAX(hold, X-s[ind]);
      else v[5-jj] = MAX(hold, s[ind]-X);
    } else
      v[5-jj] = hold;
    jj = jj + 2;
  }
}

*value = v[5];
if (greeks) {
    /* assign the value of delta (obtained from m1 = 1) */
    /* assign the value of gamma (use the values at time step m1 = 2) */
    dv1 = v[2] - v[1];
    ds1 = s[M+2] - s[M];
    dv2 = v[1] - v[0];
    ds2 = s[M] - s[M-2];
    h = 0.5*(s[M+2] - s[M-2]);
    greeks[0] = ((dv1/ds1) - (dv2/ds2))/h;
    /* assign the value of theta */
    greeks[2] = (v[1]-*value)/(2.0*dt); /* can also write: y greeks[2] = (v[1]-v[5])/(2.0*dt); */
}

Code excerpt 10.11 Function to compute the value of an option using a standard binomial lattice

The implied volatility of American options can be computed using the method outlined for European options in Section 9.3.4; however in this case the option value and Greeks are computed using a binomial lattice.

10.4.3 Binomial lattice with a control variate

The control variate technique can be used to enhance the accuracy that a standard binomial lattice gives for the value of an American vanilla option. It involves using
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the same standard binomial lattice to value of both an American option and also the equivalent European option. The Black–Scholes formula is then used to compute the accurate value of the European option. If we assume that the error in pricing the European option is the same as that for the American option we can achieve an improved estimate for the value of the American option.

When applied to the valuation of an American put option this can be expressed as follows:

European pricing error,

\[ \Delta_E = p^{BS}(S, E, \tau) - p^L(S, E, \tau) \]

American pricing error,

\[ \Delta_A = P^*(S, E, \tau) - P^L(S, E, \tau) \]

where as usual \( S \) is the current value of the asset, \( E \) is the strike price, and \( \tau \) is the maturity of the option. Also \( p^{BS}(S, E, \tau) \) is the Black–Scholes value of the European put option, \( p^L(S, E, \tau) \) is the binomial lattice estimate of the European put option, \( P^*(S, E, \tau) \) is the (unknown) accurate value of the American put option and \( P^L(S, E, \tau) \) is the binomial lattice estimate of the American put option.

Letting \( \Delta_E = \Delta_A \) we then have

\[ p^{BS}(S, E, \tau) - p^L(S, E, \tau) = P^*(S, E, \tau) - P^L(S, E, \tau) \]

which on rearrangement yields:

\[ P^*(S, E, \tau) = p^{BS}(S, E, \tau) - p^L(S, E, \tau) + P^L(S, E, \tau) \]

We thus use \( P^*(S, E, \tau) \) as the improved, control variate estimate, for the value of American put option. Of course exactly the same approach can be used to obtain an improved estimate for the value of an American call.

Code excerpt 10.12 shows the use of the control variate technique in a standard binomial lattice to provide improved estimates for both the value and the hedge statistics of an American option.

/* Set up the arrays as in the standard lattice */

for (i = 0; i < (M+1)/2; ++i) { /* Calculate the option values at maturity */
    if (put){
        a_v[M-i] = MAX(X-s[P1], zero);
        a_v[i] = MAX(X-s[P2], zero);
    }
    else {
        a_v[M-i] = MAX(s[P1]-X, zero);
        a_v[i] = MAX(s[P2]-X, zero);
    }
    e_v[i] = a_v[i];
    e_v[M-i] = a_v[M-i];
    P1 = P1 - 2;
    P2 = P2 + 2;
}

/* now work backwards through the lattice to calculate the current option value */
P2 = 0;
for (m1 = M-1; m1 >= 2; --m1) {
    P2 = P2 + 1;
    P1 = P1 - 2;
    for (n = 0; n <= m1; ++n){
        if ((a_v[n] == zero) && (a_v[n+1] == zero))
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hold = zero;
else
    hold = p_d*a_v[n] + p_u*a_v[n+1];
if (put)
    a_v[n] = MAX(hold, X-e_v[n]);
else
    a_v[n] = MAX(hold, s[n]);
if ((e_v[n] == zero) && (e_v[n+1] == zero))
    e_v[n] = zero;
else
    e_v[n] = p_d*e_v[n] + p_u*e_v[n+1];
P1 = P1 + 2;
}

/* The American values are stored in the array a_v, and the European values in the array e_v. The array indexing is the same as for the standard lattice */
jj = 2;
for (m1 = 2; m1 > 1; /C0/C0)
    ind = M/m1 + 1;
    for (n = 0; n < m1; ++n) {
        hold = p_d*a_v[5-jj-m1] + p_u*a_v[5-jj-m1];
        if (put)
            a_v[5-jj] = MAX(hold, X-s[ind]);
        else
            a_v[5-jj] = p_d*a_v[5-jj-m1] + p_u*a_v[5-jj-m1];
        e_v[5-jj] = p_d*e_v[5-jj-m1] + p_u*e_v[5-jj-m1];
        jj;   
        ind = ind + 2;
    }

/* v1 = American binomial approximation, v2 = European Binomial approximation, temp = exact (European) Black–Scholes value */
black_scholes(&temp, bs_greeks, S0, X, sigma, T, r, q, put, &iflagx);
/* return the control variate approximation */
if(greeks)

/* assign the value of delta (obtained from m1 = 1) */
a_delta = (a_v[4] - a_v[3])/(s[M+1] - s[M-1]);
e_delta = (e_v[4] - e_v[3])/(s[M+1] - s[M-1]);
voie[1] = a_delta - e_delta + bs_greeks[1];

/* assign the value of gamma (use the values at time step m1 = 2) */
dv1 = a_v[2] - a_v[1];
dv2 = e_v[2] - e_v[1];
ds1 = s[M+2] - s[M];
dv1 = a_v[1] - a_v[0];
ds2 = s[M] - s[M-2];
h = 0.5*(s[M+2] - s[M-2]);
a_gamma = ((dv1/ds1) - (dv2/ds2))/h;
dv1 = e_v[2] - e_v[1];
dv2 = e_v[1] - e_v[0];
e_gamme = ((dv1/ds1) - (dv2/ds2))/h;
voie[0] = (a_gamma - e_gamme) + bs_greexs[0];
/* assign the value of theta */
a_theta = (a_v[1] - a_v[5])/(2.0*dt);
e_theta = (e_v[1] - e_v[5])/(2.0*dt);
voie[2] = (a_v[0] - a_v[2]) + bs_greexs[2];
}

Code excerpt 10.12  Function to compute the value and hedge statistics of an American option using a binomial lattice with a control variate

Finally we should mention that the control variate technique does not just apply to American vanilla options. The method is quite general and can be used to obtain improved estimates for any integral (or exotic option) so long as an accurate (closed form) solution of a similar integral is known. One common use of the control variate method is to improve the accuracy of Monte Carlo estimates, see Clewlow and Strickland (1999).

10.4.4 The binomial lattice with BBS and BBSR

Here we consider the binomial Black–Scholes (BBS) method and also the binomial Black–Scholes method with Richardson extrapolation (BBSR), see Broadie and
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DeTemple (1996). As with the control variate method discussed in the previous section, both of these techniques can be used in conjunction with a standard binomial lattice to improve the computed results.

We will first discuss the BBS method.

The BBS Method

The BBS method is identical to the standard binomial lattice except that in the last time step (that is just before option maturity) the Black–Scholes formula is used to calculate the option values at maturity. For an \( n \) time step binomial lattice this involves evaluating the Black–Scholes formula at each of the \( n \) nodes in the penultimate time step, see Figure 10.1. In Code excerpt 10.13 we define the function \( bbs\_lattice \) which incorporates the BBS method into a standard binomial lattice. The reader will have noticed that \( bbs\_lattice \) is rather lax concerning the amount of storage that is required, see Section 10.4.2. It uses an array of size \( \mathcal{L}_n \) rather than \( \mathcal{L}_n \) to store the lattice asset prices; the modification to use an array of size \( \mathcal{L}_n \) is left as an exercise.

```c
void bbs_lattice(double *value, double greeks[], double S0, double X, double sigma, double T, double r, double q, Integer put, Integer M, Integer *iflag)
{
    /* Input parameters:
        S0 — the current price of the underlying asset
        X — the strike price
        sigma — the volatility
        T — the time to maturity
        r — the interest rate
        q — the continuous dividend yield
        put — if put is 0 then a call option, otherwise a put option
        M — the number of time steps
    Output parameters:
        value — the value of the option, greeks[] — the hedge statistics output as follows: greeks[0] is gamma, greeks[1] is delta, greeks[2] is theta.
        iflag — an error indicator.
    */
    dt = T/(double)M;
    t1 = sigma*sqrt(dt);
    u = exp(t1);
    d = exp(-t1);
    a = exp((r-q)*dt);
    p = (a-d)/(u-d);
    if ((p < zero) || (p > 1.0)) return; /* Invalid probability */
    discount = exp(-r*dt);
    p_u = p*discount;
    p_d = (1.0-p)*discount;
    jj = 0;
    s[0] = S0;
    /* The 'higher' the value of jj, at a given time instant,
        the lower the value of the asset price */
    for (m1 = 1; m1 < M-1; ++m1){/* Calculate asset values up to (M-1)th time step */
        ++jj;
        s[jj] = u*s[jj-m1];
    }
    ++jj;
    s[jj] = d*s[jj-m1-1];
    }*/
```
for (n = 0; n < M-1; ++n) /* Use Black–Scholes for the final step */
black_scholes(&temp, NULL, s[jj], X, sigma, dt, r, q, put, &iflagx);
for (jj = 0; jj < n; ++jj)
for (m1 = M-1; m1 > 3; --m1) /* work backwards through the lattice */
    for (n = 0; n < m1; ++n) {
        if ((v[n] == zero) && (v[n+1] == zero))
            hold = zero;
        else
            hold = p_d*v[n] + p_u*v[n+1];
        if (is_american)
            if (put)
                v[n] = MAX(hold, X-s[jj]);
            else
                v[n] = MAX(hold, s[jj]-X);
        else
            v[n] = hold;
    --jj;
    }
/* The values v[0], v[1] & v[2] correspond to the nodes for m1 = 2, v1 & v2 correspond to the nodes for m1 = 1 and the option value (*value) is the node for m1 = 0. For a given time step v[0] corresponds to the lowest asset price, v[1] to the next lowest etc. */
hold = p_d*v[0] + p_u*v[1];
if (is_american)
    if (put)
        v1 = MAX(hold, X-s[jj]);
    else
        v1 = MAX(hold, s[jj]-X);
else
    v1 = hold;
--jj;
hold = p_d*v[1] + p_u*v[2];
if (is_american)
    if (put)
        v2 = MAX(hold, X-s[jj]);
    else
        v2 = MAX(hold, s[jj]-X);
else
    v2 = hold;
--jj;
hold = p_d*v1 + p_u*v2;
if (is_american)
    if (put)
        *value = MAX(hold, X-s[0]);
    else
        *value = MAX(hold, s[0]-X);
else
    *value = hold;
if(greeks)
    /* assign the value of delta (obtained from m1 = 1) */
greeks[1] = (v2-v1)/(s[1]-s[2]);
    /* assign the value of gamma (use the values at time step m1 = 2) */
dv1 = v[2] - v[1];
ds1 = s[3] - s[4];
dv2 = v[1] - v[0];
ds2 = s[4] - s[5];
h = 0.5*(s[3] - s[5]);
greeks[0] = ((dv1/ds1) - (dv2/ds2))/h;
    /* assign the value of theta */
greeks[2] = (*value)/(2.0*dt);
}

**Code excerpt 10.13** The function bbs_lattice which incorporates the BBS method into a standard binomial lattice. The Black–Scholes formula is evaluated by using the function black_scholes, given in Section 9.3.3
The benefitsofusingtheBBSapproachtopriceanAmericancallareillustratedin Figure 10.3. Herewecomparetheresultsobtainedusingthefunction bbs_lattice withthosecomputedbythefunction standard_lattice,thestandardbinomial latticeofCodeexcerpt10.11. ItcanbeclearlyseenthatBBSmethodissignificantly moreaccuratethanthestandardbinomiallatticeapproach,inwhichoptionpricing errorexhibitspronouncedoscillations.

**The BBSR Method**

The BBSR method applies two point Richardson extrapolation to the computed BBS values, for more information concerning Richardson extrapolation see Marchuk and Shaidurov (1983). In this method the option price estimates from two BBS lattice, with differing number of time steps, are combined to form an improved estimate.

Here we use the following BBSR scheme to compute the value of an American call option

\[
C_{BBSR}(S, E, \tau, 2n) = \frac{4}{3} C_{BBS}(S, E, \tau, 2n) - \frac{1}{3} C_{BBS}(S, E, \tau, n)
\]

(10.96)

where \( S \) is the current asset value, \( E \) is the strike price, \( \tau \) is the option maturity, \( C_{BBS}(S, E, \tau, n) \) is the value of the call option computed using a BBS lattice with
\( n \) time steps, \( C_{BBS}(S, E, \tau, 2n) \) is the value of the call option computed using a BBS lattice with \( 2n \) time steps and \( C_{BBSR}(S, E, \tau, 2n) \) is the BBSR estimate. We compute the value of an American put using

\[
P_{BBSR}(S, E, \tau, 2n) = \frac{4}{3} P_{BBS}(S, E, \tau, 2n) - \frac{1}{3} P_{BBS}(S, E, \tau, n) \tag{10.97}
\]

Figure 10.4 displays the computed BBSR results for an American call option with \( S = 105.0, \tau = 1.0, E = 105.0, q = 0.02 \), and \( \sigma = 0.3 \).

In Tables 10.5 and 10.6 the errors in computing both an American put and an American call option are presented; the methods used are the standard binomial lattice, the BBS lattice and the BBSR lattice. It can be seen that the BBSR lattice gives the most accurate results. This is not surprising since, from Equations 10.96 and 10.97 we see that when we use either an \( n \) time step standard binomial lattice or an \( n \) time step BBS lattice the corresponding BBSR estimate is obtained using both an \( n \) time step BBS lattice and also a \( 2n \) time step BBS lattice. One way of checking whether Richardson extrapolation is providing increased accuracy is to compare the results for a \( 2n \) time step BBS lattice with those for an \( n \) time step BBSR lattice. Inspection of the results shows that Richardson extrapolation has in fact led to an improvement. For example in Table 10.5 the error for a 160 time step BBS lattice is

![Richardson extrapolation: American call](image)

**Figure 10.4** The error in the estimated value, \( est_val \), of an American call, using a BBSR binomial lattice. The parameters used were: \( T = 1.0, S = 105.0, E = 105.0, r = 0.1, q = 0.02, \sigma = 0.3 \). The very accurate value (\( acc_val \)) was 16.1697, and was computed using a 6000 step standard binomial lattice. The error in the estimated value was obtained as \( est_val - acc_val \)
### 158 Pricing Assets

**Table 10.5** The pricing errors for an American call option computed by a standard binomial lattice, a BBS lattice and also a BBSR lattice. The pricing error is defined as \( \frac{\text{estimated value}}{\text{accurate value}} \), where the accurate value, 16.1697, was obtained by using a 6000 step standard binomial lattice. The option parameters used were: \( T = 1.0, S = 105.0, E = 105.0, r = 0.1, q = 0.02, \) and \( \sigma = 0.3 \).

<table>
<thead>
<tr>
<th>( n ) steps</th>
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<th>BBS lattice</th>
<th>BBSR lattice</th>
</tr>
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</table>

**Table 10.6** The pricing errors for an American put option computed by a standard binomial lattice, a BBS lattice and also a BBSR lattice. The pricing error is defined as \( \frac{\text{estimated value}}{\text{accurate value}} \), where the accurate value, 9.2508, was obtained by using a 6000 step standard binomial lattice. The option parameters used were: \( T = 1.0, S = 105.0, E = 105.0, r = 0.1, q = 0.02, \) and \( \sigma = 0.3 \).

<table>
<thead>
<tr>
<th>( n ) steps</th>
<th>Standard lattice</th>
<th>BBS lattice</th>
<th>BBSR lattice</th>
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</table>
5.0869 \times 10^{-3}$, while that for an 80 time step BBSR lattice is $3.5725 \times 10^3$; in Table 10.6 the error for an 80 time step BBS lattice is $6.3858 \times 10^{-3}$ and that for a 40 time step BBSR lattice is $3.5725 \times 10^{-3}$.

10.5 IMPLIED LATTICE METHODS

It is well known that market option prices are not consistent with theoretical prices derived from the Black–Scholes formula. This has led traders to quote option prices in terms of a volatility, $\sigma_{\text{imp}}$, which makes the Black–Scholes formula value equal to the observed market price. Here we refer to $\sigma_{\text{imp}}$ as the implied volatility, to distinguish it from the theoretical constant volatility $\sigma$; essentially $\sigma_{\text{imp}}$ is another way of quoting option prices. Empirical studies have found that:

- For vanilla options of a given maturity the value of $\sigma_{\text{imp}}$ decreases with the level of the strike price, this asymmetry is termed volatility skew.
- For vanilla options of a given strike price the value of $\sigma_{\text{imp}}$ increases with maturity, this variation is called the volatility term structure.

Here we follow Derman and Kani (1994) and refer to both the volatility skew and the volatility term structure as the volatility smile. The precise shape and magnitude of the volatility smile is dependent on the nature of the option being considered. We are thus led to consider more sophisticated option pricing methods which capture the observed deviations from these Black–Scholes formula.

Instead of assuming, as in Section 8.3, that the underlying asset price $S_t$ follows GBM with constant drift and volatility, we will now consider the more general GBM process:

$$\frac{dS_t}{S_t} = \mu(t)dt + \sigma(S_t, t)dZ$$  \hspace{0.5cm} (10.98)

where $t$ is the current time, $\mu(t)$ is the time dependent risk neutral drift and $\sigma(S_t, t)$ is an unknown volatility function which depends on both the stock price and time. If we make use of Ito’s lemma, and write $S_{t+\Delta t}$ for the asset price at time $t + \Delta t$, Equation 10.84 can be expressed in discretized form as:

$$\log \left( \frac{S_{t+\Delta t}}{S_t} \right) = \left\{ \mu(t) - \frac{\sigma^2(S_t, t)}{2} \right\} \Delta t + \sigma(S_t, t)dZ$$  \hspace{0.5cm} (10.99)

or equivalently

$$\log \left( \frac{S_{t+\Delta t}}{S_t} \right) \sim N \left( \left\{ \mu(t) - \frac{\sigma^2(S_t, t)}{2} \right\} \Delta t, \, \sigma^2(S_t, t)\Delta t \right)$$  \hspace{0.5cm} (10.100)

In this section we will show how the volatility function $\sigma(S_t, t)$ can be evaluated by ensuring that the option prices calculated using this model agree with those of the smile.

The implied binomial lattice constructed using this extended model will no longer be a regular lattice (as is the case for the simple Black–Scholes model) but will have a distorted shape similar to that shown in Figure 10.5 below.

It can be seen that the lattice levels are equispaced in time and are $\Delta t$ apart. Lattice level 1, time $t_1$, corresponds to the root node $(1, 1)$ and is the current time, at which
we want to find the value of the option. Time \( t_n \), associated with lattice level \( n \), is given by:

\[
t_n = t_1 + (n - 1)\Delta t = t + (n - 1)\Delta t
\]

so \( t_n \) is \((n - 1)\Delta t\) in the future relative to the current time \( t \).

Construction of the implied lattice requires option prices for the complete range of strikes and maturities; these values can be obtained via interpolation from known option prices that are traded on the stock market.

Once the implied lattice has been created it can be used to price a range of European and American options.

Here we will describe the implied lattice technique developed by Derman and Kani (1994), and then consider the subsequent refinements proposed by Barle and Cakici (1995) and Chriss (1997). Of necessity our description of these techniques will be brief, and will mainly consist of explanatory detail and mathematical proofs that are not given in the original papers. For more information the reader should consult the original papers which are available (by kind permission of RISK Magazine) on the CD ROM which accompanies this book.

**Figure 10.5** An implied binomial lattice which incorporates the volatility smile observed in traded put and call options. The \( n \)th node at the \( n \)th lattice level is denoted by \((n, i)\). The current value (at time \( t \)) of the underlying asset is \( S \), and this is the asset value assigned to the root lattice node \((1, 1)\). The asset values at the other lattice nodes depend on the technique used to construct the implied lattice.
Before discussing the details of implied binomial lattices we will first consider the local volatility associated with a particular lattice node.

**Local volatility**

An expression for the stock volatility at the binomial lattice node \((n, i)\) will now be derived. At time instant \(t_n\), the stock value at this node is denoted by \(s_i\). After time \(\Delta t\), time instant \(t_{n+1}\), the stock price either has jumped up to \(S_u\), at lattice node \((n+1, i+1)\), or jumped down to \(S_d\), at lattice node \((n+1, i)\). Applying Equation 10.100 to node \((n, i)\) by setting \(t = t_n\) and \(S_t = s_i\) then gives

\[
v \sim N\left(\{\mu(t_n) - \sigma^2(s_i, t_n) / 2\} \Delta t, \sigma^2(s_i, t_n) \Delta t\right)
\]

(10.101)

where the variate \(v\) can only take the two values \(v_1 = \log(S_u / s_i)\), and \(v_2 = \log(S_d / s_i)\).

We will let \(p_i\) denote the probability of taking the value \(v_1\), corresponding to an up jump. The probability of \(v\) having the value \(v_2\), corresponding to a down jump, is thus \(1 - p_i\).

The quantity \(\sigma(s_i, t_n)\) will be referred to as the local volatility, \(\sigma_{loc}\), associated with the lattice node \((n, i)\), and using Equation 10.101 we can write

\[
\text{Var}(v) = \sigma^2_{loc} \Delta t
\]

(10.102)

An expression for \(\sigma_{loc}\) can then be obtained in terms of \(S_u\), \(S_d\) and \(p_i\), as follows. The variance of \(v\) is:

\[
\text{Var}(v) = E[v^2] - (E[v])^2
\]

where

\[
E[v^2] = p_i \left\{ \log\left(\frac{S_u}{s_i}\right) \right\}^2 + (1 - p_i) \left\{ \log\left(\frac{S_d}{s_i}\right) \right\}^2
\]

and

\[
(E[v])^2 = \left\{ p_i \log\left(\frac{S_u}{s_i}\right) + (1 - p_i) \log\left(\frac{S_d}{s_i}\right) \right\}^2
\]

which means that

\[
(E[v])^2 = p_i^2 \left\{ \log\left(\frac{S_u}{s_i}\right) \right\}^2 + (1 - p_i)^2 \left\{ \log\left(\frac{S_d}{s_i}\right) \right\}^2 + 2p_i(1 - p_i) \log\left(\frac{S_u}{s_i}\right) \log\left(\frac{S_d}{s_i}\right)
\]

We can therefore write the variance as

\[
\text{Var}(v) = p_i \left\{ \log\left(\frac{S_u}{s_i}\right) \right\}^2 + (1 - p_i) \left\{ \log\left(\frac{S_d}{s_i}\right) \right\}^2 - p_i^2 \left\{ \log\left(\frac{S_u}{s_i}\right) \right\}^2 - (1 - p_i)^2 \left\{ \log\left(\frac{S_d}{s_i}\right) \right\}^2 - 2p_i(1 - p_i) \log\left(\frac{S_u}{s_i}\right) \log\left(\frac{S_d}{s_i}\right)
\]
which simplifies to

\[ \text{Var}(\nu) = p_i(1 - p_i) \left[ \left\{ \log \left( \frac{S_u}{S_i} \right) \right\}^2 + \left\{ \log \left( \frac{S_d}{S_i} \right) \right\}^2 - 2 \log \left( \frac{S_u}{S_i} \right) \log \left( \frac{S_d}{S_i} \right) \right] \] (10.103)

However

\[ \log \left( \frac{S_u}{S_i} \right) - \log \left( \frac{S_d}{S_i} \right) = \log \left( \frac{S_u}{S_d} \right) \]

and

\[ \left\{ \log \left( \frac{S_u}{S_i} \right) - \log \left( \frac{S_d}{S_i} \right) \right\}^2 = \left\{ \log \left( \frac{S_u}{S_i} \right) \right\}^2 + \left\{ \log \left( \frac{S_d}{S_i} \right) \right\}^2 - 2 \log \left( \frac{S_u}{S_i} \right) \log \left( \frac{S_d}{S_i} \right) \]

Substituting this into Equation 10.103 we obtain:

\[ \text{Var}(\nu) = p_i(1 - p_i) \left\{ \log \left( \frac{S_u}{S_d} \right) \right\}^2 \] (10.104)

Therefore combining Equations 10.102 and 10.104 we have

\[ \sigma_{loc}^2 \Delta t = p_i(1 - p_i) \left\{ \log \left( \frac{S_u}{S_d} \right) \right\}^2 \]

and so the local volatility is given by:

**Binomial lattice: local volatility**

\[ \sigma_{loc} = \log \left( \frac{S_u}{S_d} \right) \sqrt{\frac{p_i(1 - p_i)}{\Delta t}} \] (10.105)

In an implied lattice the transition probabilities, \( p_i \), and the ratios \( S_u/S_d \) are (in general) different for each lattice node. This generates a volatility surface in which the local volatility \( \sigma_{loc} \) varies throughout the lattice. By contrast the CRR binomial lattice of Section 10.4.1 has the same value of \( \sigma_{loc} \) for all its lattice nodes. The reason for this that \( p_i \) and \( \Delta t \) are constants, and the up and down jumps are

\[ S_u = s_i u \quad \text{and} \quad S_d = s_i d, \quad \text{where} \quad u = \frac{1}{d} \]

This means that

\[ \frac{S_u}{S_d} = u^2 \]

and the (constant) local volatility is

\[ \sigma_{loc} = \log(u^2) \sqrt{\frac{p(1 - p)}{\Delta t}} \]
CRR binomial lattice: local volatility

\[ \sigma_{loc} = 2 \log(u) \sqrt{\frac{p(1-p)}{\Delta t}} \]  
where we have denoted the (constant) CRR up jump probability by \( p \).

10.5.1 Derman–Kani implied lattice

We now consider the paper by Derman–Kani (1994), henceforth referred to as DK, which describes an implied binomial lattice based on the market values of European put and call options.

The implied lattice (see Figure 10.5) consists of uniformly spaced levels \( \Delta t \) apart, and is built using forward iteration. To explain this technique we will assume that the first \( n \) lattice levels have been constructed and that they match the observed volatility smile for all strike prices and maturities out to time \( t_n \). The task is to determine the \( n+1 \) nodes at the \((n+1)\)th lattice level from the previously calculated \( n \) nodes at the \( n \)th lattice level.

For convenience we will now give the notation used in the formulae for constructing the lattice nodes in the \((n+1)\)th lattice level from the known lattice node values in the \( n \)th lattice level.

\[ r \] The known riskless interest rate for lattice level \((n+1)\).

\[ s_i \] The known stock price at node \((n, i)\); that is at the \( i \)th node on lattice level \( n \). We also note that \( s_i \) is the strike price for options expiring at lattice level \( n+1 \).

\[ F_i \] The known forward price at lattice level \( n+1 \) of the known price \( s_i \) at lattice level \( n \).

\[ S_i \] The unknown stock price at node \((n+1, i)\).

\[ \lambda_i \] The known Arrow-Debreu price at node \((n, i)\).

\[ p_i \] The unknown risk-neutral up jump transition probability from node \((n, i)\) to node \((n+1, i+1)\).

Here the \( i \)th node at level \( n \) has known stock price \( s_i \), and is denoted by \((n, i)\). The probability that the stock price \( s_i \) increases to \( S_{i+1} \) in lattice level \( n+1 \) is denoted by \( p_i \), whereas the probability that the stock price decreases to \( S_i \) in level \( n+1 \) is given by \( 1 - p_i \).

The forward price, \( F_i \), of \( s_i \) at lattice level \( n+1 \) is simply given by the risk neutral expected value of \( s_i \) one time step later. That is \( F_i = s_i \exp(r \Delta t) \), or in terms of the up and down jump probabilities \( p_i \) and \( 1 - p_i \) respectively we have:

\[ F_i = p_i S_{i+1} + (1 - p_i) S_i, \quad \text{for } i = 1, \ldots, n \]  

where as before \( S_{i+1} \) is the stock value at lattice level \( n+1 \) following an up jump and \( S_i \) is the stock value at lattice level \( n+1 \) following a down jump.

The Arrow-Debreu price, \( \lambda_i \), at each lattice node \((n, i)\) is defined as: the probability of reaching node \((n, i)\) from the root lattice node \((1, 1)\) discounted by the risk neutral interest rate between time \( t_1 \) and time \( t_n \).
The Arrow-Debreu price of a lattice node is thus the value of a security that pays $1 if the stock price reaches that node and zero otherwise. The value of \( \lambda_i \) corresponding to node \((n, i)\) is computed as the sum, over all paths from the root node \((1, 1)\) to node \((n, i)\), of the product of the riskless-discounted transition probabilities of nodes along each path from \((1, 1)\) to \((n, i)\). We provide more detail concerning the computation of \( \lambda_i \) in the example calculation at the end of this section, and consider the following two methods:

1. Direct calculation of the Arrow-Debreu prices in lattice level \( n + 1 \) by using all paths from the root lattice node \((1, 1)\).
2. Iterative calculation of the Arrow-Debreu prices in lattice level \( n + 1 \) from the known Arrow-Debreu prices in lattice level \( n \).

It is shown that direct calculation of the Arrow-Debreu prices becomes substantially more complicated as the number of lattice level increases. This is because the number of possible paths from the root node \((1, 1)\) to any given lattice node \((n + 1, i)\) increases dramatically with \( n \). The iterative approach is thus the most practical method for computing Arrow-Debreu prices in lattices containing more than just a few lattice levels.

Let \( C(K, t_{n+1}) \) and \( P(K, t_{n+1}) \) be the current, time \( t \), respective prices of European call and European put options with strike \( K \) and maturity corresponding to lattice level \( n + 1 \); the values \( C(K, t_{n+1}) \) and \( P(K, t_{n+1}) \) can be obtained via interpolation from the known market prices. An expression for \( C(K, t_{n+1}) \), can also be computed by using the binomial node values at lattice level \( n \), and this method yields the following equation:

\[
C(K, t_{n+1}) = \exp(-r\Delta t) \sum_{j=1}^{n} \lambda_j \{ p_j \max(S_{j+1} - K, 0) \\
+ (1 - p_j) \max(S_j - K, 0) \} \quad (10.108)
\]

where \( \max(S_j - K, 0) \) is the payout for the call at the \( j \)th lattice node on lattice level \( n + 1 \) and \( \max(S_{j+1} - K, 0) \) is the payout for the call at the \((j + 1)\)th lattice node on lattice level \( n + 1 \).

When the strike \( K \) equals \( s_i \) the above equation becomes

\[
\exp(r\Delta t)C(s_i, t_{n+1}) = \sum_{j=1}^{n} \lambda_j \{ p_j \max(S_{j+1} - s_i, 0) \\
+ (1 - p_j) \max(S_j - s_i, 0) \} \quad (10.109)
\]

Since the terms that contribute to the value of the call option, \( C(s_i, t_{n+1}) \), are those with positive payouts we only need consider \( j \) indices in the range \( i \) to \( n \), and the \( i \)th term of the summation on the right hand side of Equation 10.109 is:

\[
\lambda_i \{ p_i \max(S_{i+1} - s_i, 0) + (1 - p_i) \max(S_i - s_i, 0) \} = \lambda_i p_i (S_{i+1} - s_i) \quad (10.110)
\]

where we have used (see DK Figure 4) the following: \( S_{i+1} > s_i \) \((S_{i+1} \) is the up jump stock value from lattice level \( n \) to lattice level \( n + 1 \)) whereas \( S_i < s_i \) \((S_i \) is the down jump stock value from lattice level \( n \) to lattice level \( n + 1 \)).
This means that we can rewrite Equation 10.109 as:

\[
\exp(r\Delta t)C(s_i, t_{n+1}) = \lambda_i p_i (S_{i+1} - s_i) + \sum_{j=i+1}^{n} \lambda_j \{ p_j (S_j - s_i) 
+ (1 - p_j)(S_j - s_i) \} \tag{10.111}
\]

If we subtract the constant term \( s_i \) from both sides of Equation 10.107 we obtain:

\[
F_j - s_i = p_j (S_{j+1} - s_i) + (1 - p_j)(S_j - s_i), \quad j = 1, \ldots, n \tag{10.112}
\]

where we used \( s_j = p_j s_i + (1 - p_j)s_i \). Substituting Equation 10.112 into Equation 10.111 gives:

\[
\exp(r\Delta t)C(s_i, t_{n+1}) = \lambda_i p_i (S_{i+1} - s_i) + \Sigma \tag{10.113}
\]

where \( \Sigma = \sum_{j=i+1}^{n} \lambda_j (F_j - s_i) \). The first term in Equation 10.113 depends on the unknown values of the transition probability \( p_i \) and stock price \( S_{i+1} \). The last term \( \Sigma \) involves a summation over the known forward prices \( F_j \) and known stock prices \( s_i \) on lattice level \( n \). Since both \( F_j \) and \( C(s_i, t_{n+1}) \) are known, Equations 10.107 and 10.113 can be solved to give the following expressions for \( S_{i+1} \) and \( p_i \), in terms of \( S_i \):

\[
p_i = \frac{F_i - S_i}{S_{i+1} - S_i} \tag{10.114}
\]

and

\[
S_{i+1} = \frac{S_i \{ C(s_i, t_{n+1}) \exp(r\Delta t) - \Sigma \} - \lambda s_i (F_i - S_i)}{\{ C(s_i, t_{n+1}) \exp(r\Delta t) - \Sigma \} - \lambda_i (F_i - S_i)} \tag{10.115}
\]

We will now derive these two results.

**Proof of Equation 10.114 (DK equation 7)**

From Equation 10.107 we have:

\[
F_i = p_i S_{i+1} + (1 - p_i)S_i
\]

which gives:

\[
F_i = p_i (S_{i+1} - S_i) + S_i \quad \text{and} \quad p_i = \frac{F_i - S_i}{S_{i+1} - S_i} \quad \text{QED}
\]

**Proof of Equation 10.115 (DK equation 6)**

If we substitute the value of \( p_i \) from Equation 10.115 into Equation 10.113 we obtain:

\[
C(s_i, t_{n+1}) \exp(r\Delta t) = \frac{\lambda_i (F_i - S_i)(S_{i+1} - s_i)}{S_{i+1} - S_i} + \Sigma
\]
Multiplying both sides by $S_{i+1} - S_i$ yields:

$$\exp (r \Delta t) C\{S_i, t_{n+1}\}{S_{i+1} - S_i} = -\lambda_i F_i S_i - \lambda_i S_i S_{i+1} + \lambda_i F_i S_{i+1} + \lambda_i S_i + S_{i+1} \Sigma - S_i \Sigma$$

so

$$S_{i+1}\{C\{S_i, t_{n+1}\} \exp (r \Delta t) + \lambda_i S_i - \lambda_i F_i - \Sigma\}$$

$$= S_i C\{S_i, t_{n+1}\} \exp (r \Delta t) - \lambda_i S_i (F_i - S_i) - S_i \Sigma$$

or

$$S_{i+1}\{C\{S_i, t_{n+1}\} \exp (r \Delta t) - \Sigma - \lambda_i (F_i - S_i)\}$$

$$= S_i C\{S_i, t_{n+1}\} \exp (r \Delta t) - \Sigma\} - \lambda_i S_i (F_i - S_i)$$

and finally gives the following expression for $S_{i+1}$:

$$S_{i+1} = \frac{S_i C\{S_i, t_{n+1}\} \exp (r \Delta t) - \Sigma\} - \lambda_i S_i (F_i - S_i)}{C\{S_i, t_{n+1}\} \exp (r \Delta t) - \Sigma\} - \lambda_i (F_i - S_i)}$$

QED

If we know $S_i$ at one initial node then Equations 10.114 and 10.115 can be used to find iteratively the values of $S_{i+1}$ and $p_i$ for all $n/2 + 1$ nodes above the centre of the lattice on the $(n + 1)$th lattice level.

If $n + 1$ is odd then the initial value used for $S_i$ is the stock value associated with the central lattice node, that is $S_i = S$. On the other hand if $n + 1$ is even then we use the CRR lattice centering condition (see Section 10.4.1). Let $S_{i+1}$ denote the $(n + 1)$th level stock value for the node just above the centre of the lattice, and $S_i$ denote the $(n + 1)$th level stock value just below the centre of the lattice. For a CRR $(u = 1/d)$ lattice these values are related to the central node stock value, $S$, at lattice level $n$ by:

$$S_{i+1} = Su \quad \text{and} \quad S_i = Sd = \frac{S}{u}$$

and therefore

$$S_{i+1} S_i = S^2 \quad (10.116)$$

Substituting Equation 10.116 into Equation 10.114 gives the following formula for the stock value at the node just above the centre of the lattice when $n + 1$ is even:

$$S_{i+1} = \frac{S(P + \lambda_i S)}{\lambda_i F_i - P}, \quad \text{for} \quad i = \left(\frac{n + 1}{2}\right)$$

(10.117)

where $P = C\{S, t_{n+1}\} \exp (r \Delta t) - \Sigma$.

When $n + 1$ is even, Equation 10.117 can thus be used in conjunction with Equations 10.114 and 10.115 to iteratively compute the node values $S_{i+1}$ and probabilities $p_i$ for the $(n + 1)/2$ nodes above the lattice centre.
**Proof of Equation 10.117 (DK equation 8)**

From Equation 10.114 we have that the probability $p_i$ is given by:

$$p_i = \frac{F_i - S_i}{S_{i+1} - S_i} = \frac{(F_i - S_i)S_{i+1}}{(S_{i+1} - S_i)S_{i+1}}$$

since, in Equation 10.116, $S_iS_{i+1} = S^2$ we obtain:

$$p_i = \frac{F_i S_{i+1} - S^2}{S_{i+1}^2 - S^2} = \frac{F_i S_{i+1} - S^2}{(S_{i+1} - S)(S_{i+1} + S)}$$

However from Equation 10.113

$$\exp(r\Delta t) C(s_i, t_{n+1}) = \lambda_i p_i(S_{i+1} - s_i) + \Sigma$$

so

$$\Psi = \lambda_i p_i(S_{i+1} - s_i)$$

(10.118)

When $s_i = S$ we therefore have:

$$\Psi = \lambda_i p_i(S_{i+1} - S) = \frac{\lambda_i(F_i S_{i+1} - S^2)(S_{i+1} - S)}{(S_{i+1} - S)(S_{i+1} + S)} = \frac{\lambda_i(F_i S_{i+1} - S^2)}{S_{i+1} + S}$$

which gives:

$$\Psi S_{i+1} + \Psi S = \lambda_i F_i S_{i+1} - \lambda_i S^2$$

$$S_{i+1}(\Psi - \lambda_i F_i) = -S(\Psi + \lambda_i S)$$

and finally:

$$S_{i+1} = \frac{S(\Psi + \lambda_i S)}{\lambda_i F_i - \Psi} \quad QED$$

Similar formulae can be derived, using interpolated put prices, which enable the stock values and probabilities for nodes below the lattice centre to be computed. The formula to determine a lower node’s stock value from an upper node’s stock value is:

$$S_i = \frac{S_{i+1}\{P(s_i, t_{n+1})\exp(r\Delta t) - \Sigma^*\} - \lambda_i s_i(F_i - S_{i+1})}{\{P(s_i, t_{n+1})\exp(r\Delta t) - \Sigma^*\} - \lambda_i(F_i - S_{i+1})}$$

(10.119)

where $P(s_i, t_{n+1})$ is the interpolated price of a put with strike $s_i$ and expiry time $t_{n+1}$ and

$$\Sigma^* = \sum_{j=1}^{i-1} \lambda_j(s_i - F_j)$$

denotes the sum over all nodes below the node with stock price $s_i$ at which the put was struck.

When building the lattice the computed transition probabilities, $p_i$, for each lattice node must obey the constraint $0 \leq p_i \leq 1$. The upper limit $p_i \leq 1$ is equivalent to
requiring that the up-node stock price $S_{i+1}$ at the next level does not fall below the forward price $F_i$. This result comes from Equation 10.114

$$p_i = \frac{F_i - S_i}{S_{i+1} - S_i}$$

where it can easily be seen that if $F_i > S_{i+1}$, then $p_i > 1$. Similarly the lower limit $p_i \geq 0$ can be shown to be equivalent to requiring that the down-node stock price $S_i$ is above the forward price $F_i$. From Equation 10.114 we now have:

$$p_{i+1} = \frac{F_{i+1} - S_{i+1}}{S_{i+2} - S_{i+1}}$$

and so if $S_{i+1} > F_{i+1}$ then $p_{i+1} < 0$. We thus have:

$$F_i < S_{i+1} < F_{i+1}$$

(10.120)

which is illustrated in Figure 10.6.

---

**Figure 10.6** An implied lattice showing the position of the stock prices in relation to the forward prices, between the $n$th and $(n+1)$th lattice levels. The stock prices in lattice level $n$ are denoted by $s_i$, $s_i$, and $s_{i-1}$, while those in lattice level $n+1$ are represented by $S_{i+1}$, $S_{i+1}$, and $S_i$. The transition probabilities between lattice level $n$ and lattice level $n + 1$ are $p_{i+1}$, $p_i$, and $p_{i-1}$, and the forward prices are $F_{i+1}$, $F_i$, and $F_{i-1}$. If the computed stock value is $S_{i+1}$ then, in order to obtain valid transition probabilities, it must satisfy the constraint $F_i < S_{i+1} < F_{i+1}$.
### Bad probabilities

Figure 10.6 shows the relative positions of the computed stock values, $S_i$, at lattice level $n + 1$, and the forward prices $F_i$ computed from the stock values, $s_i$, at lattice level $n$. If a computed stock value, $S_{i+1}$, violates the constraints imposed by Equation 10.120 then it is necessary to choose an alternative value for which the transition probability $p_i$ is in the permitted range $0 < p_i < 1$. DK advocates choosing $S_{i+1}$ so that the logarithmic spacing between adjacent lattice nodes is the same as that in the previous lattice level; that is:

$$\frac{S_{i+1}}{S_i} = \frac{s_i}{s_{i-1}}$$

This means replacing the value of $S_{i+1}$ computed using Equation 10.115 with

$$S_{i+1} = S_i \left( \frac{s_i}{s_{i-1}} \right)$$

(10.121)

If this method still fails to produce a valid $p_i$ then Chriss (1997) suggests the following more drastic measure in which

$$S_{i+1} = F_i + \epsilon$$

(10.122)

where $\epsilon$ is a very small number (say $10^{-6}$). It can be seen from Equation 10.114 that the transition probability $p_i$ will then be a very small positive number.

When we remove bad probabilities in this manner the impact on the implied lattice will depend on both the Arrow-Debreu price of the node and its payout. Nodes near the top and bottom of the lattice will have small Arrow-Debreu prices because few paths lead to them, and thus removing bad probabilities from these nodes will have little impact on the lattice. When building an implied lattice it is a good idea to count how many bad nodes have been encountered; this will give some idea of the expected quality of the implied lattice that has been constructed. A more quantitative method of assessing the expected performance of an implied lattice is by checking how well it prices the put and call options that were originally used to create it.

### Example calculation

Here we provide more details concerning the example calculation given in the paper by Derman and Kani (1994). The implied lattice for this example is shown in Figures 10.7 and 10.8. It is assumed that the current stock value is 100.00, the dividend is zero, and the annually compounded riskless interest rate is 3 per cent a year for all option maturities. Since we have assumed a constant riskless interest of 3 per cent the forward price $F_i$ for any node is 1.03 times the node’s stock price, $s_i$.

### Computation of the Arrow-Debreu prices

We have already mentioned that the Arrow-Debreu price for node $(n, i)$ is computed as the sum, over all paths from the root node $(1, 1)$ to node $(n, i)$, of the product of the riskless-discounted transition probabilities of nodes along each path from $(1, 1)$
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Figure 10.7  Implied binomial lattice showing the stock values at each node; from DK Figure 6

Figure 10.8  An implied lattice showing the Arrow-Debreu prices (in bold) and also the transition probabilities between nodes (in a smaller font); from DK Figure 6.
to \((n, i)\). Here we provide more detail and show how the Arrow-Debreu prices can be computed for the first four lattice levels.

Level 1: Node \((1, 1)\) \(\lambda_1 = 1.0\).

Level 2: Node \((2, 2)\): There is only one route from node \((1, 1)\) to node \((2, 2)\) and path probability is 0.625. Discounting the path probability by the riskless rate of 3 per cent gives the Arrow-Debreu price:

\[
\lambda_2 = \frac{0.625}{1.03} = 0.6068
\]

Node \((2, 1)\): As for node \((2, 2)\) there is only one route from node \((1, 1)\) and the path probability is \(1 - 0.625 = 0.375\). Discounting by the riskless rate gives the Arrow-Debreu price:

\[
\lambda_1 = \frac{0.375}{1.03} = 0.3641
\]

Level 3: Node \((3, 3)\): There is only one route from node \((1, 1)\) to node \((3, 3)\) and the path probability is \(0.625 \times 0.625 = 0.42625\). Discounting by the riskless rate of 3 per cent over two time steps yields the Arrow-Debreu price:

\[
\lambda_3 = \frac{0.42625}{1.03 \times 1.03} = 0.40178
\]

Node \((3, 2)\): There are two ways of going from node \((1, 1)\) to node \((3, 2)\). The first way, route (a) includes the nodes \((1, 1), (2, 2),\) and \((3, 2)\); it has a path probability of \(0.625 \times 0.318 = 0.19875\). The contribution of route (a) to \(\lambda_2\), denoted as \(\lambda_a^2\), is therefore \(\lambda_a^2 = 0.19875/(1.03 \times 1.03) = 0.18734\). The second way, route (b) includes the nodes \((1, 1), (2, 1),\) and \((3, 2)\); it has a path probability of \(0.375 \times 0.671 = 0.2516\). The contribution of route (b) to \(\lambda_2\), denoted as \(\lambda_b^2\), is thus \(\lambda_b^2 = 0.2516/(1.03 \times 1.03) = 0.23718\).

The Arrow-Debreu price, \(\lambda_2\), is therefore:

\[
\lambda_2 = \lambda_a^2 + \lambda_b^2 = 0.18734 + 0.23718 = 0.4245
\]

Node \((3, 1)\): There is only one route from node \((1, 1)\) to node \((3, 1)\), and the path probability is \(0.375 \times 0.329 = 0.12337\). Discounting by the riskless rate of 3 per cent gives an Arrow-Debreu price:

\[
\lambda_1 = \frac{0.12337}{1.03 \times 1.03} = 0.11629
\]

Level 4: Node \((4, 4)\): There is only one route from node \((1, 1)\) to node \((4, 4)\), and the path probability is \(0.625 \times 0.682 \times 0.682 = 0.29070\). Discounting by the riskless rate of 3 per cent over three time steps gives an Arrow-Debreu price:

\[
\lambda_4 = \frac{0.26768}{1.03 \times 1.03 \times 1.03} = 0.2660
\]

Node \((4, 3)\): There are three ways of going from node \((1, 1)\) to node \((4, 2)\). The first way, route (a) includes the nodes \((1, 1), (2, 2), (3, 3),\) and \((4, 3)\). The second way, route (b) includes the nodes \((1, 1), (2, 2), (3, 2),\) and \((4, 3)\). Finally the third way, route (c) includes the nodes \((1, 1), (2, 1), (3, 2),\) and \((4, 2)\).
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The path probability for route (a) is \(0.625 \times 0.682 \times 0.316 = 0.1347\), and thus yields an Arrow-Debreu price \(\lambda_1^a = 0.1347 / (1.03 \times 1.03 \times 1.03) = 0.1233\). The path probability for route (b) is \(0.375 \times 0.671 \times 0.624 = 0.12402\), and gives an Arrow-Debreu price of \(\lambda_1^b = 0.12402 / (1.03 \times 1.03 \times 1.03) = 0.1135\).

Finally the path probability for route (c) is \(0.375 \times 0.671 \times 0.624 = 0.157014\), and \(\lambda_1^c = 0.157014 / (1.03 \times 1.03 \times 1.03) = 0.1437\). The Arrow-Debreu price, \(\lambda_1\), is therefore:

\[
\lambda_1 = \lambda_1^a + \lambda_1^b + \lambda_1^c = 0.1233 + 0.1135 + 0.1437 = 0.3805
\]

Node (4, 2): There are three ways of going from node (1, 1) to node (3, 2). The first way, route (a) includes the nodes (1, 1), (2, 1), (3, 2), and (4, 2). The second way, route (b) includes the nodes (1, 1), (2, 1), (3, 2), and (4, 2). Lastly, the third way, route (c) includes the nodes (1, 1), (2, 2), (3, 2), and (4, 2).

The path probability for route (a) is \(0.625 \times 0.318 \times 0.376 = 0.07473\), and yields \(\lambda_2^a = 0.07473 / (1.03 \times 1.03 \times 1.03) = 0.06838\). The path probability for route (b) is \(0.375 \times 0.671 \times 0.376 = 0.09461\), and yields \(\lambda_2^b = 0.09461 / (1.03 \times 1.03 \times 1.03) = 0.0865824\).

Finally the path probability for route (c) is \(0.375 \times 0.329 \times 0.541 = 0.06674\), and yields \(\lambda_2^c = 0.06674 / (1.03 \times 1.03 \times 1.03) = 0.06108\). The Arrow-Debreu price, \(\lambda_2\), is therefore:

\[
\lambda_2 = \lambda_2^a + \lambda_2^b + \lambda_2^c = 0.06838 + 0.0865824 + 0.06108 = 0.21604
\]

Node (4, 1): There is only one route from node (1, 1) to node (4, 1), and the path probability is \(0.375 \times 0.329 \times 0.459 = 0.05663\). This gives an Arrow-Debreu price:

\[
\lambda_3 = \frac{0.05663}{1.03 \times 1.03 \times 1.03} = 0.0518
\]

An alternative and simpler method of obtaining the Arrow-Debreu prices for the nodes in a particular lattice level is to use forward iteration. Here we can use the fact that the Arrow-Debreu prices for the nodes in a particular level are related, in the usual binomial fashion, to the values in the previous level and the set of transition probabilities between levels. Since the Arrow-Debreu price for root node (1, 1) is (by definition) 1, and we know how to compute the transition probabilities between levels, all the Arrow-Debreu prices in the \((n+1)\)th lattice level can be computed from those in the \(n\)th lattice level.

We will now illustrate this, by showing how to compute the Arrow-Debreu prices in lattice level 4, \(\lambda_i\), \(i = 1, \ldots, 4\), from the previously computed Arrow-Debreu prices \(\lambda_i^r\), \(i = 1, \ldots, 3\), in level 3.

\[
\lambda_4 = \frac{p_3 \times \lambda_3^r}{1 + r} = \frac{0.682 \times 0.402}{1.03} = 0.266
\]

\[
\lambda_3 = \frac{(1 - p_3) \times \lambda_3^r + p_2 \lambda_2^r}{1 + r} = \frac{0.402 \times 0.318 + 0.425 \times 0.624}{1.03} = 0.381
\]
\[
\lambda_2 = \frac{(1 - p_2) \times \lambda_2^2 + p_1 \lambda_1^2}{1 + r} = \frac{0.425 \times 0.376 + 0.116 \times 0.541}{1.03} = 0.216
\]

\[
\lambda_1 = \frac{(1 - p_1) \lambda_1^2}{1 + r} = \frac{0.459 \times 0.116}{1.03} = 0.052
\]

As a means of checking the computed Arrow-Debreu prices we can use the fact that, at any lattice level, the sum of the Arrow-Debreu prices inflated at the riskless interest rate to the root node is 1. That is for \( n \)th lattice level we have:

\[
(1 + r)^{n-1} \sum_{i=1}^{n} \lambda_i = 1 \tag{10.123}
\]

where \( r \) is the (constant) riskless interest rate. If we take into account finite computational precision then Equation 10.123 becomes:

\[
\text{ABS} \left\{ \left( (1 + r)^{n-1} \sum_{i=1}^{n} \lambda_i \right) - 1 \right\} \leq \text{tol} \tag{10.124}
\]

where \( \text{ABS}\{X\} \) denotes the absolute value of \( X \), and \( \text{tol} \) is a small number which reflects the computational accuracy. For nodes on level 4 we have:

\[
\text{ABS}\left\{ (1.03)^3 (0.266 + 0.381 + 0.216 + 0.052) - 1 \right\} \sim 1.5 \times 10^{-4}
\]

10.5.2 Barle–Cakici implied lattice

Here we briefly describe modifications proposed by Barle and Cakici (1995), henceforth denoted BC, to the algorithm used by Derman and Kani for constructing the implied lattice. These improvements reduce the occurrence of bad transition probabilities and thus lead to better quality lattices.

**First modification**

The first modification proposed by BC is to use \( F_i \) (the forward of \( s_i \)) for the strike price, \( K \), in Equation 10.109 instead of \( s_i \). Under these circumstances Equation 10.115 (DK equation 7) becomes:

\[
S_{i+1} = \frac{S_i \left\{ C(F_i, t_{n+1}) \exp(r \Delta t) - \Sigma^{BC} \right\} - \lambda_i F_i (F_i - S_i)}{\left\{ C(F_i, t_{n+1}) \exp(r \Delta t) - \Sigma^{BC} \right\} - \lambda_i (F_i - S_i)} \tag{10.125}
\]

where

\[
\Sigma^{BC} = \sum_{j=i+1}^{n} \lambda_j (F_j - F_i)
\]
Second modification

The second modification is to allow the central spine of the implied lattice to follow the values dictated by the prevailing interest rate. If the \((n+1)\)th lattice level is odd this involves setting the central node to \(S \exp \left( \frac{r-q}{C_0} t_{n+1} \right)\), where \(q\) is the continuous dividend yield, and the other symbols have already been defined in the previous section on the DK lattice. If the \((n+1)\)th lattice level is even then the two central nodes no longer satisfy Equation 10.116 but

\[
S_i S_{i+1} = F_i^2
\]

where \(i = (n+1)/2\).

The asset price at the lower central node \(S_i\) is then given by:

\[
S_i = \frac{F_i \left( \lambda_i F_i - \Psi^{BC} \right)}{\lambda_i F_i + \Psi^{BC}}
\]

whereas that at the upper central node \(S_{i+1}\) is:

\[
S_{i+1} = \frac{F_i \left( \Psi^{BC} + \lambda_i F_i \right)}{\lambda_i F_i - \Psi^{BC}}
\]

where \(\Psi^{BC} = C(F_i, t_{n+1}) \exp (r \Delta t) - \overline{\Sigma}^{BC}\).

Proof of Equation 10.127 (BC equation 9) and Equation 10.128

From Equation 10.114 we have that the transition probability, \(p_i\) is:

\[
p_i = \frac{F_i - S_i}{S_{i+1} - S_i}
\]

multiplying above and below by \(S_{i+1}\) then gives:

\[
p_i = \frac{(F_i - S_i) S_{i+1}}{(S_{i+1} - S_i) S_{i+1}}
\]

However since we are centering at the forward price, from Equation 10.126, we have \(S_i S_{i+1} = F_i^2\), and so

\[
p_i = \frac{F_i S_{i+1} - F_i^2}{S_{i+1}^2 - F_i^2} = \frac{F_i S_{i+1} - F_i^2}{(S_{i+1} - F_i)(S_{i+1} + F_i)}
\]

However from Equation 10.118 we have

\[
\Psi^{BC} = \lambda_i p_i (S_{i+1} - F_i)
\]

If we substitute the value of \(p_i\) from Equation 10.129 into Equation 10.130 we have

\[
\Psi^{BC} = \lambda_i p_i (S_{i+1} - F_i) = \frac{\lambda_i (F_i S_{i+1} - F_i^2)(S_{i+1} - F_i)}{(S_{i+1} - F_i)(S_{i+1} + F_i)} = \frac{\lambda_i (F_i S_{i+1} - F_i^2)}{S_{i+1} + F_i}
\]
Rearranging we obtain:

\[ \Psi^{BC} S_{i+1} + \Psi^{BC} F_i = \lambda_i F_i S_{i+1} - \lambda_i F_i^2 \]

\[ S_{i+1}(\Psi^{BC} - \lambda_i F_i) = -F_i(\Psi^{BC} + \lambda_i F_i) \]

which gives

\[ S_{i+1} = \frac{F_i(\Psi^{BC} + \lambda_i F_i)}{\lambda_i F_i - \Psi^{BC}} \quad \text{QED} \]

To prove Equation 10.127 we simply substitute \( S_{i+1} = F_i^2 \) into Equation 10.128 and obtain:

\[ S_i = \frac{F_i^2}{S_{i+1}} = \frac{F_i^2(\lambda_i F_i - \Psi^{BC})}{F_i(\Psi^{BC} + \lambda_i F_i)} \]

So

\[ S_i = \frac{F_i(\lambda_i F_i - \Psi^{BC})}{\lambda_i F_i + \Psi^{BC}} \quad \text{QED} \]

**Bad probabilities**

If bad transition probabilities occur then this can be rectified by setting \( S_{i+1} \) to any value between \( F_i \) and \( F_{i+1} \). In these circumstances Barle and Cakici suggest setting \( S_{i+1} \) to the average of \( F_i \) and \( F_{i+1} \).

### 10.5.3 Chriss implied lattice

Here we will briefly mention an implied lattice, devised by Chriss (1996), which can be built using the market values of both European and American options. This is in contrast to the algorithm of Derman–Kani which requires the market values of European options. We will not describe how to deal with American options; the reader can refer to the original paper which is available on the CD ROM. The first part of the paper, is concerned with European options and follows on from our previous discussions concerning the Derman–Kani and Barle–Cakici implied lattices.

As supplementary information we will now show how to derive equation (3) in the original paper, that is:

\[ u = \frac{\nu^{PUT} + K}{K \exp(-r\Delta t) - \nu^{PUT}} \]

(10.131)

the notation used here is the same as that in Chriss (1996).
Proof of Equation 10.131 (Chriss equation 3)
The transition probability of an up jump from $S_{i-1,j}$ to $S_{i,j+1}$ is denoted by $p_j$, and that of the corresponding down jump transition probability $1 - p_j$ by $q$. The forward for $S_{i-1,j}$ is denoted by $F_j$ and, since $S_{i-1,j} = K$, we have:

$$F_j = S_{i-1,j} \exp(r\Delta t) = K \exp(r\Delta t)$$ (10.132)

The up jump transition probability, see Equation 10.114, is

$$p_j = \frac{F_j - S_{i,j}}{S_{i,j+1} - S_{i,j}}$$

which results in a down jump probability of

$$1 - p_j = q = 1 - \frac{F_j - S_{i,j}}{S_{i,j+1} - S_{i,j}} = \frac{S_{i,j+1} - F_j}{S_{i,j+1} - S_{i,j}}$$

Multiplying top and bottom by $S_{i,j}$ we obtain

$$q = \frac{S_{i,j+1} - F_j}{S_{i,j+1} - S_{i,j}} = \frac{(S_{i,j+1} - F_j)S_{i,j}}{(S_{i,j+1} - S_{i,j})S_{i,j}} = \frac{S_{i,j+1}S_{i,j} - F_jS_{i,j}}{S_{i,j+1}S_{i,j} - S_{i,j}^2}$$

We choose to centre at the spot $S_{i,j+1}S_{i,j} = K^2$ and we have

$$q = \frac{S_{i,j+1}S_{i,j} - F_jS_{i,j}}{K^2 - S_{i,j}^2} = \frac{S_{i,j+1}S_{i,j} - F_jS_{i,j}}{(K - S_{i,j})(K + S_{i,j})} = \frac{K^2 - F_jS_{i,j}}{(K - S_{i,j})(K + S_{i,j})}$$ (10.133)

From the derivation of equation (1), on the first page of the original paper by Chriss, we have:

$$\nu_{i-1,j}^{PUT} = q(K - S_{i,j})\exp(-r\Delta t)$$ (10.134)

We now use Equation 10.133 to substitute for $q$ in Equation 10.134. This gives

$$\nu_{i-1,j}^{PUT} = q(K - S_{i,j})\exp(-r\Delta t) = \frac{\exp(-r\Delta t)(K^2 - F_jS_{i,j})}{K + S_{i,j}}$$

using $F_j = K \exp(r\Delta t)$ from Equation 10.132 results in

$$\nu_{i-1,j}^{PUT} = \frac{K(K\exp(-r\Delta t) - S_{i,j})}{K + S_{i,j}}$$ (10.135)

and multiplying both sides of Equation 10.135 by $K + S_{i,j}$ we obtain

$$\nu_{i-1,j}^{PUT}K + \nu_{i-1,j}^{PUT}S_{i,j} = K^2 \exp(-r\Delta t) - KS_{i,j}$$ (10.136)

Since we centre at the spot we have:

$$S_{i,j}S_{i,j+1} = K^2$$
so

\[ S_{t,j+1} = Ku = \frac{K^2}{S_{t,j}}, \]

which gives:

\[ u = \frac{K}{S_{t,j}} \quad (10.137) \]

Finally, from Equation 10.136, we have

\[ K^2 \exp(-r\Delta T) - Kv_{i-1,j}^{PUT} = \left( v_{i-1,j}^{PUT} + K \right) S_{t,j} \]

so

\[ \frac{1}{S_{t,j}} = \frac{v_{i-1,j}^{PUT} + K}{K^2 \exp(-r\Delta t) - Kv_{i-1,j}^{PUT}} \]

This results in

\[ \frac{K}{S_{t,j}} = u = \frac{\left( v_{i-1,j}^{PUT} + K \right) K}{K^2 \exp(-r\Delta t) - Kv_{i-1,j}^{PUT}} \]

or

\[ u = \frac{v_{i-1,j}^{PUT} + K}{K \exp(-r\Delta t) - v_{i-1,j}^{PUT}} \quad QED \]

More information concerning the Chriss implied lattice, and other types of implied lattices, can be found in Chriss (1997).

### 10.6 GRID METHODS FOR VANILLA OPTIONS

#### 10.6.1 Introduction

In Section 10.4 we discussed the use of binomial lattice methods for valuing both European and American options. The lattice methods we described have the advantage that they are fairly easy to implement and can value simple options, such as vanilla puts and calls, reasonably accurately. The use of up and down jump probabilities at the lattice nodes is also an appealing feature, since they are directly related to the stochastic process which is being modelled. However, lattice techniques have the following drawbacks:

- They require small time steps to ensure numerical stability.
- There is little control over where the lattice nodes are located. This can lead to very poor accuracy when valuing certain types of options; for example those with barriers at particular asset prices.
One method of avoiding these limitations is through the use of finite-difference grids. Although this approach no longer has the probabilistic interpretation of the binomial lattice it has the following advantages:

- Fewer time steps are required to ensure numerical stability, see Appendix L for a discussion of stability.
- There is complete control over the placement of grid lines, and their associated grid nodes.

10.6.2 Uniform grids

The Black–Scholes equation for the value of an option, \( f \) is given by:

\[
\frac{\partial f}{\partial t} + (r - q)S \frac{\partial f}{\partial S} + \frac{1}{2} \sigma^2 S^2 \frac{\partial^2 f}{\partial S^2} = rf
\]  

(10.138)

We want to solve this equation over the duration of the option, that is from the current time \( t \) to the maturity of the option at time \( t + \tau \). To do this we will use a grid in which the asset price \( S \) takes \( n_s \) uniformly spaced values, \( S_j = j\Delta S \), \( j = 0, \ldots, n_s - 1 \), where \( \Delta S \) is the spacing between grid points. If \( S_{\text{max}} \) is the maximum asset value we want to represent then the grid spacing, \( \Delta S^* \), can be simply calculated as:

\[
\Delta S^* = \frac{S_{\text{max}}}{(n_s - 1)}
\]  

(10.139)

However, since we would like to solve the option values and Greeks at the current asset price \( S_0 \) we would also like an asset grid line to coincide with the current asset price, see Andersen and Brotherton-Ratcliffe (1998). This avoids the use of interpolation which is necessary when the asset value does not correspond to a grid line. The method by which we achieve this is outlined in Code excerpt 10.12. Here the user supplies the function \( \text{opt}_\tau \text{-gfd} \) with values for \( S_{\text{max}} \) and \( n_s - 1 \) from which \( \Delta S^* \) is computed using Equation 10.139. We then find the integer, \( n_1 \), that is just below (or equal to) the value \( S_0/\Delta S^* \), and use this to obtain a new grid spacing \( \Delta S = S_0/n_1 \). This leads to the new asset price discretization \( S_j = j\Delta S \), \( j = 0, \ldots, n_s - 1 \), where we have now ensured that \( S_n = S_0 \).

The user also supplies the function \( \text{opt}_\tau \text{-gfd} \) with the number of time intervals for the grid. When there are \( n_t \) time intervals the grid has \( n_t + 1 \) uniformly spaced time instants, \( t_i = i\Delta t, i = 0, \ldots, n_t \), and the time step is simply:

\[
\Delta t = \frac{\tau}{n_t}
\]  

(10.140)

As with the binomial lattice methods of Sections 10.4 and 10.5 we will solve the equation backwards in time from maturity (at time \( t + \tau \)) to the present (time \( t \)). So as we solve the equation the time index will start at \( i = n_t \) (time \( t + \tau \)) and decrease to \( i = 0 \) (current time \( t \)).

Here we discuss the grid method of solving the Black–Scholes equation in terms of:

- The finite-difference approximation.
- The boundary conditions.
Computation of the option values at a given time instant.

Backwards iteration and early exercise.

Each of these aspects will now be considered in turn.

**The finite-difference approximation**

The option value corresponding to the grid node at which \( t_i = i \Delta t \) and \( S_j = j \Delta S \) will be denoted by \( f_{i,j} \). We will approximate the partial derivative of \( f_{i,j} \) w.r.t. time simply as:

\[
\frac{\partial f}{\partial t} = \frac{f_{i+1,j} - f_{i,j}}{\Delta t}
\]  

(10.141)

For the other terms in Equation 10.138 we will use the weighted, \( \Theta_m \), method. This technique involves selecting an appropriate choice for \( \Theta_m \) in the range \( 0 \leq \Theta_m \leq 1 \) so that the contribution from node \((i, j)\) is a weighted sum involving the values at nodes \((i, j)\) and \((i+1, j)\). For instance the term \( r f_{i,j} \) in Equation 10.138 is approximated as:

\[
rf_{i,j} = r \left( \Theta_m f_{i+1,j} + (1 - \Theta_m) f_{i,j} \right)
\]  

(10.142)

and the term \( \partial f / \partial S \) in Equation 10.138 is approximated as:

\[
\frac{\partial f}{\partial S}_{i,j} = \left\{ \Theta_m \frac{\partial f}{\partial S}_{i+1,j} + (1 - \Theta_m) \frac{\partial f}{\partial S}_{i,j} \right\}
\]  

(10.143)

Using this method we thus obtain, at node \((i, j)\), the following discretized version of Equation 10.138:

\[
\frac{f_{i+1,j} - f_{i,j}}{\Delta t} + (r - q)S_j \left\{ \Theta_m f'_{i+1,j} + \Theta_m^* f''_{i,j} \right\}
\]

\[
+ \frac{1}{2} \sigma^2 S_j^2 \left\{ \Theta_m f''_{i+1,j} + \Theta_m^* f''_{i,j} \right\} = r \left\{ \Theta_m f_{i+1,j} + \Theta_m^* f_{i,j} \right\}
\]  

(10.144)

where for compactness we have written \( \Theta_m^* = 1 - \Theta_m \), and denote the partial derivatives w.r.t. \( S \) at node \((i, j)\) as: \( f'_{i,j} = \partial f / \partial S_{i,j} \) and \( f''_{i,j} = \partial^2 f / \partial S^2_{i,j} \).

Finite-difference approximations for these derivatives can be obtained by considering a Taylor expansion about the point \( f_{i,j} \). We proceed as follows:

\[
f_{i,j+1} = f_{i,j} + f'_{i,j} \Delta S + \frac{1}{2} f''_{i,j}(\Delta S)^2
\]  

(10.145)

\[
f_{i,j-1} = f_{i,j} - f'_{i,j} \Delta S + \frac{1}{2} f''_{i,j}(\Delta S)^2
\]  

(10.146)

Subtracting Equations 10.145 and 10.146 we obtain:

\[
f_{i,j+1} - f_{i,j-1} = 2f'_{i,j} \Delta S
\]

and so

\[
f'_{i,j} = \frac{f_{i,j+1} - f_{i,j-1}}{2 \Delta S}
\]  

(10.147)
180  Pricing Assets

Adding Equations 10.145 and 10.146 we obtain:

\[ f_{i,j+1} + f_{i,j-1} = 2f_{i,j} + f''_{i,j} \Delta S^2 \]

which gives:

\[ f''_{i,j} = \frac{f_{i,j+1} - 2f_{i,j} + f_{i,j-1}}{\Delta S^2} \]  \(10.148\)

The complete finite-difference approximation to the Black–Scholes equation can then be found by substituting the approximations for the first and second partial derivatives, given in Equations 10.147 and 10.148, into Equation 10.144. We thus obtain:

\[ r\Delta t \{ \Theta_m f_{i+1,j} + \Theta^*_{m,k} f_{i,j} \} = f_{i+1,j} - f_{i,j} + \frac{(r-q)j\Delta t A_1}{2} + \frac{\sigma^2 j^2 \Delta t A_2}{2} \]  \(10.149\)

where we have used the fact that \(S_j = j\Delta S\), and for compactness have defined the terms:

\[ A_1 = \Theta_m f_{i+1,j+1} - \Theta_m f_{i+1,j-1} + \Theta^*_{m,k} f_{i,j+1} - \Theta^*_{m,k} f_{i,j-1} \]

and

\[ A_2 = \Theta_m f_{i+1,j+1} + \Theta_m f_{i+1,j-1} - 2\Theta_m f_{i+1,j} + \Theta^*_{m,k} f_{i,j+1} + \Theta^*_{m,k} f_{i,j-1} - 2\Theta^*_{m,k} f_{i,j} \]

Collecting like terms in \(f_{i,j}, f_{i+1,j}\), etc. results in:

\[ B_1 f_{i,j-1} + B_2 f_{i,j} + B_3 f_{i,j+1} + C_1 f_{i+1,j-1} + C_2 f_{i+1,j} + C_3 f_{i+1,j+1} = 0 \]  \(10.150\)

where

\[ B_1 = -\frac{\Theta^*_{m,k}(r-q)j\Delta t}{2} + \frac{\Theta^*_{m,k} \sigma^2 j^2 \Delta t}{2} \]

\[ B_2 = -1 - r\Delta t \Theta_m - \Theta^*_{m,k} \sigma^2 j^2 \Delta t \]

\[ B_3 = \frac{\Theta^*_{m,k}(r-q)j\Delta t}{2} + \frac{\Theta^*_{m,k} \sigma^2 j^2 \Delta t}{2} \]

\[ C_1 = \frac{\Theta_m \sigma^2 j^2 \Delta t}{2} - \frac{\Theta_m (r-q)j\Delta t}{2} \]

\[ C_2 = 1 - r\Delta t \Theta_m - \Theta_m \sigma^2 j^2 \Delta t \]

\[ C_3 = \frac{\Theta_m (r-q)j\Delta t}{2} + \frac{\Theta_m \sigma^2 j^2 \Delta t}{2} \]

Since we are solving the equation backwards in time we want to determine the option values at time index \(i\) from the known option values \((f_{i+1,j+1}, f_{i+1,j}) and
For each value of $j$ Equation 10.151 gives us a relationship between three option values, $f_{i+1,j-1}, f_{i+1,j}, f_{i+1,j+1}$ at time index $i+1$, and three option values $f_{i,j-1}, f_{i,j}, f_{i,j+1}$ at time index $i$.

This situation is shown in Figure 10.9 where we have labelled the grid nodes that contribute to the option value $f_{5,5}$ at grid node E. These are the known option values node A: $f_{6,6}$, node B: $f_{6,5}$ and node C: $f_{6,4}$ and the unknown option values, node D: $f_{5,6}$, node E: $f_{5,5}$ and node F: $f_{5,4}$.

Before we solve Equation 10.151 we will briefly consider its characteristics for different values of the weight parameter $\Theta_m$.

When $\Theta_m = 1$ the values of the coefficients in Equation 10.151 are $a_j = c_j = 0$ and $b_j = 1$. This means that Equation 10.151 reduces to:

$$f_{i,j} = a_{i,j}f_{i+1,j-1} + b_{i,j}f_{i+1,j} + c_{i,j}f_{i+1,j+1}$$

This is termed the explicit method, and it can be seen that the unknown option value $f_{i,j}$, at the grid node $(i,j)$ is just a weighted sum of the (known) option values $f_{i+1,j-1}, f_{i+1,j}, f_{i+1,j+1}$. This is the simplest situation to deal with and actually corresponds to a trinomial lattice. However, it has poor numerical properties and usually requires a very small step size to obtain accurate results, see Smith (1985).

When $\Theta_m \neq 1$, the unknown option value $f_{i,j}$ depends not only on the known option values $f_{i+1,j-1}, f_{i+1,j}, f_{i+1,j+1}$ (as in the explicit method above), but also on the
neighbouring unknown option values $f_{i,j-1}$ and $f_{i,j+1}$. It is now necessary to solve a set of simultaneous in order to compute the value $f_{i,j}$. This is therefore called an implicit method, see Smith (1985).

The implicit method $\Theta_m = 0$ is also called the fully implicit method, since now the unknown value $f_{i,j}$ only depends on the neighbouring values $f_{i,j-1}$, $f_{i,j+1}$ and its previous value, $f_{i+1,j}$, at time step $i + 1$. This can be shown by substituting $\Theta_m = 0$ in Equations 10.153 to 10.158. We then obtain $a_j = c_j = 0$ and $b_j = 1$, which means that Equation 10.151 reduces to:

$$a_j f_{i,j-1} + b_j f_{i,j} + c_j f_{i,j+1} = f_{i+1,j}$$

The implicit method $\Theta_m = 0.5$, is also termed the Crank–Nicolson method. This method, first used by Crank and Nicolson in 1946, see Crank and Nicolson (1947), computes $f_{i,j}$ by giving equal weight to the contributions from time step $i + 1$ and

Figure 10.9 An example uniform grid, which could be used to estimate the value of a vanilla option which matures in two years time. The grid parameters are: $n_s = n_t = 10$, $\Delta t = 0.2$, $\Delta S = 5$, and $S_{\text{max}} = 50$. The option parameters are $E = 25$, $S_0 = 20$, and $\tau = 2.0$. As usual we denote the grid node option values by $f_{i,j}$, where $i$ is the time index and $j$ is the asset index. The option values of the grid nodes at maturity for a vanilla put are thus labelled as $\text{val}$, $f_{10,i}$, $j = 0, \ldots, 10$, where $\text{val}$ is the value of the option at the node; these are shown on the right hand grid boundary. Since $E = 25$ only those nodes with $j < 5$ have nonzero option values.
time step $i$. Substituting $\Theta_m = 0.5$ in Equations 10.153 to 10.149 we obtain the following Crank–Nicolson coefficients:

$$a_j = -\bar{a}_j = \frac{\Delta t}{4} \{ (r - q) j - \sigma^2 j^2 \}$$

$$b_j = 1 + \frac{\Delta t}{2} \{ r + \sigma^2 j^2 \}$$

$$\bar{b}_j = 1 - \frac{\Delta t}{2} \{ r + \sigma^2 j^2 \}$$

$$c_j = -\bar{c}_j = -\frac{\Delta t}{4} \{ (r - q) j + \sigma^2 j^2 \}$$

We notice that since we are solving backwards in time, but index time in the forward direction, our values of $\Theta_m$ corresponding to implicit and explicit are different from those normally used. For example in Smith (1985) $\Theta_m = 0$ is the explicit method and $\Theta_m = 1$ is the implicit method; the Crank–Nicolson method is still $\Theta_m = 0.5$.

### The boundary conditions

In order to solve Equation 10.151 at time instant $i\Delta t$ we need to obtain the option values at the upper asset boundary, the lower asset boundary and the initial values that are specified at option maturity.

Here we calculate the boundary values by using the time independent payoff, $p_j$, at the $j$th asset index within the grid. If $E$ is the strike price then vanilla call options have payoffs

$$p_j = \max(j\Delta S - E, 0), j = 0, \ldots, n_{t-1}$$

and vanilla put options have payoffs

$$p_j = \max(E - j\Delta S, 0), j = 0, \ldots, n_{t-1}$$

**Upper asset boundary values.** At the upper boundary $j = n_{t} - 1$, and $(n_{t} - 1)\Delta S = S_{\text{max}}$; where we note that for the grid to be useful we require $S_{\text{max}} > E$. Here we assume that $S_{\text{max}} > E$ and so for call options

$$p_{n_{t-1}} = S_{\text{max}} - E$$

and for put options

$$p_{n_{t-1}} = 0$$

The option value at the upper boundary, denoted by $f_{BU}$, is set to $p_{n_{t-1}}$, and we have $f_{i, n_{t-1}} = f_{BU}$, $i = 0, \ldots, n_{t}$.

**Lower asset boundary values.** At the lower boundary $j = 0$, and the value of $j\Delta S$ is zero. So for call options

$$p_0 = 0$$
and for put options

\[ p_0 = E \]

The option value at the lower boundary, denoted by \( f_{BL} \), is set to \( p_0 \), and we have \( f_{i,0} = f_{BL}, \ i = 0, \ldots, n_t \).

**Boundary values at option maturity.** At option maturity \((i = n_t)\) the initial option (boundary) values are the previously mentioned payouts. If \( E \) is the strike price then for vanilla call options

\[ f_{n_t,j} = \max(j\Delta S - E, 0), \quad j = 0, \ldots, n_{s-1} \]

and for vanilla put options

\[ f_{n_t,j} = \max(E - j\Delta S, 0), \quad j = 0, \ldots, n_{s-1} \]

This is illustrated in Figure 10.9 for a vanilla put option with current asset value \( S_0 = 20 \), strike, \( E = 25 \) and maturity \( \tau = 2 \). The grid asset price spacing is \( \Delta S = 5 \), and the time increment is \( \Delta t = 0.2 \). At option maturity, corresponding to time index \( i = 10 \), the value of the put option is zero for all asset indices \( j \geq 5 \).

**Computation of the option values at a given time instant**

Having found the option boundary values we are now in a position to solve Equation 10.151 at time instant \( t_i = i\Delta t \).

First we note that since \( f_{i,0} = f_{BL} \) and \( f_{i,n_{s-1}} = f_{BU} \), Equation 10.151 only needs to be solved for values of the asset index \( j \) in the range \( j = 1 \) to \( j = n_{s-2} \). We now deal with the following situations:

- **CASE 1**: \( j = 1 \), the asset grid line just above the lower boundary.
- **CASE 2**: \( j = n_{s-2} \), the asset grid line just below the upper boundary.
- **CASE 3**: all other asset grid lines not included in **CASE 1** or **CASE 2**.

and consider the form that Equation 10.151 takes under each condition.

**CASE 1**: \( j = 1 \)

Substituting \( j = 1 \), into Equation 10.151 we obtain:

\[ a_1 f_{i,0} + b_1 f_{i,1} + c_1 f_{i,2} = \bar{a}_1 f_{i+1,0} + \bar{b}_1 f_{i+1,1} + \bar{c}_1 f_{i+1,2} \]

Now, since \( f_{i,0} = f_{BL} \), this becomes:

\[ b_1 f_{i,1} + c_1 f_{i,2} = (\bar{a}_1 - a_1) f_{BL} + \bar{b}_1 f_{i+1,1} + \bar{c}_1 f_{i+1,2} \]

or equivalently:

\[ b_1 f_{i,1} + c_1 f_{i,2} = R_{i+1,1} \quad (10.159) \]

where

\[ R_{i+1,1} = (a_1 - \bar{a}_1) f_{BL} + \bar{b}_1 f_{i+1,1} + \bar{c}_1 f_{i+1,2} \quad (10.160) \]
CASE 2: \( j = n_s - 2 \)

Substituting \( j = n_s - 1 \) into Equation 10.151 we obtain:
\[
a_{n_s-2} f_{i,n_s-3} + b_{n_s-2} f_{i,n_s-2} + c_{n_s-2} f_{i,n_s-1} = a_{n_s-2} f_{i+1,n_s-3} + b_{n_s-2} f_{i+1,n_s-2} + c_{n_s-2} f_{i+1,n_s-1}
\]

Since \( f_{i,n_s-1} = f_{BU} \) this gives:
\[
a_{n_s-2} f_{i,n_s-3} + b_{n_s-2} f_{i,n_s-2} = a_{n_s-2} f_{i+1,n_s-3} + b_{n_s-2} f_{i+1,n_s-2} + (c_{n_s-2} - c_{n_s-1}) f_{BU}
\]
or equivalently:
\[
a_{n_s-2} f_{i,n_s-3} + b_{n_s-2} f_{i,n_s-2} = R_{i+1,n_s-2}
\] (10.161)

where
\[
R_{i+1,n_s-2} = a_{n_s-2} f_{i+1,n_s-3} + b_{n_s-2} f_{i+1,n_s-2} + (c_{n_s-2} - c_{n_s-1}) f_{BU}
\] (10.162)

CASE 3

In this case the boundary values do not enter into the expressions, and we simply restate Equation 10.151 as:
\[
a_j f_{i,j-1} + b_j f_{i,j} + c_j f_{i,j+1} = R_{i+1,j}, \quad j = 3, \ldots, n_s - 3
\] (10.163)

where as before the right hand side, \( R_{i+1,j} \), is:
\[
R_{i+1,j} = a_j f_{i+1,j-1} + b_j f_{i+1,j} + c_j f_{i+1,j+1}
\] (10.164)

We can now gather all the information in Equations 10.159 to 10.164 and represent it by the following tridiagonal system:
\[
\begin{pmatrix}
  b_1 & c_1 & 0 & 0 & 0 & 0 \\
  a_2 & b_2 & c_2 & 0 & 0 & 0 \\
  0 & 0 & \ddots & \ddots & \ddots & 0 \\
  0 & 0 & 0 & a_{n-3} & b_{n-3} & c_{n-3} \\
  0 & 0 & 0 & 0 & a_{n-2} & b_{n-2}
\end{pmatrix}
\begin{pmatrix}
  f_{i,1} \\
  f_{i,2} \\
  \vdots \\
  f_{i,n_s-3} \\
  f_{i,n_s-2}
\end{pmatrix}
= \begin{pmatrix}
  R_{i+1,1} \\
  R_{i+1,2} \\
  \vdots \\
  R_{i+1,n_s-3} \\
  R_{i+1,n_s-2}
\end{pmatrix}
\] (10.165)

In matrix notation Equation 10.165 can be written as:
\[
Ax = R
\] (10.166)

where \( A \) is the \( n_s - 2 \times n_s - 2 \) tridiagonal matrix containing the known coefficients \( a_j, j = 2, \ldots, n_s - 2 \), \( b_j, j = 1, \ldots, n_s - 2 \), and \( c_j, j = 1, \ldots, n_s - 3 \). The vector \( R \) denotes the known right hand side, \( R_{i+1,j}, j = 1, \ldots, n_s - 2 \), and the vector \( x \) contains the unknown option values that we wish to compute, \( f_{i,j}, j = 1, \ldots, n_s - 2 \).

It is well known that, if matrix \( A \) is non-singular, Equation 10.166 can be solved using an \( LU \) decomposition. Here we factorize the \( n \times n \) matrix \( A \) as:
\[
A = LU
\]
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where $L$ is an $n \times n$ lower triangular matrix with 1s on the diagonal and $U$ is an $n \times n$ upper triangular matrix. We illustrate the $LU$ decomposition for a full $4 \times 4$ matrix below:

$$
\begin{pmatrix}
  a_{1,1} & a_{1,2} & a_{1,3} & a_{1,4} \\
  a_{2,1} & a_{2,2} & a_{2,3} & a_{2,4} \\
  a_{3,1} & a_{3,2} & a_{3,3} & a_{3,4} \\
  a_{4,1} & a_{4,2} & a_{4,3} & a_{4,4}
\end{pmatrix} =
\begin{pmatrix}
  1 & 0 & 0 & 0 \\
  l_{2,1} & 1 & 0 & 0 \\
  l_{3,1} & l_{3,2} & 1 & 0 \\
  l_{4,1} & l_{4,2} & l_{4,3} & 1
\end{pmatrix}
\begin{pmatrix}
  u_{1,1} & u_{1,2} & u_{1,3} & u_{1,4} \\
  0 & u_{2,2} & u_{2,3} & u_{2,4} \\
  0 & 0 & u_{3,3} & u_{3,4} \\
  0 & 0 & 0 & u_{4,4}
\end{pmatrix}
$$

(10.167)

If $A$ is a tridiagonal matrix then the $LU$ decomposition takes the simpler form:

$$
\begin{pmatrix}
  a_{1,1} & a_{1,2} & 0 & 0 \\
  a_{2,1} & a_{2,2} & a_{2,3} & 0 \\
  0 & a_{3,2} & a_{3,3} & a_{3,4} \\
  0 & 0 & a_{4,3} & a_{4,4}
\end{pmatrix} =
\begin{pmatrix}
  1 & 0 & 0 & 0 \\
  l_{2,1} & 1 & 0 & 0 \\
  0 & l_{3,2} & 1 & 0 \\
  0 & 0 & l_{4,3} & 1
\end{pmatrix}
\begin{pmatrix}
  u_{1,1} & u_{1,2} & 0 & 0 \\
  0 & u_{2,2} & u_{2,3} & 0 \\
  0 & 0 & u_{3,3} & u_{3,4} \\
  0 & 0 & 0 & u_{4,4}
\end{pmatrix}
$$

(10.168)

where it can be seen that now both $L$ and $U$ are bidiagonal.

Once the $LU$ decomposition of $A$ has been found it is possible to solve for $x$ in Equation 10.166 by using a two stage method (see for example Golub and Van Loan (1989)). Here forward elimination is used to solve $Ly = R$, and then back-substitution is applied to $UX = y$. We can thus write the procedure as:

$$AX = (LU)x = L(Ux) = Ly = R$$

We will now provide code excerpts which show how to solve the $n_{s-2} \times n_{s-2}$ tridiagonal system represented by Equation 10.166. These excerpts are in fact contained within the larger Code excerpt 10.18, which displays the complete C code for the option pricing function opt_gfd. If the reader requires more detail concerning the precise code used for option pricing then this code should be consulted. (It should be noted that in Code excerpt 10.18, time is indexed using $j$ and asset price using index $i$. We have modified the indices for the smaller code excerpts given below so that, as might be expected, time is indexed using $i$, and asset price using $j$. The author apologizes for any inconvenience this may cause.) Here, for brevity, we will assume that all the required arrays have already been allocated and loaded with the relevant information.

First we need to compute the $LU$ decomposition of the tridiagonal matrix $A$. The code to achieve this is given in Code excerpt 10.14 below. Here we use the following three arrays to store the elements of the tridiagonal matrix $A$: array $b$ contains the diagonal elements, array $c$ contains the upper diagonal elements, and array $a$ holds the lower diagonal elements.

```c
u[1] = b[1];
if (u[1] == 0.0) printf("ERROR in array u \n\n");
for (j=2; j<=ns-2; ++j) {
    if (u[j] == 0.0) printf("ERROR in array u \n\n");
    u[j] = b[j] - a[j]*c[j-1]/u[j-1];
}
```

Code excerpt 10.14 Computer code which calculates the diagonal elements of the matrix $U$, in an $LU$ decomposition of a tridiagonal matrix, $A$. The elements of matrix $A$ are stored in the following arrays: array $b$ contains the diagonal elements, array $c$ contains the upper diagonal elements, and array $a$ holds the lower diagonal elements. The diagonal elements of $U$ are stored in the array $u$ for later use, in Code excerpts 10.15 and 10.16.
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It should be noted we do not explicitly compute the elements of the matrix $L$. This is because all the diagonal elements of $L$ are known to be 1, and the sub-diagonal elements of $L$ can be computed from the diagonal elements of $U$ by using $1[j] = a[j]/u[j-1]$. Also we do not need to compute the upper diagonal elements of $U$ since they are known to be the same as the upper diagonal elements of the original matrix $A$, and are contained in the array $c$, see for example Hager (1988).

Having computed the $LU$ decomposition we can now solve the lower triangular system $Ly = R$ using forward elimination, this is shown in Code excerpt 10.15.

```
work[1] = rhs[1];
for (j=2; j<=ns-2; ++j) {
    work[j] = rhs[j] - a[j]*work[j-1]/u[j-1];
}
```

**Code excerpt 10.15** Computer code which uses forward elimination to solve the lower triangular system $Ly = R$, where $y$ is stored in the array `work`

In Code excerpt 10.15 we make use of the following two arrays: the array `rhs` which is used to store the elements of the right hand side $R$, and the array `work` which is both used as workspace and to store the computed solution vector $y$. As previously mentioned the sub-diagonal elements of $L$ are given by $1[j] = a[j]/u[j-1]$. This means that in Code excerpt 10.15, the line:

```
work[j] = rhs[j] - a[j]*work[j-1]/u[j-1];
```

is in fact be equivalent to:

```
work[j] = rhs[j] - 1[j]*work[j-1];
```

where $1[j]$, $j = 2 \ldots ns-2$ contains the sub-diagonal elements of $L$, if we had (needlessly) decided to allocate space for an extra array called $1$.

We are now in a position to solve the triangular system $Ux = y$ by using back-substitution. The code to achieve this is given in Code excerpt 10.16. Here the array `work` contains the previously computed values of $y$, the diagonal elements of $U$ are contained in the array `u`, and (as previously mentioned) the upper diagonal elements of $U$ are stored in the array `a`.

```
opt_vals[ns-2] = work[ns-2]/u[ns-2];
for (j = ns-2; j>=1; --j)
    opt_vals[j] = (work[j] - c[j]*opt_vals[j+1])/u[j];
```

**Code excerpt 10.16** Computer code which uses back-substitution to solve the upper triangular system $Ux = y$. At time instant $t_i = i\Delta t$, the elements of $x$ are the calculated option values $f_{i,j}$, $i = 1, \ldots, n_s-2$.

In Code excerpt 10.16 the array `opt_vals` contains the solution vector $x$. As its name suggests the contents of the array `opt_vals` are in fact the computed option values, $f_{i,j}$, $j = 1, \ldots, n_s-2$, in Equation 10.6.2 and represent the solution of the
Black–Scholes partial differential equation at time instant $t_i = i \Delta t$; based on the previously computed option values $f_{i+1,j}, j = 1, \ldots, n_s-2$.

### Backwards iteration and early exercise

The Black–Scholes equation can be solved over the time interval $t$ to $t+\tau$ by iteratively solving Equation 10.6.2. We iterate backwards in time by solving Equation 10.6.2 at the $i$th time step and then using the computed values to solve Equation 10.6.2 for the $(i-1)$th time step. The option values at current time $t$ are obtained when time index $i = 0$ is reached. It can be seen that the grid method yields $n_s-2$ option values, $f_{0,j}, j = 1, \ldots, n_s-2$, which correspond to the current asset prices $S^0_j = j \Delta S, j = 1, \ldots, n_s-2$.

As previously mentioned the asset price $S^0$ coincides with grid index $j = n_1$. Therefore $S^0 = S^0_{n_1}$, and the option value for the current asset price $S^0$ is given by $f_{0,n_1}$.

This is in contrast to the lattice methods discussed in Section 10.4 which yield a single option value corresponding to the root node.

The option values obtained using the grid methods we have just described are for vanilla European options. However, vanilla European options can be more accurately valued by using the Black–Scholes option pricing formula discussed in Section 9.3.3. The importance of finite-difference grids is that, by slightly modifying our backward iterative method, we can take into account the possibility of early exercise, and thus price American options.

This can be achieved by using Code excerpt 10.17 to modify the option prices contained in the array opt_vals as follows:

```c
if (put) {/* a put */
    for(j=1; j<n_s-2; ++j)
        opt_vals[j] = MAX(opt_vals[j], E-s[j]);
} else {/* a call */
    for(j=1; j<n_s-2; ++j)
        opt_vals[j] = MAX(opt_vals[j], s[j]-E);
}
```

**Code excerpt 10.17** Computer code which modifies the computed option values contained in array opt_vals to include the possibility of early exercise; this is required if we are to determine the value of American options. Here $s[j]$ contains the asset value at asset index $j$, opt_vals[j] contains the option value (computed by Code excerpt 10.16) at asset index $j$, and $E$ is the strike price.

Now we know how to solve the Black–Scholes equation; it is possible to include, without much difficulty, more exotic features such as lock out periods, barriers, rebates, etc.

The routine opt_gfd solves the Black–Scholes equation using a uniform grid. The asset price is set to one of the grid lines, which means that interpolation is not required.
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/* Input parameters:
theta_m — the value of theta used for the finite difference method,
asset_price — the current price of the underlying asset,
sigma — the volatility,
r — the interest rate,
T — the time to maturity,
strike — the strike price,
is_american — if is_american is 0 then a European option, otherwise an American option,
put — if put is 0 then a call option, otherwise a put option,
q — the continuous dividend yield,
nt — the number of time intervals,
smax — the maximum asset index on the grid, corresponding to the upper boundary,
Output parameters:
option_value — the value of the option,
greeks[] — the hedge statistics output as follows: greeks[0] is gamma, greeks[1] is delta, and
iflag — an error indicator.
*/

double ds, dt;
Integer i, j;
double tmp, t2, time_2mat;
Integer n1, n2, ind;
sig2, temp[4];
if (asset_price > smax) printf ("ERROR asset price > smax");
n1 = floor((asset_price/smax)*(double)pns);
n2 = pns / n1;
ds = asset_price/(double)n1;
dt = T/(double)nt; /* time interval size */
ns = n1 + n2 + 1;
/* Note: Now nps = ns - 1. Since we define asset grid lines 0...ns, this is the maximum grid line;
corresponding to the upper boundary. The lower boundary is at the asset grid line 0, and we solve for
option values between the asset grid line 1 and the asset grid line ns - 2 */
/* Allocate (all size ns+1) the arrays: a, b, c, a1, b1, c1, opt_vals, vals, rhs, s, work and u */
a[0] = 0.0;
s[0] = asset_price;
for(i = 1; i < n1 - 1; ++i) /* set prices below asset_price */
s[i] = (double)i * ds;
for(i = 1; i < n2 + 1; ++i) /* set prices above asset_price */
s[n1+i] = asset_price + (double)i * ds;
/* Set up the RHS and LHS coefficients a[], b[] and c[] are the LHS coefficients for the unknown option
values (time step j), a1[] and b1[] are the values of the RHS coefficients for the known option
prices (time step j+1). Note: a1, b1 and c1 are used to form the RHS vector rhs[] of the tridiagonal system. */
sig2 = sigma*sigma;
t2 = 1.0/theta_m; /* 1/theta (for theta method) */
for(i = 0; i < (ns-2); ++i) /* Assign elements of the (ns-2) *(ns-2) tridiagonal matrix */
a[i] = -1*(i*sig2-3)/(t2*tmp);
a1[i] = 1*(i*sig2-3)/(t2*theta_m);
c[i] = -1*(i*sig2+(r-q))/(t2*tmp);
c1[i] = 1*(i*sig2+(r-q))/(t2*theta_m);
b[i] = 1.0+"dt*tmp(1+i*sig2)*dt*tmp;
b1[i] = 1.0/(1+i*sig2)*dt*theta_m;
/* Perform LU decomposition of the tridiagonal matrix with:
diagonal elements contained in the array b[], upper diagonal elements contained in the array c[]
and lower diagonal elements in the array a[]. Store the elements of U but not those of L
(which will be computed from U) */
Matrix U: The diagonal elements of U are stored in the array u[] and the upper diagonal elements of U
are just c[]. Matrix L: For the lower triangular matrix L, the diagonal elements are 1 and the lower diagonal
elements are 1 + a[i]/u[i-1], where u[] is the upper diagonal of U. */
u[i] = b[i];
if (u[i] == 0.0) printf ("ERROR in array u \n");
for(i = 1; i < ns - 2; ++i)
u[i] = b[i] - a[i]*c[i-1]/u[i-1];
if (u[i] == 0.0) printf ("ERROR in array u \n");

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/* Set option values at maturity. Note: opt_vals[0] and opt_vals[ns-1] are the lower and upper
 (put/call) option price boundary values. */
if (put){ /* a call */
   for (i=0; i<ns; ++i)
     opt_vals[i] = MAX(s[i] - strike, 0.0);
}
else /* a put */
   for (i=0; i<ns; ++i)
     opt_vals[i] = MAX(strike - s[i], 0.0);

/* From the option values at maturity (t = nt*dt) calculate values at earlier times (nt-1)*dt etc.. */
for (j=nt-1; j>= -2; -j) {/* Go two steps past current time (0) so that can evaluate theta */
time_max = T-j*dt;
for (i=2; i<ns-1; ++i) /* set up the rhs of equation for Crank-Nicolson method */
  rhs[j] = -a[i]*opt_vals[i-1] + b[i]*opt_vals[i] + c[i] * opt_vals[i+1];
}

/* Incorporate the boundary conditions at the upper/lower asset value boundaries */
rhs[ns-2] = a[ns-2]*opt_vals[ns-3] + b[ns-2]*opt_vals[ns-2] + (c[ns-2] - c[ns-1])*opt_vals[ns-1];

/* Solve the lower triangular system Lx = b, where y is stored in array work[].
Compute the elements of L from those of U, 1[i] = a[i]/u [i-1]. */
work[1] = rhs[1];
for (i=2; i<ns-2; ++i) {
  work[i] = rhs[i] - a[i]*work[i-1];
}

/* Solve the upper (ns-2)* (ns-2) triangular system Ux = y (where x = opt_vals) */
opt_vals[ns-2] = work[ns-2]/u[ns-2];
for (i = ns-2; i>= 1; -i)
  opt_vals[i] = (work[i] - c[i]*opt_vals[i+1])/u[i];
if (is_american){ /* take into account early exercise for american options */
  if (put) {/* a put */
    for (i=1; i<ns-2; ++i)
      opt_vals[i] = MAX(opt_vals[i], strike - s[i]);
  }
  else {/* a call */
    for (i=1; i<ns-2; ++i)
      opt_vals[i] = MAX(opt_vals[i], s[i] - strike);
  }
}

if (j==0){
  for (i=0; i<ns; ++i)
    vals[i] = opt_vals[i];
}
if ((j==1)||(j==2)||(j==3)&&(j==4)) {/* Store option values so that can compute theta */
  temp[ind] = opt_vals[ind];
  ++ind;
}
}

if (greeks) {
 /* Compute gamma (4th order accuracy) */
greeks[0] = (-vals[n1+2]+16.0*vals[n1+1] - 30.0*vals[n1] + 16.0*vals[n1-1] - vals[n1-2])/(12.0*ds*ds);
/* Compute delta (4th order accuracy) */
greeks[1] = (-vals[n1+2]+8.0*vals[n1+1] - 8.0*vals[n1-1] + vals[n1-2])/(12.0*ds);
/* Compute theta (4th order accuracy) */
greeks[2] = (-temp[0]*8.0*temp[1]-8.0*temp[2]+temp[3])/(12.0*dt);
/* Note: could also compute theta as greeks[2] = (-temp[0]+4.0*temp[1]-3.0*vals[n1])/(2.0*dt); */
}

/*option_value = vals[n1]; */ /* Return option value */


Code excerpt 10.18  Function to compute the value of a vanilla option using a uniform grid

10.6.3 Nonuniform grids

In the previous section we showed how to solve the Black–Scholes equation using a uniform grid. Although this approach will provide satisfactory solutions to many option pricing problems, there are situations in which it is important to be able to place grid lines at locations which do not correspond to those available in a uniform
grid. Increasing the density of grid lines in regions of interest can lead to improved accuracy in both the estimated option values and also the estimates of the hedge statistics (the Greeks).

Here we provide an example which illustrates the benefits of using nonuniform grids in the evaluation of down and out call barrier options. Later on in Section 10.6.6 we give a further example which shows the use of nonuniform grids to evaluate double barrier options.

The purpose of this section is to show how to discretize the Black–Scholes equation using a nonuniform grid, and to derive an expression, see Equation 10.176, that is equivalent to Equation 10.151. Although the tridiagonal system of equations we have to solve in this section will be different from that in Section 10.6.2, the solution method is exactly the same. This means that once we have derived Equation 10.151 all the other information which we require to evaluate both European and American options is available in Section 10.6.2 under the headings:

- The boundary conditions.
- Computation of the option values at a given time instant.
- Backwards iteration and early exercise.

We will now consider the finite-difference approximation for a nonuniform grid, and then show how to value the down and out call barrier option.

### The finite-difference approximation

Here we consider how to discretize the Black–Scholes equation using a nonuniform grid, in which both the asset price interval $\Delta S$ and the time step $\Delta t$ are not constant but can vary throughout the grid.

Allowing for a nonconstant time step is quite simple. The time step occurs in both the first derivative of $f_{i,j}$, see Equation 10.141, and in the finite-difference equations, see Equations 10.153 to 10.157, as the constant $\Delta t$. To incorporate a varying time step, $\Delta t_i$, $i = 0, n_t$, thus only requires setting $\Delta t = \Delta t_i$, at the $i$th time step and then continue with the solution method outlined in Section 10.6.2.

The incorporation of nonconstant asset price intervals requires more work. This is because the finite-difference approximations to the first and second derivatives $f'_{i,j}$ and $f''_{i,j}$, in Equations 10.147 and 10.148, are based on a Taylor expansion about the point $f_{i,j}$.

We will now derive expressions for these derivatives. If we let $\Delta X_j^- = S_j - S_{j-1}$ and $\Delta X_j^+ = S_{j+1} - S_j$ and then using a Taylor expansion about the point $f_{i,j}$ we have

$$f_{i+1,j+1} = f_{i+1,j} + f'_{i+1,j} \Delta X_j^+ + \frac{1}{2} f''_{i+1,j} (\Delta X_j^+)^2$$

(10.169)

and also

$$f_{i+1,j-1} = f_{i+1,j} - f'_{i+1,j} \Delta X_j^- + \frac{1}{2} f''_{i+1,j} (\Delta X_j^-)^2$$

(10.170)
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Multiplying Equation 10.169 by \(\Delta X_i\) and adding it to \(\Delta X_i\) times, Equation 10.170 gives

\[
\Delta X_i f_{i+1,j-1} + \Delta X_i f_{i+1,j+1} = \Delta X_i f_{i+1,j} + \Delta X_i f_{i+1,j} \\
+ \frac{1}{2} f''_{i+1,j} \left\{ (\Delta X_i)^2 \Delta X_i + (\Delta X_i)^2 \Delta X_i \right\}
\]

Therefore

\[
\frac{1}{2} f''_{i+1,j} = \frac{\Delta X_i f_{i+1,j-1} + \Delta X_i f_{i+1,j+1} - \Delta X_i f_{i+1,j} - \Delta X_i f_{i+1,j}}{(\Delta X_i)^2 \Delta X_i + (\Delta X_i)^2 \Delta X_i}
\]

So

\[
f''_{i+1,j} = \frac{2 \left\{ \Delta X_i f_{i+1,j-1} + \Delta X_i f_{i+1,j+1} - f_{i+1,j}(\Delta X_i + \Delta X_i) \right\}}{(\Delta X_i)^2 \Delta X_i + (\Delta X_i)^2 \Delta X_i}
\] (10.171)

To calculate \(f''_{i+1,j}\) we rearrange Equation 10.170 to obtain

\[-f''_{i+1,j}\Delta X_i = f_{i+1,j-1} - f_{i+1,j} - \frac{1}{2} f''_{i+1,j}(\Delta X_i)^2\]

and

\[
f''_{i+1,j} = \frac{f_{i+1,j} - f_{i+1,j-1}}{\Delta X_i} + \frac{1}{2} f''_{i+1,j} \Delta X_i
\] (10.172)

If we now substitute for \(f''_{i+1,j}\) from Equation 10.171, into Equation 10.172 we have

\[
f''_{i+1,j} = \frac{f_{i+1,j} - f_{i+1,j-1}}{\Delta X_i} + \frac{\left\{ \Delta X_i f_{i+1,j-1} - (\Delta X_i + \Delta X_i) f_{i+1,j} + \Delta X_i f_{i+1,j+1} \right\}}{(\Delta X_i)^2 \Delta X_i + (\Delta X_i)^2 \Delta X_i} \Delta X_i
\]

which simplifies to give

\[
f''_{i+1,j} = \frac{(\Delta X_i)^2 (f_{i+1,j} - f_{i+1,j-1}) - (\Delta X_i)^2 f_{i+1,j} + (\Delta X_i)^2 f_{i+1,j+1}}{(\Delta X_i)^2 \Delta X_i + (\Delta X_i)^2 \Delta X_i}
\]

so that we finally have

\[
f''_{i+1,j} = \frac{(\Delta X_i)^2 f_{i+1,j+1} + ((\Delta X_i)^2 - (\Delta X_i)^2) f_{i+1,j} - (\Delta X_i)^2 f_{i+1,j-1}}{(\Delta X_i)^2 \Delta X_i + (\Delta X_i)^2 \Delta X_i}
\] (10.173)
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As in Section 10.6.2, we can now substitute the expressions for \( f'_{i+1,j} \) and \( f''_{i+1,j} \) given in Equations 10.173 and 10.171, into the Equation 10.144; the discretized Black–Scholes equation. If we let \( D = (\Delta X_j^+)^2 + (\Delta X_j^-)^2 + (\Delta X_j^+)^2 \) we then obtain

\[
r\Delta t(\Theta_m f_{i+1,j} + \Theta_m^* f_{i,j}) = f_{i+1,j} - f_{i,j} + \frac{(r-q)S_j \Delta t A_1}{D} + \frac{\sigma^2 S_j^2 \Delta t A_2}{D}
\]

(10.174)

where \( \Theta_m^* = 1 - \Theta_m \), and

\[
A_1 = \Theta_m [f_{i+1,j+1}(\Delta X_j^+)^2 - f_{i+1,j-1}(\Delta X_j^-)^2 - f_{i+1,j}\{(\Delta X_j^+)^2 - (\Delta X_j^-)^2\}]
\]

+ \( \Theta_m^*[f_{i+1,j}(\Delta X_j^+)^2 - f_{i,j}(\Delta X_j^-)^2 - f_{i,j}\{(\Delta X_j^-)^2 - (\Delta X_j^+)^2\}] \)

and

\[
A_2 = \Theta_m [f_{i+1,j+1}\Delta X_j^+ + f_{i+1,j-1}\Delta X_j^- - f_{i+1,j}\{\Delta X_j^- + \Delta X_j^+\}]
\]

+ \( \Theta_m^*[f_{i+1,j}\Delta X_j^- + f_{i+1,j-1}\Delta X_j^+ - f_{i,j}\{\Delta X_j^- + \Delta X_j^+\}] \)

Collecting like terms we obtain:

\[
B_1 f_{i,j-1} + B_2 f_{i,j} + B_3 f_{i,j+1} + C_1 f_{i+1,j-1} + C_2 f_{i+1,j} + C_3 f_{i+1,j+1} = 0
\]

(10.175)

where

\[
B_1 = \frac{-\Theta_m^* (r-q)S_j \Delta t (\Delta X_j^+)^2 + (1-\theta)\sigma^2 S_j^2 \Delta t \Delta X_j^+}{D}
\]

\[
B_2 = \frac{-1 - r\Delta t \Theta_m^* - \Theta_m^* \sigma^2 S_j^2 \Delta t (\Delta X_j^- + \Delta X_j^+)}{D} - \frac{\Theta_m^* (r-q)S_j \Delta t \{(\Delta X_j^-)^2 - (\Delta X_j^+)^2\}}{D}
\]

\[
B_3 = \frac{\Theta_m^* (r-q)S_j \Delta t (\Delta X_j^-)^2 + \Theta_m^* \sigma^2 S_j^2 \Delta t \Delta X_j^-}{D} - \frac{\Theta_m (r-q)S_j \Delta t (\Delta X_j^+)^2}{D}
\]

\[
C_1 = \frac{\Theta_m \sigma^2 S_j^2 \Delta t \Delta X_j^+ + \Theta_m (r-q)S_j \Delta t (\Delta X_j^+)^2}{D}
\]

\[
C_2 = \frac{1 - r\Delta t \Theta_m - \Theta_m (r-q)S_j \Delta t \{(\Delta X_j^-)^2 - (\Delta X_j^+)^2\} - \Theta_m \sigma^2 S_j^2 \Delta t \{\Delta X_j^- + \Delta X_j^+\}}{D}
\]

\[
C_3 = \frac{\Theta_m (r-q)S_j \Delta t (\Delta X_j^-)^2 + \Theta_m \sigma^2 S_j^2 \Delta t \Delta X_j^-}{D}
\]

Since we are solving the Black–Scholes equation backwards in time we will rearrange Equation 10.175 as:

**Finite-difference scheme for a nonuniform grid**

\[
a_{i+1,j} - b_{i+1,j} + c_j = R_{i+1,j}
\]

(10.176)

where the right hand side \( R_{i+1,j} \) is:

\[
R_{i+1,j} = a_j f_{i+1,j-1} + b_j f_{i+1,j} + c_j f_{i+1,j+1}
\]

(10.177)
and the coefficients are

\[ a_j = \Theta_m \Delta t \left\{ \frac{(r - q) S_j (\Delta X_j^+)^2}{D} - \frac{\sigma^2 S_j^2 \Delta X_j^+}{D} \right\} \tag{10.178} \]

\[ b_j = 1 + \Delta t \Theta_m \left\{ \frac{\sigma^2 S_j^2 (\Delta X_j^- + \Delta X_j^+)}{D} \right. \]
\[ + \left. \frac{(r - q) S_j (\Delta X_j^-)^2 - (\Delta X_j^+)^2}{D} \right\} \tag{10.179} \]

\[ c_j = \Theta_m \Delta t \left\{ \frac{-(r - q) S_j (\Delta X_j^-)^2}{D} - \frac{\sigma^2 S_j^2 \Delta X_j^-}{D} \right\} \tag{10.180} \]

\[ \bar{a}_j = \Theta_m \Delta t \left\{ \frac{\sigma^2 S_j^2 \Delta X_j^+}{D} - \frac{(r - q) S_j (\Delta X_j^+)^2}{D} \right\} \tag{10.181} \]

\[ \bar{b}_j = 1 - \Theta_m r \Delta t \]
\[ - \Theta_m \Delta t \left\{ \frac{(r - q) S_j (\Delta X_j^-)^2}{D} - \frac{\sigma^2 S_j^2 (\Delta X_j^- + \Delta X_j^+)}{D} \right\} + \frac{\sigma^2 S_j^2 \Delta X_j^+}{D} \] \tag{10.182}

\[ \bar{c}_j = \Theta_m \Delta t \left\{ \frac{(r - q) S_j (\Delta X_j^-)^2}{D} + \frac{\sigma^2 S_j^2 \Delta X_j^-}{D} \right\} \tag{10.183} \]

Here Equation 10.176, as is the case for Equation 10.151 in Section 10.6.2, provides the relationship between the three option values \( f_{i+1,j-1}, f_{i+1,j}, f_{i+1,j+1} \) at time index \( i + 1 \), and the three option values \( f_{i,j-1}, f_{i,j}, f_{i,j+1} \) at time index \( i \). It can also be seen that Equation 10.176 is the nonuniform grid equivalent of Equation 10.151 given in Section 10.6.2. We will now show that Equations 10.176 and 10.151 are identical when a uniform grid is used, that is \( \Delta X_j^+ = \Delta X_j^- \). We proceed as follows:

Let \( \Delta X_j^+ = \Delta X_j^- = \Delta S \), and \( S_j = j\Delta S \)

so

\[ D = (\Delta X_j^+)^2 \Delta X_j^- + (\Delta X_j^-)^2 \Delta X_j^+ = 2(\Delta S)^3 \]

\[ \frac{(\Delta X_j^+)^2}{D} = \frac{(\Delta X_j^-)^2}{D} = \frac{(\Delta S)^2}{2(\Delta S)^3} = \frac{1}{2\Delta S} \]
If we substitute the above values into Equations 10.178 to 10.183 we obtain the following expressions for the coefficients in Equation 10.176.

\[
a_j = (1 - \Theta_m) \Delta t \left\{ \frac{(r - q) S_j}{2 \Delta S^2} - \frac{\sigma^2 S_j^2}{2 \Delta S^2} \right\} = \frac{(1 - \Theta_m)}{2} \left\{ (r - q) j - \sigma^2 j^2 \right\}
\]

\[
b_j = 1 + \Delta t (1 - \Theta_m) \left\{ \frac{r + \sigma^2 S_j^2}{2 \Delta S^2} \right\} = 1 + (1 - \Theta_m) \Delta t \left\{ r + \sigma^2 j^2 \right\}
\]

\[
c_j = \left(1 - \Theta_m\right) \Delta t \left\{ -\frac{(r - q) S_j}{2 \Delta S^2} - \frac{\sigma^2 S_j^2}{2 \Delta S^2} \right\} = -\left(1 - \Theta_m\right) \frac{\Delta t}{2} \left\{ (r - q) j + \sigma^2 j^2 \right\}
\]

\[
a_j = \Theta_m \Delta t \left\{ \frac{\sigma^2 S_j^2}{2 \Delta S^2} - \frac{(r - q) S_j}{2 \Delta S^2} \right\} = -\Theta_m \Delta t \left\{ (r - q) j - \sigma^2 j^2 \right\}
\]

\[
b_j = 1 - \Theta_m r \Delta t - \frac{\Theta_m \sigma^2 S_j^2 \Delta t}{\Delta S^2} = 1 - \Theta_m \Delta t \left\{ r + \sigma^2 j^2 \right\}
\]

\[
c_j = \Theta_m \Delta t \left\{ \frac{(r - q) S_j}{2 \Delta S^2} + \frac{\sigma^2 S_j^2}{\Delta S^2} \right\} = -\Theta_m \frac{\Delta t}{2} \left\{ (r - q) j + \sigma^2 j^2 \right\}
\]

It can be seen that these coefficients are identical to those given in Section 10.6.2, Equations 10.153 to 10.158.

We now provide examples of using nonuniform grids to evaluate European down and out call options.

**Valuation of a down and out call option**

Here the improved accuracy that can be achieved by using nonuniform grids instead of uniform grids is illustrated in Figures 10.11 and 10.12. The uniform grids are constructed using the method outlined in Section 10.6.2 and Code excerpt 10.18. That is an asset grid line is set to coincide with the current asset price \( S_0 \), and the other grid lines are positioned above and below \( S_0 \) with a uniform spacing of \( \Delta S \). The disadvantage of this approach is that there will be an unspecified pricing error that depends on the distance, \( d_s \), of the barrier level, \( B \), to the the nearest asset grid line. Furthermore, as the number of asset points, \( n_s \), increases the magnitude of \( d_s \) will oscillate within the range 0 to \( \Delta S/2 \).

When \( d_s \sim 0 \) the grid will be accurate, but when \( |d_s| \sim \Delta S/2 \) there will be a large pricing error. This gives rise to the oscillating pricing errors shown in Figures 10.11 and 10.12.
The nonuniform grids are constructed using the techniques mentioned earlier in this section, and also Code excerpt 10.19. We now, irrespective of $n_s$, arrange for one asset grid line to coincide with the current asset value, $S_0$, and another asset grid line to coincide with $B$, the barrier asset price. In Figure 10.10 this corresponds to setting $B_L$ to $B$ and not using $B_U$.

It can be seen in Figures 10.11 and 10.12 that in this case the pricing error is very much less, and also doesn’t exhibit the pronounced oscillations that are produced by a uniform grid. In Code excerpt 10.19 below, we give the computer program which was used to obtain the nonuniform grid values for the down and out call options presented in Figures 10.11 and 10.12. Although this program only deals with European options it can easily be altered, using the same techniques as in Code excerpt 10.18, to deal with American style options; this is left as an exercise for the reader.

```c
void barrier_downout(double barrier_level, double theta_m, double asset_price, double sigma, double r,
    double T, double strike, Integer put, double *option_value, double greeks[], double q, Integerns,
    Integer nt, double smax, Integer *ifail)
{
    /* ns - the number of asset intervals
       nt - the number of time intervals
     */
```
double ds, time_step;
Integer i, j, barrier_index;
double tmp, t2, time_2mat, zero = 0.0;
Integer n1, n2, ind = 0, ns1;
double sig2, temp[4], ds_plus, ds_minus, temp1, temp2, temp3;
Double D;

n1 = floor((asset_price/smax)*(double)ns);
if (n1 < 3){
    printf (''increase the number of asset points 
'');
}

n2 = ns / C0

ds = asset_price/(double)n1;
time_step = T/(double)nt;/* time interval size */
ns1 = n1 + n2 + 2; /* number of nodes 
C0 including extra grid line*/
/*allocate the required arrays (all of size ns1+1):a,b,c,a1,b1,c1,opt_vals,vals,rhs,s,work,u */
/* set prices below asset_price */
s[0] = zero;
s[n1] = asset_price;
for(i=1; i < n1; ++i)
    s[i] = (double)i * ds;
/* set prices above asset_price */
for(i=1; i< n2+2; ++i){
    s[n1+i] = asset_price + (double)i * ds;
}
/*find out the index corresponding to barrier_level */
barrier_index = 0;
while(barrier_level > s[barrier_index]){
    ++barrier_index;
}

Figure 10.11  The absolute error in the estimated values for a European down and out call barrier option $(B < E)$ as the number of asset grid points, $n_s$, is varied. Here we show a comparison of the results obtained using both uniform and nonuniform grids; logarithmic transformations were not employed. The algorithm for the uniform grid is described in Section 10.6.2, and that for the nonuniform grid is outlined in Section 10.6.3. The Crank–Nicolson method ($\theta_m = 0.5$) was used and the other parameters were $E = 50.0, B = 47.5, S_0 = 55.0, S_{max} = 300.0, T = 0.5, \sigma = 0.2, r = \log(1.1), q = 0.0, n_t = 100$. The correct option value was 7.6512 which was obtained using the analytic formulae given in Section 9.4 and Code excerpt 9.6
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Figure 10.12 The absolute error in the estimated values for a European down and out call barrier option ($E < B$) as the number of asset grid points, $n_s$, is varied. Here we show a comparison of the results obtained using both uniform and nonuniform grids; logarithmic transformations are not employed. The algorithm for the uniform grid is described in Section 10.6.2 and that for the nonuniform grid is outlined in Section 10.6.3. The Crank–Nicolson method ($\Theta_m = 0.5$) was used and the other parameters were $E = 50.0$, $B = 52.5$, $S_0 = 65.0$, $S_{\text{max}} = 300.0$, $T = 0.5$, $\sigma = 0.2$, $r = \log(1.1)$, $q = 0.0$, $n_t = 100$. The correct option value was $17.0386$ which was obtained using the analytic formulae given in Section 9.4 and Code excerpt 9.6.

```c
if (barrier_level != s[barrier_index]) {/* decrement barrier index */
    -- barrier_index;
} 
if (s[barrier_index] != barrier_level) {/* then barrier does not correspond to an existing grid line so create another one*/
    for (i = 1; i < ns1 - barrier_index; ++i) {
        s[barrier_index + i + 1] = s[barrier_index] + (double)i * d_s;
    }
    ++barrier_index;
    s[barrier_index] = barrier_level;
    if (n1 > barrier_index) {
        ++n1;
    }
}

/* set up the RHS and LHS coefficients a[], b[] and c[] are the LHS coefficients for the unknown values (time step j) a1[], b1[] and c1[] are the values of the RHS coefficients for the known option prices (time step j+1). Note: a1, b1 and c1 are used to form the RHS vector rhs[] of the tridiagonal system. */
#2 = sigma*sigma;
t2 = time_step/2.0;
tmp = 1.0 - theta_m; /* 1 - theta (for theta method) */
/* assign elements of the (ns1-2)*(ns1-2) tridiagonal matrix */
for (i = 1; i < ns1 - 2; ++i) {
    ds_plus = s[i + 1] - s[i];
    ds_minus = s[i] - s[i - 1];
    D = ((ds_plus*ds_plus*ds_minus) + (ds_minus*ds_minus*ds_plus));
    temp1 = tmp*time_step/D;
```
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\begin{verbatim}
al[i] = -(temp1*(r-q)*s[i]*ds_plus*ds_plus) - temp1* da_plus* (s[i]*s[i]*sig2));
temp1 = (ds_minus*da_minus)/D;
temp2 = da_minus*ds_minus/D;
c[i] = -time_step*temp1*(s[i] - r + (sig2*s[i]*s[i]*temp2));
c[i] = time_step*theta_m*(temp1*s[i] - r + (sig2*s[i]*s[i]*temp2));
temp1 = (da_minus*ds_minus - (ds_plus*da_plus))/D;
temp2 = da_minus*ds_plus)/D;
b[i] = 1.0*time_step*temp1*(r + (r-q)*s[i]*temp1) + (s[i]*s[i]*sig2)*temp2); 
b[i] = 1.0*time_step*theta_m*(r + (r-q)*s[i]*temp1) + (s[i]*s[i]*sig2)*temp2);

/* Perform LU decomposition of the tridiagonal matrix with: diagonal elements contained in the array b[],
   upper diagonal elements contained in the array c[] and lower diagonal elements in the array a[].
   Store the elements of U but not those of L (they will be computed from U)
Matrix U: The diagonal elements of U are stored in the array u[] and the upper diagonal elements of U are
just c[].
Matrix L: For the lower triangular matrix L, the diagonal elements are i and the lower diagonal elements
are l[i] = a[i]/u[i-1], where u[] is the upper diagonal of U. */
u[i] = b[i];
if (u[i] == zero) printf("error in array u \n");
for (i = 1; i < ns1-2; ++i)
u[i] = b[i] - a[i]*c[i-1]/u[i-1];
if (u[i] == zero) printf("error in array u \n");

/* Set option values at maturity, Note: opt_vals[0] and opt_vals[ns1-1] are the lower and upper (put/call)
option price boundary values. */
for (i = 0; i < ns1; ++i)
opt_vals[i] = MAX(s[i] - strike, zero);
/* now modify option values to include the barrier */
for (i = 0; i <= barrier_index; ++i)
opt_vals[i] = zero;
else /* a put */
for (i = 0; i < ns1; ++i)
opt_vals[i] = MAX(strike - s[i], zero);

/* From the option values at maturity, t = nt*time_step, compute
the values at times (nt-1)*time_step to 0 (current time) */
for (j = nt-1; j >= 0; --j) /* go two steps past current time so that can evaluate theta */
time_mat = T-j*time_step;
/* set up the rhs of equation for the Theta method */
for (i = 1; i < ns1-1; ++i)
_rhs[i] = a[i]*opt_vals[i-1] + b[i]*opt_vals[i] + c[i]*opt_vals[i+1];
/* incorporate the boundary conditions at the upper/lower asset value boundaries */
orhs[ns1-2] = a[ns1-2]*opt_vals[ns1-3] + b[ns1-2]*opt_vals[ns1-2] +
(c[ns1-2]*c[ns1-2]*opt vals[ns1-1]);
/* Solve the lower triangular system Uy = b, where y is stored in array work[].
Compute the elements of b from those of u, b[i] = a[i]/u[i-1] */
work[1] = rhs[1];
for (i = 2; i < ns1-2; ++i)
work[i] = rhs[i] - a[i]*work[i-1]/u[i-1];
/* Solve the upper (ns1-2)*(ns1-2) triangular system Ux = y (where x = opt_vals) */
opt_vals[ns1-2] = work[ns1-2]/u[ns1-2];
for (i = ns1-2; i > 1; --i)
opt_vals[i] = (work[i] - c[i]*opt vals[i+1])/u[i];
if (j == 0)
for (i = 0; i < ns1; ++i)
val[i] = opt_vals[i];
/* store option values so that can compute theta */
if (((j == 1) && (j == 2)) || (j == -1)) || (j == -2))
{ temp[ind] = opt_vals[ind];
++ind;
/* now modify for barrier */
for (i = 0; i <= barrier_index; ++i)
opt_vals[i] = zero;
if (greeks) /* assume an irregular grid */
da_minus = s[ns1-1] - s[ns1-2];
da_plus = s[ns1-1] - s[ns1];
D = (ds_minus*ds_minus + ds_plus*ds_plus) + (ds_plus*ds_plus + da_minus);
temp1 = da_minus*da_minus;
\end{verbatim}
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\[ \text{temp2} = \text{ds}_+ \cdot \text{ds}_+; \]
\[ \text{temp3} = \text{temp1} / \text{C0} \cdot \text{temp2}; \]

/* GAMMA */
\[ \text{greeks}[0] = (\text{ds}_- \cdot \text{vals}[n1+1] + \text{ds}_+ \cdot \text{vals}[n1-1] - \text{vals}[n1] \cdot (\text{ds}_+ + \text{ds}_-)) / (0.5 \cdot \text{D}); \]

/* DELTA */
\[ \text{greeks}[1] = (\text{temp0} \cdot \text{vals}[n1+1] - \text{temp2} \cdot \text{vals}[n1-1] - \text{vals}[n1] \cdot \text{temp3}) / \text{D}; \]

/* THETA */
\[ \text{greeks}[2] = (\text{temp0} + 4.0 \cdot \text{temp1} - 3.0 \cdot \text{vals}[n1]) / (2.0 \cdot \text{time_step}); \]

} /* option_value */

10.6.4 The log transformation and uniform grids

Up to this point we have been dealing with the standard Black–Scholes equation, which is

\[
\frac{\partial f}{\partial t} + (r - q)S \frac{\partial f}{\partial S} + \frac{\sigma^2 S^2}{2} \frac{\partial^2 f}{\partial S^2} = rf \quad (10.184)
\]

However, if we introduce the change of variable \( Z = \log S \), we obtain the following equation:

\[
\frac{\partial f}{\partial t} + b \frac{\partial f}{\partial Z} + \frac{\sigma^2}{2} \frac{\partial^2 f}{\partial Z^2} = rf \quad (10.185)
\]

where \( b = r - q - (\sigma^2 / 2) \). This has beneficial numerical properties since it does not contain the original Black–Scholes terms in \( S \) and \( S^2 \).

Derivation of Equation 10.185

We will now derive an expression for the logarithmic Black–Scholes equation, and show that it agrees with Equation 10.185.

Since \( Z = \log S \) we have \( \frac{\partial Z}{\partial S} = 1/S \). This gives:

\[
\frac{\partial f}{\partial S} = \frac{\partial f}{\partial Z} \cdot \frac{\partial Z}{\partial S} = \frac{1}{S} \frac{\partial f}{\partial Z}
\]

and

\[
\frac{\partial^2 f}{\partial S^2} = \frac{\partial}{\partial S} \left( \frac{\partial f}{\partial Z} \right) = \frac{1}{S^2} \frac{\partial f}{\partial Z} + \frac{1}{S} \frac{\partial}{\partial S} \left( \frac{\partial f}{\partial Z} \right) = -\frac{1}{S^2} \frac{\partial f}{\partial Z} + \frac{1}{S} \frac{\partial Z}{\partial S} \frac{\partial f}{\partial Z}
\]

\[
\frac{\partial^2 f}{\partial S^2} = -\frac{1}{S^2} \frac{\partial f}{\partial Z} + \frac{1}{S^2} \frac{\partial^2 f}{\partial Z^2}
\]

Substituting the above values into Equation 10.184

\[
\frac{\partial f}{\partial t} + \frac{(r - q)S}{S} \frac{\partial f}{\partial Z} - \frac{\sigma^2 S^2}{2S^2} \frac{\partial f}{\partial Z} + \frac{\sigma^2 S^2}{2S^2} \frac{\partial^2 f}{\partial Z^2} = rf
\]
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and setting $b = r - q - \sigma^2/2$ we obtain:

$$\frac{\partial f}{\partial t} + b \frac{\partial f}{\partial Z} + \frac{\sigma^2}{2} \frac{\partial^2 f}{\partial Z^2} = rf \quad QED$$

(10.186)

We will now consider the finite-difference discretization of Equation 10.185.

**The finite-difference method**

Application of the finite difference method to the log transformed Black–Scholes equation is very similar to that already outlined in Sections 10.6.2 and 10.6.3.

Use of the $\Theta_m$ method on Equation 10.185 results in:

$$\frac{f_{i+1,j} - f_{i,j}}{\Delta t} + b \left\{ \Theta_m f_{i+1,j} + \Theta_m^* f_{i,j} \right\} + \frac{1}{2} \sigma^2 \left\{ \Theta_m f_{i+1,j}'' + \Theta_m^* f_{i,j}'' \right\} = r \left\{ \Theta_m f_{i+1,j} + \Theta_m^* f_{i,j} \right\}$$

where $\Theta_m^* = 1 - \Theta_m$. Applying a uniform discretization at node $(i, j)$ we obtain:

$$f_{i+1,j} - f_{i,j} + \frac{b \Delta t A_1}{2 \Delta Z} + \frac{\sigma^2 \Delta t A_2}{2 \Delta Z^2} = r \Delta t \left\{ \Theta_m f_{i+1,j} + \Theta_m^* f_{i,j} \right\}$$

(10.187)

where

$$A_1 = \Theta_m \left\{ f_{i+1,j+1} - f_{i+1,j-1} \right\} + \Theta_m^* \left\{ f_{i,j+1} - f_{i,j-1} \right\}$$

$$A_2 = \Theta_m \left\{ f_{i+1,j+1} - 2 f_{i+1,j} + f_{i+1,j-1} \right\} + \Theta_m^* \left\{ f_{i,j+1} - 2 f_{i,j} + f_{i,j-1} \right\}$$

Collecting like terms obtain:

$$B_1 f_{i,j-1} + B_2 f_{i,j} + B_3 f_{i,j+1} + C_1 f_{i+1,j-1} + C_2 f_{i+1,j} + C_3 f_{i+1,j+1} = 0$$

where

$$B_1 = -\Theta_m^* b \Delta t + \Theta_m^* \sigma^2 \Delta t$$

$$B_2 = -1 - r \Delta t \Theta_m - \frac{\Theta_m^* \sigma^2 \Delta t}{\Delta Z^2}$$

$$B_3 = \Theta_m \sigma^2 \Delta t + \Theta_m^* \sigma^2 \Delta t$$

$$C_1 = \Theta_m \sigma^2 \Delta t - \Theta_m b \Delta t$$

$$C_2 = -1 - r \Delta t \Theta_m - \frac{\Theta_m \sigma^2 \Delta t}{\Delta Z^2}$$

$$C_3 = \Theta_m b \Delta t + \Theta_m \sigma^2 \Delta t$$
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If we rearrange we have the following equation:

Finite-difference scheme for a uniform grid and log transformation

\[ a_j f_{i,j-1} + b_j f_{i,j} + c_j = \hat{a}_j f_{i+1,j-1} + \hat{b}_j f_{i+1,j} + \hat{c}_j f_{i+1,j+1} \] (10.188)

where:

\[
\begin{align*}
  a_j &= \frac{(1 - \Theta_m)\Delta t}{2\Delta Z^2} \{ b\Delta Z - \sigma^2 \} \\
  b_j &= 1 + (1 - \Theta_m)\Delta t \left\{ r + \frac{\sigma^2}{\Delta Z^2} \right\} \\
  c_j &= -\frac{(1 - \Theta_m)\Delta t}{2\Delta Z^2} \{ b\Delta Z + \sigma^2 \}
\end{align*}
\]

\[
\begin{align*}
  \hat{a}_j &= -\frac{\Theta_m\Delta t}{2\Delta Z^2} \{ b\Delta Z - \sigma^2 \} \\
  \hat{b}_j &= 1 - \Theta_m\Delta t \left\{ r + \frac{\sigma^2}{\Delta Z^2} \right\} \\
  \hat{c}_j &= \frac{\Theta_m\Delta t}{2\Delta Z^2} \{ b\Delta Z + \sigma^2 \}
\end{align*}
\]

It can be seen that, unlike in Section 10.6.2, the coefficients in Equations 10.188 to 10.194 are independent of the asset price index \( j \).

When \( \Theta_m = 0.5 \) (the Crank–Nicolson method) we have the following coefficients:

\[
\begin{align*}
  a_j &= -\hat{a}_j = \frac{\Delta t}{4\Delta Z^2} \{ b\Delta Z - \sigma^2 \} \\
  b_j &= 1 + \frac{\Delta t}{2} \left\{ r + \frac{\sigma^2}{\Delta Z^2} \right\} \\
  c_j &= -\hat{c}_j = -\frac{\Delta t}{4\Delta Z^2} \{ b\Delta Z + \sigma^2 \} \\
  \hat{b}_j &= 1 - \frac{\Delta t}{2} \left\{ r + \frac{\sigma^2}{\Delta Z^2} \right\}
\end{align*}
\]

The method of using the finite-difference grid to compute option prices is identical to that already outlined in Section 10.6.2, which solves the standard (nonlogarithmic) Black–Scholes equation. Table 10.7 compares the results obtained with and without a logarithmic transformation. It is shown in Appendix L.2 that the implicit method, \( \theta_m = 0 \), is unconditionally stable.

10.6.5 The log transformation and nonuniform grids

In the previous section we considered the use of a uniform grid to discretize the logarithmically transformed Black–Scholes equation.

\[
\frac{\partial f}{\partial t} + b \frac{\partial f}{\partial Z} + \frac{\sigma^2}{2} \frac{\partial^2 f}{\partial Z^2} = rf 
\] (10.195)
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**Table 10.7**  Valuation results and pricing errors for a vanilla American put option using a uniform grid with and without a logarithmic transformation; the implicit method and Crank–Nicolson method are used. The accurate values (obtained using a logarithmic transformed grid with \( \Theta_m = 0.0 \), \( n_s = 1000 \) and \( n_t = 1000 \)) are presented in the column labelled ‘Value’. The absolute pricing errors (ABS (accurate value – estimated value)) are presented in the column labelled BS were obtained using a standard uniform grid (as outlined in Section 10.6.2), and those in the column labelled Log BS use a uniform grid and logarithmic transformation as explained in this section. The maturity of the option was varied from 0.1 to 1.5 years, the other parameters were: \( S = 9.0 \), \( X = 9.7 \), \( r = 0.1 \), \( q = 0.0 \), \( \sigma = 0.30 \), \( S_{\text{max}} = 100.0 \), \( n_s = 50 \), and \( n_t = 50 \)

<table>
<thead>
<tr>
<th>Time</th>
<th>Value</th>
<th>BS</th>
<th>Log BS</th>
<th>BS</th>
<th>Log BS</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.1</td>
<td>0.7598</td>
<td>1.5142 \times 10^{-2}</td>
<td>7.7803 \times 10^{-3}</td>
<td>1.5077 \times 10^{-2}</td>
<td>7.6165 \times 10^{-3}</td>
</tr>
<tr>
<td>0.2</td>
<td>0.8334</td>
<td>4.6192 \times 10^{-2}</td>
<td>1.2924 \times 10^{-2}</td>
<td>4.5935 \times 10^{-2}</td>
<td>1.1892 \times 10^{-2}</td>
</tr>
<tr>
<td>0.3</td>
<td>0.8920</td>
<td>6.4526 \times 10^{-2}</td>
<td>1.4125 \times 10^{-2}</td>
<td>6.3969 \times 10^{-2}</td>
<td>1.2426 \times 10^{-2}</td>
</tr>
<tr>
<td>0.4</td>
<td>0.9401</td>
<td>7.4973 \times 10^{-2}</td>
<td>1.6559 \times 10^{-2}</td>
<td>7.4030 \times 10^{-2}</td>
<td>1.4483 \times 10^{-2}</td>
</tr>
<tr>
<td>0.5</td>
<td>0.9810</td>
<td>8.0546 \times 10^{-2}</td>
<td>1.8471 \times 10^{-2}</td>
<td>7.9155 \times 10^{-2}</td>
<td>1.5842 \times 10^{-2}</td>
</tr>
<tr>
<td>0.6</td>
<td>1.0164</td>
<td>8.3022 \times 10^{-2}</td>
<td>1.9125 \times 10^{-2}</td>
<td>8.1141 \times 10^{-2}</td>
<td>1.5845 \times 10^{-2}</td>
</tr>
<tr>
<td>0.7</td>
<td>1.0477</td>
<td>8.3496 \times 10^{-2}</td>
<td>1.8959 \times 10^{-2}</td>
<td>8.1098 \times 10^{-2}</td>
<td>1.5029 \times 10^{-2}</td>
</tr>
<tr>
<td>0.8</td>
<td>1.0755</td>
<td>8.2672 \times 10^{-2}</td>
<td>1.8408 \times 10^{-2}</td>
<td>7.9743 \times 10^{-2}</td>
<td>1.3894 \times 10^{-2}</td>
</tr>
<tr>
<td>0.9</td>
<td>1.1006</td>
<td>8.1012 \times 10^{-2}</td>
<td>1.7756 \times 10^{-2}</td>
<td>7.7547 \times 10^{-2}</td>
<td>1.2736 \times 10^{-2}</td>
</tr>
<tr>
<td>1.0</td>
<td>1.1234</td>
<td>7.8827 \times 10^{-2}</td>
<td>1.7138 \times 10^{-2}</td>
<td>7.4829 \times 10^{-2}</td>
<td>1.1695 \times 10^{-2}</td>
</tr>
<tr>
<td>1.1</td>
<td>1.1442</td>
<td>7.6332 \times 10^{-2}</td>
<td>1.6643 \times 10^{-2}</td>
<td>7.1807 \times 10^{-2}</td>
<td>1.0855 \times 10^{-2}</td>
</tr>
<tr>
<td>1.2</td>
<td>1.1633</td>
<td>7.3671 \times 10^{-2}</td>
<td>1.6290 \times 10^{-2}</td>
<td>6.8631 \times 10^{-2}</td>
<td>1.0217 \times 10^{-2}</td>
</tr>
<tr>
<td>1.3</td>
<td>1.1810</td>
<td>7.0946 \times 10^{-2}</td>
<td>1.6092 \times 10^{-2}</td>
<td>6.5404 \times 10^{-2}</td>
<td>9.7921 \times 10^{-3}</td>
</tr>
<tr>
<td>1.4</td>
<td>1.1973</td>
<td>6.8227 \times 10^{-2}</td>
<td>1.6042 \times 10^{-2}</td>
<td>6.2196 \times 10^{-2}</td>
<td>9.5649 \times 10^{-3}</td>
</tr>
<tr>
<td>1.5</td>
<td>1.2126</td>
<td>6.5559 \times 10^{-2}</td>
<td>1.6128 \times 10^{-2}</td>
<td>5.90565 \times 10^{-2}</td>
<td>9.5098 \times 10^{-3}</td>
</tr>
</tbody>
</table>

where

\[
b = r - q - \frac{\sigma^2}{2} \quad \text{and} \quad Z = \log S
\]

Here we will generalize these results and use a nonuniform grid to solve Equation 10.195.

Our description will be very brief since most of the details have already been discussed in previous sections. Here we are only concerned with the finite-difference approximation and derive the equations that need to be solved at each time step. Later, in Section 10.6.6, we will apply our results to solving a European double knockout barrier option.

**The finite-difference approximation**

At the grid node \((i, j)\) we have

\[
\Delta Z_j^- = Z_j - Z_{j-1} \quad \text{and} \quad \Delta Z_j^+ = Z_j + 1 - Z_j
\]

Following Section 10.6.3 the first and second derivatives of \( f \) w.r.t. \( Z \) are

\[
f_{i+1,j}'' = \frac{2 \left\{ \Delta Z_j^+ f_{i+1,j-1} + \Delta Z_j^- f_{i+1,j+1} - \Delta Z_j^- f_{i+1,j} - \Delta Z_j^+ f_{i+1,j} \right\}}{\left( \Delta Z_j^+ \right)^2 \Delta Z_j^- + \left( \Delta Z_j^- \right)^2 \Delta Z_j^+}
\]
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and

\[
\frac{f'_{i+1,j}}{f_{i,j}} = \frac{(\Delta Z_j^-)^2 f_{i+1,j+1} + ((\Delta Z_j^-)^2 - (\Delta Z_j^+)^2) f_{i+1,j} - (\Delta Z_j^+)^2 f_{i+1,j-1}}{(\Delta Z_j^-)^2 \Delta Z_j^- + (\Delta Z_j^+)^2 \Delta Z_j^+}
\]

Then discretizing Equation 10.195 in the usual manner we obtain

\[
\frac{f_{i+1,j} - f_{i,j}}{\Delta t} + b\left\{ \Theta_m f'_{i+1,j} + \Theta_m^* f'_{i,j} \right\} + \frac{\sigma^2}{2} \left\{ \Theta_m f''_{i+1,j} + \Theta_m^* f''_{i,j} \right\} = r \left\{ \Theta_m f_{i+1,j} + \Theta_m^* f_{i,j} \right\}
\]

where \( \Theta_m^* = 1 - \Theta_m \). Letting \( D = (\Delta Z_j^+)^2 \Delta Z_j^- + (\Delta Z_j^-)^2 \Delta Z_j^+ \) we obtain

\[
r \Delta t (\Theta_m f_{i+1,j} + \Theta_m^* f_{i,j}) = f_{i+1,j} - f_{i,j} + \frac{b \Delta t A_1}{D} + \frac{\sigma^2 \Delta t A_2}{D}
\]

where

\[
A_1 = \Theta_m \left[ f_{i+1,j+1}(\Delta Z_j^-)^2 - f_{i+1,j-1}(\Delta Z_j^+)^2 - f_{i+1,j}\{(\Delta Z_j^-)^2 - (\Delta Z_j^+)^2\} \right]
+ \Theta_m^* \left[ f_{i+1,j}(\Delta Z_j^-)^2 - f_{i,j-1}(\Delta Z_j^+)^2 - f_{i,j}\{(\Delta Z_j^-)^2 - (\Delta Z_j^+)^2\} \right]
\]

\[
A_2 = \Theta_m \left[ f_{i+1,j+1}\Delta Z_j^- + f_{i+1,j-1}\Delta Z_j^+ - f_{i+1,j}\{\Delta Z_j^- + \Delta Z_j^+\} \right]
+ \Theta_m^* \left[ f_{i+1,j}\Delta Z_j^- + f_{i,j-1}\Delta Z_j^+ - f_{i,j}\{\Delta Z_j^- + \Delta Z_j^+\} \right]
\]

Collecting like terms obtain:

\[
B_1 f_{i,j-1} + B_2 f_{i,j} + B_3 f_{i,j+1} + C_1 f_{i+1,j-1} + C_2 f_{i+1,j} + C_3 f_{i+1,j+1} = 0
\]

where

\[
B_1 = -\frac{\Theta_m^* b \Delta t (\Delta Z_j^+)^2}{D} + \frac{\Theta_m^* \sigma^2 \Delta t \Delta Z_j^+}{D}
\]

\[
B_2 = -\frac{\Theta_m^* \sigma^2 \Delta t (\Delta Z_j^- + \Delta Z_j^+)}{D} - \frac{\Theta_m^* b \Delta t \{(\Delta Z_j^-)^2 - (\Delta Z_j^+)^2\}}{D}
\]

\[
B_3 = \frac{\Theta_m^* b \Delta t (\Delta Z_j^+)^2}{D} + \frac{\Theta_m^* \sigma^2 \Delta t \Delta Z_j^-}{D}
\]

\[
C_1 = \frac{\Theta_m \sigma^2 \Delta t \Delta Z_j^+}{D} - \frac{\Theta_m b \Delta t (\Delta Z_j^+)^2}{D}
\]

\[
C_2 = -\frac{\Theta_m \sigma^2 \Delta t (\Delta Z_j^-)^2 - (\Delta Z_j^+)^2}{D} - \frac{\Theta_m \sigma^2 \Delta t \{(\Delta Z_j^- + \Delta Z_j^+)^2\}}{D}
\]

\[
C_3 = \frac{\Theta_m b \Delta t (\Delta Z_j^-)^2}{D} + \frac{\Theta_m \sigma^2 \Delta t \Delta Z_j^-}{D}
\]
If we rearrange we have the following equation:

**Finite-difference scheme for a nonuniform grid and log transformation**

\[
\begin{align*}
    a_j f_{i,j-1} + b_j f_{i,j} + c_j &= a_j f_{i+1,j-1} + b_j f_{i+1,j} + c_j f_{i+1,j+1} \\
    (10.197)
\end{align*}
\]

where:

\[
\begin{align*}
    a_j &= (1 - \Theta_m) \Delta t \left\{ \frac{b(\Delta Z_j^+)^2}{D} - \frac{\sigma^2 \Delta Z_j^+}{D} \right\} \\
    (10.198)
\end{align*}
\]

\[
\begin{align*}
    b_j &= 1 + \Delta t(1-\Theta_m) \left\{ r - \frac{\sigma^2(\Delta Z_j^+ + \Delta Z_j^-)}{D} \\
    &\quad - \frac{b((\Delta Z_j^-)^2 - (\Delta Z_j^+)^2)}{D} \right\} \\
    (10.199)
\end{align*}
\]

\[
\begin{align*}
    c_j &= (1 - \Theta_m) \Delta t \left\{ \frac{-b(\Delta Z_j^-)^2}{D} - \frac{\sigma^2 \Delta Z_j^-}{D} \right\} \\
    (10.200)
\end{align*}
\]

\[
\begin{align*}
    \bar{a}_j &= \Theta_m \Delta t \left\{ \frac{\sigma^2 \Delta Z_j^+}{D} - \frac{b(\Delta Z_j^+)^2}{D} \right\} \\
    (10.201)
\end{align*}
\]

\[
\begin{align*}
    \bar{b}_j &= 1 - \Theta_m r \Delta t - \Theta_m \Delta t \left\{ \frac{b((\Delta Z_j^-)^2 - (\Delta Z_j^+)^2)}{D} \\
    &\quad + \frac{\sigma^2 \{\Delta Z_j^- + \Delta Z_j^+\}}{D} \right\} \\
    (10.202)
\end{align*}
\]

\[
\begin{align*}
    \bar{c}_j &= \Theta_m \Delta t \left\{ \frac{b(\Delta Z_j^-)^2}{D} + \frac{\sigma^2 \Delta Z_j^-}{D} \right\} \\
    (10.203)
\end{align*}
\]

The incorporation of boundary conditions and the solution of Equation 10.197 is similar in manner to that already discussed in Section 10.6.2. If further details are required Code excerpt 10.19, which uses a nonuniform grid to solve the log transformed Black–Scholes equation, can be consulted.

When a uniform grid is used \(\Delta Z_j^+ = \Delta Z_j^- = \Delta Z\) and therefore

\[
D = (\Delta Z_j^+)^2 \Delta Z_j^- + (\Delta Z_j^-)^2 \Delta Z_j^+ = 2(\Delta Z)^3
\]
In these circumstances

\[ a_j = \frac{(1 - \Theta_m) \Delta t}{2 \Delta Z^2} \{ b \Delta Z - \sigma^2 \} \]
\[ b_j = 1 + \Delta t (1 - \Theta_m) \left\{ r - \frac{\sigma^2}{\Delta Z^2} \right\} \]
\[ c_j = (1 - \Theta_m) \Delta t \left\{ -b \frac{\Delta Z}{2 \Delta Z^2} - \frac{\sigma^2}{2 \Delta Z^2} \right\} \]
\[ \bar{a}_j = -\frac{\Theta_m \Delta t}{2 \Delta Z^2} \{ b \Delta Z - \sigma^2 \} \]
\[ \bar{b}_j = 1 - \Theta_m \Delta t \left\{ r + \frac{\sigma^2}{\Delta Z^2} \right\} \]
\[ \bar{c}_j = \frac{\Theta_m \Delta t}{2 \Delta Z^2} \{ b \Delta Z + \sigma^2 \} \]

which are the same as Equations 10.188 to 10.194 in Section 10.6.4.

10.6.6 The double knockout call option

The purpose of this section is to provide an example which illustrates the benefits to be gained from using both the log transformed Black–Scholes equation and also a nonuniform grid.

The problem we will consider is the European double knockout call option with strike price \( E \), and expiry date \( T \). This is a barrier option with both an upper barrier at \( B_U \) and a lower barrier at \( B_L \). If, during the life of the option, the asset price either goes above the upper barrier or below the lower barrier then the option becomes worthless. If, on the other hand, the asset price stays between the barriers then the option has value \( \max(\frac{S_T}{E}, 0) \), where \( S_T \) is the asset price at time \( T \).

This problem has been previously investigated by Boyle and Tian (1998), henceforth referred to as BT, who used an explicit finite-difference method based on a modified trinomial lattice. The method we use here is based on the finite-difference equations given in Section 10.6.5, and all the results in Tables 10.8 to 10.12 were obtained by using the function \( \text{dko\_call} \) which is provided in Code excerpt 10.19.
void dko_call(double lower_barrier, double upper_barrier, double theta_m,
  double S0, double sigma_array[], double sigma_times[], Integer n_sigma, double r,
  double opt_mat, double X, double* option_value, double greeks[], double q,
  Integer ns_below_S0, Integer ns_above_S0, Integer nt, Integer *iflag)
{
  /* Input parameters:
  lower_barrier — the asset price corresponding to the lower barrier,
  upper_barrier — the asset price corresponding to the upper barrier,
  theta_m — the value of theta used for the finite difference method,
  S0 — the current price of the underlying asset,
  sigma_array[] — an array containing values of the volatility: sigma_array[0] is the first value of
  the volatility, sigma_array[1] is the second value of the volatility, etc.,
  sigma_times[] — an array containing the times for different volatilities: sigma_times[0] is the time
  corresponding to the first volatility, sigma_times[1] is the time corresponding to the
  second volatility, etc.,
  n_sigma — the number of elements in sigma_array[], and sigma_times [],
  r — the interest rate,
  opt_mat — the time to maturity,
  X — the strike price,
  q — the continuous dividend yield,
  ns_below_S0 — the number of asset intervals below the current price S0,
  ns_above_S0 — the number of asset intervals above the current price S0,
  nt — the number of time intervals.
  Output parameters:
  option_value — the value of the option,
  greeks[] — the hedge statistics output as follows: greeks[0] is gamma, greeks[1] is delta,
  and greeks[2] is theta,
  iflag — an error indicator.
  */
  double dt, dz, dz1, dz2, zmax, zmin;
  Integer i, j;
  double tmp, t2, t4, dt2;
  Integer ind=0, n1, n2, ns1;
  double ds, log_asset,sig2, alpha, v2, b_fac, temp[4];
  double zero = 0.0;
  Integer barrier_index, ind2;
  double dz_shift, time_step, log_barrier_level1, log_barrier_level2;
  double temp1, temp2, ds_plus, ds_minus, bb, D;
  double curr_time;
  if (S0 > upper_barrier) printf (''ERROR current asset price is greater than upper_barrier
'');
  if (lower_barrier > S0) printf(''ERROR lower barrier is greater than current asset price 
'');
  if (S0 < zero) printf (''ERROR asset price is not > 0 
'');
  if (upper_barrier < lower_barrier) printf (''ERROR upper_barrier must be > lower_barrier
'');
  log_asset = log(S0);
  log_barrier_level1 = log(lower_barrier);
  log_barrier_level2 = log(upper_barrier);
  ds1 = (log_asset-log_barrier_level1)/(double) ns_below_S0;
  n1 = ns_below_S0;
  /* Include 5 extra points above the asset price so that don't get discontinuity in grid spacing
  which may adversely affect the computation of the greeks */
  n2 = ns_above_S0 + 5;
  dz_shift = ds1*5.0; /* shift caused by extra 5 grid points */
  dz2 = (log_barrier_level2-log_asset-dz_shift)/(double) ns_above_S0;
  dt = opt_mat/(double)nt; /* time interval size */
  time_step = dt;
  -n2;
  n1 = n1+n2+2;
  /* Set up the RHS and LHS coefficients a[], b[] and c[] are the LHS coefficients for the unknown option values
  (time step j) a[i1], b[i1] and c[i1] are the values of the RHS coefficients for the known option prices (time step j+1). Note: a[i1], b[i1] and c[i1] are used to form the RHS vector rhs[] of the tridiagonal system. */
  /* Allocate the required arrays (all of size (ns1+2): a, b, c, a1, b1, c1, opt_vals, vals, rhs, z, delta, gamma, work, u */
  /* Set up the RHS and LHS coefficients a[], b[] and c[] are the LHS coefficients for the unknown option values
  (time step j) a[i1], b[i1] and c[i1] are the values of the RHS coefficients for the known option prices (time step j+1). Note: a[i1], b[i1] and c[i1] are used to form the RHS vector rhs[] of the tridiagonal system. */
  /* Set grid line asset values, set one grid spacing to align with the asset price, then don't have to
  interpolate to get the option value */
  z[n1] = log_asset;
  for (i=1; i<n1; ++i) /* This should be the fine mesh */
  z[n1-i] = log_asset - (double)i*dz1;
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for (i=1; i<5; ++i) /* Include 5 extra fine mesh points here */
  z[n1+i] = log_asset + (double)i*ds2;
for (i=6; i<n2+2; ++i) /* The coarse mesh */
  j = i-5;
  z[n1+i] = z[n1+5] + (double)*ds2;

/* Set option values at maturity (for a call). Note : opt_vals[0] and opt_vals[ns1-1] are the lower and upper (put/call) option price boundary values. */
for (i=1; i<ns1; ++i)
  opt_vals[i] = MAX(exp(z[i])-X, zero);

opt_vals[0] = zero;
opt_vals[ns1-1] = zero;
tmp = 1.0-theta_m; /* 1 - theta (for theta method) */
curr_time = -1.0;
ind2 = n*sigma - 1;
for (j=nt-1; j>=2; --j) /* Iterate from maturity to current time */
  if ((ind2 > 0) && (curr_time <= sigma_times[ind2])){
    sig2 = sigma_array[ind2]*sigma_array[ind2];
t2 = time_step/2.0;
    bb = r - q - (sig2/2.0);
    --ind2;
    for (i=1; j<ns1-2; ++i)="/Assign elements of the (ns1-2)*(ns1-2) tridiagonal matrix "/
      ds_plus = z[i+1]-z[i];
      ds_minus = z[i] - z[i-1];
      D = ((ds_plus*ds_plus*ds_minus) + (ds_minus*ds_minus*ds_plus));
      tmp = tmp*time_step/D;
      a[i] = tmp1*bb*ds_plus - tmp1*ds_plus*(sig2);  
      c[i] = -time_step*tmp*(temp1*bb+(sig2*temp2));
      b[i] = time_step*theta_m*(temp1*bb+(sig2*temp2));

    for (i=1; i<n2-2; ++i)="/Solve the lower triangular system Ly = b, where y is stored in array work[]. Compute the elements of y from those of u. 1[i] = a[i]/u[i-1]. */
      for (i=2; i<n2-1;++i)
        u[i] = b[i] - a[i]*u[i-1]/u[i-1];
    if (u[1] == zero) printf ("ERROR in array u \n");
    for (i=2; i<n2-2; ++i) u[i] = b[i] - a[i]*u[i-1]/u[i-1];
    if (u[1] == zero) printf ("ERROR in array u \n");
  }

curr_time = j*dt;
/* Set up the rhs of equation for the theta method */
for (i=2; i<n2-3; ++i)
  rhs[i] = a[i]*opt_vals[i-1]+b[i]*opt_vals[i]+c[i]*opt_vals[i+1];
/* Incorporate the boundary conditions at the upper/lower asset value boundaries */
  rhs[1] = (a[1]-a[1])*opt_vals[0]+b[1]*opt_vals[1]+c[1]*opt_vals[2];
  rhs[ns1-2] = a[n2-2]*opt_vals[ns1-2]+b[n2-1]*opt_vals[ns1-3]+c[n2-2]*opt_vals[ns1-1];
/* Solve the lower triangular systemLy = b, where y is stored in array work[]. Compute the elements of y from those of u. 1[i] = a[i]/u[i-1]. */
      for (i=2; i<n2-1;++i)
        work[i] = rhs[i] - a[i]*work[i-1]/u[i-1];
    if (u[1] == zero) printf ("ERROR in array u \n");
/* Solve the upper (ns1-2)*(ns1-2) triangular systemUx = y (where x = void) */
  opt_vals[ns1-2] = work[ns1-2]/u[ns1-2];
for (i=ns1-2; i>=1; --i)
  opt_vals[i] = (work[i] - c[i]*opt_vals[i+1])/u[i];
if (j==0){
    for (i=0; i<ns1; ++i)
      val[i] = opt_vals[i];
  }
/* Store option values so that can compute theta */
if ((j==1) || (j==2) || (j==4) || (j==8))
  temp[ind] = opt_vals[ind];
++ind;
}
if (greeks){
/* Compute gamma and delta (4th order accuracy) */
Numeric methods and single asset American options

\[
greeks[1] = (-vals[n+2]+8.0*vals[n+1]-vals[n-1]+vals[n-2])/(12.0*dz1);
\]

/* Compute gamma (4th order accuracy) — use chain rule to obtain derivative wrt S */
\[
greeks[0] = (-vals[n+2]+16.0*vals[n+1]-30.0*vals[n]+16.0*vals[n-1]-vals[n-2])/(12.0*dz1*dz1);
\]

/* Compute theta (4th order accuracy) */
\[
greeks[2] = (-temp[0]+8.0*temp[1]-8.0*temp[2]+temp[3])/(12.0*dt);
\]

/* could also compute theta as: greeks[2] = (-temp[0]+4.0*temp[1]-3.0*vals[n])/2.0*dt; */

*option_value = vals[n];

Code excerpt 10.19  Function to compute the value and Greeks of a European double knock out call option using a nonuniform grid and a logarithmic transformation

Inspection of the results shows that that the finite-difference grid method has both greater accuracy and faster convergence than the method proposed by BT. The key to the accuracy achieved by dko_call is a combination of:

- The logarithmic transformation of the Black–Scholes equation.
- The ability to place a grid line at both the upper barrier \( B_U \), and also at the lower boundary \( B_L \).
- The use of a weighted \( \Theta_m \) finite-difference scheme, \( 0 \leq \Theta_m \leq 1 \), instead of the numerically unstable explicit finite-difference method used by a trinomial lattice; which in our notation (see Section 10.6.2) is equivalent to \( \Theta_m = 1 \).

Table 10.8  Estimated value of a European double knock out call option. The values in column two were computed by the function dko_call, and those in column three are the results reported in Table 2 of Boyle and Tian (1998). The model parameters were: current asset price \( S = 95.0 \), exercise price \( E = 100.0 \), volatility \( \sigma = 0.25 \), maturity \( \tau = 1.0 \), interest rate \( r = 0.1 \), dividend yield \( q = 0.0 \). The upper barrier level is set at 140.0 and the lower barrier is set at 90.0. The other parameters used by the function dko_call were: \( nt = n \), \( ns\_below\_S0 = n/2 \), \( ns\_above\_S0 = n/2 \), and \( \Theta_m = 0.5 \) (i.e. the Crank–Nicolson method)

<table>
<thead>
<tr>
<th>Time steps (n)</th>
<th>Estimated value</th>
<th>Boyle and Tian (1998)</th>
</tr>
</thead>
<tbody>
<tr>
<td>50</td>
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</tr>
<tr>
<td>600</td>
<td>1.4584</td>
<td>1.4557</td>
</tr>
<tr>
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<tr>
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<tr>
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<tr>
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<tr>
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</tr>
<tr>
<td>5000</td>
<td>1.4584</td>
<td>1.4580</td>
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</table>
Table 10.9 The estimated values of European down and out call options calculated by the function dko\_call. The fixed model parameters were: exercise price $E = 100.0$, volatility $\sigma = 0.25$, maturity $\tau = 1.0$, interest rate $r = 0.1$, dividend yield $q = 0.0$ and the lower barrier is set at 90.0.

The other parameters used by the function dko\_call were: $nt = n$, ns\_below\_S0 = $n/2$, ns\_above\_S0 = $n/2$, upper\_barrier = 1000.0, lower\_barrier = 90.0, and $\Theta_m = 0.5$

(i.e. the Crank–Nicolson method)

<table>
<thead>
<tr>
<th>Stock price</th>
<th>Time steps</th>
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<th>91</th>
<th>90.5</th>
<th>90.4</th>
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<td></td>
</tr>
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<td>0.6425</td>
<td>0.5149</td>
<td>0.3868</td>
<td>0.2583</td>
<td></td>
</tr>
<tr>
<td>800</td>
<td>2.5065</td>
<td>1.2739</td>
<td>0.6424</td>
<td>0.5148</td>
<td>0.3868</td>
<td>0.2583</td>
<td></td>
</tr>
<tr>
<td>900</td>
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<td>1.2739</td>
<td>0.6424</td>
<td>0.5148</td>
<td>0.3868</td>
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<tr>
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<tr>
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<td>1.2738</td>
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<tr>
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</tr>
</tbody>
</table>

Table 10.10 The estimated values of European down and out call options as calculated by the function dko\_call. The fixed parameters used were: exercise price $E = 100.0$ volatility $\sigma = 0.25$, maturity $\tau = 1.0$, interest rate $r = 0.1$, dividend yield $q = 0.0$, and the lower barrier is set at 90.0. The other parameters used by the function dko\_call were: $nt = n$, ns\_below\_S0 = $n/2$, ns\_above\_S0 = $n/2$, upper\_barrier = 1000.0, lower\_barrier = 90.0, and $\Theta_m = 0.0$ (i.e. the implicit method)

<table>
<thead>
<tr>
<th>Stock price</th>
<th>Time steps</th>
<th>92</th>
<th>91</th>
<th>90.5</th>
<th>90.4</th>
<th>90.3</th>
<th>90.2</th>
</tr>
</thead>
<tbody>
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<td></td>
</tr>
<tr>
<td>400</td>
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<tr>
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<tr>
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<tr>
<td>700</td>
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<tr>
<td>800</td>
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<td>0.2583</td>
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<tr>
<td>900</td>
<td>2.5060</td>
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<tr>
<td>1000</td>
<td>2.5060</td>
<td>1.2737</td>
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<tr>
<td>2000</td>
<td>2.5061</td>
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<td></td>
</tr>
<tr>
<td>Closed form</td>
<td>2.5063</td>
<td>1.2738</td>
<td>0.6424</td>
<td>0.5148</td>
<td>0.3868</td>
<td>0.2583</td>
<td></td>
</tr>
</tbody>
</table>
It should be mentioned that the function `dko_call` could, without much difficulty, be modified to deal with:

- American double knockout call options
- European double knockout call options
- American double knockout put options

and also a range of other variations which may include lockout periods, rebates, etc. In particular, options with time varying barrier levels can be dealt with by using grid lines to locate the barrier position at each time instant.
10.7 PRICING AMERICAN OPTIONS USING A STOCHASTIC LATTICE

In this section, we consider the use of Monte Carlo simulation and stochastic lattices to price American options. Information on the use of Monte Carlo simulation to value both single asset and multiasset European options is provided in Sections 11.1 and 12.3. The main difficulty in using simulation to value American options is the need to incorporate optimal early exercise policies. The standard simulation algorithms for valuing European contracts are forward in time. That is each price path, which contributes to the value of the option, is generated by stepping forward from current time, \( t \), to option maturity, \( t + \tau \), where \( \tau \) is the duration of the option. For instance if there are \( n \) equispaced time steps of size \( \Delta t \), and only one underlying asset then we use the asset values \( S_i \), \( i = 0, \ldots, n \), where \( S_i \) corresponds to the asset value at the \( i \)th time instant, \( t_i \), and \( t_0 = t \). Here \( S_{i+1} \) is generated from the previous asset value \( S_i \) as follows:

\[
\frac{S_{i+1}}{S_i} = dS_i, \quad \text{for} \quad i = 0, \ldots, n - 1
\]

where \( dS_i \) is a random variate taken from a given distribution. When \( S_i \) follows GBM we have from Equation 12.5 that:

\[
\frac{S_{i+1}}{S_i} = \exp\left\{ (r - \sigma_i^2/2)\Delta t + \sigma_i dX_i \right\}, \quad i = 0, \ldots, n - 1
\]

where \( dX_i \sim N(0, \Delta t) \) and the usual definitions are used for \( \sigma_i \) and \( r \).

For European exotic options (such as time dependent barrier options) the value of a particular price path will depend on the asset values \( S_i \), \( i = 0, \ldots, n \). This is not true of European vanilla options whose value only depends on \( S_n \), the underlying asset price at option maturity. The Monte Carlo approximation to the value of a European option is thus:

\[
f = \sum_{j=1}^{nsim} p_j(n_j) / nsim
\]

where \( nsim \) is the number of simulations used, \( n_j \) is the number of time steps associated with the \( j \)th price path, and \( p_j(n_j) \) is the value of the \( j \)th price path. In the case of European vanilla options we can use \( n_j = 1, j = 1, \ldots, nsim \); the accuracy obviously improves with increasing \( nsim \).

The valuation of American style options, which include the possibility of early exercise, is more complicated. In Section 10.4, we described the use of binomial lattices to price American options when the underlying asset price process is GBM. Dynamic programming was used and the option prices were computed by working backwards in time through the lattice. The application of Monte Carlo methods for pricing American options is described in Fu et al. (2001), Tilley (1993), Barraquand and Martineau (1995) and also Boyle et al. (1997). Here we will outline the stochastic lattice approach discussed in Broadie and Glasserman (1997), where both a high estimator and a low estimator of the American option value are calculated. Since both of these biased estimators converge (with increasing number of simulations and lattice nodes) to the

Pricing Assets
true option value we will only consider how to compute the high estimator, \( \theta_H \). We summarize the approach as follows:

- Set the parameters
- Generate the lattice asset prices
- Compute the lattice option prices
- Compute the Monte Carlo estimate.

We will now consider each of these steps in more detail.

**Set the parameters**

First we set the simulation parameters, that is: \( n_{sim} \) is the number of lattice simulations, \( b \) is the number of branches per lattice node and \( d \) is the number of time instants in the lattice. Note: This definition of \( d \) here is different from that used in the original paper by Broadie and Glasserman (1997) where \( d \) is defined as the number of time steps in the lattice.

**Generate the lattice asset prices**

Next we generate the asset prices for the \( p \)th stochastic lattice. This is done forwards in time by using a modified version of Equation 10.205. Since the lattice is nonrecombining at the \( i \)th lattice time instant there are \( b^i \) nodes/asset prices. This contrasts with the binomial lattice of Section 10.4 where the asset prices at a given time step are arranged in ascending order, that is \( S_j^i \) increases with increasing \( j \). We will denote the \( j \)th value at the \( i \)th time step by \( S_j^i \). For example in Figure 10.13, where \( b = 3 \) and \( d = 3 \), we have for the first time step

\[
S_1^1 = 115, \quad S_2^1 = 60, \quad \text{and} \quad S_3^1 = 114
\]

and for the second time step

\[
S_1^2 = 116, \quad S_2^2 = 90, \quad S_3^2 = 149, \ldots, \quad S_2^3 = 102, \quad S_2^4 = 88, \quad S_2^5 = 80
\]

The \( k \)th asset price at the \( i \)th time step, \( S_k^i \) then generates the following asset prices at the \((i + 1)\)th time step:

\[
\frac{S_{k+1}^{(k-1)b+j}}{S_k^i} = dS_j^i, \quad j = 1, \ldots, b, \quad k = 1, \ldots, b^i
\]

where \( dS_j^i \) is, as before, a random variate from a given distribution. When \( S_j \) follows GBM we therefore have:

\[
\frac{S_{k+1}^{(k-1)b+j}}{S_k^i} = \exp\{ (r - \sigma^2_t/2)\Delta t + \sigma_d X_t \}, \quad j = 1, \ldots, b, \quad k = 1, \ldots, b^i
\]

**Compute the lattice option prices**

The method used to compute the option values is similar to that used by the binomial lattice. The main difference is that there are now \( b \) branches per node
instead of two. The option values are computed by starting at the lattice terminal nodes and then iterating backwards. Here we denote the \( k \)th option value at the \( i \)th time step by \( f^k_i \).

The option values at the terminal nodes, time instant \( t_{d-1} \), are computed in the usual manner. For a put we have:

\[
f^k_{d-1} = \max(E - S^k_{d-1}, 0), \quad k = 1, \ldots, b^{d-1}
\]

where \( E \) is the exercise price.

The option values at the \((i - 1)\)th time step are computed from those at the \( i \)th time step as follows:

\[
f^{k}_{i-1} = \max(g^k_i, h^k_i)
\]

where

\[
h^k_{i-1} = \frac{\exp(-r\Delta t)}{b} \sum_{j=1}^{b} f^k_{j-1} b^{j-1}
\]
and

\[ g^k_{t-1} = \max(E - S^k_{t-1}, 0) \]

The option value for the \( p \)th stochastic lattice is therefore:

\[ \theta^p_H = f^0_0 = \frac{\exp(-r\Delta t)}{b} \sum_{j=1}^{b} f^j_1 \]

Figure 10.14 shows the option values for an American call with strike price \( E = 100 \) and interest rate \( r = 0 \), when the lattice asset prices in Figure 10.13 are been used. To make things as clear as possible we will show how the value of each node is computed.

**Terminal nodes**

The option values at the terminal nodes are:

\[
\begin{align*}
    f^1_2 &= \max(116 - 100, 0) = 16, \\
    f^2_2 &= \max(90 - 100, 0) = 0, \\
    f^3_2 &= \max(149 - 100, 0) = 49
\end{align*}
\]

![Figure 10.14](image-url)
Pricing Assets

\[
\begin{align*}
    f_2^4 &= \max(32 - 100, 0) = 0, \\
    f_2^5 &= \max(50 - 100, 0) = 0, \\
    f_2^6 &= \max(48 - 100, 0) = 0 \\
    f_2^7 &= \max(102 - 100, 0) = 2, \\
    f_2^8 &= \max(88 - 100, 0) = 0, \\
    f_2^9 &= \max(80 - 100, 0) = 0
\end{align*}
\]

**Time step 1**

Here we have:

\[
\begin{align*}
    g_1^1 &= \max(115 - 100, 0) = 15, \quad g_1^2 = \max(60 - 100, 0) = 0, \\
    g_1^3 &= \max(114 - 100, 0) = 14
\end{align*}
\]

Since \( r = 0 \) we have \( \exp(-r\Delta t) = 1 \) which gives:

\[
\begin{align*}
    h_1^1 &= \frac{1}{3} \{ f_2^4 + f_2^5 + f_2^6 \} = \frac{1}{3} \{ 16 + 0 + 49 \} = 21.7 \\
    h_1^2 &= \frac{1}{3} \{ f_2^4 + f_2^5 + f_2^6 \} = \frac{1}{3} \{ 0 + 0 + 0 \} = 0 \\
    h_1^3 &= \frac{1}{3} \{ f_2^4 + f_2^5 + f_2^6 \} = \frac{1}{3} \{ 2 + 0 + 0 \} = 0.66
\end{align*}
\]

The option values are then computed as follows:

\[
\begin{align*}
    f_1^1 &= \max(h_1^1, g_1^1) = \max(21.7, 15) = 21.7 \\
    f_1^2 &= \max(h_1^2, g_1^2) = \max(0, 0) = 0 \\
    f_1^3 &= \max(h_1^3, g_1^3) = \max(0.66, 14.0) = 14.0
\end{align*}
\]

**Time step 0**

Here

\[
\begin{align*}
    g_0^1 &= \max(101 - 100, 0) = 1, \quad \text{and} \\
    h_0^1 &= \frac{1}{3} \{ f_1^1 + f_1^2 + f_1^3 \} = \frac{1}{3} \{ 21.7 + 0 + 0.66 \} = 11.9
\end{align*}
\]

The final value of the option for this particular lattice is therefore:

\[
    f_1^1 = \max(h_0^1, g_0^1) = \max(11.9, 1) = 11.9
\]

**Compute the Monte Carlo estimate**

The Monte Carlo estimate, \( \theta_H \), is computed as the average of \( \theta_H^p, p = 1, \ldots, n_{sim} \), where \( n_{sim} \) is the number of simulations.

\[
\theta_H = \frac{1}{n_{sim}} \sum_{i=1}^{n_{sim}} \theta_H^i
\]
Below, in Code excerpt 10.20, we provide a computer program which prices single asset American put and call options using a stochastic lattice. The method used by the program is the depth-first procedure outlined in Broadie and Glasserman (1997), which has the advantage that the memory requirements are only of order \( b \times d \); as before \( b \) is the number of branches per node and \( d \) is the number of time intervals.

Here it is assumed the underlying asset following GBM and the NAG function \( \text{g05ddc}(M, S) \) is used to generate a normal distribution with mean \( M \) and standard deviation \( S \). We can therefore check the accuracy of the simulation with that obtained by a closed form solution which assumes a lognormal asset distribution, in this case the formula in Geske and Johnson (1984).

However, the real power of this method is when the underlying asset follows a more realistic process which is non-Gaussian and time varying. The only modification to the code is to replace the call to \( \text{g05ddc} \) with that of another probability distribution and supply the time varying parameters to it.

```c
#include <nag.h>
#include <stdio.h>
#include <nag_stdlib.h>
#include <math.h>
// Stochastic lattice for computing the value of American and European options via Monte Carlo simulation.
// Here we assume that the asset prices have a lognormal distribution, and so generate normal variates; this assumption can easily be removed.

void__cdecl main()
{
  long i, j, jj, is_put, is_american, w[200], num_simulations, b, d, seed;
  double T, time_step, sqrt_time_step, opt_value, pay_off, log_fac, asset_price;
  double temp, opt_val, hold, sum_opt_val, disc;
  double tot_opt_vals, X, drift_term, std_term, S0, q, r, sigma, zero = 0.0;
  double v[200][60], opt_v[200][60];

  printf("Stochastic lattice for pricing European and American options \n");
  is_put = 1; // If is_put = 0 then a call option, otherwise a put option
  T = 1.0; // The time to maturity of the option
  is_american = 1; // If is_american = 0 then an European option, otherwise an American option
  sigma = 0.2; // The volatility of the underlying asset
  X = 110.0; // The strike price
  S0 = 100.0; // The current price of the underlying asset
  r = 0.1; // The risk free interest rate
  q = 0.05; // The continuous dividend yield
  d = 4; // The number of time steps, the number time intervals = d - 1
  b = 50; // The number of branches per node in the lattice

  time_step = T/(double)(d - 1); // time step = T/(number of time intervals)
  sqrt_time_step = sqrt(time_step);
  disc = exp(-r*time_step); // The discount factor between time steps
  std_term = sigma*sqrt(time_step); // The standard deviation of each normal variate generated
  drift_term = (r - q)*sigma*sigma*0.5*time_step; // The mean value of each normal variate generated
  seed = 111; // The seed for the random number generator
  g05cbc(seed);
  tot_opt_vals = zero;
  num_simulations = 100;
  for (jj = 1; jj <= num_simulations; ++jj) {
    v[1][1] = 20;
    w[1] = 1;
    asset_price = S0;
    for (j = 2; j <= d; ++j) {
      log_fac = g05ddc(drift_term, std_term); // A normal variate: mean=drift_term, standard deviation=std_term
      asset_price = asset_price*exp(log_fac); // Compute the new asset price: assuming a lognormal distribution
      v[1][j] = asset_price;
    }
    j = d;
  }
}
```
while \((j > 0)\) { 
  if \((j == d) \&\& (w[j] < b)\) {// CASE 1::Terminal node, set asset prices for b branches, and option values
    for \(b-1\) branches
      if (is_put) {
        pay_off = MAX \((X-v[w[j]][j], zero)\); //CASE1::Terminalnode,setassetpricesforbbranches,andoptionvalues
      } else {
        pay_off = MAX \((v[w[j]][j]-X, zero)\);
      } 
      opt_v[w[j]][j] = pay_off;
      asset_price = v[w[j-1]][j-1];
      log_fac = g05ddc(drift_term, std_term);
      v[w[j]+1][j] = asset_price*exp(log_fac);
      w[j] = w[j] + 1;
    }
  } else if \((j == d) \&\& (w[j] == b)\) {// CASE 2::Terminal node, set option value for last branch
    if (is_put) {
      pay_off = MAX \((X-v[w[j]][j], zero)\);
    } else {
      pay_off = MAX \((v[w[j]][j]-X, zero)\);
    } 
    opt_v[w[j]][j] = pay_off;
    w[j] = 0;
    j = j - 1;
  } else if \((j < d) \&\& (w[j] < b)\) {//CASE3::Internal node,
    //calculate option value for node (parent wrt to cases 1 & 2)
    sum_opt_val = zero; // Also generate a new terminal node and set asset values.
    for \((i = 1; i < b; ++i)\) {
      sum_opt_val += opt_v[i][j+1];
    }
    temp = sum_opt_val/(double)b;
    hold = temp*disc;
    if (is_american) { // An American option
      if (is_put) {
        pay_off = MAX \((X-v[w[j]][j], zero)\); // pay off for a put option
      } else {
        pay_off = MAX \((v[w[j]][j]-X, zero)\); // pay off for a call option
      } 
      opt_val = MAX(pay_off, hold);
    } else { // A European option
      opt_val = hold;
    } 
    opt_v[w[j]][j] = opt_val;
  } elseif \((j < d) \&\& (w[j] == b)\) {//CASE4::Internalnode,calculatetheoptionvalueforthelastbranch
    sum_opt_val = zero;
    for \((i = 1; i < b; ++i)\) {
      sum_opt_val += opt_v[i][j+1];
    }
    temp = sum_opt_val/(double)b;
    hold = temp*disc;
    if (is_american) { // An American option
      if (is_put) {
        pay_off = MAX \((X-v[w[j]][j], zero)\); // pay off for a put option
      } else {
        pay_off = MAX \((v[w[j]][j]-X, zero)\); // pay off for a call option
      } 
      opt_val = MAX(pay_off, hold);
    } else { // A European option
      opt_val = hold;
    } 
    opt_v[w[j]][j] = opt_val;
  } while \((j > 0)\);
**Numeric methods and single asset American options**

```c
pay_off = MAX(v[w[j]][j]-X, zero); // pay off for a call option
}
opt_val = MAX(pay_off, hold);
else { // A European option
    opt_val = hold;
}
opt_v[w[j]][j] = opt_val;
w[j] = 0;
j = j + 1;
}
tot_opt_vals = tot_opt_vals + opt_v[1][1]; // Sum the option values for each simulation
}
opt_value = tot_opt_vals/(double)num_simulations; // Compute the average option value
printf (''The estimated option value = %12.4f''
'', opt_value);
```

**Code excerpt 10.20** A computer program which uses a stochastic lattice to value American and European options

In Table 10.13 below we present computed values of an American put option with maturity $\tau$, that can only be exercised at the following four times: $t$, $t + \tau/3$, $t + 2\tau/3$, and $t + \tau$, where $t$ is the current time.

The column labelled $MC_{50}^{100}$ presents the results obtained using 100 simulations of a stochastic lattice with 50 branches per node, and the column labelled $MC_{250}^{100}$ presents the values computed using a single stochastic lattice with 250 branches per node. These values demonstrate that one high accuracy stochastic lattice can give better results than using the average of 100 lower accuracy lattices. In the last two columns we present the computed binomial lattice values for the American put and also the corresponding European put. The binomial lattice had 6000

**Table 10.13** American put options values, computed using the stochastic lattice given in Code excerpt 10.20, with four the exercise times $t$, $t + \tau/3$, $t + 2\tau/3$, and $t + \tau$. The option parameters used were: $r = 0.1$, $q = 0.05$, $\tau = 1.0$, $\sigma = 0.2$, $S = 100.0$ and $E$, the strike price, is varied from 70 to 130. The column labelled $MC_{50}^{100}$ refers to the results obtained using $d = 4$, $b = 50$, $num_{simulations} = 100$, and the column labelled $MC_{250}^{100}$ refers to the results obtained using $d = 4$, $b = 50$, $num_{simulations} = 1$. The *true* values are those given in Broadie and Glasserman (1997), and were computed with the formula in Geske and Johnson (1984). The absolute error, $ABS(stochastic\_lattice\_value-true\_value)$, is given in brackets. The last two columns are the computed results using an accurate (6000 time step) binomial lattice; the column labelled $BL_{A}$ contains the American put option values, and the column labelled $BL_{E}$ contains the European put option values. It can be seen that in all cases the American put option has a significant early exercise premium.

<table>
<thead>
<tr>
<th>E</th>
<th>$MC_{50}^{100}$</th>
<th>$MC_{250}^{100}$</th>
<th>True</th>
<th>$BL_{A}$</th>
<th>$BL_{E}$</th>
</tr>
</thead>
<tbody>
<tr>
<td>70</td>
<td>0.118 (0.003)</td>
<td>0.123 (0.002)</td>
<td>0.121</td>
<td>0.126</td>
<td>0.120</td>
</tr>
<tr>
<td>80</td>
<td>0.663 (0.007)</td>
<td>0.672 (0.002)</td>
<td>0.670</td>
<td>0.696</td>
<td>0.654</td>
</tr>
<tr>
<td>90</td>
<td>2.317 (0.014)</td>
<td>2.307 (0.004)</td>
<td>2.303</td>
<td>2.389</td>
<td>2.198</td>
</tr>
<tr>
<td>100</td>
<td>5.830 (0.099)</td>
<td>5.720 (0.011)</td>
<td>5.731</td>
<td>5.928</td>
<td>5.301</td>
</tr>
<tr>
<td>110</td>
<td>11.564 (0.223)</td>
<td>11.361 (0.020)</td>
<td>11.341</td>
<td>11.770</td>
<td>10.155</td>
</tr>
<tr>
<td>120</td>
<td>20.205 (0.205)</td>
<td>20.000 (0.000)</td>
<td>20.000</td>
<td>20.052</td>
<td>16.547</td>
</tr>
<tr>
<td>130</td>
<td>30.054 (0.054)</td>
<td>30.000 (0.000)</td>
<td>30.000</td>
<td>30.000</td>
<td>24.065</td>
</tr>
</tbody>
</table>
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time steps and it was possible to exercise the option at every time step. It can be seen that the computed binomial option values for the American put are higher than the true values, which only permit the option to be exercised at four distinct times. This is in agreement with the extra flexibility present in the binomial lattice. Inspection of the computed European put and American put binomial option values also reveals that the American put option has a significant early exercise premium.
Chapter 11

Monte Carlo simulation

11.1 INTRODUCTION

Monte Carlo simulation and random number generation are techniques that are widely used in financial engineering as a means of assessing the level of exposure to risk. Typical applications include the pricing of financial derivatives and scenario generation in portfolio management. In fact many of the financial applications that use Monte Carlo simulation involve the evaluation of various stochastic integrals which are related to the probabilities of particular events occurring.

For instance in Section 9.1 we gave the value of a European call option as:

\[ c(S, E, \tau) = \exp\{-r\tau\} \int_{-\infty}^{\infty} p(S_T) \max(E - S_T, 0) dS_T \]

and that of a put as:

\[ p(S, E, \tau) = \exp\{-r\tau\} \int_{-\infty}^{\infty} p(S_T) \max(E - S_T, 0) dS_T \]

where \( E \) is the strike price, \( T \) is the expiry date, \( t \) is the current time, \( \tau = T - t \), \( r \) is the riskless interest rate and \( p(S_T) \) is the probability that the asset will have market value \( S_T \) at maturity.

In many cases however, the assumptions of constant volatility and a lognormal distribution for \( S_T \) are quite restrictive. Real financial applications may require a variety of extensions to the standard Black–Scholes model. Common requirements are for: nonlognormal distributions, time varying volatilities, caps, floors, barriers, etc. In these circumstances it is often the case that there is no closed form solution to the problem. Monte Carlo simulation can then provide a very useful means of evaluating the required integrals.

When we evaluate the integral of a function, \( f(x) \), in the dimensional unit cube, \( I^S \), by the Monte Carlo method we are in fact calculating the average of the function at a set of randomly sampled points. This means that each point adds linearly to the accumulated sum that will become the integral and also linearly to the accumulated sum of squares that will become the variance of the integral.

When there are \( N \) sample points the integral is:

\[ \nu = \frac{1}{N} \sum_{i=1}^{N} f(x^i) \]  
(11.1)
where \( \nu \) is used to denote the approximation to the integral and \( x^1, x^2, \ldots, x^N \) are the \( N \), \( s \)-dimensional, sample points. If a pseudorandom number generator is used the points \( x^i \) will be (should be) independently and identically distributed. From standard statistical results we can then estimate the expected error of the integral as shown below.

If we set \( \chi^i = f(x^i) \) then since \( x^i \) is independently and identically distributed \( \chi^i \) is also independently and identically distributed. The mean of \( \chi^i \) is \( \nu \) and we will denote the variance as \( \text{Var}(\nu) = \Delta^2 \). It is a well-known statistical property that the variance of \( \nu \) is given by \( \text{Var}(\nu) = N^{-1} \Delta^2 \), see Appendix F.1 for further details. We can therefore conclude that the estimated integral \( \nu \) has a standard error of \( N^{-1/2} \). This means that the estimated error of the integral will decrease at the rate of \( N^{-1/2} \).

It is possible to achieve faster convergence than this if the sample points are chosen to lie on a Cartesian grid. If we sample each grid point exactly once then the Monte Carlo method effectively becomes a deterministic quadrature scheme, whose fractional error decreases at the rate of \( N^{-1} \) or faster. The trouble with the grid approach is that it is necessary to decide in advance how fine it should be, and all the grid points need to be used. It is therefore not possible to sample until some convergence criterion has been met.

Quasirandom number sequences seek to bridge the gap between the flexibility of pseudorandom number generators and the advantages of a regular grid. They are designed to have a high level of uniformity in multidimensional space, but unlike pseudorandom numbers they are not statistically independent.

### 11.2 PSEUDORANDOM AND QUASIRANDOM SEQUENCES

Here we consider the generation of multidimensional pseudorandom and quasirandom sequences to approximate the multidimensional uniform distribution over the interval \([0, 1]\), that is the distribution \( U(0,1) \).

Quasirandom numbers are also called low discrepancy sequences. The discrepancy of a sequence is a measure of its uniformity and is defined below.

Given a set of points \( x^1, x^2, \ldots, x^N \in I^S \) and a subset \( G \subset I^S \), define the counting function \( S_N(G) \) as the number of points \( x^i \in G \). For each \( x = (x_1, x_2, \ldots, x_s) \in I^S \), let \( G_x \) be the rectangular \( s \)-dimensional region \( G_x = [0, x_1) \times [0, x_2) \times \cdots \times [0, x_s) \), with volume \( x_1, x_2, \ldots, x_n \). Then the discrepancy of the points \( x^1, x^2, \ldots, x^N \) is given by:

\[
D_N(x^1, x^2, \ldots, x^N) = \sup_{x \in I^S} |S_N(G_x) - N x_1, x_2, \ldots, x_s|
\]

The discrepancy is therefore computed by comparing the actual number of sample points in a given volume of multidimensional space with the number of sample points that should be there assuming a uniform distribution.

It can be shown that the discrepancy of the first terms of quasirandom sequence has the form:

\[
D_N(x^1, x^2, \ldots, x^N) \leq C_s (\log N)^S + O((\log N)^{S-1})
\]

for all \( N \geq 2 \).

The principal aim in the construction of low-discrepancy sequences is thus to find sequences in which the constant is as small as possible. Various sequences have been

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constructed to achieve this goal. Here we consider the following quasirandom sequences proposed by Niederreiter (1992), Sobol (1967), and Faure (1982).

The results of using various random number generators are shown below. Figures 11.1 to 11.3 illustrate the visual uniformity of the sequences. They were created by generating one thousand, sixteen dimensional $U(0,1)$ sample points, and then plotting the 4th dimension component of each point against its 5th dimension component.

In Figure 11.1, it can be seen that the pseudorandom sequence exhibits clustering of points, and there are regions with no points at all.

Visual inspection of Figures 11.2 and 11.3 show that both the Sobol and Niederreiter quasirandom sequences appear to cover the area more uniformly.

It is interesting to note that the Sobol sequence appears to be a structured lattice which still has some gaps. The Niederreiter sequence on the other hand appears to be more irregular and covers the area better. However, we cannot automatically conclude from this that the Niederreiter sequence is the best. This is because we have not considered all the other possible pairs of dimensions.

Perhaps the easiest way to evaluate the random number sequences is to use them to calculate an integral.

In Figure 11.4 Monte Carlo results are presented for the calculation of the six-dimensional integral:

$$I = \int_0^1 \int_0^1 \int_0^1 \int_0^1 \int_0^1 \int_0^1 \cos(ix_1)dx_1dx_2dx_3dx_4dx_5dx_6$$

![Pseudorandom sequences](image_url)

*Figure 11.1* The scatter diagram formed by one thousand points from a sixteen dimensional $U(0,1)$ pseudorandom sequence. For each point the 4th dimension component is plotted against the 5th dimension component.
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Figure 11.2  The scatter diagram formed by one thousand points from a sixteen dimensional \( U(0, 1) \) Sobol sequence. For each point the 4th dimension component is plotted against the 5th dimension component.

Figure 11.3  The scatter diagram formed by one thousand points from a sixteen dimensional \( U(0, 1) \) Niederreiter sequence. For each point the 4th dimension component is plotted against the 5th dimension component.
The exact value of this integral is:

\[ I = \prod_{i=1}^{6} \sin(i) \]

which for \( i = 6 \), gives \( I = -0.0219 \).

It can be seen that the pseudorandom sequence gives the worst performance. But as the number of points increases its approximation to the integral improves. Of the quasirandom sequences it can be seen that the Faure sequence has the worst performance, whilst both the Sobol and Niederreiter sequences give rapid convergence to the solution.

Finance literature contains many references to the benefits of using quasirandom numbers for computing important financial integrals. For instance Brotherton-Ratcliffe (1994) discusses the use of Sobol sequences for the valuation of geometric mean stock options, and provides results which show that the root mean squared pricing error obtained using quasirandom numbers is considerably less than that computed with pseudorandom numbers. Another financial application of quasirandom numbers is the efficient pricing mortgage backed securities, Caflisch et al. (1997). Here Brownian bridge techniques are employed to reduce the effective dimension of the problem and thus provide greater pricing accuracy than if pseudorandom numbers were used.

11.2.1 Portfolio allocation

In this example quasirandom numbers are applied to a Markowitz style portfolio allocation problem, see Markowitz (1989, 1994). It should be mentioned that many
portfolio problems can be solved very efficiently using Newton (gradient based) numerical optimization software to minimize a given object function subject to certain constraints. However, this approach fails if the gradient of the objective function is discontinuous; this is not the case when (quasi) random numbers are used.

We will start with an initial portfolio and use quasirandom numbers to plot out the feasible region in which portfolios must lie in order to satisfy the portfolio constraints and transaction costs. The asset vector $X$ specifies the amount of each asset in a given portfolio, and the initial portfolio allocation is denoted by the asset vector $X_I$. In particular we would like to be able to identify efficient portfolios, that is those which for a given portfolio return minimize the portfolio risk. The problem of determining efficient portfolios can be expressed as follows:

$$\text{minimize } V = X^T C X$$

subject to the following constraints:

$$\sum_{i=1}^{n} X_i = 1, \quad L_i < X_i < U_i, \quad i = 1, \ldots, n$$

and

$$E = \mu X - \sum_{i=1}^{n} \phi_i |X_i^I - X_i|$$

where $E$ is the expected portfolio return, $V$ is the portfolio risk, $\mu$ is the vector of expected asset returns, $C$ is the covariance matrix of the assets, $X$ is an asset vector which specifies the amount of each asset $ABS(X)$ is the absolute value of $X$, and $L_i$, $U_i$ are the respective lower and upper bounds on the $i$th asset.

The transaction costs, $\phi_i$, that are used in equation are $\phi_i = \phi_s$ when $X_i^I > X_i$, and $\phi_i = \phi_b$ when $X_i^I < X_i$, where $\phi_s$ is the cost of selling shares and $\phi_b$ is the cost of buying shares.

Here we consider a twenty asset portfolio, $n = 20$, with either no transaction costs or $\phi_b = 0.07$ and $\phi_s = 0.04$. The initial asset vector $X_I$ is such that there are equal amounts of each asset, that is

$$X_i^I = \frac{1}{20}, \quad i = 1, \ldots, 20$$

```vba
Private Sub Command2_Click()
    Dim quasi(50), fcall, method1, n As Variant
    Dim i, j, k, X, Y, num As Long
    Dim XI(100), XP(100), V, E As Double
    Dim Ret(100), C(50, 50) As Double
    Dim sum As Double
    Dim buy_cost, sell_cost As Double
    Dim count, maxcount As Long
    Dim max_holding(50), min_holding(50) As Double
    Picture1.Cls
    Picture1.DrawWidth = 4
    n = 20
    For i = 0 To n - 1  ' set up the expected asset returns
        Ret(i) = 0.008 * CDbl(i)
    Next i
```
**Monte Carlo simulation**

\[
\text{Ret}(n-1) = 0.06 \\
\text{For } i = 0 \text{ To } n - 1 \quad \text{set up the initial portfolio} \\
\text{XI}(i) = \text{Ret}(i) / \text{CBd}(n) \\
\text{Next } i \\
\text{For } i = 0 \text{ To } n - 1 \quad \text{set up the covariance matrix} \\
\text{For } j = 0 \text{ To } n - 1 \quad \text{C}(i, j) = 0.01 \times \text{CBd}(i + j) \\
\text{If } (i = j) \text{ Then} \\
\quad \text{C}(i, j) = \text{CBd}(i) \times 0.6 \\
\text{End If} \\
\text{Next } j \\
\text{Next } i \\
\text{C}(6, 4) = 0.4 \\
\text{C}(4, 6) = \text{C}(6, 4) \\
\text{C}(18, 10) = -0.8 \\
\text{C}(10, 18) = \text{C}(18, 10) \\
f\text{call} = 1 \quad \text{Use Sobol sequences} \\
\text{COMP11.generate fcall, n, method1, quasi(0)} \\
\text{MsgBox "Starting quasi-random generation"} \\
f\text{call} = 0 \\
buy\_cost = 0\# \quad \text{set the transaction costs} \\
sell\_cost = 0\# \\
'b\text{buy\_cost} = 0.07 \\
's\text{ell\_cost} = 0.04 \\
\text{For } i = 0 \text{ To } n - 1 \quad \text{set the maximum and minimum constraints} \\
\text{max\_holding}(i) = 0.1 \\
\text{min\_holding}(i) = 0.005 \\
\text{Next } i \\
\text{max\_holding}(0) = 0.4 \\
\text{max\_holding}(1) = 0.4 \\
\text{max\_holding}(2) = 0.1 \\
\text{max\_holding}(18) = 0.7 \\
\text{max\_holding}(19) = 0.8 \\
c\text{ount} = 0 \\
c\text{max\_count} = 500000 \\
\text{Do While (count < max\_count)} \\
\text{COMP11.generate fcall, n, method1, quasi(0)} \\
\text{sum = 0\#} \\
\text{For } j = 0 \text{ To } n - 2 \\
\quad \text{XP}(j) = \text{quasi}(j) \times (\text{max\_holding}(j) - \text{min\_holding}(j)) + \text{min\_holding}(j) \\
\quad \text{sum = sum + XP}(j) \\
\text{Next } j \\
\text{If (sum < -1) Then} \\
\quad \text{XP}(n - 1) = -1\# - sum \\
\quad E = \# \\
\quad \text{For } j = 0 \text{ To } n - 1 \\
\quad \quad E = E + \text{Ret}(j) \times \text{XP}(j) \\
\quad \text{Next } j \\
\text{For } j = 0 \text{ To } n - 1 \quad \text{transaction costs} \\
\text{If (XP}(j) > \text{XI}(j)) \text{ Then} \\
\quad E = E - \text{buy\_cost} \times (\text{XP}(j) - \text{XI}(j)) \\
\quad \text{End If} \\
\text{If (XP}(j) < \text{XI}(j)) \text{ Then} \\
\quad E = E - \text{sell\_cost} \times (\text{XI}(j) - \text{XP}(j)) \\
\quad \text{End If} \\
\text{Next } j \\
\text{V = 0\#} \\
\text{For } j = 0 \text{ To } n - 1 \\
\quad \text{For } k = 0 \text{ To } n - 1 \\
\quad \quad \text{V = V + C}(j, k) \times \text{XP}(j) \times \text{XP}(k) \\
\quad \text{Next } k \\
\text{Next } j \\
\text{Y = 5000 - E \times 4000 \times 8} \\
\text{X = V \times 3000} \\
\text{Picture1.PSet (X, Y), RGB(0, 0, 255)} \\
\text{End If} \\
\text{count = count + 1} \\
\text{Loop} \\
\text{End Sub}

**Code excerpt 11.1** Visual Basic code which uses a twenty-dimensional quasirandom Sobol sequence to plot the feasible region of a constrained portfolio consisting of twenty assets, and possible transaction costs.
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The basic method is very simple, and full details can be found in Code excerpt 11.1. We generate a quasirandom asset vector \( X \), and then check that its elements satisfy the constraints given in Equation 11.3. If they do not then we reject the asset vector \( X \) and generate another one. If the asset vector \( X \) does satisfy the constraints in Equation 11.3 we use Equation 11.2 to calculate the portfolio risk, \( V \), and Equation 11.4 to calculate the portfolio return, \( E \). The point \( E, V \) is then plotted on the diagram. This process is repeated a specified number of times.

In Code excerpt 11.1 we generate 500,000 vectors \( Q \) from a \( U(0, 1) \) twenty-dimensional quasirandom Sobol sequence, and the elements of each vector satisfy \( 0 \leq Q_i \leq 1 \), for \( i = 1, \ldots, 20 \). In order to ensure that not too many vectors get rejected we generate the portfolio allocation vector by using the following transformation:

\[
X_i = Q_i(U_i - L_i) + L_i, \quad i = 1, \ldots, 20
\]

where \( L_i \) and \( U_i \) have already been mentioned in Equation 11.3.

The resulting return/risk plots for the portfolios are shown in Figures 11.5 and 11.6. In both cases the efficient frontier is clearly visible and, as expected, the return in Figure 11.5 without transaction costs is higher than in Figure 11.6 where transaction costs are included. Furthermore, by examining the components of the asset vectors \( X \) on the efficient boundary we can find the optimal (minimum risk) portfolio composition for a given portfolio return.

![Figure 11.5](image)

Figure 11.5 Illustrating the use of a quasirandom Sobol sequence to plot the feasible region of a constrained portfolio containing twenty assets, with the transaction costs set to zero. The plot was generated by the Visual Basic Code excerpt 11.1
11.3 GENERATION OF MULTIVARIATE DISTRIBUTIONS: INDEPENDENT VARIATES

In this section we show how to generate independent variates from multivariate distributions; that is the variates have zero correlation.

11.3.1 Normal distribution

The most fundamental distribution is the univariate standard normal distribution, $N(0, 1)$, with zero mean and unit variance. In the case of $p$ independent variates this takes the form of a $p$ variate independent normal distribution $N(0, I_p)$ with zero mean and $p \times p$ unit covariance matrix $I_p$.

First we will quote a result concerning multivariate probability density functions, see Press et al. (1992). If $x_1, x_2, \ldots$ are random variates with a joint probability density function $p(x_1, x_2, \ldots)$, and if there are an equal number of $y$ variates $y_1, y_2, \ldots$ that are functions of the $x$’s, then the joint probability density function of the $y$ variates, $p(y_1, y_2, \ldots)$ is given by the following expression:

$$p(y_1, y_2, \ldots)dy_1 dy_2 \ldots = p(x_1, x_2, \ldots)J_{x,y}dy_1 dy_1$$

(11.5)

where $J_{x,y}$ is the Jacobian determinant of the $x$’s with respect to the $y$’s.

An important application of this result is the Box Muller transformation, see Box and Muller (1958), in which a $p$ variate independent normal distribution $N(0, I_p)$ is generated from a $p$ variate uniform distribution $U(0, 1)$.

*Figure 11.6* Illustrating the use of a quasirandom Sobol sequence to plot the feasible region of a constrained portfolio containing twenty assets, with transaction costs for buy and sell set to 0.07 and 0.04 respectively. The plot was generated by the Visual Basic Code excerpt 11.1.
The method works as follows: Consider two independently distributed $N(0,1)$ variables $x$ and $y$, and use the polar transformation to obtain:

$$x = r \cos \theta, \quad y = r \sin \theta, \quad \text{and} \quad r^2 = x^2 + y^2$$

From Equation 11.5 the joint probability density functions $f(r, \theta)$ and $f(x, y)$ obey the equation

$$f(r, \theta) \, dr \, d\theta = f(x, y)J_{xy,\theta} \, dr \, d\theta$$

where the Jacobian is

$$J_{xy,\theta} = \begin{vmatrix} \cos \theta & \sin \theta \\ -r \sin \theta & r \cos \theta \end{vmatrix} = r$$

We therefore have

$$f(r, \theta) = rf(x, y)$$

Furthermore since $x$ and $y$ are independent $N(0,1)$

$$f(x, y) = f(x) \cdot f(y), \quad \text{where} \quad f(x) = \frac{e^{-x^2/2}}{\sqrt{2\pi}} \quad \text{and} \quad f(y) = \frac{e^{-y^2/2}}{\sqrt{2\pi}}$$

Therefore:

$$f(r, \theta) = rf(x)f(y) = r \frac{e^{-x^2/2} e^{-y^2/2}}{\sqrt{2\pi} \sqrt{2\pi}}$$

which gives

$$f(r, \theta) = \frac{r}{2\pi} e^{(-x^2+y^2)/2} = \frac{1}{2\pi} f(\theta)f(r)$$

where $f(\theta) = 1/2\pi$, $f(r) = re^{-r^2/2}$ are independent probability density functions.

The corresponding cumulative probability distribution functions $F(\theta)$ and $F(r)$ can be found by evaluating the following integrals:

$$F(\theta) = \frac{1}{2\pi} \int_0^\theta d\theta = \frac{\theta}{2\pi}$$

and

$$F(r) = \int_0^r re^{-r^2/2} \, dr = \left[-e^{-r^2/2}\right]_0^r = 1 - e^{-r^2/2}$$

We can now use the result, see for example Evans et al. (2000), that any variate $x$ with a probability density function $f(x)$, has a cumulative distribution function $F(x) = \int_\infty^x f(x) \, dx$, which is $F(x) \sim U(0, 1)$, where $U(0,1)$ is the uniform distribution between 0 and 1.

The variables $V'_1 = F(r) = 1 - e^{-r^2/2}$ and $V'_2 = F(\theta) = \theta/2\pi$ are therefore uniformly distributed on the interval $(0, 1)$.

For convenience we will define the, $U(0,1)$, variables

$$V_1 = 1 - V'_1 = e^{-r^2/2} \quad \text{and} \quad V_2 = V'_2$$
So we have:

\[ V_1 = e^{-r^2/2}, \quad V_2 = \frac{\theta}{2\pi} \]

Therefore

\[ \log V_1 = -r^2/2, \quad r = (-2 \log V_1)^{1/2}, \quad \theta = 2\pi V_2 \]

Substituting these results into Equation 11.6 gives

\[ x = (-2 \log V_1)^{1/2} \cos 2\pi V_2, \quad y = (-2 \log V_1)^{1/2} \sin 2\pi V_2 \quad (11.9) \]

where \( x \) and \( y \) are \( N(0,1) \).

The Box–Muller method is contained in Equation 11.9, which shows that the \( N(0,1) \) variates are generated in pairs from the uniform distribution \( U(0,1) \) variates \( V_1 \) and \( V_2 \).

Since the \( N(0,1) \) variates are created two at a time, if we want to generate a normal distribution with an odd number of dimensions, \( n_{\text{odd}} \), it is necessary to generate \( n_{\text{odd}} + 1 \) dimensions and discard one of the dimensions.

It is easy to modify Equation 11.9 so that we can specify the means (\( \mu_1 \) and \( \mu_2 \)) and variances (\( \sigma_1^2 \) and \( \sigma_2^2 \)) of the generated variates \( x \) and \( y \); this is accomplished as follows:

**The Box–Muller method**

\[ x = \sigma_1 (-2 \log V_1)^{1/2} \cos 2\pi V_2 + \mu_1, \quad y = \sigma_2 (-2 \log V_1)^{1/2} \sin 2\pi V_2 + \mu_2 \quad (11.10) \]

where the distributions of \( x \) and \( y \) are:

\[ x \sim N(\mu_1, \sigma_1^2) \quad \text{and} \quad y \sim N(\mu_2, \sigma_2^2) \]

\( V_1 \) and \( V_2 \) are independent variates from the uniform distribution \( U(0,1) \).

Code excerpt 11.2 illustrates how to generate quasirandom normal variates with given means and standard deviations.

```c
long Quasi_Normal_Independent(long fcall, long seq, double xmean[], double std[], long idim, double quasi[]) {
    /* Input parameters:
        fcall — if fcall == 1 then it is an initialisation call, if fcall == 0 then a continuation call
        seq — if seq == 0 then a Faure sequence, if seq == 1 then a Niederreiter sequence,
              if seq == 2 then a Sobol sequence
        xmean[] — the means of the independent normal variates
        std[] — the standard deviations of the independent normal variates
        idim — the number of independent normal variates, idim must be less than 40
    Output parameters:
        quasi[] — the elements quasi[0], .. quasi[idim-1] contain the independent normal variates
    */
    long ierr, i, j;
    double twopi, v1, v2, pi;
```
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long ind1, ind2;
#define QUASI(I) quasi[(I)−1]
#define STD(I) std[(I)−1]
#define XMEAN(I) xmean[(I)−1]

if ((idim / 2) * 2 != idim){
    printf("Error on entry, idim is not an even number: idim = %ld\n", idim);
    return 1;
} else if (idim > 40){
    printf("On entry, idim > 40: idim = %ld\n", idim);
    return 1;
}

for (i = 1; i < idim; ++i){
    printf("On entry, the standard deviation is not greater than zero: STD(%ld) = %12.4f\n", i, STD(i));
    return 1;
}

pi = 4.0 * atan(1.0);
if (fcall){/* first call for initialisation */
    if (seq == 0){
        Generate_Faure_Sequence(fcall, idim, &QUASI(1));
    } else if (seq == 1){
        Generate_Niederreiter_Sequence(fcall, idim, &QUASI(1));
    } else if (seq == 2){
        Generate_Sobol_Sequence(fcall, idim, &QUASI(1));
    }
} else {/* a continuation call */
    if (seq == 0){
        Generate_Faure_Sequence(fcall, idim, &QUASI(1));
    } else if (seq == 1){
        Generate_Niederreiter_Sequence(fcall, idim, &QUASI(1));
    } else if (seq == 2){
        Generate_Sobol_Sequence(fcall, idim, &QUASI(1));
    }
    for (i = 1; i < indim/2; ++i) {/* generate the normal variates */
        ind1 = i * 2 − 1;
        ind2 = i * 2;
        twopi = pi * 2.0;
        v1 = sqrt(log(QUASI(ind1)) * −2.0);
        v2 = twopi * QUASI(ind2);
        QUASI(ind1) = XMEAN(ind1) + STD(ind1) * v1 * cos(v2);
        QUASI(ind2) = XMEAN(ind2) + STD(ind2) * v1 * sin(v2);
    }
    return 0;
}

Code excerpt 11.2 Generating quasirandom normal variates using the Box–Muller transformation

11.3.2 Lognormal distribution

The lognormal distribution can be generated from the normal distribution discussed in the previous section by means of a simple transformation. Here we denote a lognormal distribution with mean \( \bar{m} \) and variance \( s^2 \) by \( \Lambda(\bar{m}, s^2) \), and if a variate \( \ell \sim \Lambda(\bar{m}, s^2) \), then \( \log(\ell) \sim N(\mu, \sigma^2) \), where values for \( \mu \) and \( \sigma^2 \) are given below.

The lognormal density function, see Aitchison and Brown (1966), is:

\[
f(x) = \frac{1}{x \sigma (2\pi)^{1/2}} \exp \left( -\frac{(\log x - \mu)^2}{2\sigma^2} \right)
\]  

(11.11)
If \( z_i, i = 1, \ldots, p \) are independent normal variates \( N(\mu_i, \sigma_i^2), i = 1, \ldots, p \) then lognormal variates \( \ell_i, i = 1, \ldots, p \) can be generated using the transformation:

\[
\ell_i = \exp(z_i), \quad i = 1, \ldots, p
\]

where the mean of the \( i \)-th lognormal variate is

\[
\bar{m}_i = \exp\left(\mu_i + \frac{\sigma_i^2}{2}\right)
\]

and the variance is

\[
\sigma_i^2 = \exp(2\mu_i + \sigma_i^2)(\exp(\sigma_i^2) - 1)
\]

The ratio of variance to the mean squared is therefore

\[
\frac{\sigma_i^2}{\bar{m}_i^2} = \exp(\sigma_i^2) - 1
\]

or equivalently

\[
\sigma_i^2 = \log\left(1 + \frac{\sigma_i^2}{\bar{m}_i^2}\right)
\]

A lognormal distribution consisting of \( p \) independent variates with means \( \bar{m}_i, i = 1, \ldots, p \) and variances \( \sigma_i^2, i = 1, \ldots, p \) can thus be generated using the following procedure.

First generate the \( p \) independent normal variates

\[
\quad z_i \sim N(\mu_i, \sigma_i^2), \quad i = 1, \ldots, p
\]

where

\[
\mu_i = \log(\bar{m}_i) - \frac{\sigma_i^2}{2}
\]

and

\[
\sigma_i^2 = \log\left(1 + \frac{\sigma_i^2}{\bar{m}_i^2}\right)
\]

Then create the independent lognormal variates using

\[
\ell_i = \exp(z_i), \quad i = 1, \ldots, p
\]

### 11.3.3 Student's t distribution

If \( S_t(\mu, \nu) \) represents the Student's t distribution with mean \( \mu \) and number of degrees of freedom \( \nu \), then variates \( X \sim S_t(0, \nu) \) can be generated as follows:

\[
X \sim \frac{Z}{\sqrt{Y/\nu}}
\]
where $Z \sim N(0, 1)$, and $Y \sim \chi^2_\nu$. The variance of $X$ is:

$$E[X^2] = \frac{\nu}{\nu - 2}$$

Variaties $X'$ from a Student's $t$ distribution having $\nu$ degrees of freedom with mean $\mu$ and variance $s$ can be generated by modifying Equation 11.19 as follows:

$$X' \sim \mu + \frac{s^{1/2} Z}{\sqrt{\nu/(\nu - 2)} \sqrt{Y/\nu}} \tag{11.20}$$

The probability density function, $f(x)$, for $X'$ is:

$$f(x) = \frac{\Gamma((\nu + 1)/2)(\nu - 2)^{-1/2}s^{-1/2}}{\pi^{1/2}\Gamma(\nu/2)} \left[ 1 + \frac{(x - \mu)^2}{s(\nu - 2)} \right]^{-(\nu + 1)/2} \tag{11.21}$$

where $\nu > 2$.

11.4 GENERATION OF MULTIVARIATE DISTRIBUTIONS: CORRELATED VARIATES

In this section we will show how to generate multivariate distributions with known mean and covariance matrix. We will see later that variates from these distributions are important in Monte Carlo option pricing methods.

Multivariate generalization of univariate distributions, see for example Mardia et al. (1988).

11.4.1 Normal distribution

Here we consider how to generate a $p$ variate normal distribution with a given mean and covariance matrix.

We will denote the vector containing the variates of the $i$th observation from a $p$ variate zero mean normal distribution by $Z_i$; that is we write a sample of $n$ observations as

$$Z_i \sim N(0, C), \quad i = 1, \ldots, n \tag{11.22}$$

where $C$ is the $p \times p$ covariance matrix.

Further $Z_{i,k}$ is used to denote the $k$th element of $Z_i$, which contains the value of the $k$th variate for the $i$th observation.

From a computational point of view we can then consider a sample of $n$ observations to be represented by the $n \times p$ matrix $Z$. The $i$th row of $Z$ contains the values for $i$th observation, and the $k$th column of the $i$th row, $Z_{i,k}$, contains the value of the $k$th variate for the $i$th observation.

Also, since the distribution has zero mean, the sample covariance matrix is given by $C = ZZ^T$. To generate variates with covariance matrix $C$ we can use the fact that, if the matrix $C$ is positive definite, a Cholesky factorisation exists in which:

$$C = AA^T \tag{11.23}$$

where $A$ is lower triangular.
We can therefore generate $p$ variates which have a covariance matrix $C$ as follows. First generate, by (for example) using the Box Muller method described in Section 11.3.1, the independent normal variates:

$$X \sim N(0, I_p)$$

where the vector $X$ contains the $p$ variates, $I_p$ is the unit matrix, and $XX^T = I_p$.

Then, using the Cholesky factorisation of Equation 11.23, form

$$Y = AX$$  \hspace{1cm} (11.24)

where $Y$ is a $p$ element vector.

Now since $YY^T = AX(AX)^T = A(XX^T)A^T = AA^T = C$, we have that

$$Y \sim N(0, C)$$

Variates that have nonzero means $\mu_k, k = 1, \ldots, p$ can be obtained by simply modifying Equation 11.24 to:

$$Y' = AX + \mu$$  \hspace{1cm} (11.25)

where $Y'$ is a $p$ variate vector that is distributed as $N(\mu, C)$, and the $p$ elements of vector $\mu$ contain the means of the variates $Y'_k, k = 1, \ldots, p$.

The problem with this approach is that if the matrix $C$ is not positive definite (this could be caused by highly correlated variates or by rounding errors, etc.) then it is not possible to compute the Cholesky decomposition.

An alternative method is to use the spectral decomposition of the covariance matrix $C$,

$$C = V \Sigma V^T$$

where $\Sigma$ is a $p \times p$ diagonal matrix of eigenvalues $\lambda_i, i = 1, \ldots, p$ and the columns of the $p \times p$ matrix $V$ are the corresponding eigenvectors.

We can therefore write

$$C = V \Sigma^{1/2} \Sigma^{1/2} V^T = AA^T$$

where $A = V \Sigma^{1/2}$, and $\Sigma^{1/2} = \sqrt{\lambda_i}$, is the square root of the $i$th eigenvalue.

Equation 11.24 is then:

$$Y = V \Sigma^{1/2} X$$  \hspace{1cm} (11.26)

and Equation 11.25 is

$$Y' = V \Sigma^{1/2} X + \mu$$  \hspace{1cm} (11.27)

If the matrix $C$ is not positive definite then some (say $p - r$) of the eigenvalues will be negative. We can construct an approximation to the covariance matrix $C_r \sim C$ using only the $r$ positive eigenvalues as follows:

$$C_r = V_r \Sigma_r V_r^T = V_r \Sigma_r^{1/2} \Sigma_r^{1/2} V_r^T$$

where $C_r$ is a $p \times p$ matrix, $V_r$ is a $p \times r$ matrix and $\Sigma_r$ is a $r \times r$ matrix.
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Under these circumstances the $p$ element vectors $Y$ and $Y'$ are generated using the following modified versions of Equations 11.26 and 11.27

$$Y = V_r \Sigma_r^{1/2} X_r \quad \text{and} \quad Y' = V_r \Sigma_r^{1/2} X_r + \mu$$

where the $r$ element vector $X_r$ is just a subset of the $p$ element vector $X$. A function to generate correlated normal and lognormal variates is given in Code excerpt 11.3.

```cpp
long Quasirandom_Normal_LogNormal_Correlated(long fcall, long seq, long lnorm, double means[], long n, double c[], long tdc, double tol, long *irank, double x[], double work[], long lwk)
{
    /* Input parameters:
    fcall — if fcall == 1 then it is an initialisation call, if fcall == 0 then a continuation call
    seq — if seq == 0 then a Faure sequence, if seq == 1 then a Niederreiter sequence,
    if seq == 2 then a Sobol sequence
    lnorm — if lnorm == 1 then it is a lognormal distribution, if lnorm == 0 then a normal distribution
    n — the number of variates, n must be less than 40
    c[] — a matrix which contains the required covariance matrix, C
    tdc — the second dimension of the matrix C
    tol — the tolerance used for calculating the rank of the covariance matrix C
    means[] — the means of the independent normal variates
    std[] — the standard deviations of the independent normal variates
    lwk — the size of the work array, work
    Output parameters:
    rank — the computed rank of the covariance matrix C
    x[] — the elements $x[0]$, .. $x[n-1]$ contain the variates
    Input/Output parameters:
    work — a work array
    */
    double zero = 0.0, one = 1.0, two = 2.0;
    long nl, i, j, kk;
    double mtol, alpha;
    long ptrc, ptre, ptrv, ptrw, ptrw0, ptrw1;
    #define C(I,J) c[((I)/1) * tdc + ((J)/1)]
    #define MEANS(I) means[(I)/1]
    #define X(I) x[(I)/1]
    #define WORK(I) work[(I)/1]
    if (lwk < (2 + 3*n + 2*n*n + 3)) {
        printf("Error lwk is too small \n");
        return 1;
    }
    ptre = 2;
    ptrv = n+2;
    ptrw = n*n + n + 2;
    /* add extra 1 to allow for odd values of n */
    ptrw0 = ptrw + 1 + n;
    ptrwl = ptrw0 + 1 + n;
    nl = n;
    if (((n/2)*2) != n) {/* test for odd n */
        nl = n + 1;
    }
    if (fcall) /* first call for initialisation */
        if (lnorm) {/* lognormal distribution */
            for (i = 1; i <= n; ++i) {
                WORK(pw + (i-1)*n+i+1) = log(one + C(i,i)/(MEANS(i)*MEANS(i)));  
            }
        }
    else{/* normal distribution */
            for (i = 1; i <= n; ++i) {
                WORK(pw + (i-1)*n+i+1) = C(i,i);
            }
        }
    for (i = 1; i <= n; ++i) {
        for (j = 1; j <= n; ++j) {
            if (i == j) {
                WORK(pwr + (i-1)*n+i+j) = log(one + C(i,j)/(MEANS(i)*MEANS(j)));
            }
        }
    }
    return nl;
}
```
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/* calculate the eigenvalues and eigenvector of the matrix that has been loaded into WORK */
calc_eigvals_eigvecs (n,&WORK(ptrc),n,&WORK(ptre), &WORK(ptrv),n); /* The code uses NAG routine f02abc */

/* irank = 0; */

/* printf (' 'The eigenvalues are 
'');
for (j=n; j > 1; --j){
printf (''\%12.5f 
'', WORK(ptre + j-1));
}
*/

for (j=n; j > 1; --j){ /* use the eigenvalues to calculate the rank of the matrix */
if (WORK(ptre + j-1) < tol) goto L24;
*irank = *irank + 1;
}

printf (''*irank = \%ld 
'',*irank);

L24:
mtol = 1/tol;
if (WORK(ptre) < mtol){
printf (''Warning there is an eigenvalue less than \%12.4f 
'',mtol);
}

for (j=1; j < *irank; ++j){ /* ensure that all eigenvectors have the same sign on different machines */
kk = 1;
for (k=1; k < n; ++k){
if(WORK(ptrv+n(j-1)) != zero) goto L28;
kk = kk + 1;
}
}
L28:
/* ensure that all eigenvectors have the same sign on different machines */
alpha = sqrt(WORK(ptre + j-1));
if (WORK(ptrv+n(j-1)) < zero) alpha = -sqrt(WORK(ptre + j-1));
for (i=1; i < n; ++i){
WORK(ptrv+(i-1)*n) = WORK(ptrv+(j-1)+(i-1)*n)* alpha;
}

/* printf (' 'The eigenvectors are 
'');
for (j=1; j < *irank; ++j){
for (i=1; i < n; ++i){
printf (''\%10.5f '', WORK(ptrv + j-1)+(i-1)*n));
}
printf (''
'');
}
*/

for (i=1; i < n; ++i){ /* store a vector of ones and zeros for generating the quasi-random numbers */
WORK(ptrw0+i-1) = zero;
WORK(ptrw1+i-1) = one;
}
for (i=n; i < n1; ++i){
WORK(ptrw0+i-1) = zero;
WORK(ptrw1+i-1) = one;
}

L24;     /* end of first call section */

/* generate a vector of n1 random variables from a standard normal distribution, zero mean and unit variance */
Quasi_Normal_Independent(fcall, seq, &WORK(ptrw0), &WORK(ptrw1), n1, &WORK(ptrw));

/* printf (''The quasi random numbers are:\n''); for (i=1; i < n; ++i){
printf (''\%12.4f 
'', WORK(ptrw+(i-1)));
} */

/* Now generate variates with the specified mean and variance */
if (lnorm){ /* a lognormal distribution */
for (i=1; i < n; ++i){
X(i) = log(MEANS(i)) - WORK(ptrc+(i-1)*n+i-1)/two;
for (k=1; k < *irank; ++k){
X(i) = X(i) + WORK(ptrv+k-1) + (i-1)*n) * WORK(ptrw+k-1);
}
for (i=1; i < n; ++i){
X(i) = exp(X(i));
}
} else{ /* a normal distribution */
for (i=1; i < n; ++i){
X(i) = MEANS(i);
for (k=1; k < *irank; ++k){
X(i) = X(i) + WORK(ptrv+k-1) + (i-1)*n) * WORK(ptrw+k-1);
}
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```c
/* printf ('\n'\n'\n'\n');
for (i = 1; i <= n; ++i){
    printf ('\n%12.4f
', X(i));
}
*/
return 0;
```

**Code excerpt 11.3** The functions Quasirandom_Normal_LogNormal_Correlated which generates correlated quasirandom normal variates and correlated quasirandom lognormal variates

In order to visualize the effect of the covariance matrix we will display the results of using function Quasirandom_Normal_LogNormal_Correlated to generate the following variates:

- A vector of three normal independent variates with covariance matrix:
  
  \[
  C_1 = \begin{pmatrix}
  1.0 & 0.0 & 0.0 \\
  0.0 & 1.0 & 0.0 \\
  0.0 & 0.0 & 1.0 \\
  \end{pmatrix}
  \]

- A vector of three normal variates in which the elements of the covariance matrix are all positive; the covariance matrix is:
  
  \[
  C_2 = \begin{pmatrix}
  1.0 & 0.8 & 0.8 \\
  0.8 & 1.0 & 0.8 \\
  0.8 & 0.8 & 1.0 \\
  \end{pmatrix}
  \]

- A vector of three normal variates in which two elements of the covariance matrix are negative; the covariance matrix is:
  
  \[
  C_3 = \begin{pmatrix}
  1.0 & -0.7 & 0.2 \\
  -0.7 & 1.0 & 0.2 \\
  0.2 & 0.2 & 1.0 \\
  \end{pmatrix}
  \]

In all cases the mean vector is given by:

\[
\mu = \begin{pmatrix}
2.0 \\
2.0 \\
2.0 \\
\end{pmatrix}
\]

The results are displayed in Figures 11.7 to 11.9.

**11.4.2 Lognormal distribution**

The multivariate lognormal distribution is important because it is the asset returns distribution assumed by the Black–Scholes equation. We will denote a \( p \) variate vector \( L \) which has a lognormal distribution with \( p \) element mean vector \( \bar{m} \) and \( p \times p \) covariance matrix \( S \) as:

\[
L \sim \Lambda(\bar{m}, S)
\]
Figure 11.7 Scatter diagram for a sample of 3000 observations \((Z_i, i = 1, \ldots, 3000)\) generated from a multivariate normal distribution consisting of three variates with covariance matrix \(C_1\) and mean \(\mu\). Here we plot the values of the first variate against the values of the second variate. If we use the notation of Equation 11.22, then the \((x, y)\) co-ordinates for the points are \(x_i = Z_{i,1}, i = 1, \ldots, 3000\) and \(y_i = Z_{i,2}, i = 1, \ldots, 3000\).

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Figure 11.8 Scatter diagram for a sample of 3000 observations \((Z_i, i = 1, \ldots, 3000)\) generated from a multivariate normal distribution consisting of three variates with covariance matrix \(C_2\) and mean \(\mu\). Here we plot the values of the first variate against the values of the second variate. If we use the notation of Equation 11.22, then the \((x, y)\) co-ordinates for the points are \(x_i = Z_{i,1}, i = 1, \ldots, 3000\) and \(y_i = Z_{i,2}, i = 1, \ldots, 3000\).
This means that:

\[ \log(L)/C24_N(//C22; //C6) \]

where \( //C22 \) is a \( p \) element vector and \( //C6 \) is a \( p \times p \) matrix. It can be shown that

\[ \Sigma_{i,j} = \log\left(1 + \frac{S_{i,j}}{m_i m_j}\right) \] (11.29)

and

\[ \mu_i = \log(\bar{m}_i) - \frac{\Sigma_{i,i}}{2}, \quad i = 1, \ldots, p, \quad \text{and} \quad j = 1, \ldots, p \] (11.30)

For the case of independent variates we then have:

\[ \mu_i = \log(\bar{m}_i) - \frac{\sigma^2_i}{2}, \quad i = 1, \ldots, p \]

and

\[ \Sigma_{i,i} = \sigma^2_i = \log\left(1 + \frac{s^2_i}{m^2_i}\right), \quad i = 1, \ldots, p, \quad \text{and for} \ i \neq j, \ \Sigma_{i,j} = 0 \]

which are just Equations 11.17 and 11.18 given in Section 11.3.2.
Code excerpt 11.4 shows how to generate a multivariate lognormal distribution with a given mean $\mu$ and covariance matrix $S$. More complete information can be found in the function Quasirandom_Normal_LogNormal_Correlated which is provided in Code excerpt 11.3.

```c
double sig[40][40], s[40][40]; /* limit of 40 */
double means[40], x[40], lx[40], tmp;
#define S(I,J) s[(I)1][(J)1]
#define SIG(I,J) sig[(I)1][(J)1]
#define MEANS(i) means[(I)1] /* the means of the lognormal distribution */
#define X(I) x[(I)1] /* normal variates */
#define LX(I) lx[(I)1] /* lognormal variates */

/* obtain the Gaussian covariance matrix SIG, that corresponds to the lognormal covariance matrix S. */
for (i = 1; i < m; ++i) {
    for (j = 1; j < m; ++j) {
        tmp = MEANS(i) * MEANS(j);
        SIG(i,j) = log(1 + (S(i,j)/tmp));
    }
}

/* Generate multivariate Gaussian variates X(i), i = 1,...,m, with zero mean and covariance matrix SIG, using section */
/* Generate normal variates with the correct mean */
for (i = 1; i < m; ++i) {
    X(i) = X(i) + log(MEANS(i)) - SIG(i,i)/2;
}
/* Now exponentiate to create lognormal lognormal variates with mean XMEAN, and covariance matrix S */
for (i = 1; i < m; ++i) {
    LX(i) = exp(X(i));
}
```

**Code excerpt 11.4** Illustrating how to generate variates from a lognormal distribution with a given mean and covariance matrix

### 11.4.3 Student’s $t$ distribution

See Dickey (1967), Anderson (1984), and also Glasserman et al. (2000). Here we show how to generate observations from a multivariate Student’s $t$ distribution.

The probability density function, $f(x)$, for the $p$ variate multivariate Student’s $t$ distribution with covariance matrix $C$ is:

$$f(x) = \frac{\Gamma((m + \nu)/2)(\nu - 2)^{-1/2}C^{-1/2}x^{1/2}}{(\nu\pi)^{m/2}\Gamma(\nu/2)} \left[ 1 + \frac{x^TC^{-1}x}{(\nu - 2)} \right]^{-(m+\nu)/2}$$

(11.31)

where $C$ represents the determinant of $C$, and $\nu > 2$.

Let $\Lambda$ be a matrix with spectral decomposition $\Lambda = V\Sigma V^T$ and $T^*$ be a vector of $p$ independent Student’s $t$ variates, each with $\nu$ degrees of freedom. Then the vector $T' = V\Sigma^{1/2}T^*$ has a multivariate Student’s $t$ distribution with zero mean and a covariance matrix of $C = \nu/(\nu - 2)\Lambda$. So if we want to generate a $p$ variate vector
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$T$ from a multivariate Student’s $t$ distribution with mean vector $\mu$ and covariance matrix $C$ we do the following:

Create a scaled covariance matrix

$$B = C^{(\nu - 2) / \nu}$$

Perform the spectral decomposition

$$B = V \Sigma V^T$$

Then use the results of Section 11.3.3 to obtain a $p$ variate vector $T^*$ of independent Student’s $t$ variates and generate the required vector as

$$T = \mu + V^{1/2} T^*$$  \hspace{1cm} (11.32)

A multivariate sample of $n$ observations will be denoted by $T_i, i = 1, \ldots, n$, and the value of the $k$th variate for the $i$th observation will be denoted by $T_{i,k}$.

Of course, as in Section 11.4.1, we can if required choose to use only $r$ eigenvalues and eigenvectors. In these circumstances the Equation 11.32 becomes:

$$T = \mu + V_r^{1/2} T_r^*$$  \hspace{1cm} (11.33)

where $V_r$ is a $p \times r$ matrix, $\Sigma_r^{1/2}$ is an $r \times r$ diagonal matrix and the $r$ element vector $T_r^*$ is just a subset of vector $T^*$.
Code excerpt 11.5

The Fortran 77 function `STUDENT` which generates correlated variates from a Student’s $t$ distribution

```
RWSAV(3) = 0.000
RWSAV(4) = 0.000
RWSAV(5) = 0.000
RWSAV(6) = 0.000
RWSAV(7) = 0.000
RWSAV(8) = 0.000
RWSAV(9) = 0.000

FAC = (DF - TWO)/DF

* SCALE THE COVARIANCE MATRIX BY FAC TO PRODUCE THE EQUIVALENT SIGMA MATRIX
DO 10 I = 1, N
   DO 11 J = 1, N
      WORK(PTRC+(I-1)*N+J-1) = C(I,J)*FAC
11   CONTINUE
10   CONTINUE

CALL F02ABZ(WORK(PTRC),N,N,WORK(PTRE),WORK(PTRV),N,WORK(PTRW),IFLAGX)

* PRINT*,'THE EIGENVALUES ARE:'
* DO 3323 J = N, 1, -1
*    PRINT*, J, WORK(PTRE+(J-1)*I-1)
3323 CONTINUE

IRANK = 0
DO 23 J = N, 1, -1
   IF(WORK(PTRE+(J-1)).LT.TOL) GOTO 24
   IRANK = IRANK + 1
23 CONTINUE
24 CONTINUE

* PRINT*,'POINTA THE EIGENVECTORS:'
* DO 627 J = 1, IRANK
*    WRITE(*,'(10F10.5)') (WORK(PTRV+(J-1)*I-1),I = 1,N)
627 CONTINUE

MTOL = TOL
IF (WORK(PTRE).LT.MTOL) THEN
   PRINT*,'WARNING THERE IS AN EIGENVALUE LESS THAN ',MTOL
END IF

DO 25 J = 1, IRANK
   KK = 1
   DO 27 K = 1, N
      IF(WORK(PTRV+K-1+(J-1)*I-1).NE.ZERO) GOTO 28
      KK = KK + 1
27 CONTINUE
28 CONTINUE

* ENSURE THAT ALL EIGENVECTORS HAVE THE SAME SIGN ON DIFFERENT MACHINES
   ALPHA = SQRT(WORK(PTRE+(J-1))
   IF (WORK(PTRV+(J-1)*I-1).LT.ZERO)
      ALPHA = - SQRT(WORK(PTRE+(J-1))
   DO 29 I = 1, N
      WORK(PTRV+((J-1)*I-1)+I-1) = WORK(PTRV+((J-1)*I+I-1)) *ALPHA
29 CONTINUE

DO 625 J = 1, IRANK
   WRITE*('*(1OF10.5)') (WORK(PTRV+(J-1)*I-1)+I-1),I = 1,N
625 CONTINUE

END IF

* GENERATE A VECTOR OF N1 INDEPENDENT RANDOM VARIABLES FROM A STUDENT`ST DISTRIBUTION AND STORE THEN IN
VECTOR WORK(PTRW)
IFLAGX = 0
DO 222 I = 0, N1 - 1
   WORK(PTRW+I) = G05Khw(DF,IGEN,ISEED,RWSAV,IFLAGX)
222 CONTINUE

* PRINT*,'WORKPTRW+I) = WORK(PTRW+I)

DO 133 I = 1, N
   X(I) = MEANS(I)
133 CONTINUE
```

Monte Carlo simulation

The Fortran 77 function `STUDENT` which generates correlated variates from a Student’s $t$ distribution
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In order to visualize the effects of both the covariance matrix and the number
of degrees of freedom, \( \nu \), we display results from using the function \( \text{STUDENT} \) to generate
the following variates:

- Three Student’s \( t \) variates with covariance matrices \( C_1, C_2, C_3 \), mean \( \mu \) and
  \( \nu = 25.5 \).
- Three Student’s \( t \) variates with covariance matrices \( C_1, C_2, C_3 \), mean \( \mu \) and
  \( \nu = 4.5 \).

The values of \( \mu, C_1, C_2, \) and \( C_3 \) are those previously defined in Section 12.4.1.

The results are displayed in Figures 11.10 to 11.13. It can be seen that when
\( \nu = 25.5 \) the distribution of points is very similar to that for the normal distribution;
for example compare Figure 11.7 with Figure 11.3. However, for \( \nu = 4.5 \), the
Student’s \( t \) variates have more points in the tail of the distribution than the corre-
sponding normal variates. This has applications in finance where asset return dis-
tributions have been found to exhibit such effects.

Figure 11.10 Scatter diagram for a sample of 3000 observations \( (T_i, i = 1, \ldots, 3000) \) generated from a
multivariate Student’s \( t \) distribution consisting of three variates with covariance matrix \( C_3 \), number of
degrees of freedom \( \nu = 4.5 \) and mean \( \mu \), see Section 11.4.1. Here we plot the values of the first variate
against the values of the second variate. The \( (x, y) \) co-ordinates for the points are therefore
\( x_i = T_{i,1}, i = 1, \ldots, 3000 \) and \( y_i = T_{i,2}, i = 1, \ldots, 3000 \)
**Figure 11.11** Scatter diagram for a sample of 3000 observations \((T_i, i = 1, \ldots, 3000)\) generated from a multivariate Student’s *t* distribution consisting of three variates with covariance matrix \(C_1\), number of degrees of freedom \(\nu = 4.5\) and mean \(\mu\). Here we plot the values of the first variate against the values of the second variate. The \((x, y)\) co-ordinates for the points are therefore \(x_i = T_{i,1}, i = 1, \ldots, 3000\) and \(y_i = T_{i,2}, i = 1, \ldots, 3000\).

**Figure 11.12** Scatter diagram for a sample of 3000 observations \((T_i, i = 1, \ldots, 3000)\) generated from a multivariate Student’s *t* distribution consisting of three variates with covariance matrix \(C_3\), number of degrees of freedom \(\nu = 25.5\) and mean \(\mu\), see Section 11.4.1. Here we plot the values of the first variate against the values of the second variate. The \((x, y)\) co-ordinates for the points are therefore \(x_i = T_{i,1}, i = 1, \ldots, 3000\) and \(y_i = T_{i,2}, i = 1, \ldots, 3000\).
Figure 11.13 Scatter diagram for a sample of 3000 observations \((T_i, i = 1, \ldots, 3000)\) generated from a multivariate Student’s \(t\) distribution consisting of three variates with covariance matrix \(C_1\), number of degrees of freedom \(\nu = 25.5\) and mean \(\mu\), see Section 11.4.1. Here we plot the values of the first variate against the values of the second variate. The \((x, y)\) co-ordinates for the points are therefore \(x_i = T_{i,1}, i = 1, \ldots, 3000\) and \(y_i = T_{i,2}, i = 1, \ldots, 3000\)
Chapter 12

Multiasset European and American options

12.1 INTRODUCTION

In this section we consider the valuation of multiasset, basket, options within the Black–Scholes pricing framework. Here we will show how to price options on the maximum and minimum value of the assets in a basket using:

- Analytic methods
- Monte Carlo methods
- Multidimensional lattices.

Analytic methods can be useful for pricing multiasset European options which have a known closed form solution. They are particularly appropriate for low dimensional European options, when the closed form expressions are not too difficult to evaluate.

Monte Carlo methods have the advantage that they can easily compute the value of multiasset European options, but have difficulty including the possibility of early exercise; this is required for American style options.

On the other hand multidimensional lattice techniques allow American options to be evaluated with ease. However lattices become increasingly difficult to program as the number of dimensions increases, and the constraint of computer storage limit their use to problems involving (about) four or less assets.

12.2 THE MULTIASSET BLACK–SCHOLES EQUATION

In Section 8.3 we showed that when the price, $S$, of a single asset follows GBM the change in price, $dS$, over a time interval, $dt$, is given by:

$$dS = rSdt + \sigma SdX$$

where $r$ is the risk free interest rate, $\sigma$ is the volatility of asset $S$, and $dX$ is drawn from a normal distribution with mean zero and variance $dt$.

We also proved, see Equation 8.14 of Section 8.3, using Ito’s lemma that the process followed by $Y = \log(S)$ is:

$$dY = (r - \sigma^2/2)dt + \sigma dX$$

where $dY$ is the change in the value of $\log(S)$ over the time interval $dt$. Later on, in Section 9.3.1, we derived the (Black–Scholes) partial differential equation that is
satisfied by the value, \( V \), of an option written on a single underlying asset follows GBM; this equation is:

\[
\frac{\partial V}{\partial t} + \frac{\sigma^2 S^2 \partial^2 V}{2} + rS \frac{\partial V}{\partial S} - rV = 0
\]

The above results for a single asset can be generalized to deal with multiasset options. For \( m \) assets we have the following processes:

\[
dY_i = \left( r - \frac{\sigma_i^2}{2} \right) dt + \sigma_i dX_i, \quad i = 1, \ldots, m
\]

where the subscript \( i \) refers to the value associated with the \( i \)th asset. We can also write the above equation in vector form by introducing the \( m \) element vector \( dY \) which is normally distributed as:

\[
dY \sim N(\mu, C)
\]

where \( \mu \) is the mean vector and \( C \) is the covariance matrix. The elements of the covariance matrix are:

\[
C_{ii} = \sigma_i^2 dt, \quad i = 1, \ldots, m, \\
C_{ij} = \sigma_i \sigma_j \rho_{ij} dt, \quad i \neq j, i = 1, \ldots, m, \quad j = 1, \ldots, m
\]

where \( \rho_{ij} \) is the correlation coefficient between assets \( i \) and \( j \). The elements of the mean vector \( \mu \) are:

\[
\mu_i = r - \frac{\sigma_i^2}{2}, \quad i = 1, \ldots, m
\]

The value \( V \) of an option written on \( n \) assets satisfies the following partial differential equation:

\[
\frac{\partial V}{\partial t} + \frac{1}{2} \sum_{i=1}^{m} \sum_{j=1}^{m} \sigma_i \sigma_j \rho_{ij} S_i S_j \frac{\partial^2 V}{\partial S_i \partial S_j} + r \sum_{i=1}^{m} S_i \frac{\partial V}{\partial S_i} - rV = 0
\]

For a European call on the maximum of \( m \) assets the payoff \( P_c^{\text{MAX}} \) at maturity (time \( \tau \)) is given by \( P_c^{\text{MAX}} = \max(\max(S_1^\tau, S_2^\tau, \ldots, S_m^\tau) - E, 0) \), where \( S_i^\tau, i = 1, \ldots, m \) denotes the value of the \( i \)th asset at maturity, and \( E \) represents the strike price. Similarly a European put option on the minimum of \( m \) assets has a payoff, \( P_p^{\text{MIN}} \), at time \( \tau \), given by \( P_p^{\text{MIN}} = \max(E - \min(S_1^\tau, S_2^\tau, \ldots, S_m^\tau), 0) \).

### 12.3 MULTIDIMENSIONAL MONTE CARLO METHODS

We have already mentioned that Monte Carlo simulation can easily price European multiasset options (also sometimes referred to as basket options or rainbow options) involving a large number of assets (say 20 or more).
Multiasset European and American options

In addition Monte Carlo simulation can also include the following features into an option without much difficulty:

- Non-Gaussian distribution of stock returns; distributions with heavy tails are usually of interest because they more accurately represent what is observed in the financial markets.
- Options with path dependency (such as barrier options, etc.); these are known as exotic options.
- Complex time dependency (e.g. ARMA, GARCH, or Levy processes) of model parameters such as interest rates, asset prices, etc.

The main drawbacks with Monte Carlo simulation are:

- It is difficult to compute the value of American style options.
- It is difficult (or impossible) to achieve the same accuracy that can be obtained using finite-difference methods.

In a different section of this book we will show how Monte Carlo simulation can be used to price American options by using a hybrid Monte Carlo lattice approach originally developed by Boyle et al. (1997).

In Chapter 11 we showed that when pseudorandom numbers are used the standard errors of integrals computed via Monte Carlo simulation decrease at the rate $N^{1/2}$, where $N$ is the number of simulations. This means that it can require hundreds of thousands of simulations just to achieve an accuracy of $10^{-1}$ or $10^{-2}$ in the estimated option price. It is because of this that various Monte Carlo variance reduction techniques are used to increase the accuracy of the computed integral.

In this section we show how to price a three asset basket option using Monte Carlo simulation; the accuracy of the results obtained with quasirandom numbers and pseudorandom numbers is compared.

The options we consider are European put and call options on the maximum and minimum of three assets. All the options have a maturity of one year, and the other model parameters used are given in Tables 12.1 and 12.2.

In Code excerpt 12.1 most of the work is done by the routine Quasirandom_Normal_LogNormal_Correlated, which was described in Section 11.4.1. This generates a vector of multivariate quasirandom numbers with a particular covariance matrix. In the program the values of the assets at current time, $t$ are $S_1 = S_2 = S_3 = 100$. To compute the asset prices when the option matures, at $T = 1$, we make use of Equation 12.1.

Another way of writing Equation 12.1 is

$$dY_i = \log(S_{i,t+dt}) - \log(S_{i,t}) = (r - \sigma_i^2/2)dt + \sigma_idX_i, \quad i = 1, \ldots, m$$

where we have used the notation $S_{i,t}$ to denote the value of the $i$th asset at current time $t$, and $S_{i,t+dt}$ to denote the value of the asset at the future time $t + dt$. Simple rearrangement of the above equation gives:

$$\log\left(\frac{S_{i,t+dt}}{S_{i,t}}\right) = (r - \sigma_i^2/2)dt + \sigma_i dX_i, \quad i = 1, \ldots, m$$
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Table 12.1  The computed values and absolute errors, in brackets, for European options on the maximum of three assets. Monte Carlo simulation was used with both quasirandom (Sobol) sequences and pseudorandom sequences. The number of paths used varied from 500 to 3000. The parameters were:

\[ E = 100.0, \ S_1 = S_2 = S_3 = 100.0, \ r = 0.1, \ \tau = 1.0, \ \sigma_1 = \sigma_2 = \sigma_3 = 0.2, \ \rho_{12} = \rho_{13} = \rho_{23} = 0.5, \]

\[ q_1 = q_2 = q_3 = 0.0. \]  The accurate values were 0.936 for a put and 22.672 for a call, see Table 12.7 of Section 12.6 and Table 2 of Boyle, Evnine, and Gibbs (1989)

<table>
<thead>
<tr>
<th>nsim</th>
<th>Quasi</th>
<th>Pseudo</th>
<th>Quasi</th>
<th>Pseudo</th>
</tr>
</thead>
<tbody>
<tr>
<td>500</td>
<td>0.890 (4.5948 × 10^{-2})</td>
<td>1.1044 (1.6839 × 10^{-1})</td>
<td>22.629 (4.3231 × 10^{-2})</td>
<td>22.4089 (2.6312 × 10^{-1})</td>
</tr>
<tr>
<td>1000</td>
<td>0.924 (1.1354 × 10^{-2})</td>
<td>1.0193 (8.3297 × 10^{-2})</td>
<td>22.683 (1.1306 × 10^{-2})</td>
<td>22.3520 (3.1998 × 10^{-1})</td>
</tr>
<tr>
<td>1500</td>
<td>0.919 (1.6807 × 10^{-2})</td>
<td>0.8957 (4.0344 × 10^{-2})</td>
<td>22.670 (2.2954 × 10^{-3})</td>
<td>22.6346 (3.7430 × 10^{-2})</td>
</tr>
<tr>
<td>2000</td>
<td>0.932 (4.3221 × 10^{-3})</td>
<td>0.8995 (3.6488 × 10^{-2})</td>
<td>22.685 (1.3299 × 10^{-2})</td>
<td>22.7675 (9.5491 × 10^{-2})</td>
</tr>
<tr>
<td>2500</td>
<td>0.932 (3.5698 × 10^{-3})</td>
<td>0.8886 (4.7352 × 10^{-2})</td>
<td>22.670 (1.6619 × 10^{-3})</td>
<td>22.9326 (2.6058 × 10^{-1})</td>
</tr>
<tr>
<td>3000</td>
<td>0.937 (1.1376 × 10^{-3})</td>
<td>0.9025 (3.3548 × 10^{-2})</td>
<td>22.679 (7.2766 × 10^{-3})</td>
<td>22.8050 (1.3301 × 10^{-1})</td>
</tr>
</tbody>
</table>

Table 12.2  The computed values and absolute errors, in brackets, for European options on the minimum of three assets. Monte Carlo simulation was used with both quasirandom (Sobol) sequences and pseudorandom sequences. The number of paths used varied from 500 to 3000. The parameters were:

\[ E = 100.0, \ S_1 = S_2 = S_3 = 100.0, \ r = 0.1, \ \tau = 1.0, \ \sigma_1 = \sigma_2 = \sigma_3 = 0.2, \ \rho_{12} = \rho_{13} = \rho_{23} = 0.5, \]

\[ q_1 = q_2 = q_3 = 0.0. \]  The accurate values were 7.403 for a put and 5.249 for a call, see Table 12.8 of Section 12.6 and Table 2 of Boyle, Evnine, and Gibbs (1989)

<table>
<thead>
<tr>
<th>nsim</th>
<th>Quasi</th>
<th>Pseudo</th>
<th>Quasi</th>
<th>Pseudo</th>
</tr>
</thead>
<tbody>
<tr>
<td>500</td>
<td>7.365 (3.8122 × 10^{-2})</td>
<td>7.6760 (2.7298 × 10^{-1})</td>
<td>5.312 (6.3431 × 10^{-2})</td>
<td>5.3086 (5.9591 × 10^{-2})</td>
</tr>
<tr>
<td>1000</td>
<td>7.425 (2.1554 × 10^{-2})</td>
<td>7.7607 (3.5772 × 10^{-1})</td>
<td>5.293 (4.3958 × 10^{-2})</td>
<td>5.4376 (1.8857 × 10^{-1})</td>
</tr>
<tr>
<td>1500</td>
<td>7.408 (5.1223 × 10^{-3})</td>
<td>7.5654 (1.6240 × 10^{-1})</td>
<td>5.253 (4.0761 × 10^{-3})</td>
<td>5.4121 (1.6307 × 10^{-1})</td>
</tr>
<tr>
<td>2000</td>
<td>7.399 (3.6364 × 10^{-3})</td>
<td>7.4820 (7.8995 × 10^{-2})</td>
<td>5.266 (1.7236 × 10^{-2})</td>
<td>5.4029 (1.5390 × 10^{-1})</td>
</tr>
<tr>
<td>2500</td>
<td>7.407 (4.1463 × 10^{-3})</td>
<td>7.3592 (4.3754 × 10^{-2})</td>
<td>5.267 (1.7707 × 10^{-2})</td>
<td>5.4690 (2.2005 × 10^{-1})</td>
</tr>
<tr>
<td>3000</td>
<td>7.400 (2.7166 × 10^{-3})</td>
<td>7.3997 (3.3236 × 10^{-3})</td>
<td>5.245 (3.5024 × 10^{-3})</td>
<td>5.4331 (1.8407 × 10^{-1})</td>
</tr>
</tbody>
</table>

Taking exponentials of both sides we obtain:

\[
\frac{S_{i,t+dt}}{S_{i,t}} = \exp\left\{ (r - \sigma_i^2/2)dt + \sigma_i dX_i \right\}, \quad i = 1, \ldots, m
\]

which is equivalent to:

\[
S_{i,t+dt} = S_{i,t} \exp\left\{ (r - \sigma_i^2/2)dt + \sigma_i dX_i \right\}
\]

(12.5)
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double T, the_max, the_min, E, ST1, ST2, ST3, S1, S2, S3;
double disc, sumit_max_put, sumit_max_call, sumit_min_put, sumit_min_call;
double *rvec = (double *)0, rho_12, rho_13, rho_23;
double *c3, *z, *means, tol, *work, tmp1, tmp2, sigma1, sigma2, sigma3;
long lnorm, seq, fcall;
#define MEANS(I) means[(I)-1]
#define WORK(I) work[(I)-1]
#define Z(I) z[(I)-1]
#define C3(I,J) c3[((I)-1)*3+(J)-1]
m = 3; // the number of assets
lwk = 100000;
c3 = (double*)malloc((size_t)(sizeof(double)*3*3));
means = (double *)malloc((size_t)(sizeof(double)*3));
z = (double *)malloc((size_t)(sizeof(double)*3));
work = (double*)malloc((size_t)(sizeof(double)*lwk));
if (!means || !z || !work) {
    printf("Allocation error 
");
}
T = 1.0; // the maturity of the options
r = 0.1; // the riskless interest rate
sqrt_T = sqrt(T);
disc = exp(-r*T);
tol = 1.0e-8;
skip = 1000;
sigma1 = 0.2; // the volatility of asset 1
sigma2 = 0.2; // the volatility of asset 2
sigma3 = 0.2; // the volatility of asset 3
S1 = 100.0; // the current price of asset 1
S2 = 100.0; // the current price of asset 2
S3 = 100.0; // the current price of asset 3
E = 100.0; // the strike price
rho_12 = 0.5; // the correlation coefficient between asset 1 and asset 2
rho_13 = 0.5; // the correlation coefficient between asset 1 and asset 3
rho_23 = 0.5; // the correlation coefficient between asset 2 and asset 3
C3(1,1) = sigma1*sigma1*T; // set the elements of the covariance matrix
C3(2,2) = sigma2*sigma2*T;
C3(3,3) = sigma3*sigma3*T;
C3(1,2) = sigma1*sigma2*T*rho_12;
C3(2,3) = sigma2*sigma3*T*rho_23;
C3(1,3) = sigma1*sigma3*T*rho_13;
C3(2,1) = C3(1,2);
C3(3,1) = C3(1,3);
C3(3,2) = C3(2,3);
MEANS(1) = (r - sigma1*sigma1*half)*T;
MEANS(2) = (r - sigma2*sigma2*half)*T;
MEANS(3) = (r - sigma3*sigma3*half)*T;
printf("THREE ASSET OPTIONS USING QUASIRANDOM NUMBERS 
");
fcall = 1; // initialisation call
seq = 2; // use Sobol sequences
lnorm = 0; // generate a normal distribution
Quasirandom_Normal_LogNormal_Correlated(fcall, seq, lnorm, &MEANS(1), m, &C3(1,1), m, tol, &irank, &E(1), &WORK(1), lwk);
fcall = 0; // continuation call
sumit_max_put = zero;
sumit_max_call = zero;
sumit_min_put = zero;
sumit_min_call = zero;
num_simulations = 3000; // the number of simulations to use
for (i = 1; i <= num_simulations; i++) {
Quasirandom_Normal_LogNormal_Correlated(fcall, seq, lnorm, &MEANS(1), m, &C3(1,1), m, tol, &irank, &E(1), &WORK(1), lwk);
ST1 = S1*exp(Z(1)); // the price of asset 1 at option maturity
ST2 = S2*exp(Z(2)); // the price of asset 2 at option maturity
ST3 = S3*exp(Z(3)); // the price of asset 3 at option maturity
// options on the maximum
tmp2 = MAX(ST1, ST2);
the_max = MAX(tmp2, ST3);
tmp1 = the_max - E;
opt_val = MAX(tmp1, zero);
sumit_max_call += opt_val*disc;
tmp1 = E - the_max;
opt_val = MAX(tmp1, zero);
sumit_max_put += opt_val*disc;
// options on the minimum
tmp2 = MIN(ST1, ST2);
the_min = MIN(tmp2, ST3);
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tmp1 = the_min - E;
opt_val = MAX(tmp1, zero);
sumit_min_call += opt_val*disc;
tmp1 = E - the_min;
opt_val = MAX(tmp1, zero);
sumit_min_put += opt_val*disc;
}

opt_val = sumit_max_put/(double)num_simulations;
printf("MAX:PUT = \%12.4f < \%8.4e \(0.936\)\n", opt_val, FABS(opt_val - 0.936));
opt_val = sumit_max_call/(double)num_simulations;
printf("MAX:CALL = \%12.4f < \%8.4e \(22.672\) \n", opt_val, FABS(opt_val - 22.672));

opt_val = sumit_min_put/(double)num_simulations;
printf("MIN:PUT = \%12.4f < \%8.4e \(7.403\) \n", opt_val, FABS(opt_val - 7.403));
opt_val = sumit_min_call/(double)num_simulations;
printf("MIN:CALL = \%12.4f < \%8.4e \(5.249\) \n", opt_val, FABS(opt_val - 5.249));

.. Initialisation of model parameters etc the same as for quasirandom code..

Code excerpt 12.1 A Monte Carlo simulation computer program, using quasirandom numbers, for estimating the value of European put and call options on the maximum and minimum of three underlying assets. The results are presented in Tables 12.1 and 12.2.

Code excerpt 12.2 A Monte Carlo simulation computer program, using pseudorandom numbers, for estimating the value of European put and call options on the maximum and minimum of three underlying assets. It can be seen that, apart from code concerned with calling the random number generator, the program is identical to that given in Code excerpt 12.1 above. The results are presented in Tables 12.1 and 12.2.
12.4 MULTIDIMENSIONAL LATTICE METHODS

Finite-difference lattices can be used to value options on up to about four assets before they require impossibly large amounts of computer memory. The main advantage of finite-difference method is that they are able to easily cater for American style early exercise facilities within the option. This is not true of Monte Carlo methods. They can easily model complex European options, but have difficulty modelling American style options.

In this section we use the approach of Kamrad and Ritchken (1991), and Boyle, Evnine and Gibbs (1989), which we will call the BEGKR method, to price multiasset options. We first derive expressions for the jump size and jump probabilities for a single asset, and show that these are equivalent to those of the Cox, Ross, and Rubinstein binomial lattice (CRR lattice) discussed in Section 10.4.1. We will then give a expression for the jump sizes and jump probabilities of a general multiasset option.

Finally there will be a brief discussion of two lattice techniques, namely truncated lattices and recursive lattices, that the author has found useful in computing multi-asset option values.

To derive the BEGKR equations for one asset we first assume that the asset follows a lognormal processes with drift $\mu = r - \sigma^2/2$, where $r$ is the riskless interest rate and $\sigma$ is the instantaneous volatility.

Therefore if $S_t$ is the price of the asset at time $t$, and $S_{t+\Delta t}$ is the price at time instant $t+\Delta t$, we then have the following equations:

$$\log(S_{t+\Delta t}) = \log(S_t) + \epsilon_t, \quad \epsilon_t \sim N(\mu \Delta t, \sigma^2 \Delta t)$$

or equivalently

$$\log \left( \frac{S_{t+\Delta t}}{S_t} \right) \sim N(\mu \Delta t, \sigma^2 \Delta t)$$

where $\epsilon_t$ represents a random variable and as usual $N(\mu \Delta t, \sigma^2 \Delta t)$ denotes a Gaussian with mean $\mu \Delta t$ and variance $\sigma^2 \Delta t$.

We will now consider how to construct a binomial lattice by only allowing $\epsilon_t$ to jump up or down by an amount $\nu = \sigma \sqrt{\Delta t}$ at each lattice node. This means that:

For an up jump

$$\log \left( \frac{S_{t+\Delta t}}{S_t} \right) = \sigma \sqrt{\Delta t}, \quad \text{or} \quad S_{t+\Delta t} = S_t \exp(\sigma \sqrt{\Delta t}) \quad (12.6)$$

For a down jump

$$\log \left( \frac{S_{t+\Delta t}}{S_t} \right) = -\sigma \sqrt{\Delta t}, \quad \text{or} \quad S_{t+\Delta t} = S_t \exp(-\sigma \sqrt{\Delta t}) \quad (12.7)$$

The reader will notice that these expressions are the same as those for the nodes of the CCR lattice described in Section 10.4.1. That is: for an up jump $S_{t+\Delta t} = S_t u$, for a down jump $S_{t+\Delta t} = S_t d$, and $u = 1/d = \exp(\sigma \sqrt{\Delta t})$. 

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The probability of undergoing either an up or down jump occurring can be found by matching the mean and variance of \( \epsilon_t \).

From the mean:

\[
E[\epsilon_t] = \nu(p_u - p_d) = \mu \Delta t
\]

and from the variance:

\[
Var[\epsilon_t] = \nu^2(p_u + p_d) = \sigma^2 \Delta t
\]

Eliminating \( p_d \) from Equations 12.8 and 12.9 gives

\[
\nu \mu \Delta t + \sigma^2 \Delta t = 2\nu^2 p_u
\]

and so

\[
p_u = \frac{1}{2} \left\{ \frac{\sigma^2 \Delta t}{\nu^2} + \frac{\mu \Delta t}{\nu} \right\}
\]

which on substituting \( \nu = \sigma \sqrt{\Delta t} \) yields

\[
p_u = \frac{1}{2} \left\{ 1 + \frac{\mu \sqrt{\Delta t}}{\sigma} \right\}
\]

\[
p_d = 1 - p_u = \frac{1}{2} \left\{ 1 - \frac{\mu \sqrt{\Delta t}}{\sigma} \right\}
\]

We shall now show that, to first order, the jump probabilities in Equations 12.10 and 12.11 are the same as those for the CRR lattice.

For the CRR lattice (Section 10.4.1, Equation 10.89) we have:

\[
p_u = \frac{\exp(r \Delta t) - d}{u - d}
\]

expanding \( \exp(r \Delta t) \), \( u \) and \( d \) to order \( \Delta t \) we obtain

\[
\exp(r \Delta t) \sim 1 + r \Delta t
\]

\[
u = \exp(\sigma \sqrt{\Delta t}) \sim 1 + \sigma \sqrt{\Delta t} + \frac{\sigma^2}{2} \Delta t
\]

\[
d = \exp(\sigma \sqrt{\Delta t}) \sim 1 - \sigma \sqrt{\Delta t} + \frac{\sigma^2}{2} \Delta t
\]

so \( \exp(r \Delta t) - d \sim r \Delta t + \sigma \sqrt{\Delta t} - \frac{\sigma^2 \Delta t}{2} \)

and \( u - d \sim 2\sigma \sqrt{\Delta t} \)

So

\[
p_u = \frac{\exp(r \Delta t) - d}{u - d} \sim \frac{r \Delta t + \sigma - \sigma^2/2 \Delta t}{2\sigma \sqrt{\Delta t}}
\]
which simplifies to

\[ p_u = \frac{1}{2} \left( 1 + \frac{\mu \sqrt{\Delta t}}{\sigma} \right) \]

and therefore

\[ p_d = 1 - p_u = \frac{1}{2} \left( 1 - \frac{\mu \sqrt{\Delta t}}{\sigma} \right) \]

which are the expressions for \( p_u \) and \( p_d \) given in Equations 12.10 and 12.11 respectively. So we have shown that, to first order in \( \Delta t \), both the size of the jump and the probability of the jump are the same as the CRR binomial lattice.

The attractive feature of the BEGKR binomial lattice model is that it can easily be generalized to describe a model consisting of \( k \) assets. Here we will merely quote the results in Kamrad and Ritchken (1991). As before, it is assumed that the asset prices follow a multivariate lognormal distribution. Let \( \mu_i = r - \sigma_i^2/2 \), and \( \sigma_i \) be the instantaneous mean and variance respectively \((i = 1, 2, \ldots, k)\) and let \( \rho_{ij} \) be the correlation between assets \( i \) and \( j \). There are now \( 2^k \) different jumps from each lattice node over the time interval \( \Delta t \), and

\[ x_{im} = 1 \text{ if asset } i \text{ has an up jump in state } m, \quad x_{im} = -1 \text{ if asset } i \text{ has a down jump in state } m. \]

In addition \( x_{ij}^m = 1 \) if assets \( i \) and \( j \) have jumps in the same direction in state \( m \), and \( x_{ij}^m = -1 \) if assets \( i \) and \( j \) have jumps in the opposite direction in state \( m \).

### The jump probabilities for a \( k \)-asset binomial lattice: Kamrad and Ritchken (1991)

The \( 2^k \) jump probabilities, \( p_m \), \( m = 1, \ldots, 2^k \), for each lattice node are:

\[ p_m = \frac{1}{2^k} \left\{ 1 + \sqrt{\Delta t} \sum_{i=1}^{k} x_{im} \left( \frac{\mu_i}{\sigma_i} \right) + \sum_{i=1}^{k-1} \sum_{j=i+1}^{k} (x_{ij}^m \rho_{ij}) \right\}, \]

\[ m = 1, 2, \ldots, 2^k, \quad k \geq 2 \]

(12.12)

where \( x_{im}^1 = 1 \) if asset \( i \) has an up jump in state \( m \), and \( x_{im}^{-1} = 1 \) if asset \( i \) has a down jump in state \( m \). In addition \( x_{ij}^m = 1 \) if assets \( i \) and \( j \) have jumps in the same direction in state \( m \), and \( x_{ij}^m = -1 \) if assets \( i \) and \( j \) have jumps in the opposite direction in state \( m \).

#### 12.4.1 Truncated lattices

The truncated lattice makes use of the fact that not all of the lattice will contribute significantly to the value of the option. This can be seen by merely considering the probability of undergoing \( n \) jumps in a given direction. It can be seen from Equation 12.12 that, for a \( k \) asset lattice, each of the \( 2^k \) jumps from an individual lattice node has a probability \( p \sim 1/2^k \). The probability of undergoing \( n \) jumps in a given direction is \( p^n \), and since \( p < 1 \), it follows that \( p^n \sim 0 \) for large \( n \). This means that the probability of attaining the very high or very low asset values which occur in the wings of the lattice is extremely small. This approach is similar to that used in the Hull and White interest rate model, see Hull and White (1994).
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12.4.2 Recursive lattices

The recursive lattice used here is a multiasset extension of the BBS binomial lattice described in Section 10.4.4, where the analytic Black–Scholes formula was used to compute the option values of the last lattice step. If we want to use exactly the same technique then we would need to use some complicated expression involving multidimensional cumulative normal distribution functions, see Section 12.5 where these are given for two asset options. One way round this problem is to approximate the analytic solution by using a higher accuracy lattice to compute the last step. This can be achieved by a recursive call to the original lattice as shown in the code excerpt below (the complete code for a two-dimensional recursive lattice is given in Appendix D.2).

```c
void RECURSIVE_2D_binomial(double *value, double S1, double S2, double X,
                           double sigma1, double sigma2, double rho, double T, double r, double q1, double q2,
                           Integer put, Integer M, Integer opt_type, Integer is_american, Integer recc, Integer *iflag)
{
    if (recc == 0) { /* called without recursion, assign terminal nodes as for a standard two dimensional lattice */
        ...
    }
}
```

Figure 12.1 Diagram illustrating a one-dimensional truncated binomial lattice in which max\_index = 4. This means that there are only nine different asset values in the lattice: the current asset price, four above the current asset value and four below the current asset value.
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```c
else { /* called with recursive last step */
    P1 = 1;
    for (i = 0; i <= M - 1; ++i) {
        P2 = 1;
        for (j = 0; j <= M - 1; ++j) {
            loc_T = dt;
            loc_M = 10;
            loc_recc = 0;
            loc_iflag = 0;
            loc_is_american = is_american;
            recursive_2D_binomial(&hold, s1[P1], s2[P2], X, sigma1, sigma2, rho,
                               loc_T, r, q1, q2, put, loc_M, opt_type, loc_is_american, loc_recc, &loc_iflag);
            if (is_american) { /* An american option so use */
                hold, s1[P1], and s2[P2] to calculate the option value */
                ...
            }
            else {
                V(i, j) = hold;
            }
            P2 = P2 + 2;
        }
        P1 = P1 + 2;
    }
    for (m1 = M - 1 - recc; m1 > 0; --m1) { /* work backwards through the lattice to calculate the option value */
        P1 = M - m1;
        /* Identical code to the equivalent loop of the standard 2 dimensional binomial lattice
         * see code excerpt 3.11 */
        ...
    }
    *value = V(0, 0);
}
```

Code excerpt 12.3 Code excerpt showing the recursive calculation for the last time step, using a ten step lattice over the time interval $dt$

In Sections 12.5 and 12.6 we present results showing the benefits of using a recursive lattice for options on the maximum or minimum of two and three assets.

### 12.5 TWO ASSET OPTIONS

Here we consider options based on the underlying prices of two assets, $S_1$ and $S_2$. We give analytic formulae to value European options based on the maximum and minimum of two assets and also show how two-dimensional binomial lattices can be constructed to value American style options.

#### 12.5.1 European options

We begin by presenting results from Stulz (1982) and Johnson (1987) concerning the value of European call option on the maximum and minimum of two assets.

**Call options on the maximum and minimum of two assets**

Let the value of a European call option on the minimum of two assets, $S_1$ and $S_2$, with strike price $E$, maturity $\tau$, and correlation coefficient $\rho$, be denoted by $\text{c}_{\text{min}}$. The value of the corresponding call option on the maximum of these assets will be represented by $\text{c}_{\text{max}}$. 
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Then, following Stulz (1982) and Johnson (1987), we have:

\[ c_{\text{max}} = S_1 N_2(d_1(S_1, E, \sigma_1^2), d_1'(S_1, S_2, \sigma_2^2), \rho_1) + S_2 N_2(d_1(S_2, E, \sigma_2^2), d_1'(S_2, S_1, \sigma_1^2), \rho_2) \]

\[ -E \exp(-r\tau) \{ 1 - N_2(-d_2(S_1, E, \sigma_1^2), -d_2(S_2, E, \sigma_2^2), \rho) \} \]

(12.13)

and

\[ c_{\text{min}} = S_1 N_2(d_1(S_1, E, \sigma_1^2), -d_1'(S_1, S_2, \sigma_2^2), -\rho_1) \]

\[ + S_2 N_2(d_1(S_2, E, \sigma_2^2), -d_1'(S_2, S_1, \sigma_1^2), -\rho_2) \]

\[ -E \exp(-r\tau) N_2(d_2(S_1, E, \sigma_1^2), d_2(S_2, E, \sigma_2^2), \rho) \]

(12.14)

where \( N_2(a_1, b_1, c_1) \) is the bivariate cumulative normal with \ldots, this can for instance be computed using the NAG routine g01hac. The other symbols are defined as follows:

\[ \sigma_*^2 = \sigma_1^2 - 2\rho \sigma_1 \sigma_2 + \sigma_2^2 \]

\[ d_1(S_i, E, \sigma_i^2) = \frac{\log(S_i/E) + (r + \sigma_i^2/2)\tau}{\sigma_i \sqrt{\tau}} \]

\( i = 1, 2 \)

\[ d_2(S_i, E, \sigma_i^2) = \frac{\log(S_i/E) + (r - \sigma_i^2/2)\tau}{\sigma_i \sqrt{\tau}} \]

\( i = 1, 2 \)

\[ d_1'(S_i, S_j, \sigma_*^2) = \frac{\log(S_i/S_j) + (\sigma_*^2/2)\tau}{\sigma_* \sqrt{\tau}} \]

for \( i = 1, j = 2 \), or \( i = 2, j = 1 \)

and

\[ \rho_1 = \frac{\sigma_1 - \rho \sigma_2}{\sigma_*}, \quad \rho_2 = \frac{\sigma_2 - \rho \sigma_1}{\sigma_*} \]

It can also be shown that:

\[ c_{\text{max}}(S_1, S_2, E, \tau) + c_{\text{min}}(S_1, S_2, E, \tau) = c(S_1, E, \tau) + c(S_2, E, \tau) \]

(12.15)

where \( c(S, E, \tau) \) is the value of a vanilla European call. We will now derive expression for the value of the corresponding European put options.

Put options on the minimum of two assets

It will now be shown that the price of a European put option on the minimum of two assets, \( p_{\text{min}}(S_1, S_2, E, \tau) \) is:

\[ p_{\text{min}}(S_1, S_2, E, \tau) = E \exp(-r\tau) - c_{\text{min}}(S_1, S_2, 0, \tau) + c_{\text{min}}(S_1, S_2, E, \tau) \]

(12.16)

where the meaning of the symbols has been previously defined. This result can be proved by considering the following two investments:

Portfolio A

Purchase one put option on the minimum of \( S_1 \) and \( S_2 \) with exercise price \( E \).
Portfolio B

Purchase one discount bond which pays $E$ at maturity. Write (that is sell) one option on the minimum of $S_1$ and $S_2$ with an exercise price of zero. Purchase one option on the minimum of $S_1$ and $S_2$ with exercise price $E$.

We now consider the values of these portfolios at option maturity, time $\tau$.

**If** $\min(S_1, S_2) \geq E$

- **Portfolio A:** Pays zero
- **Portfolio B:** Pays $E - \min(S_1, S_2) + \min(S_1, S_2) - E = 0$

**If** $\min(S_1, S_2) = S_1 < E$

- **Portfolio A:** Pays $E - S_1$
- **Portfolio B:** Pays $E - S_1 + 0 = E - S_1$

**If** $\min(S_1, S_2) = S_2 < E$

- **Portfolio A:** Pays $E - S_2$
- **Portfolio B:** Pays $E - S_2 + 0 = E - S_2$

We have therefore shown that, under all possible circumstances, Portfolio A has the same value as Portfolio B. This means that Equation 12.16 is true.

**Put options on the maximum of two assets**

It will now be shown that the price of a European put option on the maximum of two assets, $p_{\max}(S_1, S_2, E, \tau)$ is:

$$p_{\max}(S_1, S_2, E, \tau) = E \exp(-r\tau) - c_{\max}(S_1, S_2, 0, \tau) + c_{\max}(S_1, S_2, E, \tau) \quad (12.17)$$

where, as before, the meaning of the symbols has been previously defined. This result can be proved by considering the following two investments:

**Portfolio A:**

Purchase one put option on the maximum of $S_1$ and $S_2$ with exercise price $E$.

**Portfolio B:**

Purchase one discount bond which pays $E$ at maturity. Write (that is sell) one option on the maximum of $S_1$ and $S_2$ with an exercise price of zero. Purchase one option on the maximum of $S_1$ and $S_2$ with exercise price $E$.

As before we now consider the values of these portfolios at option maturity, time $\tau$. 

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If \( \max(S_1, S_2) \geq E \)

Portfolio A: Pays zero
Portfolio B: Pays \( E - \max(S_1, S_2) + \max(S_1, S_2) - E = 0 \)

If \( \max(S_1, S_2) = S_2 < E \)

Portfolio A: Pays \( E - S_1 \)
Portfolio B: Pays \( E - S_1 + 0 = E - S_1 \)

If \( \max(S_1, S_2) = S_2 < E \)

Portfolio A: Pays \( E - S_2 \)
Portfolio B: Pays \( E - S_2 + 0 = E - S_2 \)

It therefore follows that, under all possible circumstances, Portfolio A has the same value as Portfolio B, and this means that Equation 12.17 is true.

```c
void rainbow_bs_2d(double *opt_value, double S1, double S2, double X, double sigma1,
                    double sigma2, double rho, double opt_mat, double r, Integer is_max, Integer *iflag)
{
    /* Input parameters: */
    /* S1 — the current price of the underlying asset 1, */
    /* S2 — the current price of the underlying asset 2, */
    /* X — the strike price, */
    /* sigma1 — the volatility of asset 1, */
    /* sigma2 — the volatility of asset 2, */
    /* rho — the correlation coefficient between asset 1 and asset 2, */
    /* opt_mat — the time to maturity, */
    /* r — the interest rate, */
    /* is_max — if is_max is 1 then the option is a call on the maximum of two assets, otherwise the option is a */
    /* call on the minimum of two assets. */
    
    Output parameters:
    /* opt_value — the value of the option, */
    /* iflag — an error indicator. */
    */
    double one = 1.0, two = 2.0, zero = 0.0;
    double eps,d1,d2_1,d2_2,temp,temp1,temp2,pi,np;
    double rho_112, rho_212, d1_prime;
    double sigma, term1, term2, term3;
    static NagError nagerr;
    eps = X02AJC;
    if(X < eps) printf ("ERROR the strike price is too small
'');
    if (sigma1 < eps) printf ("ERROR the volatility(sigma1) is too small 
'');
    if (sigma2 < eps) printf ("ERROR the volatility(sigma2) is too small 
'');
    if (opt_mat < eps) printf ("ERROR the time to maturity (opt_mat) is too small \n'");
    sigma = sqrt((sigma1*sigma1
                + sigma2*sigma2)/(two* sigma1*sigma2*rho));

    if (is_max == 1) /* then the maximum of two assets */
    /* calculate term1 */
    temp = log(S1/X);
    d1 = temp+(r+(sigma1*sigma1/2)*opt_mat);
    d1 = d1/(sigma1*sqrt(opt_mat));
    temp = log(S1/S2);
    d1_prime = temp+(sigma1*sigma2/2)*opt_mat;
    d1_prime = d1_prime/(sigma2*sqrt(opt_mat));
    rho_112 = (sigma1 - rho*sigma2) / sigma;
    term1 = g01hac(d1,d1_prime,rho_112,&nagerr);
    term1 = term1*S1;
    /* calculate term2 */
    temp = log(S2/X);
    d1 = temp+(r+(sigma2*sigma2/2)*opt_mat);
    d1 = d1/(sigma2*sqrt(opt_mat));
    temp = log(S2/S1);
```
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\[
d_1' = \text{temp} + (\sigma^2 \sigma / 2) \times \text{opt_mat}; \\
d_1' = d_1' / \left( \sigma \sqrt{\text{opt_mat}} \right); \\
d_2' = \text{term2} - \text{g01hac}(d_1, d_1', \text{rho}, \text{flag}); \\
\text{term2} = S_2 \times \text{g01hac}(d_1, d_1', \text{rho}, \text{flag}); \\
\text{rho} = (\sigma / \sqrt{\text{opt_mat}}); \\
\text{term3} = -\text{g01hac}(-d_2, -d_2', \text{rho}, \text{flag}); \\
\text{opt_value} = \text{term1} + \text{term2} \times \exp \left( -r \times \text{opt_mat} \right) \times \text{term3}; \\
\}
\]

else { /* the minimum of two assets */
    /* calculate term1 */
    \[
    \text{temp} = \log(S_1 / X); \\
d_1 = \text{temp} + (r - \sigma^2 \sigma / 2) \times \text{opt_mat}; \\
d_1 = d_1 / \left( \sigma \sqrt{\text{opt_mat}} \right); \\
\text{temp} = \log(S_2 / X); \\
d_2 = \text{temp} + (r - \sigma^2 \sigma / 2) \times \text{opt_mat}; \\
d_2 = d_2 / \left( \sigma \sqrt{\text{opt_mat}} \right); \\
\text{rho} = (\sigma / \sqrt{\text{opt_mat}}); \\
\text{term1} = \text{g01hac}(d_1, d_1', \text{rho}, \text{flag}); \\
\text{term1} = \text{term1} \times S_1; \\
\}
    /* calculate term2 */
    \[
    \text{temp} = \log(S_2 / X); \\
d_1 = \text{temp} + (r - \sigma^2 \sigma / 2) \times \text{opt_mat}; \\
d_1 = d_1 / \left( \sigma \sqrt{\text{opt_mat}} \right); \\
\text{temp} = \log(S_1 / X); \\
d_2 = \text{temp} + (r - \sigma^2 \sigma / 2) \times \text{opt_mat}; \\
d_2 = d_2 / \left( \sigma \sqrt{\text{opt_mat}} \right); \\
\text{term3} = \text{g01hac}(d_1, d_1', \text{rho}, \text{flag}); \\
\text{term3} = \text{g01hac}(d_2, d_2', \text{rho}, \text{flag}); \\
\text{term3} = \text{g01hac}(d_1, d_2', \text{rho}, \text{flag}); \\
\text{term3} = \text{g01hac}(d_2, d_1', \text{rho}, \text{flag}); \\
\text{opt_value} = \text{term1} + \text{term2} - X \times \exp \left( -r \times \text{opt_mat} \right) \times \text{term3}; \\
\}
return;

Code excerpt 12.4 Function to calculate the value of a European call on the maximum or minimum of two assets using the analytic result of Johnson (1987) and Stulz (1982)

\[
\text{void opt_rainbow.bs.2d(double *opt_value, double S1, double S2, double X, double sigma1,} \\
\text{double sigma2, double rho, double opt_mat, double r, Integer is_max, Integer putcall, Integer *flag)} \\
\{ \\
\} /* Input parameters: \\
\text{S1 — the current price of the underlying asset 1,} \\
\text{S2 — the current price of the underlying asset 2,} \\
\text{X — the strike price,} \\
\text{sigma1 — the volatility of asset 1,} \\
\text{sigma2 — the volatility of asset 2,} \\
\text{rho — the correlation coefficient between asset 1 and asset 2,} \\
\text{opt_mat — the time to maturity,} \\
\text{r — the interest rate,} \\
\text{is_max — if is_max is 1 then the option is on the maximum of two assets, otherwise the option is on} \\
\text{the minimum of two assets,} \\
\text{putcall — if putcall is 0 then the option is a call, otherwise the option is a put.} \\
\} /* Output parameters: \\
\text{opt_value — the value of the option,} \\
\text{iflag — an error indicator.} \\
\} */

double temp1;
double temp2;
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```
double fac:
double a_zero = 1.0e-6; /* approximate zero number to prevent overflow in rainbow_bs_2d */
if (putcall) /* a put option */
    fac = X*exp(-r*opt_mat); //todo: make sure the dimension is correct
rainbow_bs_2d(&temp1, S1, S2, a_zero, sigma1, sigma2, rho, opt_mat, r, is_max, flag);
else /* a call option */
    rainbow_bs_2d(opt_value, S1, S2, X, sigma1, sigma2, rho, opt_mat, r, is_max, flag);
}
```

Code excerpt 12.5 Function to calculate the value of a European put or call on the maximum or minimum of two assets using the analytic result of Johnson (1987) and Stulz (1982)

Option prices computed, using a two-dimensional binomial lattice and also the analytic formula of Johnson and Stulz, are presented in Tables 12.3 and 12.4.

Table 12.3 The computed values and absolute errors for European put and call options on the maximum of two assets. The results were obtained using a binomial lattice and the analytic formula (Johnson, 1987; Stulz, 1982). The time to maturity of the option is varied from 0.1 to 0.8 years. The parameters are: \( E = 44.0, S_1 = 40.0, S_2 = 50.0, r = 0.1, \sigma_1 = 0.2, \sigma_2 = 0.2, q_1 = q_2 = 0.0, \rho = 0.5, n_{steps} = 50 \)

<table>
<thead>
<tr>
<th>Time</th>
<th>Call</th>
<th>Put</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td>Analytic</td>
<td>Lattice</td>
</tr>
<tr>
<td>0.1</td>
<td>6.45320</td>
<td>6.45245</td>
</tr>
<tr>
<td>0.2</td>
<td>6.96192</td>
<td>6.95953</td>
</tr>
<tr>
<td>0.3</td>
<td>7.49587</td>
<td>7.49376</td>
</tr>
<tr>
<td>0.4</td>
<td>8.03710</td>
<td>8.04022</td>
</tr>
<tr>
<td>0.5</td>
<td>8.57808</td>
<td>8.57916</td>
</tr>
<tr>
<td>0.6</td>
<td>9.11529</td>
<td>9.10809</td>
</tr>
<tr>
<td>0.7</td>
<td>9.64700</td>
<td>9.64838</td>
</tr>
<tr>
<td>0.8</td>
<td>10.17238</td>
<td>10.17663</td>
</tr>
</tbody>
</table>

Table 12.4 The computed values and absolute errors for European put and call options on the minimum of two assets. The results were obtained using a binomial lattice and the analytic formula (Johnson, 1987; Stulz, 1982). The time to maturity of the option is varied from 0.1 to 0.8 years. The parameters are: \( E = 44.0, S_1 = 40.0, S_2 = 50.0, r = 0.1, \sigma_1 = 0.2, \sigma_2 = 0.2, q_1 = q_2 = 0.0, \rho = 0.5, n_{steps} = 50 \)

<table>
<thead>
<tr>
<th>Time</th>
<th>Call</th>
<th>Put</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td>Analytic</td>
<td>Lattice</td>
</tr>
<tr>
<td>0.1</td>
<td>0.10810</td>
<td>0.10753</td>
</tr>
<tr>
<td>0.2</td>
<td>0.40862</td>
<td>0.40781</td>
</tr>
<tr>
<td>0.3</td>
<td>0.74162</td>
<td>0.73418</td>
</tr>
<tr>
<td>0.4</td>
<td>1.06989</td>
<td>1.07299</td>
</tr>
<tr>
<td>0.5</td>
<td>1.38675</td>
<td>1.38909</td>
</tr>
<tr>
<td>0.6</td>
<td>1.69203</td>
<td>1.69025</td>
</tr>
<tr>
<td>0.7</td>
<td>1.98691</td>
<td>1.96399</td>
</tr>
<tr>
<td>0.8</td>
<td>2.27276</td>
<td>2.26274</td>
</tr>
</tbody>
</table>
12.5.2 American options

We assume that the prices of assets 1 and 2 follow a lognormal process with drift terms of $\mu_1 = r - \sigma_1^2/2$, and $\mu_2 = r - \sigma_2^2/2$ respectively. As before $r$ is the riskless interest rate and $\sigma_1$ and $\sigma_2$ are the instantaneous volatilities of assets 1 and 2.

If we let $S_{1,t}$ and $S_{2,t}$ denote the respective prices of assets 1 and 2 at time $t$, then we can write:

$$\log(S_{1,t+\Delta t}) = \log(S_{1,t}) + \epsilon_{1,t}$$  \hspace{1cm} (12.18)

and

$$\log(S_{2,t+\Delta t}) = \log(S_{2,t}) + \epsilon_{2,t}$$  \hspace{1cm} (12.19)

where $\epsilon_{1,t}$ is a random normal variable with mean $\mu_1 \Delta t$ and variance $\sigma_1^2 \Delta t$, and $\epsilon_{2,t}$ is a random normal variable with mean $\mu_2 \Delta t$ and variance $\sigma_2^2 \Delta t$.

In the binomial lattice model, over the time interval $\Delta t$, the variate $\log(S_{1,t})$ is only allowed to jump up or down by an amount $\nu_1 = \sigma_1 \sqrt{\Delta t}$, and similarly the variate $\log(S_{2,t})$ is only permitted to jump up and down by the amount $\nu_2 = \sigma_2 \sqrt{\Delta t}$. We will denote the probability of both $\log(S_{1,t})$ and $\log(S_{2,t})$ having an up jump over time interval $\Delta t$ by $p_{uu}$, and the probability of $\log(S_{1,t})$ having an up jump and $\log(S_{2,t})$ having a down jump by $p_{ud}$, etc.

The mean values in Equations 12.18 and 12.19 then give

$$E[\epsilon_{1,t}] = \mu_1 (p_{uu} + p_{ad} - p_{dd} - p_{du}) = \mu_1 \Delta t$$  \hspace{1cm} (12.20)

$$E[\epsilon_{2,t}] = \mu_2 (p_{uu} + p_{ad} - p_{dd} - p_{du}) = \mu_2 \Delta t$$  \hspace{1cm} (12.21)

and the variance/covariance terms yields

$$Var[\epsilon_{1,t}] = \nu_1^2 (p_{uu} + p_{ad} + p_{dd} + p_{du}) = \sigma_1^2 \Delta t$$  \hspace{1cm} (12.22)

$$Var[\epsilon_{2,t}] = \nu_2^2 (p_{uu} + p_{ad} + p_{dd} + p_{du}) = \sigma_2^2 \Delta t$$  \hspace{1cm} (12.23)

$$E[\epsilon_{1,t}\epsilon_{2,t}] = \nu_1 \nu_2 (p_{uu} - p_{ad} + p_{dd} - p_{du}) = \rho \sigma_1 \sigma_2 \Delta t$$  \hspace{1cm} (12.24)

where $\rho$ is the correlation coefficient between $\epsilon_{1,t}$ and $\epsilon_{2,t}$.

We therefore obtain:

$$p_{uu} + p_{ad} - p_{dd} + p_{du} = \frac{\mu_1 \sqrt{\Delta t}}{\sigma_1}$$

$$p_{uu} - p_{ad} - p_{dd} + p_{du} = \frac{\mu_2 \sqrt{\Delta t}}{\sigma_2}$$

$$p_{uu} + p_{ad} + p_{dd} - p_{du} = 1$$

$$p_{uu} - p_{ad} + p_{dd} - p_{du} = \rho$$
These lead to the following jump probabilities:

\[
p_{uu} = \frac{1}{4} \left( 1 + \Delta t \left( \frac{\mu_1}{\sigma_1} + \frac{\mu_2}{\sigma_2} \right) + \rho \right)
\]

\[
p_{ul} = \frac{1}{4} \left( 1 + \Delta t \left( \frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} \right) - \rho \right)
\]

\[
p_{dl} = \frac{1}{4} \left( 1 + \Delta t \left( -\frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} \right) + \rho \right)
\]

\[
p_{du} = \frac{1}{4} \left( 1 + \Delta t \left( -\frac{\mu_1}{\sigma_1} + \frac{\mu_2}{\sigma_2} \right) - \rho \right)
\]

In Code excerpt 12.6, we provide the computer code for a standard binomial lattice which prices options on the maximum and minimum of two assets.

The parameter \( M \) is the number of time steps used, and the lattice is constructed under the assumption that \( M \) is even.

```c
void standard_2D_binomial(double *value, double S1, double S2, double X, double sigma1, double sigma2,
                          double rho, double T, double r, double q1, double q2, Integer put, Integer M, Integer opt_type, Integer
                          is_american, Integer *iflag)
{
    /* Input parameters: 
       S1 — the current price of the underlying asset 1 
       S2 — the current price of the underlying asset 2 
       X — the strike price 
       sigma1 — the volatility of asset 1 
       sigma2 — the volatility of asset 2 
       rho — the correlation coefficient between asset 1 and asset 2 
       T — the time to maturity 
       r — the interest rate 
       q1 — the continuous dividend yield for asset 1 
       q2 — the continuous dividend yield for asset 2 
       put — if put is 0 then a call option, otherwise a put option 
       M — the number of time steps, the zeroth time step is the root node of the lattice 
       opt_type — if opt_type is 0 then an option on the maximum of two asset otherwise an option 
                     on the minimum of two assets 
       is_american — if is_american is 0 then a European option, otherwise an American option 
    Output parameters: 
       value — the value of the option, 
       iflag — an error indicator. */

double discount, dt, d1, d2, u1, u2;
Integer i, j, m, iflagx, jj, ind;
double zero=0.0, hold;
double temp, ds1, ds2, dv1, dv2, h, tmp;
double *s1, *s2, *v;
double p[4];
Integer P1, P2, tdv;
double sqrt_dt, t, mu1, mu2, jp1, jp2;
double one=1.0, half=0.5, quarter=0.25;
Integer vi;

if (((M+1)/2 == N/2)) {
    printf("ERROR THE NUMBER OF TIME STEPS IS NOT EVEN \n");
    return;
}

    tdv = M + 1;
```
#define V(I,J) v[(I) * tdv + (J)]
#define UU 0
#define UD 1
#define DD 2
#define DU 3
dt = T/(double)M;
sqrt_dt = sqrt(dt);
jp1 = sigma1*sqrt_dt;
jp2 = sigma2*sqrt_dt;
u1 = r - q1 - sigma1*sigma1*half;
u2 = r - q2 - sigma2*sigma2*half;
ui = exp(jp1); /* assign the jump sizes */
ud = exp(jp2);
d1 = exp(-jp1);
d2 = exp(-jp2);
p[UU] = quarter*(one + sqrt_dt*((mu1/sigma1) + (mu2/sigma2)) + rho); /* setup the jump probabilities */
p[UD] = quarter*(one + sqrt_dt * ((mu1/sigma1) / C0 * (mu2/sigma2)) / C0 * rho);
p[DD] = quarter*(one + sqrt_dt * ((mu1/sigma1) / C0 * (mu2/sigma2)) + rho);
p[DU] = quarter*(one + sqrt_dt * ((mu1/sigma1) / C0 * (mu2/sigma2)) / C0 * rho);

for (i = 0; i < M; ++i) {
  if ((p[i] < zero) || (p[i] >1.0)) printf (''ERROR p out of range
'');
}
discount = exp(-r*dt);
for (i = 0; i < M; ++i) {
p[i] = p[i]*discount;
}

/* Allocate the arrays v[(M+1)*(M+1)], s1[2*M+1] and s2[2*M+1] */
s1[M] = S1; /* assign the 2*M+1 asset values for s1 */
for (i = 1; i <= M; ++i) {
s1[M+i] = u1*s1[M+i-1];
s1[M-i] = d1*s1[M-i+1];
}
s2[M] = S2; /* assign the 2*M+1 asset values for s2 */
for (i = 1; i <= M; ++i) {
s2[M+i] = u2*s2[M+i-1];
s2[M-i] = d2*s2[M-i+1];
}
P1 = 0;

for (i = 0; i <= M; ++i) /* Calculate the option values at maturity */
P2 = 0;
for (j = 0; j <= M; ++j) {
  if (is_american) { /* An American option */
    if (opt_type == 0) { /* Maximum of two assets */
      if (put) {
        V(i,j) = MAX(X - MAX(s1[P1],s2[P2]),zero);
      } else {
        V(i,j) = MAX(MAX(s1[P1],s2[P2]) - X,zero);
      }
    } else {
      if (put) { /* Minimum of two assets */
        V(i,j) = MAX(X-MIN(s1[P1],s2[P2]),zero);
      } else {
        V(i,j) = MAX(MIN(s1[P1],s2[P2]) - X,zero);
      }
    }
  } else { /* Minimum of two assets */
    V(i,j) = MAX(s1[P1],s2[P2]) - X;
  }
  P2 = P2 + 2;
  P1 = P1 + 2;
}

for (m1 = M-1; m1 > 0; --m1) /* work backwards through the lattice to calculate option value */
P1 = M-m1;
for (i = 0; i <= m1; ++i) {
P2 = M-m1;
for (j = 0; j <= m1; ++j) {
  hold = p[UD]*V(i+1,j+1) + p[UU]*V(i+1,j+1) + p[DD]*V(i+1,j+1) + p[DU]*V(i,j);
  if (is_american) { /* An American option */
    if (opt_type == 0) { /* Maximum of two assets */
      if (put) {
        V(i,j) = MAX(hold, X-MAX(s1[P1],s2[P2]));
      } else {
        V(i,j) = MAX(hold, MAX(s1[P1], s2[P2]) - X);
      }
    } else { /* Minimum of two assets */

```
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```cpp
if (put)
    V(i,j) = MAX(hold, X - MIN(s1[F1], s2[F2]));
else
    V(i,j) = MAX(hold, MIN(s1[F1], s2[F2]) - X);
}
else {
    V(i,j) = hold;
}
```

*value = V(0,0);*

**Code excerpt 12.6** Function to calculate the value of a European put or call on the maximum or minimum of two assets using a standard binomial lattice

The computer code for a truncated two-dimensional binomial lattice is given in Appendix D.1.

**Table 12.5** The computed values and absolute errors for an American put option on the maximum of two assets. A truncated binomial lattice was used and we show how the accuracy depends on the value of `max_index`. The parameters are: \(E = 100.0, S_1 = S_2 = 100.0, r = 0.2, \sigma_1 = \sigma_2 = 0.2, \rho = 0.5, q_1 = q_2 = 0.0.\) The first column gives the value of `max_index`, the second column the computational time in milliseconds, the third column the computed value of the option, and the last column the absolute error. The accurate value took 280 ms to compute and was obtained using a standard lattice with `n_steps = 200`.

<table>
<thead>
<tr>
<th>max_index</th>
<th>Time (ms)</th>
<th>Value</th>
<th>Error</th>
</tr>
</thead>
<tbody>
<tr>
<td>22</td>
<td>10.0</td>
<td>1.8478</td>
<td>6.3912 \times 10^3</td>
</tr>
<tr>
<td>26</td>
<td>20.0</td>
<td>1.8525</td>
<td>1.7179 \times 10^{-3}</td>
</tr>
<tr>
<td>40</td>
<td>30.0</td>
<td>1.8542</td>
<td>1.277 \times 10^{-5}</td>
</tr>
<tr>
<td>48</td>
<td>40.0</td>
<td>1.8542</td>
<td>5.0338 \times 10^{-7}</td>
</tr>
<tr>
<td>52</td>
<td>50.0</td>
<td>1.8542</td>
<td>8.6936 \times 10^{-8}</td>
</tr>
<tr>
<td>58</td>
<td>60.0</td>
<td>1.8542</td>
<td>5.1993 \times 10^{-9}</td>
</tr>
</tbody>
</table>

**Table 12.6** The computed values and absolute errors for an American put option on the minimum of two assets. A truncated binomial lattice was used and we show how the accuracy depends on the value of `max_index`. The parameters are: \(E = 100.0, S_1 = S_2 = 100.0, r = 0.2, \sigma_1 = \sigma_2 = 0.2, \rho = 0.5, q_1 = q_2 = 0.0.\) The first column gives the value of `max_index`, the second column the computational time in milliseconds, the third column the computed value of the option, and the last column the absolute error. The accurate value took 311 ms to compute and was obtained using a standard lattice with `n_steps = 200`.

<table>
<thead>
<tr>
<th>max_index</th>
<th>Time (ms)</th>
<th>Value</th>
<th>Error</th>
</tr>
</thead>
<tbody>
<tr>
<td>26</td>
<td>10.0</td>
<td>4.7008</td>
<td>3.8500 \times 10^{-2}</td>
</tr>
<tr>
<td>32</td>
<td>20.0</td>
<td>4.7301</td>
<td>9.2094 \times 10^{-3}</td>
</tr>
<tr>
<td>40</td>
<td>30.0</td>
<td>4.7383</td>
<td>1.0258 \times 10^{-3}</td>
</tr>
<tr>
<td>44</td>
<td>40.0</td>
<td>4.7390</td>
<td>2.9951 \times 10^{-4}</td>
</tr>
<tr>
<td>48</td>
<td>50.0</td>
<td>4.7392</td>
<td>7.9631 \times 10^{-5}</td>
</tr>
<tr>
<td>52</td>
<td>60.0</td>
<td>4.7393</td>
<td>1.9230 \times 10^{-5}</td>
</tr>
</tbody>
</table>
### 12.6 THREE ASSET OPTIONS

For three assets we have the following jump probabilities:

\[
p_{uuu} = \frac{1}{8} \left\{ 1 + \sqrt{\Delta t} \left( \frac{\mu_1}{\sigma_1} + \frac{\mu_2}{\sigma_2} + \frac{\mu_3}{\sigma_3} \right) + \rho_{12} + \rho_{13} + \rho_{23} \right\}
\]

\[
p_{uud} = \frac{1}{8} \left\{ 1 + \sqrt{\Delta t} \left( \frac{\mu_1}{\sigma_1} + \frac{\mu_2}{\sigma_2} - \frac{\mu_3}{\sigma_3} \right) + \rho_{12} - \rho_{13} - \rho_{23} \right\}
\]

\[
p_{udu} = \frac{1}{8} \left\{ 1 + \sqrt{\Delta t} \left( \frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} + \frac{\mu_3}{\sigma_3} \right) - \rho_{12} + \rho_{13} - \rho_{23} \right\}
\]

\[
p_{udd} = \frac{1}{8} \left\{ 1 + \sqrt{\Delta t} \left( \frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} - \frac{\mu_3}{\sigma_3} \right) - \rho_{12} - \rho_{13} + \rho_{23} \right\}
\]

\[
p_{duu} = \frac{1}{8} \left\{ 1 + \sqrt{\Delta t} \left( -\frac{\mu_1}{\sigma_1} + \frac{\mu_2}{\sigma_2} + \frac{\mu_3}{\sigma_3} \right) - \rho_{12} - \rho_{13} + \rho_{23} \right\}
\]

\[
p_{dud} = \frac{1}{8} \left\{ 1 + \sqrt{\Delta t} \left( -\frac{\mu_1}{\sigma_1} + \frac{\mu_2}{\sigma_2} - \frac{\mu_3}{\sigma_3} \right) - \rho_{12} + \rho_{13} - \rho_{23} \right\}
\]

\[
p_{ddu} = \frac{1}{8} \left\{ 1 + \sqrt{\Delta t} \left( -\frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} + \frac{\mu_3}{\sigma_3} \right) + \rho_{12} - \rho_{13} - \rho_{23} \right\}
\]

\[
p_{ddd} = \frac{1}{8} \left\{ 1 + \sqrt{\Delta t} \left( -\frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} - \frac{\mu_3}{\sigma_3} \right) + \rho_{12} + \rho_{13} + \rho_{23} \right\}
\]

The computer code for a standard three-dimensional lattice is given in Code excerpt 12.7 below. Code for truncated and recursive lattices is supplied on the CD ROM.

```c
void standard_3D_binomial (double *value, double S1, double S2, double S3, double X, double sigma1,
    double sigma2, double sigma3, double rho_12, double rho_13, double rho_23, double T, double r,
    Integer put, Integer M, Integer opt_type, Integer is_american, Integer *iflag)
{
    /* Input parameters:
    S1 — the current price of the underlying asset 1
    S2 — the current price of the underlying asset 2
    S3 — the current price of the underlying asset 3
    X — the strike price
    sigma1 — the volatility of asset 1
    sigma2 — the volatility of asset 2
    sigma3 — the volatility of asset 3
    rho_12 — the correlation coefficient between asset 1 and asset 2
    rho_13 — the correlation coefficient between asset 1 and asset 3
    rho_23 — the correlation coefficient between asset 2 and asset 3
    T — the time to maturity
    r — the interest rate
```
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put — if put is 0 then a call option, otherwise a put option
M — the number of time steps, the zeroth time step is the root node of the lattice
opt_type — if opt_type is 0 then an option on the maximum of two asset otherwise an option on the minimum of two assets
is_american — if is_american is 0 then a European option, otherwise an American option.

Output parameters:
value — the value of the option,
iflag — an error indicator.

```
double discount, t1, dt, d1, d2, d3, u1, u2, u3;
Integer i, j, k, m1, n, iflagx, jj, ind;
double zero = 0.0, hold;
double temp, ds1, ds2, dv1, dv2, h, tmp, tmp1, tmp2;
double *s1, *s2, *s3, *v;
double p[9];
Integer P1, P2, P3, tdv, tdv2;
double sqrt_dt, t, mu1, mu2, mu3, j1, j2, j3;
double one = 1.0, half = 0.5, eighth = 0.125;
Integer v1;
if (!((M + 1)/2 == M/2)) {
    printf ('ERROR THE NUMBER OF TIME STEPS IS NOT EVEN 
');
    return;
}
tdv = M + 1;
tdv2 = tdv * tdv;
#define V(I, J, K) v[(I) * tdv2 + (J) * tdv + (K)]
#define UUU 0
#define UUD 1
#define UDU 2
#define UDD 3
#define DUU 4
#define DUD 5
#define DDU 6
#define DDD 7
dt = T/(double)M;
sqrt_dt = sqrt(dt);
j1 = sigma1*sqrt_dt;
j2 = sigma2*sqrt_dt;
j3 = sigma3*sqrt_dt;
mu1 = r/C0*sigma1*sigma1*half;
mu2 = r/C0*sigma2*sigma2*half;
mu3 = r/C0*sigma3*sigma3*half;
for (i = 0; i < 8; ++i) {
    p[i] = exp(j1); /* assign the jump sizes */
    u2 = exp(j2);
    u3 = exp(j3);
    d1 = exp(-j1);
    d2 = exp(-j2);
    d3 = exp(-j3);
    /* set up the jump probabilities */
    p[UUU] = eighth'((one + sqrt_dt * ((mu1/sigma1) + (mu2/sigma2) + (mu3/sigma3)) + rho_12 + rho_13 + rho_23);
    p[UUD] = eighth'((one + sqrt_dt * ((mu1/sigma1) + (mu2/sigma2) - (mu3/sigma3)) + rho_12 - rho_13 + rho_23);
    p[UDU] = eighth'((one + sqrt_dt * ((mu1/sigma1) - (mu2/sigma2) + (mu3/sigma3)) - rho_12 + rho_13 - rho_23);
    p[DDD] = eighth'((one + sqrt_dt * ((mu1/sigma1) + (mu2/sigma2) + (mu3/sigma3)) - rho_12 + rho_13 - rho_23);
    for (i = 0; i < 8; ++i) {
        if ((p[i] < zero) || (p[i] > 1.0)) printf ("ERROR p[%d] = %2.4f out of range\n", i, p[i]);
    }
    discount = exp(-*dt);
    for (i = 0; i < 8; ++i) {
        p[i] = p[i]*discount;
    }
    /* Allocate the arrays v[(M+1)*(M+1)], s1[2*M+1], s2[2*M+1], and s3[2*M+1] */
    s1[M] = s1;
    for (i = 1; i <= M; ++i) {/* assign the 2*M+1 asset values for s1 */
```
Multiasset European and American options

\[
s_1[M+1] = u_1*s_1[M+1-1];
\]
\[
s_1[M-1] = d_1*s_1[M-1+1];
\]
\[
s_2[M] = s_2;
\]
for (i = 1; i <= M; ++i) { /* assign the 2*M+1 asset values for s_2 */
  s_2[M+1] = u_2*s_2[M+1-1];
  s_2[M-1] = d_2*s_2[M-1+1];
}
\[
s_3[M] = s_3;
\]
for (i = 1; i <= M; ++i) { /* assign the 2*M+1 asset values for s_3 */
  s_3[M+1] = u_3*s_3[M+1-1];
  s_3[M-1] = d_3*s_3[M-1+1];
}
/* Calculate the option values at maturity */
P_1 = 0;
for (i = 0; i <= M; ++i) {
  P_2 = 0;
  for (j = 0; j <= M; ++j) {
    P_3 = 0;
    for (k = 0; k <= M; ++k) {
      hold = p[UU]*V(i+1, j+1, k+1) + p[UD]*V(i+1, j+1, k) + p[DD]*V(i+1, j+1, k+1) +
             p[UU]*V(i+1, j, k+1) + p[UD]*V(i+1, j+1, k) + p[DD]*V(i+1, j, k);
      if (is_american) {
        if (put) {
          if (opt_type == 0) { /* Maximum of 3 assets */
            tmp = MAX(s_1[i], s_2[j]);
            V(i, j, k) = MAX(X - MAX(tmp, s_3[k]), zero);
          }
        }
      }
    }
  }
}
for (m_1 = M-1; m_1 >= 0; --m_1) { /* work backwards through the lattice to calculate the option value */
P_1 = M-m_1;
for (i = 0; i <= m_1; ++i) {
  P_2 = M-m_1;
  for (j = 0; j <= m_1; ++j) {
    P_3 = M-m_1;
    for (k = 0; k <= m_1; ++k) {
      if (put) {
        if (opt_type == 0) { /* Maximum of 3 assets */
          tmp = MAX(s_1[i], s_2[j]);
          tmp1 = MAX(tmp, s_3[k]);
          V(i, j, k) = MAX(tmp1, zero);
        }
      }
    }
  }
}
if (put) {
  if (opt_type == 0) { /* Maximum of 3 assets */
    tmp = MAX(s_1[i], s_2[j]);
    tmp1 = MAX(tmp, s_3[k]);
    V(i, j, k) = MAX(tmp1, zero);
  }
}
Pricing Assets

\[
P_1 = P_1 + 2;
\]

\}

'value = V(0, 0, 0);
}

Code excerpt 12.7  Standard three-dimensional binomial lattice

The results of pricing three asset options, in which \( \rho_{13} = \rho_{23} = 0.5 \), are given in Tables 12.7 to 12.9; standard, truncated, and recursive lattices are used.

<table>
<thead>
<tr>
<th>( n ) steps</th>
<th>Standard lattice</th>
<th>Recursive lattice</th>
</tr>
</thead>
<tbody>
<tr>
<td>10</td>
<td>0.9112 (2.485 \times 10^{-2})</td>
<td>0.9617 (2.574 \times 10^{-2})</td>
</tr>
<tr>
<td>20</td>
<td>0.9192 (1.678 \times 10^{-2})</td>
<td>0.9463 (1.030 \times 10^{-2})</td>
</tr>
<tr>
<td>30</td>
<td>0.9232 (1.276 \times 10^{-2})</td>
<td>0.9416 (5.640 \times 10^{-3})</td>
</tr>
<tr>
<td>40</td>
<td>0.9254 (1.056 \times 10^{-2})</td>
<td>0.9394 (3.370 \times 10^{-3})</td>
</tr>
<tr>
<td>50</td>
<td>0.9268 (9.180 \times 10^{-3})</td>
<td>0.9380 (2.025 \times 10^{-3})</td>
</tr>
<tr>
<td>60</td>
<td>0.9278 (8.236 \times 10^{-3})</td>
<td>0.9371 (1.135 \times 10^{-3})</td>
</tr>
<tr>
<td></td>
<td>21.8601 (8.119 \times 10^{-1})</td>
<td>22.2488 (4.232 \times 10^{-1})</td>
</tr>
<tr>
<td></td>
<td>22.2807 (3.913 \times 10^{-1})</td>
<td>22.4640 (2.080 \times 10^{-1})</td>
</tr>
<tr>
<td></td>
<td>22.4137 (2.583 \times 10^{-1})</td>
<td>22.5339 (1.381 \times 10^{-1})</td>
</tr>
<tr>
<td></td>
<td>22.4792 (1.928 \times 10^{-1})</td>
<td>22.5686 (1.034 \times 10^{-1})</td>
</tr>
<tr>
<td></td>
<td>22.5182 (1.538 \times 10^{-1})</td>
<td>22.6033 (6.875 \times 10^{-2})</td>
</tr>
</tbody>
</table>

Table 12.7  The computed values and absolute errors for European options on the maximum of three assets. A binomial lattice was used and we show how the accuracy of the results depends on the number of time steps. The parameters are: \( E = 100.0, S_1 = S_2 = S_3 = 100.0, r = 0.1, \tau = 1.0, \sigma_1 = \sigma_2 = \sigma_3 = 0.2, \rho_{12} = \rho_{13} = \rho_{23} = 0.5, q_1 = q_2 = q_3 = 0.0. \) The accurate values are 0.936 for a put and 22.672 for a call, see Table 2 Boyle, Evnine, and Gibbs (1989)

<table>
<thead>
<tr>
<th>( n ) steps</th>
<th>Standard lattice</th>
<th>Recursive lattice</th>
</tr>
</thead>
<tbody>
<tr>
<td>10</td>
<td>27.0759 (3.271 \times 10^{-1})</td>
<td>7.3658 (3.723 \times 10^{-2})</td>
</tr>
<tr>
<td>20</td>
<td>7.2402 (1.628 \times 10^{-1})</td>
<td>7.3685 (1.653 \times 10^{-2})</td>
</tr>
<tr>
<td>30</td>
<td>7.2953 (1.077 \times 10^{-1})</td>
<td>7.3931 (9.926 \times 10^{-2})</td>
</tr>
<tr>
<td>40</td>
<td>7.3229 (8.015 \times 10^{-2})</td>
<td>7.3963 (6.676 \times 10^{-3})</td>
</tr>
<tr>
<td>50</td>
<td>7.3394 (6.357 \times 10^{-2})</td>
<td>7.3983 (4.741 \times 10^{-3})</td>
</tr>
<tr>
<td>60</td>
<td>7.3505 (5.251 \times 10^{-2})</td>
<td>7.3995 (3.459 \times 10^{-3})</td>
</tr>
<tr>
<td></td>
<td>5.2072 (4.176 \times 10^{-2})</td>
<td>5.2359 (1.312 \times 10^{-2})</td>
</tr>
<tr>
<td></td>
<td>5.2263 (2.269 \times 10^{-2})</td>
<td>5.2406 (8.414 \times 10^{-3})</td>
</tr>
<tr>
<td></td>
<td>5.2334 (1.560 \times 10^{-2})</td>
<td>5.2429 (6.060 \times 10^{-3})</td>
</tr>
<tr>
<td></td>
<td>5.2371 (1.192 \times 10^{-2})</td>
<td>5.2443 (4.749 \times 10^{-3})</td>
</tr>
<tr>
<td></td>
<td>5.2393 (9.665 \times 10^{-2})</td>
<td>5.2451 (3.922 \times 10^{-3})</td>
</tr>
<tr>
<td></td>
<td>5.2409 (8.143 \times 10^{-3})</td>
<td>5.2456 (3.353 \times 10^{-3})</td>
</tr>
</tbody>
</table>

Table 12.8  The computed values and absolute errors for European options on the minimum of three assets. A binomial lattice was used and we show how the accuracy of the results depends on the number of time steps. The parameters are: \( E = 100.0, S_1 = S_2 = S_3 = 100.0, r = 0.1, \tau = 1.0, \sigma_1 = \sigma_2 = \sigma_3 = 0.2, \rho_{12} = \rho_{13} = \rho_{23} = 0.5, q_1 = q_2 = q_3 = 0.0. \) The accurate values are 7.403 for a put and 5.249 for a call, see Table 2 Boyle, Evnine, and Gibbs (1989)
The result of pricing three asset options, in which $\rho_{12} = -0.5$, and $\rho_{23} = 0.5$, are given in Tables 12.10 to 12.12.

Table 12.9  The computed values and absolute errors for a European put option on the maximum of three assets. A truncated binomial lattice was used and we show how the accuracy depends on the value of $\text{max\_index}$. The parameters are: $E = 0.00, S_1 = S_2 = S_3 = 0.00, r = 0.1, \tau = 1.0$, $\sigma_1 = \sigma_2 = \sigma_3 = 0.2$, $\rho_{12} = \rho_{13} = -0.5, \rho_{23} = 0.5, q_1 = q_2 = q_3 = 0.0$. The first column gives the value of $\text{max\_index}$, the second column the computational time in milliseconds and the third column the computed value of the option and also the absolute error in brackets. The accurate value is 0.9290 (took 1633 ms to compute) and was obtained using a standard lattice with $n\_steps = 80$.

<table>
<thead>
<tr>
<th>max_index</th>
<th>Time (ms)</th>
<th>Value (error)</th>
</tr>
</thead>
<tbody>
<tr>
<td>14</td>
<td>20</td>
<td>0.3464 (5.8254 $\times 10^{-1}$)</td>
</tr>
<tr>
<td>16</td>
<td>30</td>
<td>0.4976 (4.3142 $\times 10^{-1}$)</td>
</tr>
<tr>
<td>18</td>
<td>50</td>
<td>0.6344 (2.9458 $\times 10^{-1}$)</td>
</tr>
<tr>
<td>20</td>
<td>60</td>
<td>0.7432 (1.8581 $\times 10^{-1}$)</td>
</tr>
<tr>
<td>22</td>
<td>81</td>
<td>0.8204 (1.0858 $\times 10^{-1}$)</td>
</tr>
<tr>
<td>24</td>
<td>100</td>
<td>0.8700 (5.8935 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>26</td>
<td>121</td>
<td>0.8992 (2.9782 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>28</td>
<td>140</td>
<td>0.9149 (1.4036 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>30</td>
<td>181</td>
<td>0.9228 (6.1763 $\times 10^{-3}$)</td>
</tr>
<tr>
<td>32</td>
<td>220</td>
<td>0.9264 (2.5939 $\times 10^{-3}$)</td>
</tr>
<tr>
<td>34</td>
<td>270</td>
<td>0.9280 (9.7559 $\times 10^{-4}$)</td>
</tr>
<tr>
<td>36</td>
<td>320</td>
<td>0.9286 (3.5018 $\times 10^{-4}$)</td>
</tr>
<tr>
<td>38</td>
<td>391</td>
<td>0.9289 (1.1735 $\times 10^{-4}$)</td>
</tr>
<tr>
<td>40</td>
<td>441</td>
<td>0.9289 (3.6673 $\times 10^{-5}$)</td>
</tr>
<tr>
<td>42</td>
<td>521</td>
<td>0.9290 (1.0672 $\times 10^{-5}$)</td>
</tr>
</tbody>
</table>

Table 12.10  The computed values and absolute errors for European options on the maximum of three assets. A binomial lattice was used and we show how the accuracy depends on the number of time steps. The parameters are: $E = 100.0, S_1 = S_2 = S_3 = 100.0, r = 0.1, \tau = 1.0, \sigma_1 = \sigma_2 = \sigma_3 = 0.2$, $\rho_{12} = -0.5, \rho_{13} = -0.5, \rho_{23} = 0.5, q_1 = q_2 = q_3 = 0.0$. The accurate values are 0.0526 for a put and 27.8271 for a call, and were computed using Monte Carlo simulation with $10^7$ paths.

<table>
<thead>
<tr>
<th>n steps</th>
<th>Put Standard lattice</th>
<th>Put Recursive lattice</th>
<th>Call Standard lattice</th>
<th>Call Recursive lattice</th>
</tr>
</thead>
<tbody>
<tr>
<td>10</td>
<td>0.0122 (4.041 $\times 10^{-2}$)</td>
<td>0.0273 (2.531 $\times 10^{-2}$)</td>
<td>27.3180 (5.091 $\times 10^{-1}$)</td>
<td>27.5666 (2.605 $\times 10^{-1}$)</td>
</tr>
<tr>
<td>20</td>
<td>0.0295 (2.314 $\times 10^{-2}$)</td>
<td>0.0396 (1.301 $\times 10^{-2}$)</td>
<td>27.5743 (2.528 $\times 10^{-1}$)</td>
<td>27.6963 (1.306 $\times 10^{-1}$)</td>
</tr>
<tr>
<td>30</td>
<td>0.0366 (1.600 $\times 10^{-2}$)</td>
<td>0.0438 (8.770 $\times 10^{-3}$)</td>
<td>27.6589 (1.682 $\times 10^{-1}$)</td>
<td>27.7396 (8.745 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>40</td>
<td>0.0404 (1.221 $\times 10^{-2}$)</td>
<td>0.0460 (6.618 $\times 10^{-3}$)</td>
<td>27.7010 (1.261 $\times 10^{-1}$)</td>
<td>27.7614 (6.569 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>50</td>
<td>0.0427 (9.868 $\times 10^{-3}$)</td>
<td>0.0473 (5.316 $\times 10^{-3}$)</td>
<td>27.7263 (1.008 $\times 10^{-1}$)</td>
<td>27.7745 (5.258 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>60</td>
<td>0.0443 (8.280 $\times 10^{-3}$)</td>
<td>0.0482 (4.444 $\times 10^{-3}$)</td>
<td>27.7431 (8.396 $\times 10^{-2}$)</td>
<td>27.7833 (4.383 $\times 10^{-2}$)</td>
</tr>
</tbody>
</table>
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Table 12.11  The computed values and absolute errors for European options on the minimum of three assets. A binomial lattice was used and we show how the accuracy depends on the number of time steps. The parameters are: $E = 100.0$, $S_1 = S_2 = S_3 = 100.0$, $r = 0.1$, $\tau = 1.0$, $\sigma_1 = \sigma_2 = \sigma_3 = 0.2$, $\rho_{12} = -0.5$, $\rho_{13} = -0.5$, $\rho_{23} = 0.5$, $q_1 = q_2 = q_3 = 0.0$. The accurate values are 9.2776 for a put and 1.5847 for a call, and were computed using Monte Carlo simulation with $10^7$ paths.

<table>
<thead>
<tr>
<th>$n$ steps</th>
<th>Put</th>
<th>Call</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td>Standard lattice</td>
<td>Recursive lattice</td>
</tr>
<tr>
<td>10</td>
<td>8.9646 (3.130 $\times 10^{-1}$)</td>
<td>9.2791 (1.457 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>20</td>
<td>9.1231 (1.545 $\times 10^{-1}$)</td>
<td>9.2796 (1.979 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>30</td>
<td>9.1749 (1.027 $\times 10^{-1}$)</td>
<td>9.2792 (1.594 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>40</td>
<td>9.2007 (7.694 $\times 10^{-2}$)</td>
<td>9.2789 (1.299 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>50</td>
<td>9.2161 (6.151 $\times 10^{-2}$)</td>
<td>9.2787 (9.336 $\times 10^{-3}$)</td>
</tr>
<tr>
<td>60</td>
<td>9.2264 (5.123 $\times 10^{-2}$)</td>
<td>9.2785 (9.336 $\times 10^{-4}$)</td>
</tr>
</tbody>
</table>

Table 12.12  The computed values and absolute errors for a European put option on the maximum of three assets. A truncated binomial lattice was used and we show how the accuracy depends on the value of $\max_{\text{index}}$. The parameters are: $E = 100.0$, $S_1 = S_2 = S_3 = 100.0$, $r = 0.1$, $\tau = 1.0$, $\sigma_1 = \sigma_2 = \sigma_3 = 0.2$, $\rho_{12} = -0.5$, $\rho_{13} = -0.5$, $\rho_{23} = 0.5$, $q_1 = q_2 = q_3 = 0.0$. The first column gives the value of $\max_{\text{index}}$, the second column the computational time in milliseconds and the third column the computed value of the option and also the absolute error in brackets. The accurate value is 0.0463 (took 1632 ms to compute) and was obtained using a standard lattice with $n_{\text{steps}} = 80$.

<table>
<thead>
<tr>
<th>$\max_{\text{index}}$</th>
<th>Time (ms)</th>
<th>Value (error)</th>
</tr>
</thead>
<tbody>
<tr>
<td>14</td>
<td>20</td>
<td>0.0328 (1.3545 $\times 10^{-2}$)</td>
</tr>
<tr>
<td>16</td>
<td>30</td>
<td>0.0397 (6.6100 $\times 10^{-3}$)</td>
</tr>
<tr>
<td>18</td>
<td>41</td>
<td>0.0434 (2.8917 $\times 10^{-3}$)</td>
</tr>
<tr>
<td>20</td>
<td>60</td>
<td>0.0452 (1.1577 $\times 10^{-3}$)</td>
</tr>
<tr>
<td>22</td>
<td>70</td>
<td>0.0459 (4.2916 $\times 10^{-4}$)</td>
</tr>
<tr>
<td>24</td>
<td>100</td>
<td>0.0462 (1.4810 $\times 10^{-4}$)</td>
</tr>
<tr>
<td>26</td>
<td>120</td>
<td>0.0463 (4.7659 $\times 10^{-5}$)</td>
</tr>
<tr>
<td>28</td>
<td>150</td>
<td>0.0463 (1.4299 $\times 10^{-5}$)</td>
</tr>
</tbody>
</table>

12.7  FOUR ASSET OPTIONS

The jump probabilities for a binomial lattice which computes options on four assets is given in Appendix D.3, and computer code is available on the CD ROM. The results of using a four-dimensional binomial lattice to price options are presented in Tables 12.13 and 12.14.
Table 12.13 The computed values and absolute errors for European options on the maximum of four assets. A binomial lattice was used and we show how the accuracy depends on the number of time steps. The parameters are: $E = 100.0$, $S_1 = S_2 = S_3 = S_4 = 100.0$, $r = 0.1$, $\tau = 1.0$, $\sigma_1 = \sigma_2 = \sigma_3 = \sigma_4 = 0.2$, $\rho_{12} = 0.5$, $\rho_{13} = 0.5$, $\rho_{23} = 0.5$, $q_1 = q_2 = q_3 = q_4 = 0.0$. The accurate values are 0.6309 for a put and 25.2363 for a call, and were computed using Monte Carlo simulation with $10^7$ paths.

<table>
<thead>
<tr>
<th>$n$ steps</th>
<th>Put Estimated value</th>
<th>Error</th>
<th>Call Estimated value</th>
<th>Error</th>
</tr>
</thead>
<tbody>
<tr>
<td>4</td>
<td>0.6548</td>
<td>$2.386 \times 10^{-2}$</td>
<td>22.1403</td>
<td>3.096</td>
</tr>
<tr>
<td>8</td>
<td>0.6268</td>
<td>$4.129 \times 10^{-3}$</td>
<td>23.8640</td>
<td>1.372</td>
</tr>
<tr>
<td>12</td>
<td>0.6246</td>
<td>$6.275 \times 10^{-3}$</td>
<td>24.3630</td>
<td>$8.733 \times 10^{-1}$</td>
</tr>
<tr>
<td>16</td>
<td>0.6251</td>
<td>$5.836 \times 10^{-3}$</td>
<td>24.5934</td>
<td>$6.429 \times 10^{-1}$</td>
</tr>
<tr>
<td>20</td>
<td>0.6257</td>
<td>$5.167 \times 10^{-3}$</td>
<td>24.7270</td>
<td>$5.093 \times 10^{-1}$</td>
</tr>
<tr>
<td>24</td>
<td>0.6263</td>
<td>$4.570 \times 10^{-3}$</td>
<td>24.8144</td>
<td>$4.219 \times 10^{-1}$</td>
</tr>
<tr>
<td>28</td>
<td>0.6268</td>
<td>$4.047 \times 10^{-3}$</td>
<td>24.8762</td>
<td>$3.601 \times 10^{-1}$</td>
</tr>
<tr>
<td>32</td>
<td>0.6272</td>
<td>$3.665 \times 10^{-3}$</td>
<td>24.9222</td>
<td>$3.141 \times 10^{-1}$</td>
</tr>
</tbody>
</table>

Table 12.14 The computed values and absolute errors for European options on the minimum of four assets. A binomial lattice was used and we show how the accuracy depends on the number of time steps. The parameters are: $E = 100.0$, $S_1 = S_2 = S_3 = S_4 = 100.0$, $r = 0.1$, $\tau = 1.0$, $\sigma_1 = \sigma_2 = \sigma_3 = \sigma_4 = 0.2$, $\rho_{12} = 0.5$, $\rho_{13} = 0.5$, $\rho_{23} = 0.5$, $q_1 = q_2 = q_3 = q_4 = 0.0$. The accurate values are 8.5394 for a put and 4.0662 for a call, and were computed using Monte Carlo simulation with $10^7$ paths.

<table>
<thead>
<tr>
<th>$n$ steps</th>
<th>Put Estimated value</th>
<th>Error</th>
<th>Call Estimated value</th>
<th>Error</th>
</tr>
</thead>
<tbody>
<tr>
<td>4</td>
<td>7.8274</td>
<td>$7.120 \times 10^{-1}$</td>
<td>3.5676</td>
<td>$4.986 \times 10^{-1}$</td>
</tr>
<tr>
<td>8</td>
<td>8.1571</td>
<td>$3.823 \times 10^{-1}$</td>
<td>3.8528</td>
<td>$2.134 \times 10^{-1}$</td>
</tr>
<tr>
<td>12</td>
<td>8.2794</td>
<td>$2.600 \times 10^{-1}$</td>
<td>3.9300</td>
<td>$1.362 \times 10^{-1}$</td>
</tr>
<tr>
<td>16</td>
<td>8.3429</td>
<td>$1.965 \times 10^{-1}$</td>
<td>3.9659</td>
<td>$1.003 \times 10^{-1}$</td>
</tr>
<tr>
<td>20</td>
<td>8.3815</td>
<td>$1.579 \times 10^{-1}$</td>
<td>3.9868</td>
<td>$7.944 \times 10^{-2}$</td>
</tr>
<tr>
<td>24</td>
<td>8.4075</td>
<td>$1.319 \times 10^{-1}$</td>
<td>4.0004</td>
<td>$6.577 \times 10^{-2}$</td>
</tr>
<tr>
<td>28</td>
<td>8.4262</td>
<td>$1.132 \times 10^{-1}$</td>
<td>4.0101</td>
<td>$5.612 \times 10^{-2}$</td>
</tr>
<tr>
<td>32</td>
<td>8.4402</td>
<td>$9.920 \times 10^{-2}$</td>
<td>4.0173</td>
<td>$4.894 \times 10^{-2}$</td>
</tr>
</tbody>
</table>
Chapter 13

Dealing with missing data

13.1 INTRODUCTION

So far in all our discussions we have assumed that there are no missing values in the financial data that is used to estimate the asset volatility. In practice this is rarely the case. Reasons for this include: the variation in trading times of the stock exchanges across the world, technical problems with storing/retrieving the data, and disruptive social/economic events in various countries.

Some very simplistic approaches to dealing with missing data include:

- Replacing the missing value by the preceding (known) value.
- Excluding all the data collected at a given time if it contains at least one missing value.

The last method could result in a large amount of very useful information being ignored. For example, just because no data is available from one country’s stock exchange does not mean the data collected from all the other financial markets is not useful.

Accurately replacing missing data is of great importance if good estimates of the volatility and covariance of assets are to be obtained, and used as input parameters to the option pricing and portfolio models that are discussed here.

Here we will consider data with values missing at random, that is when the probability of a missing variate value is not related to the values of other variates. The data is assumed to consist of \( n \) observations (rows) and \( p \) variates (columns), and takes the form of an \( n \times p \) matrix. We will let the \( p \) element vector \( \bar{X}_i \) denote the \( i \)th observation and the row vector \( \bar{X} \) denote the mean variate values. The \( k \)th element of \( \bar{X}_i, \bar{X}_{i,k} \), represents the value of the \( k \)th variate for the \( i \)th observation, and the \( k \)th element of \( \bar{X}, \bar{X}_k \) represents the mean value of the \( k \)th variate. If there are no missing values then we calculate the \( p \times p \) covariance matrix as follows:

\[
C_{jk} = \frac{1}{n} \sum_{i=1}^{n} (X_{i,j} - \bar{X}_j)(X_{i,k} - \bar{X}_k)
\]

or in more compact notation:

\[
C = \frac{1}{n} \sum_{i=1}^{n} (\bar{X}_i - \bar{X})(\bar{X}_i - \bar{X})^T
\]
where

$$\bar{X} = \sum_{i=1}^{p} X_i$$

However, when the data contains missing values it is not possible to calculate the covariance matrix in this manner.

In this section two methods of filling in missing data are considered, they are:

- Iterative multivariate regression, which we call \textit{MREG}.
- The \textit{EM} algorithm.

We will now describe both of these approaches in more detail.

### 13.2 ITERATIVE MULTIPLE LINEAR REGRESSION, \textit{MREG}

This method fills in missing values by performing multiple linear regression on the columns of the data matrix; see Beale and Little (1975), Orchard and Woodbury (1972), and Little and Rubin (1987). We will denote the \( n \) element column vector containing the observations of the \( k \)th variate by \( y^k \).

The procedure is as follows:

\textit{Step 1}

Replace any missing values in column vectors \( y^k \), \( k = 1, \ldots, p \) by the corresponding mean value. That is if the \( k \)th column vector \( y^k \) contains \( n_m \) missing values then these are replaced by \( \bar{y}^k \), which is calculated as

$$\bar{y}^k = \frac{1}{(n - n_m)} \sum_{i=1}^{n} X_{i,k}$$

where \( X_{i,k} \) is taken to be zero if it is missing.

\textit{Step 2}

Starting with the first column as the dependent variable perform a multiple linear regression for each variable on the remaining \( p - 1 \) independent variables. After each regression update the missing values in the dependent variable with those values obtained from the regression.

This works as follows: with the first column as the dependent variable the \( n \) element regression vector \( \hat{y}^1 \) can be written as:

$$\hat{y}^1 = C + \beta_2 y^2 + \beta_3 y^3 + \cdots + \beta_p y^p$$

where \( C \) is a vector of size \( n \) with all elements \( C_i, i = 1, \ldots, n = c \), and the scalars \( \beta_i, i = 2, \ldots, p \) are the regression coefficients.

Here we assume that the data has been centred about the origin and use

$$\hat{y}^1 = \beta_2 y^2 + \beta_3 y^3 + \cdots + \beta_p y^p$$
We denote the $i$th element of $\hat{y}^1$ by $\hat{y}^1_i$, and if the $i$th element of $y^1$ is a missing value, which is the data element $X_{i,1}$, we update using:

$$X_{i,1} = \hat{y}^1_i$$

Similarly when the $k$th column is the dependent variable we have:

$$\hat{y}^k = \beta Y$$

where $Y$ is a $(p - 1) \times n$ matrix which contains all the column vectors $y^j$, $j = 1, \ldots, p$, except $y^k$, and the $(p - 1)$ element vector $\beta$ contains the regression coefficients. If the $i$th element of $y^k$, denoted by $y^k_i$, is a missing value, then we update the data using:

$$X_{i,k} = \hat{y}^k_i$$

**Step 3**

We now compute the current mean values of each variable. The mean value of the $k$th variable is computed as:

$$\hat{y}^k = \frac{1}{n} \sum_{i=1}^{n} X_{i,k}$$

If the difference between the current mean values and the previous mean values is greater than a specified tolerance we repeat step 2. If step 2 has been repeated more than a specified number of times we stop. The function $MREG$, which implements this method, is given in Code excerpt 13.1 below.

```c
void MREG(double x[], long num, long m, double tol, long max_cycle, long *iflag)
{
    /* Input parameters:
       x[] — if put is 0 then a call option, otherwise a put option
       num — the number of time steps, the zeroth time step is the root node of the lattice
       m — the number of time steps, the zeroth time step is the root node of the lattice
       tol — if opt_type is 0 then an option on the maximum of two assets otherwise an option on the minimum
             of two assets
       max_cycle — if is_american is 0 then a European option, otherwise an American option.
    
    Output parameters:
       x[] — the value of the option,
       *iflag — an error indicator.
    */

    double rss;
    long row_ptr, col_ptr, i, jj, ip, rank, j, k;
    double df, zero = 0.0;
    Boolean svd;
    Nag_IncludeMean mean;
    double loc_tol, tmp, sum, *b;
    double *wptr = (double *)0;
    long *sx, dep_var, tdq, count;
    NagError loc_fail;
    Boolean terminate;
    long num_missing, *missing_row, *missing_column;
```
Dealing with missing data

#define Y(I) y[(I)
#define MISSING_ROW(I) missing_row[(I)
#define MISSING_COLUMN(I) missing_column[(I)
#define SX(I) sx[(I)
#define MEANS(I) means[(I)
#define NMEANS(I) nmeans[(I)
#define X(I,J) x[((I)) * m + ((J))]
#define WX(I,J) wx[((I)) * m + ((J))]
#define RES(I) res[(I)]
#define B(I) b[(I)]

mean = 286; /* There is no constant term in the regression, it will pass through the origin */
ip = m - 1;
tdq = ip + 1;
/* Allocate arrays: cov(ip*(ip+1)/2), b(ip), se(ip), res(num), sx(ip), h(num), y(num), q(num*tq),
p((2*ip+1)*ip), com_ar (5*(ip+ip)+ip*ip), means(m), nmeans(m) */
/* Initial processing of the missing data set all missing values to the variable means */
iflag = 0;
num_missing = 0;
for (j = 1; j <= m; jj++) {
    for (i = 1; i <= num; ii++) {
        if (X(i,j) != 999.0) {
            sum = sum + X(i,j);
        } else {
            ++num_missing;
        }
    }
    MEANS(j) = sum/(double)(num - num_missing);
}
/* Allocate arrays: missing_row(num_missing+1), missing_column(num_missing+1) */
/* For the handling of the missing data set here (as opposed to the EM) we use column order
addressing - since the algorithm is column based. */
num_missing = 0;
for (j = 1; j <= m; jj++) {
    for (i = 1; i <= num; ii++) {
        if (X(i,j) == -999.0) {
            num_missing = 1;
            MISSING_ROW(num_missing) = i;
            MISSING_COLUMN(num_missing) = j;
        }
    }
}

dep_var = 1;
count = 0;
terminate = FALSE;
loc_tol = 1.0e-8;
while ((!terminate) && (count <= max_cycle)) { /* Outer cycle */
    /* Replace missing variable values with their means */
    col_ptr = 1;
    row_ptr = 1;
    for (j = 1; j <= m; jj++) {
        for (i = 1; i <= num; ii++) {
            while ((col_ptr <= num_missing) && (MISSING_COLUMN(col_ptr) == j)) {
                i = MISSING_ROW(row_ptr);
                X(i,j) = MEANS(j);
                ++row_ptr;
                ++col_ptr;
            }
        }
        /* if (col_ptr > num_missing) printf("\n") ; */
    }
    col_ptr = 1;
    row_ptr = 1;
    for (jj = 1; jj <= m; jj++) { /* Loop over all the variables selecting one as the dependent */
        SX(jj) = 1;
    }
    SX(dep_var) = 0;
    for (i = 1; i <= num; ii++) { /* Load the dependent variable into the vector y */
        Y(i) = X(i,dep_var);
    }
g02dac(mean, num, &SX(1), m, n, &SX(1), ip, &Y(1), wtptr, &se, &df, &B(1), &SEV, &rank, p, loc_tol, com_ar, &loc_fail);
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    /* load the estimated values back into the data matrix WX */
    for (i = 1; i <= num; ++i) {
        if (MISSING_COLUMN(col_ptr) == dep_var) && (MISSING_ROW(row_ptr) == i)) {
            k = MISSING_ROW(row_ptr);
            ++row_ptr;
            ++col_ptr;
            X(k, dep_var) = Y(i) / C0_RES(i);
        }
    } 
    if (loc_fail.code != NE_NOERROR) printf("ERROR in routine \n");
    sum = zero;
    for (i = 1; i <= num; ++i) { /* calculate the new means */
        sum += X(i, dep_var);
    }
    NMEANS(dep_var) = sum/(double)num;
    dep_var = dep_var + 1;
    if (dep_var > m) dep_var = 1;

    /* now check for the termination criterion */
    terminate = TRUE;
    for (j = 1; j <= m; ++j) {
        tmp = FABS(MEANS(j)/C0_NMEANS(j));
        if (tmp > tol) {
            terminate = FALSE;
        }
    }
    for (j = 1; j <= m; ++j) {
        MEANS(j) = NMEANS(j);
    }
    if (terminate) printf("Stop iterating \n");
    count = count + 1;

Code excerpt 13.1  Function MREG, which uses iterative multiple linear regression to fill in missing values

13.3  THE EM ALGORITHM

The EM algorithm is an iterative method and involves both an Estimation step (or E-step) and a Prediction step, also known as a Maximum likelihood step (or M-step), see Dempster et al. (1977) and Little and Rubin (1987).

Here we assume that the incomplete \( n \times p \) data matrix has been generated by a \( p \) variate normal distribution and we would like to estimate the mean and covariance matrix of this distribution. Code excerpt 13.2 provides an implementation of the EM algorithm which uses a one-pass updating technique for both the mean and covariance matrix, see West (1979) and Chan et al. (1982).

The steps are explained below.

The estimation step

Here we estimate the sample mean \( \bar{X} \) (also denoted \( \mu \)), and sample covariance matrix \( \bar{\Sigma} \) using the current sample data values. The estimates, of the mean, sum of squares about the mean, and the covariance matrix, based on the first \( i \) observations are denoted by \( \bar{X}_{[i]} \), \( \bar{S}_{[i]}^2 \) and \( P_{[i]} \) respectively.

The updating equations, see Appendix F.5, for these quantities are, for the mean

\[
\bar{X}_{[i]} = \bar{X}_{[i-1]} + \frac{1}{i} (X_i - \bar{X}_{[i-1]})
\]  \hspace{1cm} (13.1)
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the sum of squares about the mean is updated as
\[ S_2[i] = S_2[i-1] + \left( \frac{i-1}{i} \right) (X_i - \bar{X}_{[i-1]}) (X_i - \bar{X}_{[i-1]}) \]  
(13.2)

and the sum of cross products about the mean is updated as
\[ P_2[i] = P_2[i-1] + \left( \frac{i-1}{i} \right) (X_i - \bar{X}_{[i-1]}) (Y_i - \bar{Y}_{[i-1]}) \]  
(13.3)

The estimated covariance matrix \( \tilde{\Sigma} \), based on the first \( i \) observations, is then obtained using
\[ \tilde{\Sigma} = \frac{P_2[i]}{i-1} \]  
(13.4)

The prediction step

For each vector \( X_j \) with missing values let \( x_j^{(1)} \) denote the missing components and \( x_j^{(2)} \) denote those components which are known. Thus we have: \( X_j = [x_j^{(1)}, x_j^{(2)}] \) and \( \mu = [\mu_j^{(1)}, \mu_j^{(2)}] \). Given the estimates \( \tilde{\mu} \) and \( \tilde{\Sigma} \) from the E-step we use the mean of the conditional normal distribution of \( x_j^{(1)} \), given \( x_j^{(2)} \), to predict the missing values. That is:
\[ \tilde{x}_j^{(1)} = \mathbb{E}(x_j^{(1)}|x_j^{(2)}; \tilde{\mu}, \tilde{\Sigma}) = \tilde{\mu}^{(1)} + \tilde{\Sigma}_{12} \Sigma_{22}^{-1} (x_j^{(2)} - \tilde{\mu}^{(2)}) \]  
(13.5)

In Equation 13.5 we have used the result, see Appendix E, that if:
\[ X_j \sim N(\mu, \Sigma) \]
with \( \mu = \mu_j^{(1)}/\mu^{(2)} \) and \( \Sigma = (\Sigma_{11}|\Sigma_{12})/(\Sigma_{21}|\Sigma_{22}) \), and \( |\Sigma_{22}| > 0 \) then:
\[ x_j^{(1)} \sim N(\mu', \Sigma') \]
where
\[ \mu' = \mu^{(1)} + \Sigma_{12} \Sigma_{22}^{-1} (x_j^{(2)} - \mu^{(2)}) \]
and
\[ \Sigma' = \Sigma_{11} - \Sigma_{12} \Sigma_{22}^{-1} \Sigma_{21} \]

It can thus be seen that the covariance matrix does not depend on the value of the conditioning variable, \( x_j^{(2)} \).
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```c
long row_ptr, col_ptr, ii, num_missing, id;
double d1;
long *missing_index, *known_index, n_missing, n_known, count, ind, *missing_row, *missing_column;
NagError loc_fail;
Boolean terminate;
/* define for easy referencing of vectors and matrices */
#define MISSING_INDEX(I) missing_index[(I)]
#define MISSING_ROW(I) missing_row[(I)]
#define SIGMA_UU(I,J) sigma_uu[((I) * m + (J))]
/* Allocate arrays: sigma(m*m), nsigma(m*m), sigma_kk(m*m), sigma_uu(m*m), means(m), nmeans(m),
work_mat(m*m), work_vec(m), missing_index(m), known_index(m) */
/* initial processing of the missing data set all missing values to the variable means */
num_missing = 0;
for (j = 1; j <= m; ++j){
    sum = zero;
    count = 0;
    for (i = 1; i <= num; ++i){
        if(X(i,j) != -999.0) {
            sum = sum + X(i,j);
            ++count;
        } else {
            ++num_missing;
        }
    }
    MEANS(j) = sum/(double)count; /* calculate the overall means */
}
/* Allocate arrays: missing_row(num_missing+1), missing_column (num_missing+1) */
/* Set the indices for the missing values */
num_missing = 0;
for (i = 1; i <= num; ++i){
    for (j = 1; j <= m; ++j){
        if(X(i,j) == -999.0) {
            ++num_missing;
            MISSING_ROW(num_missing) = i;
            MISSING_COLUMN(num_missing) = j;
            X(i,j) = MEANS(j);
        } else {
            X(i,j) = X(i,j);
        }
    }
}
/* Initialise data matrix */
row_ptr = 1;
col_ptr = 1;
for (i = 1; i <= num; ++i){ /* Set missing values to the appropriate variate mean */
    while ((row_ptr <= num_missing) && (MISSING_ROW(row_ptr) == i)){
        j = MISSING_COLUMN(col_ptr);
        X(i,j) = MEANS(j);
        ++row_ptr;
        ++col_ptr;
    }
}
for (i = 1; i <= m; ++i){
    for (j = 1; j <= m; ++j){
        SIGMA(i,j) = zero;
    }
}
/* Estimate the initial matrix SIGMA */
for (i = 1; i <= m; ++i){ /* Estimate the initial matrix SIGMA */
    for (j = 1; j <= m; ++j){
        tmp1 = zero;
        for (k = 1; k <= num; ++k){
            SIGMA(i,j) = SIGMA(i,j) + (X(k,i) - MEANS(i))*(X(k,j) - MEANS(j));
        }
        SIGMA(i,j) = (SIGMA(i,j)/(double)(num));
    }
}
count = 0;
terminate = FALSE;
loc_tol = 1.0e-8;
while ((!terminate) && (count <= max_cycle)) { /* Outer cycle loop */
    for (j = 1; j <= m; ++j) { /* Initialize NMEANS */
        ...
    }
    ...
```
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NMEANS(j) = zero;
} /* Initialize NSIGMA, it will be used to provide an estimate for a new value of SIGMA, based on West's updating method */
for (i = 1; i <= m; ++i) {
  for (j = 1; j <= n; ++j) {
    NSIGMA(i,j) = zero;
  }
}
row_ptr = 1;
col_ptr = 1;
for (ii = 1; ii <= num; ++ii) { /* Loop over all observations */
  n_missing = 0;
  n_known = 0;
  while (row_ptr <= num_missing && (MISSING_ROW(row_ptr) == ii)) {
    j = MISSING_COLUMN(col_ptr);
    ++row_ptr;
    ++col_ptr;
    ++n_missing;
    MISSING_INDEX(n_missing) = j;
  }
  k = 1;
  for ((j = 1; j <= n; ++j) {
    if ((k < n_missing) && (MISSING_INDEX(k) == j)) {
      ++k;
    } else {
      ++n_known;
      KNOWN_INDEX(n_known) = j;
    }
  }
  if (n_missing > 0) { /* Are there missing values? */
    if (n_missing == m) { /* deal with the special case in which all the observation is missing */
      for (i = 1; i <= m; ++i) {
        X(ii,i) = MEANS(i);
      }
    } else { /* Form the partial covariance matrices SIGMA_UU, SIGMA_KK and SIGMA_KU */
      for (i = 1; i <= n_missing; ++i) { /* SIGMA_UU */
        p = MISSING_INDEX(i);
        for (j = 1; j <= n_missing; ++j) {
          q = MISSING_INDEX(j);
          SIGMA_UU(i,j) = SIGMA(p,q);
        }
      }
      for (i = 1; i <= n_known; ++i) { /* SIGMA_KK */
        p = KNOWN_INDEX(i);
        for (j = 1; j <= n_known; ++j) {
          q = KNOWN_INDEX(j);
          SIGMA_KK(i,j) = SIGMA(p,q);
        }
      }
      for (i = 1; i <= n_known; ++i) { /* SIGMA_KU */
        p = KNOWN_INDEX(i);
        for (j = 1; j <= n_missing; ++j) {
          q = MISSING_INDEX(j);
          SIGMA_KU(i,j) = SIGMA(p,q);
        }
      }
      /* Obtain INVERSE(SIGMA_KK) * SIGMA_KU by solving SIGMA_KK * X = SIGMA_KU */
      /* Can use cholesky factorisation since SIGMA_KK is positive definite */
      f03aee(n_known, &SIGMA_KK(1,1), m, &WORK_VEC(1), &d1, &id, &loc_fail);
      if (loc_fail.code != NE_NOERROR) {
        printf("Cholesky factorisation error/n");
        return;
      } else { /* solve the equation */
        f04agc(n_known, n_missing, &SIGMA_KK(1,1), m, &WORK_VEC(1), &d1, &id, &loc_fail);
      }
      /* Predict the mean values of the missing data in the current observation */
      /* These values are stored in the array WORK_VEC */
      for (i = 1; i <= n_missing; ++i) {
        WORK_VEC(i) = zero;
        sum = zero;
      }
    } // end if(n_missing > 0)
  } // end if(n_missing == m)
} // end for(ii = 1; ii <= num; ++ii)
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```
for (k = 1; k <= n_known; ++k){
p = KNOWN_INDEX(k);
sum = sum + WORK_MAT(k,i) * (X(ii,p) - MEANS(p));
}
q = MISSING_INDEX(i);
WORK_VEC(i) = MEANS(q) + sum;
X(ii,q) = WORK_VEC(i); /* store the new estimates */
}
q = MISSING_INDEX(i);
WORK_VEC(i) = MEANS(q) + sum;
X(ii,q) = WORK_VEC(i); /* store the new estimates */
}
/* end of else clause */
/* end of n_missing > 0 clause */
/* Use West's (1979) algorithm to update the means and predicted the covariance terms corresponding
to the cross product X_U * X_K^T in NSIGMA (Use West's (1979) algorithm) */
fac = (double)(ii-1)/(double)ii;
fac = fac/(double)num;
for (i = 1; i <= n_missing; ++i){
p = MISSING_INDEX(i);
for (j = 1; j <= n_known; ++j){
q = KNOWN_INDEX(j);
NSIGMA(p,q) = NSIGMA(p,q) + (X(ii,p) - MEANS(p))*(X(ii,q) - MEANS(q))*fac;
}
}
for (i = 1; i <= n_missing; ++i){
p = MISSING_INDEX(i);
for (j = 1; j <= n_known; ++j){
q = KNOWN_INDEX(j);
NSIGMA(p,q) = NSIGMA(p,q) + (X(ii,p) - MEANS(p))*(X(ii,q) - MEANS(q))*fac;
}
}
/* Now update the covariance matrix using the known values, using West's (1979) algorithm. */
for (i = 1; i <= n_known; ++i){
p = KNOWN_INDEX(i);
for (j = 1; j <= n_known; ++j){
q = KNOWN_INDEX(j);
NSIGMA(p,q) = NSIGMA(p,q) + (X(ii,p) - MEANS(p))*(X(ii,q) - MEANS(q))*fac;
}
}
for (i = 1; i <= n_known; ++i){
for (j = 1; j <= n_known; ++j){
NMEANS(i) = NMEANS(i) + ((X(ii,1) - MEANS(1))/(double)ii);
}
for (j = 1; j <= n_known; ++j){
NMEANS(1) = NMEANS(1) + ((X(ii,1) - MEANS(1))/(double)ii);
}
}
/* end of the observation loop (ii) */
/* Now check for the termination criterion */
terminate = TRUE;
for (j = 1; j <= m; ++j){
tmp = FABS(MEANS(j) - NMEANS(j));
if (tmp > tol){
terminate = FALSE;
}
}
/* get ready for the next iteration through the data */
for (j = 1; j <= m; ++j){
MEANS(j) = NMEANS(j);
}
for (i = 1; i <= m; ++i){
for (j = 1; j <= m; ++j){
SIGMA(i,j) = NSIGMA(i,j);
}
if (terminate) printf ("will terminate /n");
count = count + 1;
/* count loop */
}
```

**Code excerpt 13.2** Function which uses the EM algorithm to fill in missing values
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To test the accuracy of the missing data algorithms we generate sample data with known mean $\mu$ and known covariance matrix $C$. We then remove, at random, a given percentage of the data, and try to reconstruct the original data using either the EM algorithm or function MREG, which uses iterative multiple linear regression. If $\hat{C}$ is the sample covariance matrix of the reconstructed data then the quality of this estimate can be quantified using the following distance:

$$D^* = \left\{ \sum_{i=1}^{p} \sum_{j=1}^{p} (C_{ij} - \hat{C}_{ij})^2 \right\}^{1/2}$$

(13.6)

or in more compact notation:

$$D^* = ||C - \hat{C}||$$

(13.7)

The sample data was generated from a normal distribution with mean $\mu$ and covariance matrix $C$ given by:

$$C_{ij} = \frac{i \times j}{10}, \quad i = 1, \ldots, p, \text{ for } i \neq j$$

(13.8)

$$C_{ii} = 1.1i, \quad i = 1, \ldots, p$$

(13.9)

and let the variate means be defined by:

$$\mu_i = 2i, \quad i = 1, \ldots, p$$

(13.10)

Inspection of the results in Tables 13.1 to 13.3 shows that the time taken by the EM algorithm increases as the amount of missing data is raised from 5 per cent to 25 per cent; this is in contrast to the time taken by MREG which is almost

| $n$     | MREG | | EM |
|---------|------| |----|
|         | Time (ms) | Distance, $D^*$ | Time (ms) | Distance, $D^*$ |
| 100     | 80   | 0.6528 | 30   | 0.6980 |
| 400     | 100  | 0.6293 | 60   | 0.7306 |
| 5000    | 2523 | 0.6193 | 540  | 0.6404 |
| 10000   | 5448 | 0.6299 | 1031 | 0.6642 |
| 100000  | 5561 | 0.6394 | 9674 | 0.6990 |
| 300000  | 165157 | 0.6396 | 26758 | 0.7048 |
It can be seen that, for less than 400 observations, the performance of MREG is similar to that of the EM algorithm. However, as the number of observations is increased the speed of MREG decreases dramatically; although the accuracy of achieved is still similar to that obtained by the EM algorithm.
Part III

Financial Econometrics
Chapter 14

Introduction

Here we are concerned with modelling financial returns, see Section 14.1, which are generated from share prices, stock market indices, or currency exchange rates.

Here we describe the financial returns data using regression-based models of the form:

\[ y_i = X_i^T \beta + \epsilon_i, \quad i = 1, \ldots, n \]  

(14.1)

where \( n \) is the length of the time series, \( y_i \) is the \( i \)th return, \( X_i \) is a vector of size \( k \), \( \beta \) is a vector of \( k \) regression coefficients and \( \epsilon_i \) are the residuals. The variance \( \sigma_i^2 \) of the \( i \)th residual is thus given by \( \sigma_i^2 = E[\epsilon_i^2] \). In finance literature the term volatility depends on context, and refers either to the variance \( \sigma_i^2 \) or the standard deviation \( \sigma_i \). Equation 14.1 looks deceptively simple and it hides the fact that we are really interested in determining the characteristics of \( \epsilon_i \) so that we can model the volatility.

Empirical studies suggest that financial returns have the following characteristics:

(i) Large returns occur more frequently than expected for a Gaussian distribution. This means that the unconditional probability distribution for \( \epsilon_i \) has fatter tails (and therefore a larger unconditional kurtosis) than that of a Gaussian distribution.

(ii) The variance (volatility) of the returns exhibit clustering. There are periods of high volatility separated by regions of low volatility.

(iii) When bad news occurs it is often followed by high volatility. That is negative stock market returns are usually followed by high volatility. For exchange rate returns data it is not clear what constitutes bad news, since a large fall in the exchange rate may be good or bad depending on your point of view.

(iv) In stock market data large negative returns (corresponding to bad news) occur more frequently than large positive returns. This means that the unconditional probability distribution of \( \epsilon_i \) is asymmetric about zero, and the probabilities on the negative side of the distribution are higher than the probabilities on the positive side. This asymmetry can be measured in terms of skewness. We thus state that stock market data has been found to exhibit negative skewness. Again there is no reason why exchange rate returns should have any particular sign associated with the skewness.

In this part of the book we show how points (i) to (iv) can be modelled by using symmetric and asymmetric GARCH models with the conditional probability
distribution of the residuals, $\epsilon_i$, having a Gaussian distribution with time varying conditional variance $h_i$. The standardized residuals $Z_i = \epsilon_i / \sqrt{h_i}$ should then be distributed as $NID(0, 1)$, and so have a kurtosis of 3. However, it has also been found that the standardized residuals are non-Gaussian and so other conditional probability distributions for the residuals such as the Student’s $t$ distribution and the Generalized Error distribution are also considered.

Estimates of return volatility are used to assess the level of risk associated with many financial products. Accurate measures and reliable forecasts of volatility are crucial for option pricing techniques as well as trading and hedging strategies that arise in portfolio allocation problems.

We assume minimal prior knowledge of statistics and aim to provide mathematical details and proofs that may be taken for granted or omitted from more advanced econometric literature. This is especially the case for the information provided concerning the properties of various statistical distributions. Here the expected values of the distribution are derived from first principles using integration, rather than the more usual approach of either quoting standard results or using moment generating functions. Here we concentrate on standard linear and nonlinear univariate GARCH processes. However, information is also provided concerning other models such as component GARCH, stochastic volatility models and Levy processes. The testing of GARCH software is covered and comprehensive information is supplied concerning the calculation of the first and second order derivatives of the log likelihood function.

Before embarking on a detailed study of time series methods and applications to forecast the volatility of financial assets we will quote from a recent article by Granger (2002). There reference is made to a survey of 40 papers which compare the forecasting ability of techniques such as: historical and implied volatility (see Part II Section 9.3.4), stochastic volatility (SV), and GARCH. It is stated that:

1. Five papers find that GARCH beats HISTORICAL.
2. Five papers find that HISTORICAL beats GARCH.
3. Only three papers consider SV forecasts; one finds SV better than GARCH, one finds GARCH better than SV, and a third paper finds SV better than GARCH for stocks but the reverse for currencies.
4. Thirteen papers compare IMPLIED with HISTORICAL, with twelve preferring IMPLIED.
5. Fourteen papers compare IMPLIED with GARCH; all but one find that IMPLIED provides better forecasts. One paper also finds that IMPLIED performs better than SV.

Granger concludes that:

Overall, IMPLIED seems to be the superior technique with GARCH and HISTORICAL roughly equal second. The result is not really surprising as the IMPLIED forecasts are based on a wider information set than the alternatives, not just depending on the past returns but also on using option prices. On the other hand, suitable options may not always be available and so these forecasts cannot be used on many occasions.
14.1 ASSET RETURNS

The return can be defined in several different ways.

If we let $P_t$ denote the price (or index) at time $t$, and for simplicity assume a series of $n$ values $P_t, t = 1, \ldots, n$ in which the sampling period is the unit time interval, then the Simple net return, $R_t$, between instant $t - 1$ and instant $t$, is:

$$SR_t = \frac{P_t - P_{t-1}}{P_{t-1}} = \frac{P_t}{P_{t-1}} - 1$$

(14.2)

the Gross return, $R_t$, is defined as:

$$R_t = \frac{P_t}{P_{t-1}}$$

(14.3)

The gross return compounded over $k$ periods takes the form:

$$R_t(k) = \frac{P_{t+k}}{P_{t-1}} = \left( \frac{P_t}{P_{t-1}} \right) \left( \frac{P_{t+1}}{P_t} \right) \left( \frac{P_{t+2}}{P_{t+1}} \right) \ldots \left( \frac{P_{t+k-1}}{P_{t+k-2}} \right) \left( \frac{P_{t+k}}{P_{t+k-1}} \right)$$

An alternative approach is to use the continuously compounded returns (or logarithmic returns). This is defined using:

$$r_t = \log \left( \frac{P_t}{P_{t-1}} \right) = \log(P_t) - \log(P_{t-1})$$

(14.4)

where log denotes the natural logarithm.

The return compounded over $k$ periods is:

$$r_t(k) = \log(P_{t+k}) - \log(P_{t-1})$$

$$= \log(P_{t+k}) - \log(P_{t+k-1}) + \log(P_{t+k-1}) - \log(P_{t+k-2}) - \log(P_{t+k-2}) + \log(P_{t+k-3}) - \log(P_{t+k-3}) + \ldots + \log(P_t) - \log(P_{t-1})$$

$$r_t(k) = r_t + r_{t+1} + \cdots + r_{t+k-1} + r_{t+k}$$

(14.5)

Thus unlike multiperiod gross compounding which is a multiplicative process, multiperiod continuous compounding is additive.

We also note that:

$$\log(x) = (x - 1) - \frac{1}{2}(x - 1)^2 \cdots \text{ for } 2 \geq x > 0$$

and therefore $\log(x) \sim x$ when $x \sim 1$.

Since $(P_t)/(P_{t-1}) \sim 1$, Equations 14.2 and 14.4 give: $r_t \sim R_t$. This means that the simple net return is virtually the same as the logarithmic return. It may also be convenient to create a scaled return series using:

$$r_t = \phi \{ \log(P_t) - \log(P_{t-1}) \}$$

(14.6)

where $\phi$ is the scale factor. When $\phi = 100$ the series gives the percentage logarithm returns.

If dividend payments, $D_t$, are included then Equation 14.4 takes the form:

$$r_t = \log \left( \frac{P_t + D_t}{P_t} \right)$$

(14.7)
This can be re-expressed using the following steps:

\[ r_t = \log(P_t) - \log(P_{t-1}) + \log(P_t + D_t) - \log(P_t) \]

\[ r_t = \log(P_t) - \log(P_{t-1}) + \log\left(1 + \frac{D_t}{P_t}\right) \]

\[ r_t = p_t - p_{t-1} + \log(1 + \exp(d_t - p_t)) \quad (14.8) \]

where \( p_t = \log(P_t) \) and \( d_t = \log(D_t) \). Equation 14.7 is a nonlinear function of the logarithm of the dividend to price ratio, \( \lambda_t = \log(D_t/P_t) \). If we linearize about the mean value of \( \lambda_t, \bar{\lambda} \), we then obtain:

\[ r_t = k + \rho \ p_t + (1 - \rho) \ d_t - p_{t-1} \quad (14.9) \]

where

\[ \rho = \frac{1}{1 + \exp(\bar{\lambda})} \quad \text{and} \quad k = -\log(\rho) - (1 - \rho)\left\{\log\left(\frac{1}{\rho} - 1\right)\right\} \]

It can be seen that the returns \( r_t \) are computed using a weighted sum of the logarithm of stock price and the logarithm of the dividend. Empirical studies, see Campbell et al. (1997), have found that \( \rho \) is about 0.96, which means that nearly all of the contribution to the value of returns is from the stock price.

**Proof of Equation 14.9**

Since \( \lambda_t = d_t - p_t \) we have from Equation 14.8 that:

\[ r_t = p_t - p_{t-1} + f(\lambda_t) \quad (14.10) \]

where \( f(\lambda_t) = \log(1 + \exp(\lambda_t)) \).

Using a Taylor expansion about the mean value \( \bar{\lambda} \) we have:

\[ f(\lambda_t) = f(\bar{\lambda}) + f'(\bar{\lambda})(\lambda_t - \bar{\lambda}) \quad (14.11) \]

Now from elementary calculus we have:

\[ f'(\bar{\lambda}) = \frac{\exp(\bar{\lambda})}{1 + \exp(\bar{\lambda})} \]

Substituting into Equation 14.11 we obtain:

\[ f(\lambda_t) = f(\bar{\lambda}) + \frac{\exp(\bar{\lambda})}{1 + \exp(\bar{\lambda})}(\lambda_t - \bar{\lambda}) \quad (14.12) \]

letting \( \rho = \frac{1}{1 + \exp(\bar{\lambda})} \) we have:

\[ \log(1 + \exp(\bar{\lambda})) = -\log(\rho), \quad \exp(\bar{\lambda}) = \left(\frac{1}{\rho} - 1\right) \quad \text{and} \quad \bar{\lambda} = \log\left(\frac{1}{\rho} - 1\right) \]

Therefore Equation 14.11 gives:

\[ f(\lambda_t) = -\log(\rho) + \left(\frac{1}{\rho} - 1\right)\rho(\lambda_t - \bar{\lambda}) \]

\[ f(\lambda_t) = -\log(\rho) + (1 - \rho)\lambda_t - (1 - \rho)\bar{\lambda} \]
Substituting into Equation 14.10 we have:

\[ r_t = p_t - p_{t-1} - \log(\rho) + (1 - \rho)\lambda_t - (1 - \rho)\tilde{\lambda} \]

substituting for \( \tilde{\lambda} \)

\[ r_t = -\log(\rho) - (1 - \rho)\left\{ \log\left(\frac{1}{\rho} - 1\right) \right\} - p_{t-1} + (1 - \rho)d_t + \rho p_t \]

which gives

\[ r_t = k + \rho p_t + (1 - \rho)d_t - p_{t-1} \quad QED \]

where

\[ k = -\log(\rho) - (1 - \rho)\left\{ \log\left(\frac{1}{\rho} - 1\right) \right\} \]

Empirical studies have shown that in many instances the logarithm of dividend to price ratio, \( D_t/P_t \), can be taken as a constant, and in these circumstances we have \( \lambda_t = \lambda, t = 1, \ldots, n \); where \( \lambda \) is a constant.

### 14.2 NONSYNCHRONOUS TRADING

The nonsynchronous trading effect arises when data is assumed to be recorded at certain times when in fact it is collected at other times. As an example the daily closing security prices, which give the last transaction price for each security on the previous day, do not occur at the same time each day. By referring to these values as *daily* closing prices we incorrectly assume that they occur at equally spaced 24 hour time intervals. As another example consider two stocks A and B, whose prices are independent but stock A trades less frequently than B. If stock market news arrives near the close of trade it is more likely to be reflected in the closing price of stock B than that of stock A; this is because stock A may not trade after the arrival of the information. The fact that stock A will respond to the new information after a significant time lag can induce spurious correlations between the daily returns of stocks A and B, if these are based on daily closing prices. This lagged response can also induce negative autocorrelations in the daily returns of A. This is because when A is not trading its observed return is zero, and when it does trade its returns revert to the cumulated mean return.

Lo and MacKinlay (1990) have developed a nonsynchronous trading model which captures these effects. It is assumed that a security has in each time period \( t \) an unobserved or virtual continuously compounded return \( r_t \). These virtual returns represent the changes in the true underlying value of the security; they reflect changes in value caused by both company information and general stock market information.

We suppose that at each time period there is the probability \( \gamma \) that the security does not trade; the probability that the security trades is then \( (1 - \gamma) \). The observed return, \( r_t^{v} \), depends on whether the security trades or not. If the security does not trade in period \( t \) then \( r_t^{v} = \log(P_t/P_{t-1}) = \log(1) = 0 \). If on the other hand the security trades
in period $t$, its observed return is taken as the sum of the virtual returns in period $t$ and all previous consecutive periods in which the security did not trade.

For example consider a sequence of six consecutive time periods in which the security trades in periods 1, 2, and 6, but does not trade in periods 3, 4, and 5. The nontrading model implies that the observed return in period 2 is the virtual return, $r_2^o = r_2$, the observed return in periods 3, 4, and 5 are zero, $r_3^o = r_4^o = r_5^o = 0$, and the observed return in period 6 is the sum of the virtual returns from periods 3 to 6, $r_6^o = r_3 + r_4 + r_5 + r_6$. Here the impact of news is captured in the virtual returns process and the lag caused by nontrading is modelled in the observed returns process $r_t^o$. We will now define the variable $k_t$, which is the number of past consecutive periods, at time $t$, for which the asset has not been traded. The mean and variance of $k_t$ are related to the nontrading probability, $\gamma$, in the following manner:

$$E[k_t] = \frac{\gamma}{1 - \gamma}, \quad Var[k_t] = \frac{\gamma}{(1 - \gamma)^2}$$ (14.13)

**Proof of Equation 14.13**

First we will prove the equation for the mean.

$$E[k_t] = 0(1 - \gamma) + (1 - \gamma)\gamma + 2(1 - \gamma)\gamma^2 + 3(1 - \gamma)\gamma^3 + 4(1 - \gamma)\gamma^4 + \cdots$$

$$E[k_t] = (1 - \gamma)(\gamma + 2\gamma^2 + 3\gamma^3 + 4\gamma^4 + 5\gamma^5 + \cdots)$$

$$E[k_t] = \gamma + 2\gamma^2 + 3\gamma^3 + 4\gamma^4 + 5\gamma^5 - \gamma^2 - 2\gamma^3 - 3\gamma^4 - 4\gamma^5 - \cdots$$

This is a Geometric Progression with first term $\gamma$ and common ratio $\gamma$, therefore:

$$E[k_t] = \sum_{j=1}^{\infty} \gamma^j = \frac{\gamma}{1 - \gamma} \quad QED$$

Now we consider the equation for the variance of $k_t$.

$$Var[k_t] = E[k_t^2] - (E[k_t])^2$$ (14.16)

$$E[k_t^2] = 0(1 - \gamma) + (1 - \gamma)\gamma + 4(1 - \gamma)\gamma^2 + 9(1 - \gamma)\gamma^3 + 16(1 - \gamma)\gamma^4$$

$$+ 25(1 - \gamma)\gamma^5 + \cdots$$

$$E[k_t^2] = (1 - \gamma)\{\gamma + 4\gamma^2 + 9\gamma^3 + 16\gamma^4 + 25\gamma^5 + \cdots\}$$

$$E[k_t^2] = \gamma + 4\gamma^2 + 9\gamma^3 + 16\gamma^4 + 25\gamma^5 + \cdots - \gamma^2 - 4\gamma^3 - 9\gamma^4 - 16\gamma^5 - \cdots$$

Now from Equation 14.15 we have

$$(E[k_t])^2 = (\gamma + \gamma^2 + \gamma^3 + \gamma^4 + \gamma^5 + \cdots)^2 = \gamma^2 + 2\gamma^3 + 3\gamma^4 + 4\gamma^5 + \cdots$$

So substituting into Equation 14.16

$$Var[k_t] = \gamma + 3\gamma^2 + 5\gamma^3 + 7\gamma^4 + 9\gamma^5 + \cdots - \gamma^2 - 2\gamma^3 - 3\gamma^4 - 4\gamma^5 - \cdots$$

$$Var[k_t] = \gamma + 2\gamma^2 + 3\gamma^3 + 4\gamma^4 + 5\gamma^5 + \cdots$$ (14.17)
From Equations 14.14 and 14.13 we have:

\[ E[k_i] = \frac{\gamma}{1 - \gamma} = (1 - \gamma)(\gamma + 2\gamma^2 + 3\gamma^3 + 4\gamma^4 + \cdots) \]  

(14.18)

which means that Equation 14.17 can be written as:

\[ \text{Var}[k_i] = \frac{\gamma}{(1 - \gamma)^2} \quad QED \]

Substituting into Equation 14.13 we find that if \( \gamma = 0.75 \) then the average number of consecutive periods of nontrading is three. If the asset trades on every period then \( \gamma = 0 \) and both the mean and variance of \( k_i \) are zero.

Lo and MacKinlay (1990) consider a virtual returns process of the form:

\[ r_t = \mu + \epsilon_t \]  

(14.19)

where \( \mu \) is a constant drift term, \( \epsilon_t \) is zero mean IID noise. In this case:

\[ E[r_t] = \mu \quad \text{Var}[r_t] = \sigma^2 + \frac{2\gamma}{1 - \gamma} \mu^2 \]  

(14.20)

and

\[ \text{Corr}[r_t^n, r_{t+n}] = \frac{-\mu^2 \gamma^n}{\sigma^2 + g \mu^2}, \quad n > 0 \]  

(14.21)

where \( \sigma^2 = \text{Var}[r_t] \) and \( g = 2\gamma/(1 - \gamma) \).

We thus conclude that nontrading does not affect the mean of the observed returns. However, if the expected return of the security is nonzero, then nontrading increases the observed variance of the security returns, and also induces negative serial correlation in the returns.

14.3 BID-ASK SPREAD

The presence of the bid-ask spread means that instead of one price for each asset there are now three: the bid price, the ask price and the actual transaction price which need not be either the bid or ask price. To account for the impact of the bid-ask spread Roll (1984) proposed the following model:

\[ P_t = P_t^* + I_t \frac{s}{2} \]  

(14.22)

where \( P_t \) is the observed asset price at time \( t \), \( P_t^* \) is the true asset price, \( s \) is the bid-ask spread, and the IID indicator variable \( I_t \) which takes the value +1 with probability 0.5 (to signify a buyer initiated bid) and the value −1 with probability 0.5 to indicate a seller initiated ask.

The assumption that \( P_t^* \) is the true value of the security implies that \( E[I_t] = 0 \), and hence \( \Pr(I_t = 1) = \Pr(I_t = -1) = 0.5 \).

If the true security value, \( P_t^* \), does not change with time then the process for the price observed changes is:

\[ \Delta P_t = (I_t - I_{t-1}) \frac{s}{2} \]  

(14.23)
which means that:

\[ \text{Var}[\Delta P_t] = \frac{s^2}{2} \quad (14.24) \]

\[ \text{Cov}[\Delta P_{t-1}, \Delta P_t] = -\frac{s^2}{4} \quad (14.25) \]

\[ \text{Cov}[\Delta P_{t-k}, \Delta P_t] = 0, \quad k > 1 \quad (14.26) \]

\[ \text{Corr}[\Delta P_{t-1}, \Delta P_t] = -\frac{1}{2} \quad (14.27) \]

It can be seen that despite the fact that the true value is fixed \( \Delta P_t \) has volatility and also negative correlation. This is caused by the bid-ask bounce. The reason for this is as follows: If \( P_t^* \) is fixed than the observed price can only take on two values, the bid price and the ask price. If the current price is the ask then the price change between the current price and the previous price must either be zero or \( s \), and the price change between the next price and the current price must either be zero or \( -s \), which induces negative covariance. The same is true if the current price is the bid price.

If \( P_t^* \) changes with time, and its increments are serially uncorrelated and independent of \( P_t \), then Equation 14.25 still applies. However, Equation 14.27 is no longer true, and the correlation is now given by:

\[ \text{Corr}[\Delta P_{t-1}, \Delta P_t] = -\frac{s^2/4}{s^2/2 + \sigma_p^2} \quad (14.28) \]

where \( \sigma_p^2 \) is the variance of \( \Delta P_t^* \).

The bid-ask spread \( s \) can be estimated from the covariance of the price changes using:

\[ s = 2\sqrt{-\text{Cov}[\Delta P_{t-1}, \Delta P_t]} \quad (14.29) \]

Estimating the bid-ask spread in this manner may seem rather strange when it is already available from market data. However, the quoted value can differ from the effective value, and in many cases transactions occur at prices within the bid-ask spread. This is because discounts may be given to certain customers, and also, if updating is not frequent enough, the quoted values for \( s \) may not be the actual values used.

Roll’s model assumes that the value for \( s \) is a given constant, and is independent of the value of \( P_t^* \). For a more sophisticated model of the bid-ask spread see Glosten and Milgrom (1985).

### 14.4 MODELS OF VOLATILITY

In this section we provide a brief overview of two methods that are commonly used to model volatility in finance: stochastic volatility processes and Levy processes. Here we give a short definition of each process. Section 14.5 gives more information...
on stochastic autoregressive processes and Section 14.6 provides more information on the generalized hyperbolic Levy process.

14.4.1 Stochastic volatility models

A continuous standard Brownian process \( X \) can be discretized as:

\[
X_t = \mu t + \sigma \epsilon_t, \quad \epsilon_t \sim NID(0, 1)
\]  

(14.30)

where \( X_t \) is the value of the Brownian variate at time \( t \), \( X_0 = 0 \), \( \mu \) is the constant drift, and \( \sigma \) is the constant volatility.

The stochastic volatility model, see Ghysels et al. (1996) and Taylor (1994), which permits a time varying volatility, generalizes Equation 14.30 to:

\[
X_t = \mu + \sigma \epsilon_t, \quad \epsilon_t \sim NID(0, 1)
\]

(14.31)

where the time dependent volatility, \( \sigma_t \), is termed the stochastic volatility. We will assume that the process \( \sigma_t \) has no causal relationship with the process \( \epsilon_t \). Thus it is assumed that the process \( \epsilon_t \) is not caused by the process \( \sigma_t \), and also that the process \( \epsilon_t \) does not cause the process \( \sigma_t \).

We will now consider the following two \( \sigma_t \) processes.

1. \( \sigma_t \) is an independent stochastic process

Here we take \( \sigma_t \) to be a stochastic process that is independent of the information set \( \psi_{t-1} \).

An example is the stochastic random autoregressive (ARV) model, which is discussed in Section 14.5.

The general form of an ARV(1) model is:

\[
X_t = \mu + \sigma_{t-1} \epsilon_t
\]

(14.32)

and

\[
\log(\sigma_t) = \alpha + \phi \log(\sigma_{t-1}) + \eta_t
\]

(14.33)

where \( \mu, \alpha \) and \( \phi \) are constants. The variates \( \epsilon_t \) and \( \eta_t \) are from an IID bivariate normal distribution with correlation coefficient \( \rho \).

2. \( \sigma_t \) is a deterministic function of the information set \( \psi_{t-1} \)

In this case \( \sigma_t \) is a deterministic function of previous process values, contained in the information set \( \psi_{t-1} \).

An example is the generalized autoregressive conditional heteroskeolostic GARCH(p,q) process which is defined as follows:

\[
X_t = \mu + \sigma_t \epsilon_t, \quad \epsilon_t \sim NID(0, 1)
\]

or equivalently

\[
X_t = \mu + \epsilon_t, \quad \epsilon_t \sim NID(0, \sigma_t^2)
\]
and

\[ \sigma_t^2 = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{t-j}^2 + \sum_{j=1}^{p} \beta_j h_{t-j}, \quad t = 1, \ldots, n, \quad \epsilon_t | \psi_{t-1} \sim \text{NID}(0, \sigma_t^2) \]  

(14.34)

It can be seen from Equation 14.34 that \( \sigma_t^2 \) is a weighted sum of the previous values of \( \epsilon_t \) and \( h_t \).

More information on GARCH models can be found in Chapter 15 and the following sections.

### 14.4.2 Levy processes

In contrast to Brownian motion and stochastic volatility models which describe continuous process, a Levy process \( X_t \) consists of discontinuous jumps.

If the first moment is finite then the Levy process can be represented as:

\[ X_t = Z_t + \mu t + \sigma \epsilon_t, \quad \epsilon_t \sim \text{NID}(0, 1) \]  

(14.35)

where \( \sigma \) is the volatility, \( \mu \) is a continuous drift term and \( Z_t \) is a discontinuous Martingale process, see Part II Section 8.2, independent of \( \epsilon_t \).

When the term \( Z_t \) in Equation 14.35 is set to zero we obtain the equation for continuous Brownian motion; that is:

\[ X_t = \mu t + \sigma \epsilon_t, \quad \epsilon_t \sim \text{NID}(0, 1) \]  

(14.36)

We now give a more formal definition of a Levy process. The process \( X \) is a Levy process if:

1. \( X \) has increments that are independent of the past:
   
   This means that \( X_t - X_s \) is independent of \( \mathcal{F}_s, 0 \leq s < t < \infty \), where \( \mathcal{F}_s \) denotes the history up to time \( t = s \).
2. \( X \) has stationary increments:
   
   That is \( X_t - X_s \) has the same distribution as \( X_{t-s}, 0 \leq s < t < \infty \).
3. \( X \) is continuous in probability:
   
   So \( X_t \rightarrow X_s \) as \( t \rightarrow s \).

   In Section 14.6 we consider the use of generalized hyperbolic Levy motion to model asset returns.

### 14.5 STOCHASTIC AUTOREGRESSIVE VOLATILITY, ARV

A popular form of ARV(1) model, see Taylor (1994), is:

\[ \log(P_t) = \log(P_{t-1}) + \mu + \sigma_{t-1} \epsilon_t \]  

(14.37)
\[ \log(\sigma_t) = \alpha + \phi \{ \log(\sigma_{t-1}) - \alpha \} + \theta \eta_t \]  

where \( \mu, \alpha, \phi, \) and \( \theta \) are constants. The pairs \((\epsilon_t, \eta_t)\) are IID bivariate normal and the standard normal variates \( \epsilon_t \) and \( \eta_t \) have correlation coefficient \( \rho \). The logarithm of the volatility follows a stationary AR(1) process when \(-1 < \phi < 1\).

Since the volatility appears as \( \sigma_{t-1} \) in Equation 14.37 the process is termed a lagged ARV(1) model. Another specification, see Taylor (1986), is:

\[ \log(P_t) = \log(P_{t-1}) + \mu + \sigma_t \sum_t \]  

in which case Equations 14.38 and 14.39 define a contemporaneous ARV(1) model.

The stationary ARV(1) has five parameters that need to be estimated: \( \mu, \alpha, \phi, \theta, \) and \( \rho \). Estimation of the parameter \( \phi \) is of particular interest because it provides information concerning the persistence of the volatility shocks. Various techniques have been used to estimate this parameter, including:

- Maximum-likelihood techniques, Harvey et al. (1994).

These studies have shown that the value of \( \phi \) is greater than 0.95; which means that volatility shocks have a high level of persistence.

### 14.6 GENERALIZED HYPERBOLIC LEVY MOTION

Barndorff-Nielsen (1977) introduced the generalized hyperbolic (GH) distribution and used it to model the grain size distributions of wind blown sand. It can be shown, see Barndorff-Nielsen and Halgreen (1977), that the generalized hyperbolic distribution generates a (discontinuous) Levy process with increments of length 1.

#### The generalized hyperbolic distribution

The one dimensional density function of the generalized hyperbolic (GH) distribution is:

\[
GH(x) = A \times (\delta^2 + (x - \mu)^2)^{(\lambda - 1)/2} \\
\times K_{\lambda-1/2} \left( \alpha \sqrt{\delta^2 + (x - \mu)^2} \right) \exp \left( \beta (x - \mu) \right)
\]  

where \( \alpha > 0, \ 0 \geq |\beta| < \alpha, \ \delta > 0 \) and

\[
A = \frac{(\alpha^2 - \beta^2)^{\lambda/2}}{\sqrt{2\pi} \alpha^{\lambda-1/2} \delta^{\lambda} K_{\lambda} \left( \delta \sqrt{\alpha^2 - \beta^2} \right)}
\]
and $K_\lambda$ is a modified Bessel function of the third kind with index $\nu$. The integral representation of $K_\nu$ is:

$$K_\nu(x) = \frac{1}{2} \int_0^\infty y^{\nu-1} \exp\left(-\frac{1}{2} x(y + y^{-1})\right) dy$$

For $\lambda = n + 1/2, n = 0, 2, \ldots$, the Bessel function $K_\lambda$ is:

$$K_{n+1}(x) = \frac{\pi}{2} x^{-1/2} \exp(-x) \left(1 + \sum_{i=1}^n \frac{(n + i)!}{(n - i)!} (2x)^{-i}\right)$$

Since $K_\lambda(x) = K_{-\lambda}(x)$, and $K_{1/2}(x) = K_{-1/2}(x) = \sqrt{\pi/2} x^{-1/2} \exp(-x)$; which is used below to simplify the expressions for the cases $\lambda = 1$, and $\lambda = -1/2$.

From Equation 14.40 it can easily be shown that the generalized hyperbolic log-likelihood, for $n$ independent observations, $X_i, i = 1, \ldots, n$ is:

$$\mathcal{L} = \log(A) + \left(\frac{\lambda}{2} - 1\right) \sum_{i=1}^n \log\left(\delta^2 + (X_i - \mu)^2\right)$$

$$+ \sum_{i=1}^n \left\{ \log\left(K_{\lambda-1/2} \left(\alpha \sqrt{\delta^2 + (X_i - \mu)^2} + \beta(X_i - \mu)\right)\right) \right\}$$

(14.41)

The five parameters in the GH density $\alpha, \beta, \delta, \mu$, and $\lambda$ allow much more flexibility in modelling financial data than the Gaussian distribution which only has two parameters $\mu$ and $\sigma$. Estimates for the parameter values can be obtained by using numerical optimization software to maximize the log-likelihood function for a particular set of data values.

The parameter $\alpha$ controls the shape, $\beta$ the skewness, $\delta$ the scaling (similar to $\sigma$ in the normal distribution), $\mu$ the location and $\lambda$ the heaviness of the tails. The normal distribution is obtained as a limiting case of the generalized hyperbolic distribution for $\delta \to \infty$ and $\delta/\alpha \to \sigma^2$.

The mean of GH is:

$$E[X] = \mu + \frac{\beta \delta}{\sqrt{\alpha^2 - \beta^2}} \frac{K_{\lambda+1}(\zeta)}{K_{\lambda}(\zeta)}$$

(14.42)

the variance of GH is:

$$Var[X] = \delta^2 \left(\frac{K_{\lambda+1}(\zeta)}{\zeta \, K_{\lambda}(\zeta)} + \frac{\beta^2}{\alpha^2 - \beta^2} \left\{ \frac{K_{\lambda+2}(\zeta)}{K_{\lambda}(\zeta)} - \left(\frac{K_{\lambda+1}(\zeta)}{K_{\lambda}(\zeta)}\right)^2\right\}\right)$$

(14.43)

where $\zeta = \delta \sqrt{\alpha^2 - \beta^2}$. The variance term (in large brackets) multiplied by $\delta^2$ is independent of $\mu$ and $\delta$.

When the $GH$ distribution is centred ($\mu = 0$) and symmetric ($\beta = 0$) then $\zeta = \delta \alpha$, and the mean and variance are simply:

$$E[X] = 0 \text{ and } Var[X] = \frac{\delta}{\alpha}$$

(14.44)

We will now consider two cases of special interest, namely when $\lambda = 1$ and when $\lambda = 1/2$. 
The hyperbolic distribution

This is the special case when \( \lambda = 1 \). In these circumstances the generalized hyperbolic distribution (\( GH \)) simplifies to the hyperbolic distribution (\( H \)) which has density:

\[
H(x) = \frac{\sqrt{\alpha^2 - \beta^2}}{2\delta \alpha K_1(\delta \sqrt{\alpha^2 - \beta^2})} \exp \left( -\alpha \sqrt{\delta^2 + (x - \mu)^2} + \beta (x - \mu) \right)
\]

(14.45)

where \( 0 \geq \delta \), and \( |\beta| < \alpha \).

The normal inverse Gaussian

This is the special case when \( \lambda = -1/2 \).

In these circumstances the generalized hyperbolic distribution (\( GH \)) simplifies to the normal inverse Gaussian distribution (\( NIG \)), see Barndorff-Nielsen (1998), which has the density:

\[
NIG(x) = \frac{\alpha \delta}{\pi} \exp \left( \delta \sqrt{\alpha^2 - \beta^2} + \beta (x - \mu) \right) \frac{K_1\left(\alpha \sqrt{\delta^2 + (x - \mu)^2}\right)}{\sqrt{\delta^2 + (x - \mu)^2}}
\]

(14.46)

where \( 0 \geq \delta \), and \( 0 \geq |\beta| \geq \alpha \).

When the skewness parameter \( \beta \) is zero and also the mean value \( \mu \) is zero, we have the symmetric centred \( NIG \) distribution, \( NIG_{sc} \), which has the density:

\[
NIG_{sc}(x) = \frac{\alpha \delta}{\pi} \exp(\delta \alpha) \frac{K_1(\alpha \sqrt{\delta^2 + x^2})}{\sqrt{\delta^2 + x^2}}
\]

(14.47)

An alternative parameterization of Equation 14.47, see Forsberg and Bollerslev (2002), is:

\[
NIG_{sc}(x) = \frac{\alpha^{1/2}}{\pi \sigma^*} \exp(\alpha^*) q(\frac{x}{\sigma^* \alpha^{1/2}})^{-1} K_1\left(\alpha \sigma^* \alpha^{1/2} q(\frac{x}{\sigma^* \alpha^{1/2}})\right)
\]

(14.48)

where \( q(x) = 1/(1 + x^2) \), \( \alpha^* = \alpha \delta \), and \( \sigma^* = \delta^{1/2}/\alpha^{1/2} \).

We can show this as follows. Substituting for \( \alpha^* \) and \( \sigma^* \) into Equation 14.48 we obtain:

\[
NIG_{sc}(x) = \frac{\alpha^{1/2} \delta^{1/2} \alpha^{1/2}}{\pi \delta^{1/2}} \exp(\alpha \delta) q\left(\frac{x}{\delta \alpha^{1/2}}\right)^{-1} K_1\left(\alpha \delta q\left(\frac{x}{\delta \alpha^{1/2}}\right)\right)
\]

where we have made use of the fact that \( \sigma^* \alpha^{1/2} = \delta \). Simplifying further we have:

\[
NIG_{sc}(x) = \frac{\alpha}{\pi} \exp(\alpha \delta) \frac{1}{\sqrt{1 + x^2/\delta^2}} K_1\left(\alpha \delta \sqrt{1 + x^2/\delta^2}\right)
\]
and finally:

\[ NIG_{sc}(x) = \frac{y^{\delta \alpha}}{\pi} \exp(\delta \alpha) \frac{K_1(\alpha \sqrt{\delta^2 + x^2})}{\sqrt{\delta^2 + x^2}} \quad QED \]

### 14.6.1 Modelling asset returns

The empirical distributions of financial returns data show that, compared to the normal distribution, there is: more mass near the origin, less in the flanks and considerably more in the tails. This means that tiny price movements occur with higher frequency, small- and medium-sized movements with lower frequency and big price changes are much more frequent than that predicted by a Gaussian distribution. The generalized hyperbolic distribution allows for an almost perfect statistical match to these empirical distributions, see Prause (1999), Raible (2000), and Eberlein (2001).

If there are \( n \) stock prices and they are modelled as:

\[ P_i = P_{i-1} \exp(X_i), \quad i = 1, \ldots, n \]

where \( X_{i>0} \) is generalized hyperbolic Levy motion, then

\[ \log(P_i) - \log(P_{i-1}) = X_i, \quad i = 1, \ldots, n \]

and the five parameters defining the generalized hyperbolic distribution can be estimated by maximizing Equation 14.41; the log-likelihood function.

#### GARCH–NIG model

Although the generalized hyperbolic distribution can adequately capture the fat tailed unconditional distribution of the returns, it does not take into account volatility clustering. In order to take these effects into account, Forsberg and Bollerslev (2002) proposed the following GARCH–NIG model.

\[ \epsilon_{i|t-1} \sim NIG_{sc}(\sigma_i^2, \alpha^*) \] (14.49)

\[ \sigma_i^2 = \alpha_o + \alpha_1 \epsilon_{i-1}^2 + \beta_1 \sigma_{i-1}^2 \] (14.50)

where we have written the distribution corresponding to the probability density function \( NIG_{sc}(\alpha) \) as \( NIG_{sc}(\sigma_i^2, \alpha^*) \) to show the dependence on the parameters \( \sigma_i^2 \) and \( \alpha^* \). Equations 14.49 and 14.50 describe a GARCH(1,1) model, see Section 15.2, and the parameters \( \alpha_0, \alpha_1, \beta_1, \) and \( \alpha^* \) can be estimated using maximum likelihood techniques, see Chapter 18.
Chapter 15

GARCH models

In this chapter we discuss the properties of linear GARCH models, in terms of the more fundamental AR and ARMA processes; further details can be found in the Box and Jenkins (1976), Hamilton (1994), and Engle (1995).

15.1 BOX JENKINS MODELS

This approach concerns the modelling of \( n \) observations \( y_i, i = 1, \ldots, n \) in the presence of white noise \( \nu_i, i = 1, \ldots, n \). The aim is to explain any observation \( y_i \) in terms of the current noise \( \nu_i \) and also a weighted linear sum of previous (lagged) observations and noise.

An autoregressive time series model of order \( p \) obeys the following equation:

\[
y_i = c + \sum_{j=1}^{p} \phi_j y_{i-j} + \nu_i, \quad \text{for } i = 1, \ldots, n \tag{15.1}
\]

where the \( \phi_j, j = 1, \ldots, p \) are termed the autoregressive coefficients and \( \nu_i \) is white noise satisfying:

\[
E[\nu_i] = 0, \quad E[\nu_i \nu_j] = 0, \quad i \neq j, \quad E[\nu_i^2] = \sigma_0^2 \tag{15.2}
\]

Such a process is also denoted as AR(p) and it can be shown that \( y_i \) is covariance stationary provided the roots, \( z^l \), of the polynomial,

\[
P(z) = 1 - \phi_1 z - \phi_2 z^2 - \cdots - \phi_p z^p = 0 \tag{15.3}
\]

all have modulus greater than 1, that is \( |z^l| > 1 \), for \( j = 1, \ldots, p \).

If the AR(p) process is covariance stationary then \( E[y_i] = \mu \), for all \( i \), where \( \mu \) is the unconditional mean of the sequence. Taking expectations of Equation 15.1 and using our previous results concerning \( E[y_i] \) and \( E[\nu_i] \), we have:

\[
E[y_i] = c + \sum_{j=1}^{p} \phi_j E[y_{i-j}] + E[\nu_i] \tag{15.4}
\]

\[
\mu = c + \mu \sum_{j=1}^{p} \phi_j \tag{15.5}
\]
We thus have:

The unconditional mean of an AR(p) process is:

\[
\mu = c \left\{ 1 - \sum_{j=1}^{p} \phi_j \right\}^{-1}
\]  

(15.6)

An autoregressive process can be generalized into an autoregressive moving average process by the inclusion of extra lagged terms as follows:

\[
y_i = c + \sum_{j=1}^{p} \phi_j y_{i-j} + \sum_{j=1}^{q} \theta_j \nu_{i-j} + \nu_i, \quad \text{for } i = 1, \ldots, n
\]

(15.7)

where all terms have the same meaning as before and \( \theta_j, j = 1, q \) are called the moving average coefficients.

Such a process is also denoted as ARMA(p,q) and it can be shown that the conditions for \( y_i \) to be covariance stationary are the same as those for an AR(p) process. That is the extra \( q \) moving average coefficients do not affect the conditions for the process to be covariance stationary. Taking expectations of Equation 15.7, and using our previous results concerning \( E[y_i] \) and \( E[\nu_i] \), we have:

\[
E[y_i] = c + \sum_{j=1}^{p} \phi_j E[y_{i-j}] + \sum_{j=1}^{q} \theta_j E[\nu_{i-j}] + E[\nu_i]
\]

\[
E[\nu_i] = c + \mu \sum_{j=1}^{p} \phi_j
\]

(15.8)

So

The unconditional mean of an ARMA(p,q) process is:

\[
\mu = c \left\{ 1 - \sum_{j=1}^{p} \phi_j \right\}^{-1}
\]

(15.9)

which is the same as for an AR(p) process.

\( E[y_i] \) denotes the unconditional expectation of \( y_i \) and \( E[y_i | \psi_{i-1}] \), denotes the expectation of \( y_i \) conditional on all relevant information up to instant \( i - 1 \). Since neither of these expectations is time-dependent the above process is said to be both unconditionally and conditionally homoskedastic.
15.2 GAUSSIAN LINEAR GARCH

GARCH relaxes this constraint and allows the conditional variance of \( y_t \) to vary with time. For example when \( p = 0 \) and \( c = 0 \) in Equation 15.1 we now have:

\[
y_i = \epsilon_i, \quad i = 1, \ldots, n
\]

where \( E(\epsilon_i^2) = h_i \) and \( h_i \) is the time-dependent conditional variance. However the unconditional variance of \( \epsilon_i \) is still constant,

\[
E[\epsilon_i^2] = E[E(\epsilon_i^2 | \psi_{i-1})] = E[h_i] = \sigma_0^2
\]

In a similar manner to the Box Jenkins approach described above in Section 15.1, we can define an autoregressive conditional heteroskedastic process of order \( q \), ARCH\((q)\), process, with Gaussian residuals as follows:

\[
h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n \quad \epsilon_i | \psi_{i-1} \sim NID(0, h_i)
\] (15.10)

This can then be generalized to a GARCH\((p,q)\) process in the same way that an ARMA\((p,q)\) is a generalization of an AR\((q)\) process.

In the same way that an ARMA\((p,q)\) is a generalization of an AR\((q)\) process, we can define a generalized autoregressive conditional heteroskedastic of order \((p, q)\), GARCH\((p,q)\) as follows:

\[
\text{Linear GARCH}(p,q)
\]

\[
h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n \quad \epsilon_i | \psi_{i-1} \sim NID(0, h_i)
\] (15.11)

The relationship between GARCH and ARMA processes can be illustrated as follows:

\[
h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n
\] (15.12)

If \( \epsilon_i^2 \) is added to both sides, and the zero term \( \sum_{j=1}^{p} \beta_j \epsilon_{i-j}^2 - \sum_{j=1}^{p} \beta_j \epsilon_{i-j}^2 \) is added to the right hand side we have:

\[
h_i + \epsilon_i^2 = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j \epsilon_{i-j}^2 + \epsilon_i^2 - \sum_{j=1}^{p} \beta_j (\epsilon_{i-j}^2 - h_{i-j})
\]

So

\[
\epsilon_i^2 = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j) \epsilon_{i-j}^2 + \epsilon_i^2 - \sum_{j=1}^{p} \beta_j (\epsilon_{i-j}^2 - h_{i-j})
\] (15.13)

where \( \kappa = \max(p, q) \) and we have \( \alpha_i = 0 \), for \( i > q \), and \( \beta_i = 0 \), for \( i > p \).
We notice that $h_i$ is the forecast for $e_i^2$ based on its own lagged values. The term $\nu_i = e_i^2 - h_i$ is the forecast error associated with this forecast, and is therefore a white noise process. Substituting for $\nu_i$ in Equation 15.13 we then have:

$$e_i^2 = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j) e_{i-j}^2 - \sum_{j=1}^{p} \beta_j \nu_{i-j} + \nu_i$$  \hspace{1cm} (15.14)

Comparing this with the above equation for an ARMA($p,q$) process we see that the sequence $e_i^2$ is an ARMA($\kappa,p$) process with $\kappa$ autoregressive coefficients $(\alpha_j + \beta_j)$, $j = 1, \ldots, \kappa$, and $p$ moving average coefficients $\beta_j$, $j = 1, \ldots, p$. So if the residuals $e_i$ are described by a GARCH($p,q$) process then $e_i^2$ are described by an ARMA($\kappa,p$) process, where $\kappa = \max (p,q)$.

From standard results for ARMA processes, $e_i^2$ is covariance stationary provided $\nu_i$ has finite variance and the roots, $z^*$, of the polynomial

$$P(z) = 1 - (\beta_1 + \alpha_1)z - (\beta_2 + \alpha_2)z^2 - \cdots - (\beta_\kappa + \alpha_\kappa)z^\kappa = 0$$  \hspace{1cm} (15.15)

all have modulus greater than 1, that is $|z^*| > 1$, for $j = 1, \ldots, \kappa$, Box and Jenkins (1976), and Levi (1942).

If we impose the nonnegativity requirement $\alpha_0 > 0$ and $\alpha_j \geq 0, \beta_j \geq 0$, for $j = 1, \ldots, \kappa$ then we will now show that the condition for $e_i^2$ to be covariance stationary is:

**Condition for GARCH to be covariance stationary:**

$$(\beta_1 + \alpha_1) + (\beta_2 + \alpha_2) + \cdots + (\beta_\kappa + \alpha_\kappa) < 1$$  \hspace{1cm} (15.16)

or more concisely

$$\sum_{j=1}^{\kappa} (\beta_j + \alpha_j) < 1$$

which means that $|z^*| > 1$, for $j = 1, \ldots, \kappa$.

The proof is as follows:

1. **Show that if $\sum_{j=1}^{\kappa} (\beta_j + \alpha_j) \geq 1$ then $e_i^2$ can’t be stationary**

If $\sum_{j=1}^{\kappa} (\beta_j + \alpha_j) \geq 1$ then because $\alpha_j \geq 0, \beta_j \geq 0$, $P(1) = 1 - \sum_{j=1}^{\kappa} (\beta_j + \alpha_j) < 0$, and $P(0) = 1 > 0$. Since the polynomial has changed sign between 0 and 1 this means that there is a root in this interval. So under these circumstances we must have at least one root $z_j$ with $|z^*| < 0$, which means that the process is not covariance stationary.
2. **Show that if** \( \sum_{j=1}^{\kappa} (\beta_j + \alpha_j) < 1 \) **then** \( \varepsilon_i^2 \) **must be stationary**

If \( \sum_{j=1}^{\kappa} (\beta_j + \alpha_j) < 0 \) **and there is a root of** \( P(z) \), \( z' \), **with** \( |z'| < 1 \), **then we have**

\[
P(z') = 1 - \sum_{j=1}^{\kappa} (\beta_j + \alpha_j)z' = 0\]

But since \( \alpha_j \geq 0, \beta_j \geq 0 \) **and** \( z' \leq |z'| \) **we have:**

\[
1 = |\sum_{j=1}^{\kappa} (\beta_j + \alpha_j)|z'| \leq \sum_{j=1}^{\kappa} (\beta_j + \alpha_j) < 1\]

which is inconsistent.

So if \( \sum_{j=1}^{\kappa} (\beta_j + \alpha_j) < 1 \) **then we must have** \( |z'| > 1 \), **for** \( j = 1, \ldots, \kappa \) **QED**

We will now assume that \( \varepsilon_i^2 \) **is covariance stationary and calculate its unconditional variance by taking expectations in Equation 15.14 as follows:**

\[
E[\varepsilon_i^2] = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j)E[\varepsilon_j^2] - \sum_{j=1}^{\kappa} \beta_j E[\nu_{i-j}] + E[\nu_i] \]  \hspace{1cm} (15.17)

But since \( \varepsilon_i^2 \) **is covariance stationary and** \( \nu_i \) **is white noise we have:**

\[
E[\varepsilon_i^2] = E[\varepsilon_{i-j}^2] \quad \text{and} \quad E[\nu_i] = E[\nu_{i-j}] = 0
\]

Therefore:

\[
E[\varepsilon_i^2] = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j)E[\varepsilon_j^2] \]  \hspace{1cm} (15.18)

and

**GARCH unconditional variance:**

\[
\sigma_0^2 = E[\varepsilon_i^2] = \alpha_0 \left( 1 - \sum_{j=1}^{\kappa} (\alpha_j + \beta_j) \right)^{-1} \]  \hspace{1cm} (15.19)

**15.2.1 The unconditional kurtosis of the residuals**

In Equation 15.11 the conditional distribution of the residuals, \( \varepsilon_i \), was:

\[
\varepsilon_i | \psi_{i-1} \sim NID(0, h_i)
\]

For convenience we will now rewrite this as:

\[
\varepsilon_i = \sqrt{h_i}Z_i, \quad \text{where} \quad Z_i \sim NID(0, 1)
\]

Therefore \( \varepsilon_i^4 = h_i^2 Z_i^4 \). Using the fact that \( h_i \) **and** \( Z_i \) **are independent of each other we have**

\[
\]

Jensen’s inequality (see Goldberger (1997) and Appendix F.6), states that for a random variate \( X \), \( E[X^2] \geq E[X]^2 \), since the function \( X^2 \) is convex.
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Using this result we have:

\[ E[h_i^n] \geq E[h_i] \]
which gives \( E[h_i^n] E[Z_i^n] \geq E[h_i] E[Z_i^n] \).


Therefore the unconditional kurtosis is:

\[ \kappa = \frac{E[\epsilon_i^2]}{E[\epsilon_i^2]} \geq E[Z_i^n] \]

But since \( Z_i \) comes from a standardized Gaussian distribution it has variance \( E[Z_i^2] = 1 \), and a kurtosis of 3, see Chapter 17. This means that:

\[ \frac{E[Z_i^n]}{E[Z_i]} = E[Z_i] = 3, \quad \text{so} \quad \kappa = \frac{E[\epsilon_i^2]}{E[\epsilon_i^2]} \geq 3 \]

This shows that although the residuals have a Gaussian conditional distribution their unconditional distribution is leptokurtic and therefore non-Gaussian.

In fact we can use Jensen's inequality in a similar manner to show that for any arbitrary conditional distribution \( \mathcal{R}(0, h_i) \) the unconditional kurtosis of \( \epsilon_i \) will be higher than the kurtosis of \( \mathcal{R}(0, h_0) \).

We will now derive the value of the unconditional kurtosis of an ARCH(1) process.

**Kurtosis for an ARCH(1) process**

For an ARCH(1) process we have:

\[ \epsilon_i = \left\{ \alpha_0 + \alpha_1 \epsilon_{i-1}^2 \right\}^{1/2} Z_i, \quad Z_i \sim NID(0, 1) \]

Therefore

\[ E[\epsilon_i^4] = E[(\alpha_0 + \alpha_1 \epsilon_{i-1}^2)^2 Z_i^4] = E[(\alpha_0 + \alpha_1 \epsilon_{i-1}^2)^2] E[Z_i^4] \]

But

\[ E[(\alpha_0 + \alpha_1 \epsilon_{i-1}^2)^2] = E[\alpha_0^2 + \alpha_1^2 \epsilon_{i-1}^4 + 2\alpha_0 \alpha_1 \epsilon_{i-1}^2] \]

\[ = \alpha_0^2 + \alpha_1^2 E[\epsilon_{i-1}^4] + 2\alpha_0 \alpha_1 E[\epsilon_{i-1}^2] \]

But since \( E[\epsilon_{i-1}^4] = E[\epsilon_i^4] \) and \( E[\epsilon_{i-1}^2] = E[\epsilon_i^2] \) we have:

\[ E[\epsilon_i^4] = 3 (\alpha_0^2 + \alpha_1^2 E[\epsilon_i^4] + 2\alpha_0 \alpha_1 E[\epsilon_i^2]) = 3 \alpha_0^2 + 3 \alpha_1^2 E[\epsilon_i^4] + 6 \alpha_0 \alpha_1 E[\epsilon_i^2] \]

using \( E[\epsilon_i^2] = (\alpha_0)/(1 - \alpha_1) \) we have:

\[ E[\epsilon_i^4](1 - 3 \alpha_1^2) = 3 \left( \frac{\alpha_0^2 + 2 \alpha_0 \alpha_1}{1 - \alpha_1} \right) = \frac{3 \alpha_0^2}{(1 - 3 \alpha_1^2)} \left( 1 + \alpha_1 \right) \]
which gives:

\[ \kappa = \frac{E[e_i^4]}{E[e_i^2]^2} = \frac{3\alpha_0^2}{1 - 3\alpha_i^2} \left( \frac{1 + \alpha_i}{1 - \alpha_i} \right) \left( 1 - \frac{\alpha_i^2}{\alpha_0^2} \right)^2 \]

So the kurtosis is:

\[ \kappa = \frac{3(1 + \alpha_i)(1 - \alpha_i)}{1 - 3\alpha_i^2} = \frac{3(1 - \alpha_i^2)}{1 - 3\alpha_i^2} \]

For finite values of \( E[e_i^2] \) and \( E[e_i^4] \), we require \( \alpha_1 < 1 \) and \( 3\alpha_i < 1 \) respectively. Since \( 1 - \alpha_i^2 > 1 - 3\alpha_i^2 \) we have \( \kappa > 3 \), which means that the ARCH model has heavier tails than a Gaussian distribution.

### Kurtosis for a GARCH(1,1) process

To derive the unconditional kurtosis of a GARCH(1,1) is quite complicated, so we simply present the following results, see Bollerslev (1986):

For GARCH(1,1) we have:

\[ E(e_i^4) = \frac{3\alpha_0^2(1 + (\alpha_1 + \beta_i))}{(1 - (\alpha_1 - \beta_i))(1 - \beta_i^2 - 2\alpha_1\beta_i - 3\alpha_i^2)} \]

and from Equation 15.19 the conditional variance is:

\[ E[e_i^2] = \frac{\alpha_0}{1 - (\alpha_1 + \beta_i)} \]

Therefore the unconditional kurtosis of a GARCH(1,1) process is:

\[ \kappa = \left[ \frac{3\alpha_0^2(1 + (\alpha_1 + \beta_i))}{(1 - (\alpha_1 - \beta_i))(1 - \beta_i^2 - 2\alpha_1\beta_i - 3\alpha_i^2)} \right] \left[ \frac{(1 - (\alpha_1 + \beta_i))^2}{\alpha_0^2} \right] \]

\[ = 3 + \frac{6\alpha_i^2}{(1 - \beta_i^2 - 2\alpha_1\beta_i - 3\alpha_i^2)} \]

For a finite value of \( E[e_i^4] \), we require \( 3\alpha_i^2 + 2\alpha_1\beta_i + \beta_i^2 < 1 \). When this constraint is satisfied \( \kappa > 3 \).

### 15.2.2 Forecasting and mean-reversion in a GARCH(1,1) process

Here we derive an expression for the \( T \) step ahead volatility forecast of a GARCH(1,1) process. Given the information set \( \psi_{t-1} \) the expected volatility \( E[h_i|\psi_{t-1}] \), at instant \( i \) as:

\[ E[h_i|\psi_{t-1}] = \alpha_0 + \alpha_1 e_{i-1}^2 + \beta_1 h_{i-1} \]

and at instant \( i + 1 \), \( E[h_{i+1}|\psi_{t-1}] \) is thus:

\[ E[h_{i+1}|\psi_{t-1}] = \alpha_0 + \alpha_1 E[e_i^2] + \beta_1 E[h_i|\psi_{t-1}] \]
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Now since $E[c^2_\psi|\psi_{t-1}] = E[h_t|\psi_{t-1}]$ we have:

$$E[h_{t+1}|\psi_{t-1}] = \alpha_0 + (\alpha_1 + \beta_1)E[h_t|\psi_{t-1}]$$  \hfill (15.20)

Proceeding in a similar manner we have:

$$E[h_{t+2}|\psi_{t-1}] = \alpha_0 + \alpha_1E[c^2_{\psi_{t+1}}|\psi_{t-1}] + \beta_1E[h_{t+1}|\psi_{t-1}]$$
$$E[h_{t+2}|\psi_{t-1}] = \alpha_0 + (\alpha_1 + \beta_1)E[h_{t+1}|\psi_{t-1}]$$
$$E[h_{t+2}|\psi_{t-1}] = \alpha_0 + (\alpha_1 + \beta_1)\{\alpha_0 + (\alpha_1 + \beta_1)E[h_t|\psi_{t-1}]\}$$
$$E[h_{t+2}|\psi_{t-1}] = \alpha_0 + \alpha_0(\alpha_1 + \beta_1) + (\alpha_1 + \beta_1)^2E[h_t|\psi_{t-1}]$$  \hfill (15.21)

$$E[h_{t+3}|\psi_{t-1}] = \alpha_0 + \alpha_1E[c^2_{\psi_{t+2}}|\psi_{t-1}] + \beta_1E[h_{t+2}|\psi_{t-1}]$$
$$E[h_{t+3}|\psi_{t-1}] = \alpha_0 + (\alpha_1 + \beta_1)E[h_{t+2}|\psi_{t-1}]$$
$$E[h_{t+3}|\psi_{t-1}] = \alpha_0 + (\alpha_1 + \beta_1)\{\alpha_0 + \alpha_0(\alpha_1 + \beta_1) + (\alpha_1 + \beta_1)^2E[h_t|\psi_{t-1}]\}$$
$$E[h_{t+3}|\psi_{t-1}] = \alpha_0 + \alpha_0(\alpha_1 + \beta_1) + \alpha_0(\alpha_1 + \beta_1)^2 + (\alpha_1 + \beta_1)^3E[h_t|\psi_{t-1}]$$  \hfill (15.22)

So we have:

$$E[h_{t+T}|\psi_{t-1}] = \alpha_0 + \alpha_0(\alpha_1 + \beta_1) + \alpha_0(\alpha_1 + \beta_1)^2 + \cdots + (\alpha_1 + \beta_1)^{T-1}$$
$$+ (\alpha_1 + \beta_1)^TE[h_t|\psi_{t-1}]$$  \hfill (15.23)

Equation 15.23 is the sum of $T$ terms of a Geometric Progression with first term $\alpha_0$ and common factor $(\alpha_1 + \beta_1)$, and there is also an additional term $(\alpha_1 + \beta_1)^TE[h_t|\psi_{t-1}]$. So

**GARCH(1,1) forecast:**

$$E[h_{t+T}|\psi_{t-1}] = \alpha_0 \left\{ \frac{1 - (\alpha_1 + \beta_1)^T}{1 - (\alpha_1 + \beta_1)} \right\} + (\alpha_1 + \beta_1)^TE[h_t|\psi_{t-1}]$$  \hfill (15.24)

Since $\alpha_1 + \beta_1 < 1$ for a stationary sequence, as $T \to \infty$ we have

$$E[h_{t+T}|\psi_{t-1}] = \frac{\alpha_0}{1 - (\alpha_1 + \beta_1)}$$  \hfill (15.25)

This is just the unconditional variance of the GARCH sequence. It can thus be seen from Equations 15.24 and 15.25 that the GARCH volatility forecast is mean reverting, and that the smaller the value of $\alpha_1 + \beta_1$ the faster is the reversion speed.
15.3 THE IGARCH MODEL

It has been found that the use of a GARCH(1,1) model on financial data often results in $\beta_1 > 0.7$ and $\alpha_1 \approx 1 - \beta_1$. This has motivated the integrated GARCH(p,q), also termed IGARCH(p,q), in which $\alpha_1 + \beta_1 = 1$, see Engle and Bollerslev (1986).

From Equation 15.19 it can be seen that the unconditional variance of the sequence, $E(\epsilon^2_i)$, is infinite, and from Equation 15.26 that the sequence is not covariance-stationary. However, Nelson (1990) shows that:

$$h_i = a_0 \left( \sum_{k=1}^{i-1} \prod_{j=1}^{k} (\alpha_1 \zeta_{i-k} + \beta_1) \right) + \prod_{j=1}^{i} (\alpha_1 \zeta_{i-j} + \beta_1) h_0$$

where $\zeta_k = e^2_i / h_k$, and that the sequence is strictly stationary if $E[\log(\alpha_1 \zeta_{i-j} + \beta_1)] < 0$. When this condition is satisfied the effect of the initial value $h_0$ disappears asymptotically.

15.3.1 Exponentially weighted moving average: EWMA

The exponentially weighted moving average (EWMA) method is a special case of the IGARCH(1,1) model:

$$h_i = a_0 + \alpha_1 e^2_{i-1} + (1 - \alpha_1) h_{i-1}, \quad i = 1, \ldots, n$$

In the case of EWMA we take $a_0 = 0$ and obtain the scheme:

$$h_i = \lambda e^2_{i-1} + (1 - \lambda) h_{i-1}, \quad i = 1, \ldots, n \quad (15.26)$$

where the parameter $\lambda$ is known as the weight, or decay factor. It can be seen that the value of $h_i$ is the weighted average of $e^2_{i-1}$ and $h_{i-1}$.

Risk metrics, J. P. Morgan (1996) advocate this method of modelling volatility, and selected $\lambda = 0.97$ as the optimal value to use.

15.4 THE GARCH-M MODEL

Finance theory suggests that, on average, an asset with a higher risk should have a higher return.

Engle et al. (1987) proposed the ARCH-M model to capture this effect. A simple GARCH-M model is:

$$y_i = \delta h_i + \epsilon_i \quad (15.27)$$

$$\epsilon_i | \psi_{i-1} = N(0, h_i) \quad (15.28)$$

$$h_i = a_0 + \sum_{j=1}^{q} \alpha_j \epsilon^2_{i-j} + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n \quad (15.29)$$

Here $y_i$ is the mean asset return at time $i$, and $h_i$ the variance of $\epsilon_i$, is a measure of the associated risk. It can be seen that the extra term $\delta h_i$ leads to increased returns for higher values of $h_i$. 
15.5 REGRESSION-GARCH AND AR-GARCH

Up to now we have used GARCH models with variables defined as $y_i = \epsilon_i$.

We will now include linear regression into the GARCH model.

A regression-GARCH(p,q) sequence containing $n$ terms with Gaussian shocks, $\epsilon_i$, takes the following form:

$$y_i = b_0 + X_i^T b + \epsilon_i, \quad \epsilon_i | \psi_{i-1} \sim NID(0, h_i)$$  \hspace{1cm} (15.30)

$$h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n$$  \hspace{1cm} (15.31)

This process is described by $q+1$ coefficients $\alpha_j, \; j = 0, \ldots, q$, $p$ coefficients $\beta_j, \; j = 1, \ldots, p$, mean $b_0$, $k$ linear regression coefficients $b_j, \; j = 1, \ldots, k$, endogenous/exogenous variables $y_i$ and $X_i$ respectively, shocks $\epsilon_i$, $h_i$ the conditional variance, and the set of all information up to time $i$, $\psi_i$. The conditional probability distribution of $\epsilon_i$ is denoted by $\mathcal{P}(0, h_i)$, a distribution with zero mean and time-varying variance $h_i$.

Here $X_i$ denotes the $k$ element row vector of exogenous variables at time $i$, and $b$ refers to the $k$ element column vector or regression coefficients. We also use $X_i^T$ to indicate the column vector formed by the transpose of $X_i$, and the $k$ individual elements of $X_i$ are denoted by $X_{ij}, \; j = 1, \ldots, k$.

It should be noted that the $n$ term regression-GARCH(p,q) model above can easily be used to model an $n-m$ term AR(m)–GARCH(p,q) sequence defined as follows:

$$y_i = c + \sum_{j=1}^{p} \phi_j y_{i-j} + \epsilon_i, \quad i = m + 1, \ldots, n$$  \hspace{1cm} (15.32)

$$h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = m + 1, \ldots, n$$  \hspace{1cm} (15.33)

where the terms $y_i, \; i = 1, \ldots, m$ are used as the pre-observed values for the AR(m)–GARCH(p,q) sequence.

If we let $k = m$ then the mean term $b_0$ is identified as $c$ and the $k$ time-dependent exogenous variables are replaced by the lagged values of $y_i$. That is the row vector $X_i = (y_{i-1}, y_{i-2}, \ldots, y_{i-m})$. 


Chapter 16

Nonlinear GARCH

The standard GARCH model assumes that both positive and negative shocks of equal magnitude have an identical effect on future volatility. However, empirical studies on stock returns have shown that they are characterized by increased volatility following negative shocks (bad news). This *leverage* effect was first recognized by Black (1976), who reasoned that it is connected with the way in which firms are financed. When the value of a firm’s stock decreases the debt-to-equity ratio increases, which leads to an increase in the volatility of the returns on equity. The leverage effect suggests that positive and negative shocks have an *asymmetric* impact on the conditional volatility of subsequent observations. It has been found that the returns for different asset classes display different leverage characteristics. The returns for equities and equity indices have negative leverage (negative shocks increase subsequent volatility). By contrast the returns for commodities and commodity futures exhibit both positive and negative leverage effects, McKenzie *et al.* (2001). Finally exchange rate returns, where the concept of good/bad news is less well defined, have no leverage effects at all. This is because a return series of currency $X$ in terms of currency $Y$ can be inverted (negative shocks now transformed into positive shocks) to yield a return series of currency $Y$ in terms of currency $X$.

Since linear GARCH models cannot capture these effects various nonlinear GARCH extensions have been proposed. These models include: Exponential GARCH (EGARCH) (Nelson, 1991), Asymmetric GARCH (AGARCH) (Engle and Ng, 1993), GJR–GARCH (Glosten *et al.*, 1993), Markov-Switching GARCH (MSW-GARCH) (Dueker, 1997), and Asymmetric Nonlinear Smooth Transition GARCH (ANST-GARCH) (Anderson *et al.*, 1999). Hentschel (1995) provides a more comprehensive overview of nonlinear GARCH models.

Empirical studies have also found that both the conditional and unconditional distributions of financial returns exhibit leptokurtosis (have fatter tails than a normal distribution). A popular choice, Bollerslev (1987), Engle and Gonzalez-Rivera (1991), is to assume that, instead of a Gaussian distribution, the errors $\epsilon_t$ have a Student’s $t$ distribution with $\nu$ degrees of freedom.

Here we consider asymmetric effects in AGARCH-I, AGARCH-II, and GJR–GARCH sequences, which can be modelled by the inclusion of an extra asymmetry parameter, $\gamma$. The mathematical definition of these processes is as follows:
AGARCH-I

\[ h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j (\epsilon_{i-j} + \gamma)^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n \]  

(16.1)

AGARCH-II

\[ h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j (|\epsilon_{i-j}| + \gamma \epsilon_{i-j})^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n \]  

(16.2)

GJR–GARCH

\[ h_i = \alpha_0 + \sum_{j=1}^{q} (\alpha_j + \gamma S_{i-j}) \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n \]  

(16.3)

where \( S_i = 1, \text{ if } \epsilon_i < 0 \) and \( S_i = 0, \text{ if } \epsilon_i \geq 0 \).

EGARCH

\[
\log(h_i) = \alpha_0 + \sum_{j=1}^{q} \alpha_j Z_{i-j} + \sum_{j=1}^{q} \phi(|Z_{i-j} - E[|Z_{i-j}|])
+ \sum_{j=1}^{p} \beta_j \log(h_{i-j}), \quad i = 1, \ldots, n
\]

(16.4)

where \( Z_i = \epsilon_i / \sqrt{h_i} \) and \( E[|Z_{i-j}|] \) denotes the expected value of \(|Z_{i-j}|\).

In AGARCH-I the asymmetric effects are modelled via the extra parameter. For example, in the standard GARCH(1,1) model when \( h_{i-1} \) is fixed \( h_i = h(\epsilon_{i-1}) \) is a parabola with a minimum at \( \epsilon_{i-1} = 0 \). The introduction of the additional parameter \( \gamma \) shifts the parabola horizontally so that the minimum occurs at \( \epsilon_{i-1} = -\gamma \). The conditional variance following negative shocks can therefore be enhanced by choosing \( \gamma < 0 \), so that \( h(-\epsilon_{i-1}) > h(\epsilon_{i-1}) \) for \( \epsilon_{i-1} > 0 \).

In an AGARCH-II model the inclusion of \( \gamma \) can also result in an enhancement of \( h_i \) following a negative shock \( \epsilon_{i-1} \). For a GARCH(1,1) model \( h(-\epsilon_{i-1}) > h(\epsilon_{i-1}) \) for \( \epsilon > 0 \) and \( \gamma < 0 \).

Similarly in the GJR–GARCH(1,1) model the value of \( h_i \) is increased above the symmetric case when \( \epsilon_{i-1} < 0 \) and \( \gamma > 0 \).
For EGARCH, asymmetric response arises from the term \( \sum_{j=1}^{q} \alpha_j Z_{i-j} \). In an EGARCH(1,1), if \( \alpha_1 < 0 \) then a negative shock \( \epsilon_{i-1} \) increases the value of \( h_i \), that is \( \log \{ h_i(Z_{i-1}) \} > \log \{ h_i(Z_{i-1}) \} \).

16.1 AGARCH-I

From Equation 16.1 the AGARCH-I process is defined as:

\[
h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j (\epsilon_{i-j} + \gamma)^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, \nu
\]

Since \( (\epsilon_{i-j} + \gamma)^2 = \epsilon_{i-j}^2 + 2\gamma \epsilon_{i-j} + \gamma^2 \) we have

\[
h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j} + 2\gamma \sum_{j=1}^{q} \epsilon_{i-j} + \sum_{j=1}^{q} \alpha_j \gamma^2
\]

Following the same procedure as in Section 16.2 we have

\[
\epsilon_i^2 = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j) \epsilon_{i-j}^2 - \sum_{j=1}^{p} \beta_j \nu_{i-j} + \nu_i + 2\gamma \sum_{j=1}^{q} \epsilon_{i-j} + \sum_{j=1}^{q} \alpha_j \gamma^2
\]

where \( \kappa = \max(p,q) \) and we have \( \alpha_i = 0 \), for \( i > q \) and \( \beta_i = 0 \), for \( i > p \).

Taking expectations gives:

\[
E[\epsilon_i^2] = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j) E[\epsilon_{i-j}^2] - \sum_{j=1}^{p} \beta_j E[\nu_{i-j}] + E[\nu_i] + 2\gamma \sum_{j=1}^{q} E[\epsilon_{i-j}] + \sum_{j=1}^{q} \alpha_j \gamma^2
\]

(16.5)

Now since \( E[\nu_i] = 0 \) and \( E[\epsilon_i] = 0 \) we have:

\[
E[\epsilon_i^2] = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j) E[\epsilon_{i-j}^2] + \sum_{j=1}^{q} \alpha_j \gamma^2
\]

This is an AR(\( \kappa \)) process and the condition for \( \epsilon_i^2 \) to be covariance stationary is:

\[
\sum_{j=1}^{\kappa} (\alpha_j + \beta_j) < 1
\]

(16.6)

which is the same condition as for the standard linear GARCH(p,q) process.

Assuming that \( \epsilon_i^2 \) is covariance stationary we have \( \sigma_0^2 = E[\epsilon_i^2] = E[\epsilon_{i-j}^2] \) and so

\[
E[\epsilon_i^2] = \alpha_0 + \gamma^2 \sum_{j=1}^{q} \alpha_j + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j) E[\epsilon_i^2]
\]

which results in
16.1.1 Kurtosis

We will now calculate the kurtosis for an AGARCH-I(0,1) process:

\[ \epsilon_i = \left\{ \alpha_0 + \alpha_1 (\epsilon_{i-1} + \gamma) \right\}^{1/2} Z_i, \quad Z_i \sim NID(0, 1) \]

Therefore

\[ E[\epsilon_i^4] = E[(\alpha_0 + \alpha_1 (\epsilon_{i-1} + \gamma))^2 Z_i^4] = E[(\alpha_0 + \alpha_1 (\epsilon_{i-1} + \gamma))^2] E[Z_i^4] \]

We will assume that process is covariance stationary and use the fact that the expectation of odd powers of \( \epsilon_i \) is zero and that \( E[Z_i^4] = 3 \).

\[ E[\epsilon_i^4] = 3 \left( \alpha_0^2 + 2 \alpha_0 \alpha_1 \gamma^2 + \alpha_1^2 \gamma^4 + 2 \alpha_1 \alpha_0 E[\epsilon_i^2] + 6 \alpha_1^2 \gamma^2 E[\epsilon_i^2] + \alpha_1^2 E[\epsilon_i^2] \right) \]

which gives

\[ E[\epsilon_i^4](1 - 3\alpha_1^2) = 3 \left\{ (\alpha_0 + \alpha_1 \gamma^2)^2 + E[\epsilon_i^2] (6\alpha_1^2 \gamma^2 + 2\alpha_1 \alpha_0) \right\} \]

substituting for \( E[\epsilon_i^2] \) we have:

\[ E[\epsilon_i^4](1 - 3\alpha_1^2) = 3(\alpha_0 + \alpha_1 \gamma^2)^2 \left\{ (\alpha_0 + \alpha_1 \gamma^2) + \frac{(6\alpha_1^2 \gamma^2 + 2\alpha_1 \alpha_0)}{(1 - \alpha_1)} \right\} \]

\[ = \frac{3(\alpha_0 + \alpha_1 \gamma^2)^2}{(1 - \alpha_1)} \left\{ (\alpha_0 + \alpha_1 \gamma^2)(1 - \alpha_1) + (6\alpha_1^2 \gamma^2 + 2\alpha_1 \alpha_0) \right\} \]

But

\[ (\alpha_0 + \alpha_1 \gamma^2)(1 - \alpha_1) + (6\alpha_1^2 \gamma^2 + 2\alpha_1 \alpha_0) = (\alpha_0 + \alpha_1 \gamma^2)(1 + \alpha_1) + 4\alpha_1^2 \gamma^2 \]

Therefore

\[ E[\epsilon_i^4] = \frac{3(\alpha_0 + \alpha_1 \gamma^2)^2}{(1 - 3\alpha_1^2)(1 - \alpha_1)} \left\{ (\alpha_0 + \alpha_1 \gamma^2)(1 + \alpha_1) + 4\alpha_1^2 \gamma^2 \right\} \]

So the kurtosis is:

\[ \kappa = \frac{E[\epsilon_i^4]}{(E[\epsilon_i^2])^2} = \frac{3(1 - \alpha_1^2 + \mathcal{F})}{1 - 3\alpha_1^2}, \quad \text{where} \quad \mathcal{F} = \frac{4\alpha_1^2 \gamma^2(1 - \alpha_1)}{\alpha_0 + \alpha_1 \gamma^2} \quad (16.8) \]

It is therefore evident that when \( \gamma = 0 \) we have \( \mathcal{F} = 0 \), and the kurtosis is the same as for the linear ARCH(1). However, for any non-zero value of \( \gamma \) the kurtosis will be greater than that for the standard ARCH(1). Furthermore, since \( \mathcal{F} \) increases monotonically with the absolute value of \( \gamma \), the unconditional kurtosis increases with \( \gamma \).
16.1.2 Skewness

We assume the non-negativity constraints $\alpha_0 > 0$ and $\alpha_1 > 0$.

$$\epsilon_i = \left\{ \alpha_0 + \alpha_1 (\epsilon_{i-1} + \gamma)^2 \right\}^{1/2} Z_i, \quad Z_i \sim \mathcal{R}(0, 1)$$

where $Z_i$ is an arbitrary symmetric distribution.

Therefore:

$$E[\epsilon_i^3] = E \left[ \left\{ \alpha_0 + \alpha_1 (\epsilon_{i-1} + \gamma)^2 \right\}^{3/2} Z_i^3 \right], \quad Z_i \sim \mathcal{R}(0, 1)$$

By decomposing this expectation into the part with $\epsilon_i \geq 0$, and the part with $\epsilon_i < 0$, we have:

$$E[\epsilon_i^3] = E \left[ \left\{ \alpha_0 + \alpha_1 (|\epsilon_{i-1} - \gamma)|^2 \right\}^{3/2} \right] E[Z_i^3] + E \left[ \left\{ \alpha_0 + \alpha_1 (|\epsilon_{i-1} + \gamma|^2 \right\}^{3/2} \right] E[Z_i^3]$$

where because $\mathcal{R}(0, 1)$ is symmetric we have $E[Z_i^3] = 0 = E[Z_i^3] + E[Z_i^3]$. This means that:

$$E[\epsilon_i^3] = \left\{ E \left[ \left\{ \alpha_0 + \alpha_1 (|\epsilon_{i-1} - \gamma|^2 \right\}^{3/2} \right] - E \left[ \left\{ \alpha_0 + \alpha_1 (|\epsilon_{i-1} + \gamma|^2 \right\}^{3/2} \right] \right\} E[Z_i^3]$$

Since $E[\epsilon_i^3] > 0$, the skewness is:

$$S = \frac{E[\epsilon_i^3]}{E[\epsilon_i^2]^{3/2}} \quad (16.9)$$

It can be seen that the skewness is zero for $\gamma = 0$, and becomes increasingly negative as the value of $\gamma$ is raised.

16.1.3 Forecasting and mean-reversion in an AGARCH-I(1,1) process

Here we derive an expression for the $T$ step ahead volatility forecast of an AGARCH-I(1,1) process. Given the information set $\psi_{i-1}$ we can forecast the expected volatility $E[h_i|\psi_{i-1}]$ at time instant $i$ as:

$$E[h_i|\psi_{i-1}] = \alpha_0 + \alpha_1 (\epsilon_{i-1} + \gamma)^2 + \beta_1 h_{i-1}$$

and at instant $i + 1$, $E[h_{i+1}|\psi_{i-1}]$ is:

$$E[h_{i+1}|\psi_{i-1}] = \alpha_0 + \alpha_1 E[\epsilon_i^2|\psi_{i-1}] + \beta_1 E[h_i|\psi_{i-1}] + \alpha_1 \gamma^2 + 2\gamma E[\epsilon_i|\psi_{i-1}]$$

Now since $E[\epsilon_i|\psi_{i-1}] = 0$, and $E[\epsilon_i^3|\psi_{i-1}] = E[h_i|\psi_{i-1}]$ we have:

$$E[h_{i+1}|\psi_{i-1}] = \alpha_0 + \alpha_1 \gamma^2 + (\alpha_1 + \beta_1) E[h_i|\psi_{i-1}] \quad (16.10)$$
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Proceeding in a similar manner we have:
\[
E[h_{i+2} \psi_{i-1}] = \alpha_0 + \alpha_1 E[\psi_{i-1}] + \beta_1 E[h_{i+1} \psi_{i-1}] + \alpha_1 \gamma^2 + 2\gamma E[\epsilon_{i+1} \psi_{i-1}]
\]
\[
E[h_{i+2} \psi_{i-1}] = \alpha_0 + \alpha_1 \gamma^2 + (\alpha_1 + \beta_1) E[h_{i+1} \psi_{i-1}]
\]
\[
E[h_{i+2} \psi_{i-1}] = \alpha_0 + \alpha_1 \gamma^2 + (\alpha_1 + \beta_1) \{ \alpha_0 + \alpha_1 \gamma^2 + (\alpha_1 + \beta_1) E[h_i \psi_{i-1}] \}
\]
\[
E[h_{i+2} \psi_{i-1}] = (\alpha_0 + \alpha_1 \gamma^2) \{ 1 + (\alpha_1 + \beta_1) \} + (\alpha_1 + \beta_1)^2 E[h_i \psi_{i-1}] \tag{16.11}
\]

So in general we have:
\[
E[h_{i+T} \psi_{i-1}] = (\alpha_0 + \alpha_1 \gamma^2) \left[ 1 + (\alpha_1 + \beta_1) + \cdots + (\alpha_1 + \beta_1)^{T-1} \right] + (\alpha_1 + \beta_1)^T E[h_i \psi_{i-1}] \tag{16.13}
\]

Equation 16.13 is the sum of \( T \) terms of a Geometric Progression with first term \( \alpha_0 + \alpha_1 \gamma^2 \) and common factor \( (\alpha_1 + \beta_1) \), and also additional term \( (\alpha_1 + \beta_1)^T E[h_i \psi_{i-1}] \). So

**AGARCH-I(1,1) forecast:**
\[
E[h_{i+T} \psi_{i-1}] = \frac{\{ \alpha_0 + \alpha_1 \gamma^2 - (\alpha_1 + \beta_1)^T \}}{1 - (\alpha_1 + \beta_1)} + (\alpha_1 + \beta_1)^T E[h_i \psi_{i-1}] \tag{16.14}
\]

Since \( \alpha_1 + \beta_1 < 1 \) for a stationary process, as \( T \to \infty \) we have
\[
E[h_{i+T} \psi_{i-1}] = \frac{\alpha_0 + \alpha_1 \gamma^2}{1 - (\alpha_1 + \beta_1)} \tag{16.15}
\]

which is just the unconditional variance of the GARCH sequence. It can be seen from Equations 16.13 and 16.15 that the volatility forecast is mean reverting, and that the smaller the value of \( \alpha_1 + \beta_1 \) the faster is the reversion speed.

16.2 AGARCH-II

The AGARCH-II process is defined by:
\[
h_i = \alpha_0 + \sum_{j=1}^{a} \alpha_j (\epsilon_{i-j} + \gamma \epsilon_{i-j})^2 + \sum_{j=1}^{b} \beta_j h_{i-j}, \quad i = 1, \ldots, n
\]
Following the same procedure as in Section 15.2 we have
\[ \epsilon_t^2 = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \gamma^2 \Delta_j) \epsilon_{t-j}^2 - \sum_{j=1}^{p} \beta_j \nu_{t-j} + \nu_t + 2\gamma \sum_{j=1}^{q} \alpha_j |\epsilon_{t-j}| \epsilon_{t-j} \]

where \( \kappa = \max(p, q) \) and we have \( \Delta_j = \alpha_j = 0 \), for \( j > q \), \( \beta_j = 0 \), for \( j > p \), and \( \Delta_j = 1 \) for \( j \leq q \). Taking expectations gives:
\[ E[\epsilon_t^2] = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \gamma^2 \Delta_j) E[\epsilon_{t-j}^2] \]

since \( E[\nu_t] = 0 \) and \( E[|\epsilon_t|\epsilon_t] = 0 \).

This is an AR(\( \kappa \)) process, and the condition for \( \epsilon_t^2 \) to be covariance stationary is:
\[ \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \gamma^2 \Delta_j) < 1 \]  \( \text{(16.16)} \)

Assuming that \( \epsilon_t^2 \) is covariance stationary we have \( \sigma_0^2 = E[\epsilon_t^2] = E[\epsilon_{t-j}^2] \) and so
\[ E[\epsilon_t^2] = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \gamma^2 \Delta_j) E[\epsilon_{t-j}^2] \]

which results in

<table>
<thead>
<tr>
<th>AGARCH-II unconditional variance</th>
</tr>
</thead>
<tbody>
<tr>
<td>[ \sigma_0^2 = E[\epsilon_t^2] = \frac{\alpha_0}{(1 - \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \gamma^2 \Delta_j))} ]  ( \text{(16.17)} )</td>
</tr>
</tbody>
</table>

16.3 GJR–GARCH

The GJR–GARCH(p,q) process is defined as:
\[ h_i = \alpha_0 + \sum_{j=1}^{q} (\alpha_j + \gamma S_{t-j}) \epsilon_{t-j}^2 + \sum_{j=1}^{p} \beta_j \nu_{t-j}, \quad i = 1, \ldots, n \]

where \( S_t = 1 \), if \( \epsilon_t < 0 \) and \( S_t = 0 \), if \( \epsilon_t \geq 0 \).

Following the same procedure as in Section 15.2 we have
\[ \epsilon_t^2 = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \gamma S_{t-j} \Delta_j) \epsilon_{t-j}^2 - \sum_{j=1}^{p} \beta_j \nu_{t-j} + \nu_t \]

where \( \kappa = \max(p, q) \) and we have \( \Delta_j = \alpha_j = 0 \), for \( j > q \), \( \beta_j = 0 \), for \( j > p \), and \( \Delta_j = 1 \) for \( j \leq q \). Taking expectations gives:
\[ E[\epsilon_t^2] = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + E[S_{t-j}] \gamma \Delta_j) E[\epsilon_{t-j}^2] \]
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\[ E[\epsilon_i^2] = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \frac{\gamma_j}{2}\Delta_j)E[\epsilon_{i-j}^2] \]

Since the probability distribution for \( \epsilon_i \) is symmetric about zero we have the probability for \( \epsilon_i < 0 \) is \( 1/2 \) and \( E[S_i] = 1/2 \).

This is an AR(\( \kappa \)) process, and the condition for \( \epsilon_i^2 \) to be covariance stationary is:

\[ \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \frac{\gamma_j}{2}\Delta_j) < 1 \tag{16.18} \]

Assuming that \( \epsilon_i^2 \) is covariance stationary we have \( \sigma_0^2 = E[\epsilon_i^2] = E[\epsilon_{i-j}^2] \) and so

\[ E[\epsilon_i^2] = \alpha_0 + \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \frac{\gamma_j}{2}\Delta_j)E[\epsilon_j^2] \]

We therefore have

GJR–GARCH unconditional variance

\[ \sigma_0^2 = E[\epsilon_i^2] = \alpha_0 \left\{ 1 - \sum_{j=1}^{\kappa} (\alpha_j + \beta_j + \frac{\gamma_j}{2}\Delta_j) \right\}^{-1} \tag{16.19} \]
Chapter 17

GARCH conditional probability distributions

Here we give some useful results concerning various conditional probability distributions that are commonly used in GARCH models. For each distribution we give the following information:

- The probability density function, \( f(\epsilon_i) \).
- The quantity \( \mathcal{L}_i(\theta) \), which is minus the log likelihood (see Chapter 18). Here \( \theta \) is the vector of GARCH model parameters, and the subscript \( i \) indicates the contribution from the \( i \) term in the sequence. The sample log likelihood for the complete \( n \) term GARCH sequence is \( \mathcal{L}(\theta) = -\sum_{i=1}^{n} \mathcal{L}_i(\theta) \). In this section we assume that vector \( \theta \) contains the model parameters for a non-linear regression-GARCH(p,q) process in which the residuals, \( \epsilon_i \), are described by a single asymmetry parameter, \( \gamma \), and a given conditional probability density function. Thus the parameter vector \( \theta \) given here is correct for AGARCH-I, AGARCH-II, and GJR–GARCH processes, but would require extra elements for an EGARCH process. More information concerning the use of \( \mathcal{L}_i(\theta) \) in parameter estimation can be found in Chapter 18.
- The value of \( E[|\epsilon_i|] \), which is used in the EGARCH model.
- The value of the kurtosis, which indicates how thick the tails of the distribution are.

17.1 GAUSSIAN DISTRIBUTION

17.1.1 The probability density function

The probability density function for Gaussian shocks, \( \epsilon_i \), with zero mean and variance \( h_i \) is:

**Gaussian probability density function**

\[
f(\epsilon_i) = \frac{1}{\sqrt{2\pi h_i}} \exp\left(-\frac{\epsilon_i^2}{2h_i}\right)
\]

(17.1)

17.1.2 The kurtosis

The kurtosis for a Gaussian distribution is 3. This can be proved as follows:

\[
E[\epsilon_i^2] = \frac{1}{\sqrt{2\pi h_i}} \int_{-\infty}^{\infty} \epsilon_i^2 \exp\left(-\frac{\epsilon_i^2}{2h_i}\right) d\epsilon_i = \frac{2}{\sqrt{2\pi h_i}} \int_{0}^{\infty} \epsilon_i^2 \exp\left(-\frac{\epsilon_i^2}{2h_i}\right) d\epsilon_i
\]
Using the standard integral results in Appendix K, and the substitution $a = 1/2h_i$ we have:

$$E[\varepsilon_i^2] = \frac{4h_i}{4\sqrt{2\pi h_i}} \sqrt{2\pi h_i} = h_i$$

Similarly

$$E[\varepsilon_i^4] = \frac{1}{\sqrt{2\pi h_i}} \int_{-\infty}^{\infty} \varepsilon_i^4 \exp\left(-\frac{\varepsilon_i^2}{2h_i}\right) d\varepsilon_i = \frac{2}{\sqrt{2\pi h_i}} \int_{0}^{\infty} \varepsilon_i^4 \exp\left(-\frac{\varepsilon_i^2}{2h_i}\right) d\varepsilon_i$$

$$E[\varepsilon_i^4] = \frac{2}{\sqrt{2\pi h_i}} \frac{12h_i^2}{8} \sqrt{2\pi h_i} = 3h_i^2$$

Therefore

Gaussian kurtosis

$$\kappa = \frac{E[\varepsilon_i^4]}{(E[\varepsilon_i^2])^2} = \frac{3h_i^2}{h_i^2} = 3$$  \hspace{1cm} (17.2)

17.1.3 The log likelihood

If we take the logarithm of the probability density function in Section 17.1.1 we obtain the following expression for the log likelihood:

$$L_i(\theta) = \frac{1}{2} \log(2\pi) + \frac{1}{2} \log(h_i) + \frac{e_i^2}{2h_i}$$  \hspace{1cm} (17.3)

or ignoring the constant term:

Gaussian log likelihood

$$L_i(\theta) = \frac{1}{2} \log(h_i) + \frac{e_i^2}{2h_i}$$  \hspace{1cm} (17.4)

where $\theta = (\omega^T, b_0, b^T), \omega^T = (\alpha_0, \alpha_1, \ldots, \alpha_q, \beta_1, \ldots, \beta_p, \gamma)$ and $b^T = (b_1, \ldots, b_k)$.

17.1.4 Calculation of $E[|\varepsilon_i|]$

$$E[|\varepsilon_i|] = \frac{1}{\sqrt{2\pi h_i}} \int_{-\infty}^{\infty} |\varepsilon_i| \exp\left(-\frac{\varepsilon_i^2}{2h_i}\right) d\varepsilon_i$$

$$= \frac{2}{\sqrt{2\pi h_i}} \int_{0}^{\infty} \varepsilon_i \exp\left(-\frac{\varepsilon_i^2}{2h_i}\right) d\varepsilon_i$$
Using the standard integral results given in Appendix K, and on the substitution of
\( y = \epsilon / \sqrt{2h} \) we have:
\[
E[|\epsilon|] = \frac{2h}{\sqrt{2\pi h^2}} = \sqrt{\frac{2h}{\pi}}
\]  
(17.5)

17.2 STUDENT’S t DISTRIBUTION

17.2.1 The probability density function

The probability density function for shocks \( \epsilon \) following a Student’s t distribution with \( \nu \) degrees of freedom, zero mean, and variance \( h \) is (DeGroot, 1970):

\[
f(\epsilon) = \frac{\Gamma((\nu + 1)/2)(\nu - 2)^{-1/2}h^{-1/2}}{\pi^{1/2}\Gamma(\nu/2)} \left[ 1 + \frac{\epsilon^2}{h(\nu - 2)} \right]^{-(\nu+1)/2}, \text{ where } \nu > 2
\]

(17.6)

17.2.2 The kurtosis

The kurtosis is (see Appendix J):

\[
\kappa = \frac{3(\nu - 2)}{(\nu - 4)}, \text{ where } \nu > 4
\]

(17.7)

For convenience we now tabulate the kurtosis, \( \kappa \), for different values of \( \nu \):

<table>
<thead>
<tr>
<th>( \nu )</th>
<th>( \kappa )</th>
</tr>
</thead>
<tbody>
<tr>
<td>4.2</td>
<td>33.000</td>
</tr>
<tr>
<td>5.0</td>
<td>9.000</td>
</tr>
<tr>
<td>10.0</td>
<td>4.000</td>
</tr>
<tr>
<td>20.0</td>
<td>3.375</td>
</tr>
<tr>
<td>50.0</td>
<td>3.1304</td>
</tr>
<tr>
<td>100.0</td>
<td>3.0625</td>
</tr>
</tbody>
</table>

It can be seen that \( \kappa \) is always greater than the kurtosis for a Gaussian distribution. However, for values of \( \nu \) below about 5 the tails are very thick compared to a Gaussian distribution, while when \( \nu \) is above about 20.0 they are almost identical to a Gaussian distribution.
21.11.2003 3:46PM

17.2.3 The log likelihood

The log likelihood is obtained by taking the logarithm of the probability density function given in Section 17.2.1, and is:

\[ L_i(\theta) = -\log(\Gamma((\nu + 1)/2)) + \log(\Gamma(\nu/2)) + \frac{1}{2} \log(\pi) + \frac{1}{2} \log(\nu - 2) \]

\[ + \frac{1}{2} \log(h_i) + \frac{\nu + 1}{2} \log \left(1 + \frac{\epsilon_i^2}{(\nu - 2)h_i}\right) \] \hspace{1cm} (17.8)

or ignoring the constant term:

**Student’s t distribution log likelihood**

\[ L_i(\theta) = -\log(\Gamma((\nu + 1)/2)) + \log(\Gamma(\nu/2)) + \frac{1}{2} \log(\nu - 2) \]

\[ + \frac{1}{2} \log(h_i) + \frac{\nu + 1}{2} \log \left(1 + \frac{\epsilon_i^2}{(\nu - 2)h_i}\right) \] \hspace{1cm} (17.9)

where \( \theta = (\omega^T, \nu, b_0, b^T) \), \( \omega^T = (\alpha_0, \alpha_1, \ldots, \alpha_q, \beta_1, \ldots, \beta_p, \gamma) \), and \( b^T = (b_1, \ldots, b_k) \).

17.2.4 Calculation of \( E[|\epsilon_i|] \)

As previously stated the Student’s t distribution density function is:

\[ f(\epsilon_i) = K \left[ 1 + \frac{\epsilon_i^2}{h_i(\nu - 2)} \right]^{-(\nu + 1)/2} \]

where

\[ K = \frac{\Gamma((\nu + 1)/2)(\nu - 2)^{-1/2}h_i^{-1/2}}{\pi^{1/2}\Gamma(\nu/2)} \]

we have:

\[ E[|\epsilon_i|] = K \int_{-\infty}^{\infty} \left(1 + \frac{\epsilon_i^2}{h_i(\nu - 2)}\right)^{-(\nu + 1)/2} |\epsilon_i|d\epsilon_i \]

\[ = 2K \int_0^{\infty} \frac{\epsilon_i d\epsilon_i}{\left(1 + \frac{\epsilon_i^2}{h_i(\nu - 2)}\right)^{(\nu + 1)/2}} \]

\[ = 2K(h_i(\nu - 2))^{(\nu + 1)/2} \int_0^{\infty} \frac{\epsilon_i d\epsilon_i}{(h_i(\nu - 2) + \epsilon_i^2)^{(\nu + 1)/2}} \]
Using the value of the integral \( \int_0^\infty (e^a d \epsilon_i)/(C \epsilon^b) \) in Appendix K, with \( a = 1, b = 2, c = (\nu + 1)/2 \) and \( m = (\nu - 2)h_i \) we have:

\[
\frac{m^{a+1-bc/b}}{b} = \frac{(h_i(\nu - 2))^{(1-\nu)/2}}{2}, \quad \Gamma\left(\frac{a+1}{b}\right) = \Gamma(1) = 1, \quad \Gamma\left(\frac{c-a+1}{b}\right)
\]

This gives:

\[
E[|\epsilon_i|] = 2K(h_i(\nu - 2))^{(\nu+1)/2} \frac{(h_i(\nu - 2))^{(1-\nu)/2} \Gamma((\nu - 1)/2)}{2 \Gamma((\nu + 1)/2)}
\]

Substituting for \( K \) and cancelling similar terms we obtain:

\[
E[|\epsilon_i|] = \frac{((\nu - 2)h_i)^{1/2} \Gamma((\nu - 1)/2)}{\pi^{1/2} \Gamma(\nu/2)}
\]

Using ((\nu - 1)/2) \Gamma((\nu - 1)/2) = \Gamma((\nu - 1)/2 + 1) = \Gamma((\nu + 1)/2) \) we obtain

\[
E[|\epsilon_i|] = \frac{2((\nu - 2)h_i)^{1/2} \Gamma((\nu + 1)/2)}{\pi^{1/2} \Gamma(\nu/2)(\nu - 1)} \quad (17.10)
\]

Note: This corrects an error in the literature (Taylor, 1994) which, for \( h_i = 1 \), gives the expression as:

\[
E[|\epsilon_i|] = \frac{2((\nu - 2))^{1/2} \Gamma((\nu/2 + 1))}{\pi^{1/2} \Gamma(\nu/2)(\nu - 1)}
\]

17.3 GENERAL ERROR DISTRIBUTION

This distribution is also known as: the exponential power distribution, the error distribution and the generalized error distribution. The distribution is symmetric about the mean, and the kurtosis can be varied by the altering the value of the distribution’s shape parameter.

17.3.1 The probability density function

The general error distribution function, see for example Nelson (1991), is:

\[
f(\epsilon_i) = \frac{a}{\lambda 2^{(1+1/a)\Gamma(1/a)}} \exp\left(-\frac{1}{2} \frac{\epsilon_i}{\lambda}\right)^a \quad (17.11)
\]

where \( \lambda \) is the scale factor, \( a \) is the exponent (or shape parameter), and the distribution has zero mean.
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Sometimes this equation is written in the form:

\[
f(\epsilon_i) = \frac{1}{\lambda^{2(1+1/a)} \Gamma(1+1/a)} \exp \left( -\frac{1}{2} \frac{|\epsilon_i|^a}{\lambda} \right)
\]  

(17.12)

where we have used \(1/a \Gamma(1/a) = \Gamma(1+1/a)\).

Another form, see for example Good (1979) and Tadikamalla (1980), is:

\[
f(\epsilon_i) = \frac{1}{2 \Gamma(1+1/a)} \exp(-|\epsilon_i|^a)
\]  

(17.13)

This is just Equation 17.11 with a scale factor \(\lambda = 1/2^{1/a}\).

If the variance of the distribution is \(h_i\) then we have (see Appendix I.1):

\[
\lambda = \left( \frac{2^{-2/a} \Gamma(1/a) h_i}{\Gamma(3/a)} \right)^{1/2}
\]  

(17.14)

17.3.2 The kurtosis

The kurtosis of the distribution (see Appendix I.2) is:

\[
\kappa = \frac{\Gamma(5/a) \Gamma(1/a)}{\Gamma(3/a) \Gamma(3/a)}
\]  

(17.15)

We will now illustrate how the kurtosis of the distribution changes with the shape parameter, \(a\).

When \(a = 1\), then \(f(\epsilon_i)\) becomes the Laplace distribution (double-sided exponential distribution), since:

\[
f(\epsilon_i) = \frac{1}{2\mu} \exp \left( -\frac{|\epsilon_i|}{\mu} \right)
\]

where \(\mu = 2\lambda\) is the width of the distribution, and \(\lambda = \left\{ \frac{\Gamma(1) h_i}{4\Gamma(3)} \right\}^{1/2} = \frac{1}{2} \sqrt{\frac{h_i}{6}}\).

The kurtosis of a Laplace distribution is 6. This can be verified by using \(\Gamma(n) = (n-1)!\), and substituting \(a = 1\) into Equation 17.15:

\[
\kappa = \frac{\Gamma(5) \Gamma(1)}{\Gamma(3) \Gamma(3)} = \frac{4 \times 3 \times 2 \times 1}{2 \times 2} = 6
\]
When \( a = 2 \), then \( f(\epsilon_i) \) simplifies to the Gaussian distribution:
\[
\lambda = \left\{ \frac{2^{-1/2} \Gamma(1/2) h_i}{\Gamma(3/2)} \right\}^{1/2} = \sqrt{h_i},
\]
using \( \Gamma(3/2) = (1/2) \Gamma(1/2) \) and since \( \Gamma(1/2) = \sqrt{\pi} \) we have:
\[
f(\epsilon_i) = \frac{2h_i^{-1/2}}{2^{3/2} \Gamma(1/2)} \exp \left( -\frac{\epsilon_i^2}{2h_i} \right) = \frac{1}{\sqrt{2\pi h_i}} \exp \left( -\frac{\epsilon_i^2}{2h_i} \right)
\]

The kurtosis of a Gaussian distribution is 3. This can easily be verified by using \( \Gamma(3/2) = \sqrt{\pi}/2, \Gamma(5/2) = 3\sqrt{\pi}/4 \), and substituting \( a = 2 \) into Equation 17.15:
\[
\kappa = \frac{\Gamma(5/2) \Gamma(1/2)}{\Gamma(3/2) \Gamma(3/2)} = \frac{3}{4 \sqrt{\pi}} \frac{\sqrt{\pi}}{1/2 \sqrt{\pi}} = 3
\]

When \( a \to \infty \), then we have (Nelson, 1991 and Appendix I.3):
\[
f(\epsilon_i) \to \mathcal{U}(-h_i^{-1/2}, h_i^{-1/2})
\]
where \( \mathcal{U}(a, b) \) is a uniform distribution with lower and upper limits \( a \) and \( b \) respectively, and a kurtosis of 9/5.

In summary then, when \( a < 2 \), the distribution is leptokurtic (has tails that are thicker than those for a Gaussian), and when \( a > 2 \) the distribution is platykurtic (has tails that are thinner than those for a Gaussian).

### 17.3.3 The log likelihood

The log likelihood is obtained by taking the logarithm of the probability density function given in Section 17.3.1, and is:
\[
\mathcal{L}_i(\theta) = -\log(a) + \log(\lambda) + (1 + 1/a) \log(2) + \log(\Gamma(1/a)) + \frac{1}{2} \frac{|\epsilon_i|^a}{a}
\]

or ignoring the constant term, \( \log(2) \), we have:

**General error distribution log likelihood**
\[
\mathcal{L}_i(\theta) = -\log(a) + \log(\lambda) + \frac{1}{a} \log(2) + \log(\Gamma(1/a)) + \frac{1}{2} \frac{|\epsilon_i|^a}{a}
\]  \tag{17.16}

where \( \theta = (\omega^T, a, \lambda, b_0, b^T) \), \( \omega^T = (\alpha_0, \alpha_1, \ldots, \alpha_q, \beta_1, \ldots, \beta_p, \gamma) \), and \( b^T = (b_1, \ldots, b_k) \).

### 17.3.4 Calculation of \( E[|\epsilon_i|] \)

\[
E[|\epsilon_i|] = K \int_{-\infty}^{\infty} |\epsilon_i| \exp \left( -\frac{1}{2} \frac{|\epsilon_i|^a}{a} \right) d\epsilon_i = 2K \int_{0}^{\infty} \epsilon_i \exp \left( -\frac{1}{2} \left( \frac{\epsilon_i}{\lambda} \right)^a \right) d\epsilon_i
\]

Using the standard integral results in Appendix K with \( n = 1, p = a, \) and \( b = (1/2)(1/\lambda)^a \) gives:
\[
E[|\epsilon_i|] = \frac{2K}{a} \Gamma \left( \frac{2}{a} \right) \left\{ \frac{1}{2} \left( \frac{1}{\lambda} \right)^a \right\}^{-2/a}
\]
After some simplification this yields:

\[ E[|\varepsilon_i|] = \frac{2\lambda^2 K}{a} \Gamma\left(\frac{2}{a}\right) \left(\frac{1}{2}\right)^{-2/a} \]

and substituting for \( K \) we then have:

\[ E[|\varepsilon_i|] = \frac{\Gamma(2/a)\lambda^{1/a}}{\Gamma(1/a)} \quad (17.17) \]
Chapter 18

Maximum likelihood parameter estimation

In this chapter we will discuss how the model parameter vector $\theta$ for a GARCH sequence can be estimated. For a standard linear GARCH(p,q) with regression terms we have $\theta = (\omega^T, b^T)$, where $\omega^T = (\alpha_0, \alpha_1, \ldots, \alpha_q, \beta_1, \ldots, \beta_p)$ and $b^T = (b_1, \ldots, b_k)$.

### 18.1 The Conditional Log Likelihood

Assume we have a standard linear GARCH(p,q) sequence of length $n$, in which the observations $y_i$, $i = 1, \ldots, n$ are given by:

$$y_i = b_0 + X^T_i b + \epsilon_i | \psi_{i-1} \sim \mathcal{N}(0, h_i)$$  \hspace{1cm} (18.1)

$$h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n$$  \hspace{1cm} (18.2)

The residuals, $\epsilon_i$, are independently distributed according to the arbitrary probability distribution $\mathcal{N}(0, h_i)$, which has zero mean and time-dependent variance $h_i$.

The notation $\psi_{i-1}$ has been used to denote information content up to and including time instant $i - 1$, that will affect the conditional distribution of $\epsilon_i$. In this case $\psi_{i-1}$ represents the information that affects the variance $h_i$ of $\mathcal{N}(0, h_i)$. The syntax $\epsilon_i | \psi_{i-1}$ is used to indicate that the PDF of the residual $\epsilon_i$ is conditional on $\psi_{i-1}$. For the GARCH models considered here it is only the variance $h_i$ of the PDF for $\epsilon_i$ that is affected by the information $\psi_{i-1}$. Also, since $\epsilon_i$ is independently distributed to $\epsilon_{i-1}$ we have that $E(\epsilon_i^2 | \psi_{i-1}) = 0$ and $E(\epsilon_i^2 | \psi_{i-1} \epsilon_{i-1}^2 | \psi_{i-2}) = E(\epsilon_i^2 | \psi_{i-1}) E(\epsilon_{i-1}^2 | \psi_{i-2})$.

The joint density distribution for a sample of independently distributed variables can be obtained by taking the product of the individual probability densities.

This means that the joint probability density distribution of the first two residuals in a GARCH sequence is:

$$f(\epsilon_2, \epsilon_1; \theta) = f(\epsilon_2 | \psi_1; \theta) f(\epsilon_1 | \psi_0; \theta)$$

where we have used the notation $f(\epsilon_2 | \psi_1; \theta)$ to indicate that the distribution of $\epsilon_2$ is conditional on $\psi_1$ and depends on the parameter vector $\theta$.

Similarly the joint probability density distribution of the first three residuals in a GARCH sequence is:

$$f(\epsilon_3, \epsilon_2, \epsilon_1; \theta) = f(\epsilon_3 | \psi_2; \theta) f(\epsilon_2 | \psi_1; \theta) f(\epsilon_1 | \psi_0; \theta)$$
Continuing this process for all the residuals in the sequence yields the sample joint probability density function, \( F(\theta) \), for the residuals of the complete series:

\[
F(\theta) = f(\epsilon_n, \ldots, \epsilon_1; \theta) = \prod_{i=1}^{n} f(\epsilon_i|\psi_{i-1}; \theta)
\]

Taking natural logarithms we obtain:

\[
\log(F(\theta)) = \sum_{i=1}^{n} \log(f(\epsilon_i|\psi_{i-1}; \theta))
\]

If Equation 18.2 is conditioned using known pre-observed values \( \epsilon_i, \epsilon_i^2, h_i, i \leq 0 \), (see Section 20.1 for more details) then we can use the parameter vector \( \theta \) to iteratively evaluate the time dependent variance \( h_1, \ldots, h_n \) and also determine the information content \( \psi_1, \ldots, \psi_{n-1} \). This means that we can substitute \( \epsilon_i = y_i - b_0 - X_i^T b \) into the PDF for \( R(0, h_i) \) and thus obtain the probabilities \( f(\epsilon_i|\psi_{i-1}; \theta) \).

We can then evaluate the sample log likelihood, \( \mathcal{L}(\theta) \), using:

\[
-\log(F(\theta)) = \mathcal{L}(\theta) = \sum_{i=1}^{n} \mathcal{L}_i(\theta)
\]

where \( \mathcal{L}_i(\theta) = -\log(f(\epsilon_i|\psi_{i-1}; \theta)) \), see Chapter 17.

The maximum likelihood estimator, \( \hat{\theta} \), for the parameter vector \( \theta \) is that which minimises \( \mathcal{L}(\theta) \) (see Section 18.2) and is the solution to the likelihood equations:

\[
\frac{\partial \mathcal{L}(\theta)}{\partial \theta} = 0
\]

At the minimum the Hessian \( \frac{\partial^2 \mathcal{L}(\theta)}{\partial \theta \theta^T} \) is a positive definite matrix. However, care needs to be exercised since this does not guarantee that a global minimum rather than a local minimum has been reached.

18.2 THE COVARIANCE MATRIX OF THE PARAMETER ESTIMATES

In this section we will show how the covariance matrix of the maximum likelihood parameter estimates are related to the Hessian of the log likelihood function. For convenience we have adopted the \( D \) operator convention:

\[
D \mathcal{L}(\theta) = \frac{\partial \mathcal{L}(\theta)}{\partial \theta} \quad \text{and} \quad D^2 \mathcal{L}(\theta) = \frac{\partial^2 \mathcal{L}(\theta)}{\partial \theta^2}
\]

We will assume that the log likelihood is locally well behaved about its minimum and also that the minimum is far enough away from any boundaries that have been imposed during the optimization process. If \( \theta_0 \) is the true value for the model parameter vector \( \theta \) and \( \hat{\theta} \) is the maximum likelihood estimator for \( \theta \) then we can use a Taylor expansion for the value of the log likelihood about the true value as follows:

\[
\mathcal{L}(\hat{\theta}) \approx \mathcal{L}(\theta_0) + (\hat{\theta} - \theta_0) D \mathcal{L}(\theta_0) + \frac{(\hat{\theta} - \theta_0)^2}{2} D^2 \mathcal{L}(\theta_0)
\]
Maximum likelihood parameter estimation

Where $\mathcal{D}\mathcal{L}(\theta_0)$ is the gradient evaluated at $\theta_0$ and $\mathcal{D}^2\mathcal{L}(\theta_0)$ is the Hessian evaluated at $\theta_0$. We can also expand the gradient $\mathcal{D}\mathcal{L}(\theta)$ about the true value $\theta_0$ as:

$$\mathcal{D}\mathcal{L}(\theta) \approx \mathcal{D}\mathcal{L}(\theta_0) + (\theta - \theta_0)\mathcal{D}^2\mathcal{L}(\theta_0)$$

However, at a minimum (which is a solution of the likelihood equations in Section 18.1) we must have $\mathcal{D}\mathcal{L}(\hat{\theta}) = 0$. This gives:

$$(\hat{\theta} - \theta_0)\mathcal{D}^2\mathcal{L}(\theta_0) = -\mathcal{D}\mathcal{L}(\theta_0)$$

and the estimation error of $(\hat{\theta} - \theta_0)$ is:

$$(\hat{\theta} - \theta_0) \approx -\frac{\mathcal{D}\mathcal{L}(\theta_0)}{\mathcal{D}^2\mathcal{L}(\theta_0)}$$

(18.4)

We will now assume that $\theta$ is a scalar and show how the variance of $(\hat{\theta} - \theta_0)$ is related to $\mathcal{D}^2\mathcal{L}(\theta_0)$.

For a sample of $n$ observations we must, by definition, have:

$$\int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} F d\epsilon_1, \ldots, d\epsilon_n = 1$$

where for convenience the sample joint probability density function $F(\theta)$ from Section 18.1 has been denoted by $F$.

Differentiating w.r.t $\theta$ we have:

$$\frac{\partial}{\partial \theta} \int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} F d\epsilon_1, \ldots, d\epsilon_n = \int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \frac{\partial F}{\partial \theta} d\epsilon_1, \ldots, d\epsilon_n = 0$$

Now since $(\partial \log(F))/(\partial \theta) = (\partial \log(F))/(\partial F)(\partial F/\partial \theta) = (1/F)(\partial F/\partial \theta)$ we have:

$$\int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \left( \frac{1}{F} \frac{\partial F}{\partial \theta} \right) F d\epsilon_1, \ldots, d\epsilon_n = \int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \left( \frac{\partial \log(F)}{\partial \theta} \right) F d\epsilon_1, \ldots, d\epsilon_n$$

so

$$E \left[ \frac{\partial \log(F)}{\partial \theta} \right] = \int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \left( \frac{\partial \log(F)}{\partial \theta} \right) F d\epsilon_1, \ldots, d\epsilon_n = 0$$

Differentiating again w.r.t $\theta$ we have:

$$\frac{\partial}{\partial \theta} \int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \left( \frac{1}{F} \frac{\partial F}{\partial \theta} \right) F d\epsilon_1, \ldots, d\epsilon_n = \int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \left\{ F \frac{\partial}{\partial \theta} \left( \frac{1}{F} \frac{\partial F}{\partial \theta} \right) \right\} d\epsilon_1, \ldots, d\epsilon_n + \left( \frac{1}{F} \frac{\partial F}{\partial \theta} \right) (1/F) = 0$$

But $(\partial^2 \log(F))/(\partial \theta^2) = (\partial/\partial \theta)((1/F)(\partial F/\partial \theta))$ so we have:

$$\int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \left\{ \frac{\partial^2 \log(F)}{\partial \theta^2} + \left( \frac{1}{F} \frac{\partial F}{\partial \theta} \right)^2 \right\} F d\epsilon_1, \ldots, d\epsilon_n = 0$$

which gives

$$\int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \left\{ \frac{\partial^2 \log(F)}{\partial \theta^2} + \left( \frac{\partial \log(F)}{\partial \theta} \right)^2 \right\} F d\epsilon_1, \ldots, d\epsilon_n = 0$$
So we have:

\[ \int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \left( \frac{\partial \log(F)}{\partial \theta} \right)^2 F d\epsilon_1, \ldots, d\epsilon_n = - \int_{-\infty}^{\infty} \ldots \int_{-\infty}^{\infty} \left( \frac{\partial^2 \log(F)}{\partial \theta^2} \right) F d\epsilon_1, \ldots, d\epsilon_n \]

which using Equation 18.3 gives:

\[ E \left[ \sum_{i=1}^{n} \left( \frac{\partial \log(f(\epsilon_i|\psi_{i-1}; \theta))}{\partial \theta} \right)^2 \right] = -E \left[ \sum_{i=1}^{n} \frac{\partial^2 \log(f(\epsilon_i|\psi_{i-1}; \theta))}{\partial \theta^2} \right] = \mathcal{F}_\theta \]

This can be restated as:

\[ \frac{1}{n} \left\{ \frac{\partial \mathcal{L}(\theta)}{\partial \theta} \right\}^2 = \frac{1}{n} \frac{\partial^2 \mathcal{L}(\theta)}{\partial \theta^2} = \mathcal{F}_\theta \quad \text{or equivalently:} \quad \mathcal{D}^2 \mathcal{L}(\theta) = n\mathcal{F}_\theta \quad (18.5) \]

where \( \mathcal{F}_\theta \) is the average variance of the independent random variables

\[ \frac{\partial \log(f(\epsilon_i|\psi_{i-1}; \theta))}{\partial \theta} , \quad i = 1, \ldots, n \]

If we denote the variance of the \( i \)th variable by \( \sigma_i^2 \) and the sum of these variables by \( \sigma_n^2 \) then:

\[ \sigma_n^2 = \left\{ \frac{\log(f(\epsilon_i|\psi_{i-1}; \theta))}{\partial \theta} \right\}^2 \quad \text{and} \quad \sigma_n^2 = \sum_{i=1}^{n} \sigma_i^2 = n\mathcal{F}_\theta \]

For convenience we will also use:

\[ S_n = \sum_{i=1}^{n} \frac{\log(f(\epsilon_i|\psi_{i-1}; \theta))}{\partial \theta} = -\mathcal{D}\mathcal{L}(\theta) \]

Now the generalized central limit theorem (Feller, 1971) states that as \( n \to \infty \) the variable \( S_n/\sigma_n \) becomes distributed as \( N(0, 1) \). So at \( \theta = \theta_0 \) we have:

\[ \eta = \frac{\mathcal{D}\mathcal{L}(\theta_0)}{(n\mathcal{F}_{\theta_0})^{1/2}} \quad (18.6) \]

where \( \eta \sim N(0, 1) \)

However, from Equations 18.4 and 18.5, we have:

\[ \hat{\theta} - \theta_0 = -\frac{\mathcal{D}\mathcal{L}(\theta_0)}{\mathcal{D}^2\mathcal{L}(\theta_0)/n\mathcal{F}_{\theta_0}} \quad (18.7) \]

So using Equation 18.6 to substitute for \( \mathcal{D}\mathcal{L}(\theta_0) \) in Equation 18.7 we obtain:

\[ \hat{\theta} - \theta_0 = \frac{\eta}{(n\mathcal{F}_{\theta_0})^{1/2}} \quad (18.8) \]

This means that: \( \eta' \sim N(0, n^{-1}\mathcal{F}_{\theta_0}^{-1}) \)
Maximum likelihood parameter estimation

where \( \eta' = \hat{\theta} - \theta_0 \). The maximum likelihood estimate \( \hat{\theta} \) is therefore distributed about the true value \( \theta_0 \) as:

\[
\hat{\theta} = N(\theta_0, n^{-1} \mathcal{F}_{\theta_0}^{-1})
\]

(18.9)

The value \( \mathcal{F}_{\theta_0} \) was called by Fisher (1925) the information about \( \theta_0 \), see Silvey (1975), and Cox and Hinkley (1979). The justification for this is simply that when there is more Fisher information the variance of the estimate \( \hat{\theta} \) will be lower and therefore the maximum likelihood estimate will improve.

We have just considered the estimation of a single parameter \( \theta \) and thus \( \mathcal{F}_{\theta_0} \) is a scalar. In the more general case \( \theta \) is a vector of \( N_p \) model parameters and the \( N_p \times N_p \) matrix \( \mathcal{F}_{\theta_0} \) is termed the Fisher information matrix. Under these circumstances Equation 18.5 then becomes:

\[
E \left[ \frac{\partial^2 \mathcal{L}(\theta)}{\partial \theta \partial \theta^T} \right] = \mathcal{F}_{\theta}
\]

and in Equation 18.9 \( n^{-1} \mathcal{F}_{\theta_0}^{-1} \) is the inverse of an \( N_p \times N_p \) matrix which yields the covariance matrix, \( \mathcal{C} \), of the estimated parameter vector \( \theta \).

At first sight the preceding discussion seems to have provided us with a very useful result. There is however a major problem. We don’t know the true parameter vector \( \theta_0 \), and so we can’t evaluate \( \mathcal{F}_{\theta_0} \). Indeed if we did know the value of \( \theta_0 \) it would be rather pointless computing \( \hat{\theta} \).

The only way forward is to use some kind of approximation to \( \mathcal{F}_{\theta_0} \). The most obvious is to evaluate \( \mathcal{F}_{\theta_0} \) at \( \theta = \hat{\theta} \) and then use \( \mathcal{F}_{\theta_0} \approx \mathcal{F}_{\hat{\theta}} \).

We can now rewrite Equation 18.9 in the following usable form:

\[
\hat{\theta} = N(\theta_0, n^{-1} \mathcal{F}_{\hat{\theta}}^{-1}), \quad \text{where } n^{-1} \mathcal{F}_{\hat{\theta}}^{-1} = \left[ \frac{\partial^2 \mathcal{L}(\theta)}{\partial \theta \partial \theta^T} \right]^{-1}_{\theta = \hat{\theta}}
\]

(18.11)

In the next section we will discuss numerical optimization and show how \( \mathcal{F}_{\hat{\theta}}^{-1} \) occurs naturally in the equations that are used to maximize the log likelihood.

18.2.1 The standard errors and significance

The variance of each estimated parameter is contained in the corresponding diagonal element of the covariance matrix \( \mathcal{C} \). So for a model with \( N_p \) parameters the standard errors of the estimated parameters are:

\[
\sigma_i = \sqrt{\mathcal{C}_{i,i}}, \quad i = 1, \ldots, N_p
\]

where \( \mathcal{C}_{i,i} \) is used to denote the \( i \)th diagonal element of the covariance matrix. The standardized parameter estimate, \( t \) statistic, of the estimated value is given by the estimated value divided by the estimated standard error. So for the \( i \)th estimated parameter we have a \( t \) statistic of \( t_i = \hat{\theta}_i / \sigma_i \).

We can use the value of \( t_i \) to provide evidence against the null hypothesis, \( H_0 \), that the actual parameter value is zero. That is \( H_0 \) assumes that the distribution of the \( i \)th standardized parameter estimate is \( N(0, 1) \).
To illustrate how $t_i$ can be used we will now use the following data for a standardized Gaussian distribution:

$$Pr(t_i \geq 0.52) = 0.3, \quad Pr(t_i \geq 1.64) = 0.05$$
$$Pr(t_i \geq 1.96) = 0.025, \quad Pr(t_i \geq 2.57) = 0.005$$

where $Pr(t_i \geq X)$ is the probability that the value of $t_i$ will be greater or equal to $X$.

For instance if the estimated value of $t_i$ is 2.57 then, the probability of obtaining this value or greater from $H_0$ is only 0.5 per cent. Under these conditions we should reject the null hypothesis, and the estimated parameter value $\theta_i$ is then said to be significant at the 0.5 per cent level.

If however the estimated value of $t_i$ is only 0.52 then, the probability of obtaining this value or greater from $H_0$ is 30 per cent (which is quite high). We therefore cannot reject the null hypothesis that the value of $\theta_i$ is zero. The estimated value of $\theta_i$ is then said to be not significant.

### 18.3 NUMERICAL OPTIMIZATION

The GARCH model parameters $\theta$ can be estimated by using numerical optimization to maximize the conditional log likelihood, or equivalently the value of $\theta$ which minimizes minus the log likelihood. From now on we will denote minus the log likelihood by $-L(\theta)$, and for simplicity refer to this quantity as the loglikelihood, see Section 18.1.

Most optimization procedures use gradient information (either analytic or numeric) in order to iterate to a global maximum (or minimum).

In most gradient algorithms the $k$th iteration used to minimize $L(\theta)$ takes the form:

$$\hat{\theta}^k = \hat{\theta}^{k-1} - \lambda H^{-1}DL(\hat{\theta}^{k-1})$$  \hspace{1cm} (18.12)

where $\hat{\theta}^{k-1}$ is the estimate of the parameter vector obtained after $k-1$ iterations, $H$ is some approximation to the Hessian computed at $\hat{\theta}^{k-1}$, which determines the direction of the $k$th step, $\lambda$ is a scalar which specifies the step size in the given direction and $DL(\hat{\theta}^{k-1})$ the gradient is computed at $\hat{\theta}^{k-1}$.

Some commonly used approximations to $H$ are as follows:

- The actual Hessian $\partial^2 L(\theta)/\partial \theta \partial \theta^T$.
- The conditional expectation of the Hessian.
- A positive definite matrix that is an approximation to the Hessian.
- The outer product $(\partial L(\theta)/\partial \theta)(\partial L(\theta)/\partial \theta^T)$.

When the Hessian is approximated by the outer product the method is known as the BHHH algorithm, see Berndt et al. (1974).

We note that when $\lambda = 1$ and $H$ is the actual Hessian $\partial^2 L(\theta)/\partial \theta \partial \theta^T$ then the optimization algorithm is called Newton–Raphson or simply Newton.

In maximum likelihood estimation it is often convenient to approximate the Hessian by $n \mathcal{F}_\theta$, where $\mathcal{F}_\theta$ is the Fisher information matrix. When this is done we have the method of scoring, and Equation 18.12 then becomes:

$$\hat{\theta}^k = \hat{\theta}^{k-1} - \lambda n^{-1}\mathcal{F}_\theta^{-1}DL(\hat{\theta}^{k-1})$$  \hspace{1cm} (18.13)
Maximum likelihood parameter estimation

This technique is likely to have a lower convergence rate than a straightforward Newton method because the information matrix is only an approximation to the Hessian. However, in many instances the information matrix has a simple form and is much easier to compute than the complete Hessian. Also the information matrix will always be positive definite and so its inverse can be computed, this is not necessarily the case for the actual Hessian.

Quasi-Newton methods do not require the Hessian to be explicitly evaluated (Gill et al., 1981; Murtagh and Saunders, 1983). The iterative scheme is of the form of Equation 18.12 and the matrix $H$ must be a positive definite. At each iteration $H$ is updated in such a way as to yield a series of positive definite matrices which eventually converge to the inverse of the Hessian. The initial $H$ matrix can be any positive definite matrix, and a common choice is the identity matrix.

In Chapter 21 results are presented which show the relative advantages/disadvantages of using numeric/analytic gradients during maximum likelihood optimization. These results are from GARCH software which used a general purpose quasi-Newton nonlinear optimization routine. First derivatives could be supplied either in analytic form or computed numerically by finite-difference techniques. The optimization process relied on a Hessian which was always computed internally by the nonlinear optimizer. However, it was possible to retrieve the Hessian at the solution point $\hat{\theta}$, and thus use it as an approximation to the Fisher information matrix. GARCH stationary conditions could be ensured by imposing the linear constraint $\sum_{j=1}^{q} \gamma_j + \sum_{j=1}^{p} \beta_j < 1$ during the numerical optimization.

In Chapter 21 the following approximations to the Fisher information matrix were used:

- The second-derivative estimate, based on the actual value of the Hessian at the solution point $\hat{\theta}$, that is $(\partial^2 \mathcal{L}(\theta)) / (\partial \theta \partial \theta^T)_{\hat{\theta}}$. This is calculated numerically using finite differences.
- The second-derivative estimate, based on the conditional expectation of the Hessian at the solution point $\hat{\theta}$, that is $E((\partial^2 \mathcal{L}(\theta)) / (\partial \theta \partial \theta^T))_{\hat{\theta}}$.

The difficulty of modelling a GARCH(p,q) sequence depends on both p and q and also on how much volatility memory there is in the process. Higher values of the parameters $\beta_j, j = 1, \ldots, p$, give rise to more volatility memory and are therefore harder to model accurately. Increasing the number of model parameters will also make the model more difficult to model simply because there are more variables to numerically optimize. This suggests the following order of difficulty ARCH(1), ARCH(2), ARCH(3), GARCH(1,1), GARCH(1,2), GARCH(2,2), etc.

In Chapter 19 information is given on how to compute the analytic gradients for a regression GJR–GARCH(p,q) sequence. Chapter 20 elaborates on the information in Chapter 19, and provides complete pseudocode that enables the reader to write computer programs to calculate both the conditional log likelihood and its gradients.
18.4 SCALING THE DATA

Numerical optimization procedures can have difficulty in minimizing a function in which the magnitudes of the individual variables differ by a large factor (say $10^6$ or greater). This can occur in GARCH($p,q$) processes where the parameters $\alpha_i, i = 1, \ldots, p, \beta_i, i = 1, \ldots, q$ are usually in the range 0.1 to 1, but the parameter $\alpha_0$ can be very small. In these circumstances scaling the observations, $y_i, i = 1, \ldots, n$, by $\lambda$ will result in a time series in which $\alpha_0$ is multiplied by the factor $\lambda^2$. For instance if $\alpha_0$ is $10^{-6}$ in the original sequence, then scaling the data by 100 gives a new series with $\alpha_0 = 10^{-2}$.

Here we will consider data scaling for both linear and nonlinear GARCH models, and show how the model parameters for the scaled data are related to those of the original data.

18.4.1 Scaling a linear GARCH process

Here we consider the effect of scaling the GARCH process:

$$ y_i = X_i^T b + b_0 + \epsilon_i, \quad \epsilon_i | \psi_{i-1} \sim \mathcal{N}(0, h_i) $$

$$ h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n $$

If the observations $y_i$ are scaled by the factor $\lambda$ then we have the new GARCH process:

$$ Y_i = X_i^T B + B_0 + E_i, \quad E_i | \psi_{i-1} \sim \mathcal{N}(0, H_i) $$

$$ H_i = L_0 + \sum_{j=1}^{q} \alpha_j E_{i-j}^2 + \sum_{j=1}^{p} \beta_j H_{i-j}, \quad i = 1, \ldots, n $$

where $Y_i = \lambda y_i, L_0 = \lambda^2 \alpha_0, B = \lambda b, B_0 = \lambda b_0, E_i = \lambda \epsilon_i,$ and $H_i = \lambda^2 h_i$.

The GARCH model parameter vector, $\theta$, of the scaled process is:

$$ \theta = (L_0, \alpha_i, i = 1, \ldots, q, \beta_i, i = 1, \ldots, p, B, B_0) $$

18.4.2 Scaling an AGARCH-I process

Referring to the AGARCH-I process specification in Chapter 16, and proceeding in a similar manner to Section 18.3.1, we have:

$$ Y_i = X_i^T B + B_0 + E_i, \quad E_i | \psi_{i-1} \sim \mathcal{N}(0, H_i) $$

$$ H_i = L_0 + \sum_{j=1}^{q} \alpha_j (E_{i-j} + G)^2 + \sum_{j=1}^{p} \beta_j H_{i-j}, \quad i = 1, \ldots, n $$

where $\lambda$ is the scale factor and $Y_i = \lambda y_i, H_i = \lambda^2 h_i, L_0 = \lambda^2 \alpha_0, E_i = \lambda \epsilon_i, G = \lambda \gamma, B = \lambda b,$ and $B_0 = \lambda b_0$.

The GARCH model parameter vector, $\theta$, of the scaled process is then:

$$ \theta = (L_0, \alpha_i, i = 1, \ldots, q, \beta_i, i = 1, \ldots, p, G, B, B_0) $$
18.4.3 Scaling an AGARCH-II process

Referring to the AGARCH-II process specification in Chapter 16, and proceeding in a similar manner to Section 18.3.1, we have:

\[ Y_i = X_i^T B_0 + E_i, \quad E_i|\psi_{i-1} \sim \mathcal{N}(0, H_i) \]

\[ H_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j (|E_{i-j}| + \gamma E_{i-j})^2 + \sum_{j=1}^{p} \beta_j H_{i-j}, \quad i = 1, \ldots, n \]

where \( \lambda \) is the scale factor, and \( L_0 = \lambda^2 \alpha_0, H_i = \lambda^2 h_i, E_i = \lambda \epsilon_i, B = \lambda b, \) and \( B_0 = \lambda b_0. \)

The GARCH model parameter vector, \( \theta \), of the scaled process is then:

\[ L_0, \alpha_i, i = 1, \ldots, q, \beta_i, i = 1, \ldots, p, \gamma, B \text{ and } B_0. \]

18.4.4 Scaling a GJR–GARCH process

Referring to the GJR–GARCH process specification in Chapter 16, and proceeding in a similar manner to Section 18.3.1, we have:

\[ Y_i = X_i^T B_0 + E_i, \quad E_i|\psi_{i-1} \sim \mathcal{N}(0, H_i) \]

\[ H_i = L_0 + \sum_{j=1}^{q} (\alpha_j + \gamma S_{i-j}) E_{i-j}^2 + \sum_{j=1}^{p} \beta_j H_{i-j}, \quad i = 1, \ldots, n \]

where \( \lambda \) is the scale factor and \( S_i = 1, \) if \( E_i < 0, \) and \( S_i = 0, \) if \( E_i \geq 0. \) The scaled parameters are now: \( L_0 = \lambda^2 \alpha_0, H_i = \lambda^2 h_i, E_i = \lambda \epsilon_i, B = \lambda b, \) and \( B_0 = \lambda b_0. \)

The GARCH model parameter vector, \( \theta \), of the scaled process is then: \( L_0, \alpha_i, i = 1, \ldots, q, \beta_i, i = 1, \ldots, p, \gamma, B \text{ and } B_0. \)
Chapter 19

Analytic derivatives of the log likelihood

In this chapter we show how to calculate analytic expressions for the first and second order partial derivatives of the log likelihood function. As previously mentioned in Section 18.3 these partial derivatives are used by Newton type numerical optimizers to minimize the log likelihood and thus obtain an estimate for the GARCH model parameter vector, \( \theta \). The analytic second derivative is used to as an approximation to the Fisher information matrix, and as a means of calculating the standard errors.

Information on how to compute the analytic derivatives of a standard regression-GARCH\((p,q)\) process with Gaussian residuals is available in the literature, Fiorentini et al. (1996).

In Section 19.1 we show how to compute the first derivatives of a regression-GARCH\((p,q)\) process which has either Gaussian distributed residuals or Student’s \( t \) distributed residuals.

In Section 19.2 we show how to compute the conditional expectation of the Hessian for a regression-GARCH\((p,q)\) process with Gaussian distributed residuals. This is used as an approximation for the Fisher information matrix.

The results of this section will be used in Chapter 20 to derive computational algorithms which compute the derivatives of a regression-GJR–GARCH model. The results are also used in Appendix H to compute the derivatives of a regression-AGARCH-I model.

19.1 THE FIRST DERIVATIVES

19.1.1 Gaussian distribution

Here we obtain expressions for the partial derivatives of the Gaussian log likelihood, Equation 17.4.

Partial derivatives w.r.t. the parameter vector \( \omega \):

\[
\frac{\partial L_i(\theta)}{\partial \omega} = 1 \frac{\partial (\log h_i)}{\partial \omega} \frac{\partial h_i}{\partial \omega} + \frac{\epsilon_i^2}{2} \frac{\partial (1/h_i)}{\partial \omega} \frac{\partial h_i}{\partial \omega}
\]

\[
\frac{\partial \epsilon_i^2(\theta)}{\partial \omega} = \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} - \frac{\epsilon_i^2}{2h_i^2} \frac{\partial h_i}{\partial \omega}
\]

(19.1)
Analytic derivatives of the log likelihood

Partial derivative w.r.t. the parameter $b_0$:  

$$
\frac{\partial L_0(\theta)}{\partial b_0} = \frac{1}{2} \frac{\partial \log(h_i)}{\partial b_0} + \frac{1}{2} \frac{\partial \epsilon_i^2}{\partial b_0} \frac{1}{h_i} + \frac{\epsilon_i^2}{2} \frac{\partial (1/h_i)}{\partial b_0} 
$$

$$
= \frac{1}{2h_i} \frac{\partial h_i}{\partial b_0} + \frac{1}{2h_i} \frac{\partial \epsilon_i^2}{\partial b_0} - \frac{\epsilon_i^2}{2h_i^2} \frac{\partial h_i}{\partial b_0}
$$

(19.2)

But since $\epsilon_i = y_i - X_i^T b - b_0$ we obtain:

$$
\frac{\partial \epsilon_i^2}{\partial b_0} = 2\epsilon_i \frac{\partial \epsilon_i}{\partial b_0} = 2\epsilon_i \frac{\partial (y_i - X_i^T b - b_0)}{\partial b_0} = -2\epsilon_i
$$

$$
\frac{\partial L_0(\theta)}{\partial b_0} = \frac{1}{2h_i} \frac{\partial h_i}{\partial b_0} - \frac{\epsilon_i}{h_i} \left(\frac{\epsilon_i^2}{h_i^2} - 1\right)
$$

(19.3)

Similarly we obtain the partial derivative w.r.t. the parameter vector $b$:  

$$
\frac{\partial L_i(\theta)}{\partial b} = \frac{1}{2} \frac{\partial \log(h_i)}{\partial b} + \frac{1}{2} \frac{\partial \epsilon_i^2}{\partial b} \frac{1}{h_i} + \frac{\epsilon_i^2}{2} \frac{\partial (1/h_i)}{\partial b} 
$$

$$
= \frac{1}{2h_i} \frac{\partial h_i}{\partial b} + \frac{1}{2h_i} \frac{\partial \epsilon_i^2}{\partial b} - \frac{\epsilon_i^2}{2h_i^2} \frac{\partial h_i}{\partial b}
$$

(19.4)

Since

$$
\frac{\partial \epsilon_i^2}{\partial b} = 2\epsilon_i \frac{\partial \epsilon_i}{\partial b} = 2\epsilon_i \frac{\partial (y_i - X_i^T b - b_0)}{\partial b} = -2\epsilon_i X_i
$$

$$
\frac{\partial L_i(\theta)}{\partial b} = \frac{1}{2h_i} \frac{\partial h_i}{\partial b} - \frac{\epsilon_i X_i}{h_i} \left(\frac{\epsilon_i^2}{h_i^2} - 1\right)
$$

(19.5)

In summary we have:

**Gaussian log likelihood partial derivatives**

$$
\frac{\partial L(\theta)}{\partial \omega} = -\sum_{i=1}^{n} \left[ \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} \left(\frac{\epsilon_i^2}{h_i} - 1\right) \right]
$$

(19.6)

$$
\frac{\partial L(\theta)}{\partial b_0} = -\sum_{i=1}^{n} \left[ \frac{\epsilon_i}{h_i} + \frac{1}{2h_i} \frac{\partial h_i}{\partial b_0} \left(\frac{\epsilon_i^2}{h_i} - 1\right) \right]
$$

(19.7)

$$
\frac{\partial L(\theta)}{\partial b} = -\sum_{i=1}^{n} \left[ \frac{\epsilon_i X_i}{h_i} + \frac{1}{2h_i} \frac{\partial h_i}{\partial b} \left(\frac{\epsilon_i^2}{h_i} - 1\right) \right]
$$

(19.8)
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19.1.2  Student’s t distribution

Here we obtain expressions for the partial derivatives of the log likelihood when the series shocks have a Student’s t distribution. Using Equation 17.8 we have:

Partial derivatives w.r.t. the parameter vector \( \omega \):

\[
\frac{\partial \mathcal{L}_\theta(\theta)}{\partial \omega} = \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} \left( \nu + 1 \right) \frac{\partial \log(1 + \epsilon_i^2/(\nu - 2)h_i)}{\partial \omega} \frac{1}{2} \frac{\partial (1 + \epsilon_i^2/(\nu - 2)h_i)}{\partial \omega} \\
= \left( \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} \right) \left( \frac{\nu + 1}{2(1 + \epsilon_i^2/(\nu - 2)h_i)} \right) \frac{\partial (\epsilon_i^2/(h_i(\nu - 2)))}{\partial \omega} \\
= \left( \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} \right) \left( \frac{\nu + 1}{2(1 + \epsilon_i^2/(\nu - 2)h_i)} \right) \frac{\epsilon_i}{h_i} \frac{\partial h_i}{\partial \omega} \\
= \left( \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} \right) \left( \frac{\nu + 1}{2h_i^2((\nu - 2) + \epsilon_i^2/h_i)} \epsilon_i \right) \frac{\partial h_i}{\partial \omega} \\
\frac{\partial \mathcal{L}_\theta(\theta)}{\partial \omega} = \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} - \frac{\epsilon_i^2}{2h_i^2} \frac{\partial h_i}{\partial \omega} G, \text{ where } G = \frac{(\nu + 1)}{(\nu - 2) + (\epsilon_i^2/h_i)} \tag{19.9}
\]

Similarly we obtain the partial derivative w.r.t. the mean term, \( b_0 \):

\[
\frac{\partial \mathcal{L}_\theta(\theta)}{\partial b_0} = -\frac{\epsilon_i}{h_i} G + \frac{1}{2h_i} \left( 1 - G^2 \frac{\epsilon_i^2}{h_i} \right) \frac{\partial h_i}{\partial b_0} \tag{19.10}
\]

Partial derivatives w.r.t. the parameter vector \( b \):

\[
\frac{\partial \mathcal{L}_\theta(\theta)}{\partial b} = \frac{1}{2} \frac{\partial (\log h_i)}{\partial b} \frac{\partial h_i}{\partial b} + \frac{1}{2} \frac{\partial (1 + \epsilon_i^2/(\nu - 1)h_i)}{\partial b} \\
= \frac{1}{2h_i} \frac{\partial h_i}{\partial b} + \frac{(\nu + 1)}{2h_i^2(\nu - 2)} \left( 1 + \frac{1}{1 + \epsilon_i^2/(\nu - 2)h_i} \right) \frac{\partial \epsilon_i^2}{\partial b} \\
+ \frac{(\nu + 1)}{2} \left( 1 + \frac{1}{(1 + \epsilon_i^2/(\nu - 2)h_i)} \frac{\partial (1/h_i)}{\partial b} \right) \tag{19.11}
\]

Since \( \partial (1/h_i)/\partial b = -(1/h_i^2) \partial h_i/\partial b \) we obtain:

\[
\frac{\partial \mathcal{L}_\theta(\theta)}{\partial b} = \frac{1}{2h_i} \frac{\partial h_i}{\partial b} \frac{(\nu + 1)}{2h_i^2(\nu - 2)} \left( 1 + \frac{1}{1 + \epsilon_i^2/(\nu - 2)h_i} \right) \frac{\partial h_i}{\partial b} \\
- \frac{2\epsilon_iX_i(\nu + 1)}{2h_i(\nu - 2)} \left( 1 + \frac{1}{1 + \epsilon_i^2/(\nu - 2)h_i} \right) \\
= -\frac{\epsilon_iX_i(\nu + 1)}{(\nu - 2)h_i} \left( 1 + \frac{1}{1 + \epsilon_i^2/(\nu - 2)h_i} \right) \\
+ \frac{1}{2h_i} \left( 1 - \frac{\epsilon_i^2(\nu + 1)}{h_i(\nu - 2)} \left( 1 + \frac{1}{1 + \epsilon_i^2/(\nu - 2)h_i} \right) \right) \\
\frac{\partial \mathcal{L}_\theta(\theta)}{\partial b} = -\frac{\epsilon_iX_i}{h_i} G + \frac{1}{2h_i} \left( 1 - G^2 \frac{\epsilon_i^2}{h_i} \right) \frac{\partial h_i}{\partial b} \tag{19.12}
\]
Partial derivative w.r.t. the number of degrees of freedom, $\nu$:

Since $\frac{\partial (\log \Gamma(\nu))}{\partial \nu} = \psi(\nu)$ we have the following, Abramowitz and Stegun (1968):

$$\frac{\partial (\log \Gamma(\nu + 1/2))}{\partial \nu} = \frac{1}{2} \psi \left( \frac{\nu + 1}{2} \right) \quad \text{and} \quad \frac{\partial (\log \Gamma(\nu/2))}{\partial \nu} = \frac{1}{2} \psi \left( \frac{\nu}{2} \right)$$

Using this we obtain:

$$\frac{\partial L_i(\theta)}{\partial \nu} = -\frac{1}{2} \psi \left( \frac{\nu + 1}{2} \right) + \frac{1}{2} \psi \left( \frac{\nu}{2} \right) + \frac{1}{2} \log \left( 1 + \frac{\epsilon_i^2}{(\nu - 2)h_i} \right)$$

$$\frac{\partial L_i(\theta)}{\partial \nu} = -\frac{1}{2} \psi \left( \frac{\nu + 1}{2} \right) + \frac{1}{2} \psi \left( \frac{\nu}{2} \right) + \frac{1}{2} \log \left( 1 + \frac{\epsilon_i^2}{(\nu - 2)h_i} \right)$$

$$- \frac{\nu + 1}{2(1 + \epsilon_i^2/((\nu - 2)h_i))} h_i (\nu - 2)^2$$

$$\frac{\partial L_i(\theta)}{\partial \nu} = -\frac{1}{2} \psi \left( \frac{\nu + 1}{2} \right) + \frac{1}{2} \psi \left( \frac{\nu}{2} \right) + \frac{1}{2} \log \left( 1 + \frac{\epsilon_i^2}{(\nu - 2)h_i} \right) - \frac{\epsilon_i^2}{2(\nu - 2)h_i} G$$

(19.13)

In summary we have:

**Student’s t distribution log likelihood partial derivatives**

$$\frac{\partial L(\theta)}{\partial \omega} = -\sum_{i=1}^n \left[ \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} \left( \frac{\epsilon_i^2}{h_i} G - 1 \right) \right]$$

(19.14)

$$\frac{\partial L(\theta)}{\partial \nu} = -\sum_{i=1}^n \left[ K - \frac{1}{2} \log \left( 1 + \frac{\epsilon_i^2}{(\nu - 2)h_i} \right) + \frac{\epsilon_i^2}{2(\nu - 2)h_i} G \right]$$

(19.15)

$$\frac{\partial L(\theta)}{\partial b_0} = -\sum_{i=1}^n \left[ \frac{\epsilon_i}{h_i} G + \frac{1}{2h_i} \frac{\partial h_i}{\partial b_0} \left( \frac{\epsilon_i^2}{h_i} G - 1 \right) \right]$$

(19.16)

$$\frac{\partial L(\theta)}{\partial b} = -\sum_{i=1}^n \left[ \frac{\epsilon_i X_i}{h_i} G + \frac{1}{2h_i} \frac{\partial h_i}{\partial b} \left( \frac{\epsilon_i^2}{h_i} G - 1 \right) \right]$$

(19.17)

where $G = \left( \frac{\nu + 1}{(\nu - 2) + \epsilon_i^2/h_i} \right)$ and $K = \frac{1}{2} \psi \left( \frac{\nu + 1}{2} \right) - \frac{1}{2} \psi \left( \frac{\nu}{2} \right) - \frac{1}{2(\nu - 2)}$

### 19.2 THE SECOND DERIVATIVES

As previously mentioned the Hessian of the log likelihood can be used as an approximation to the Fisher information matrix. Here we will assume that the conditional PDF of the residuals is Gaussian and calculate the conditional expectation of
the Hessian. We will use the result (Engle, 1982) that the off-diagonal block elements of this matrix are zero, and will only compute the diagonal block elements.

We will denote the standardized residuals $\epsilon_i / \sqrt{h_i}$ by $Z_i$. So we have $Z_i | \psi_{i-1} \sim NID(0, 1)$. Further we will use following results:

$$E(Z_i | \psi_{i-1}) = 0, E(Z_i^2 | \psi_{i-1}) = 1,$$

and

$$E(Z_i^2 | \psi_{i-1} - 1) = 0$$ (19.18)

We note, $\epsilon_j$ and $\epsilon_k$, $k \neq j$ are independent, and since in a GARCH(p,q) process $h_i$ only depends on past values of the residuals, $\epsilon_i, i = 1, \ldots, q$, we have that $h_i$ and $Z_i$ are independent.

Using this gives:

$$E\left(\frac{Z_i^2}{h_i}\right) = E(\frac{Z_i^2}{h_i}) E\left(\frac{1}{h_i}\right) = \frac{1}{h_i}$$ (19.19)

**Calculation of the diagonal block** $\partial^2 L_i(\theta) / \partial \omega \partial \omega^T$

Recalling from Section 19.1 that the first derivative is:

$$\frac{\partial L_i(\theta)}{\partial \omega} = -\frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} \left\{ \epsilon_i^2 \right\}$$

Taking second derivatives w.r.t. $\omega$ we have:

$$\frac{\partial^2 L_i(\theta)}{\partial \omega \partial \omega^T} = -\left\{ \epsilon_i^2 \right\} \frac{\partial}{\partial \omega^T} \left\{ \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} \right\} - \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega^T} \left\{ \epsilon_i^2 \right\} - \frac{1}{2h_i} \epsilon_i \frac{\partial h_i}{\partial \omega} \frac{\partial (1/h_i)}{\partial \omega} \frac{\partial h_i}{\partial \omega^T}$$

$$= -\left\{ \epsilon_i^2 \right\} \frac{\partial}{\partial \omega^T} \left\{ \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} \right\} + \epsilon_i \frac{\partial h_i}{\partial \omega} \frac{\partial h_i}{\partial \omega^T}$$

which expressed using standardized residuals is:

$$\frac{\partial^2 L_i(\theta)}{\partial \omega \partial \omega^T} = -\{ Z_i^2 - 1 \} \cdot \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} + Z_i^2 \frac{\partial h_i}{\partial \omega} \frac{\partial h_i}{\partial \omega^T}$$

Therefore the conditional expectation of the block at time instant $i$ is:

$$E\left(\frac{\partial^2 L_i(\theta)}{\partial \omega \partial \omega^T}\right) = E\left(\{ Z_i^2 - 1 \} \cdot \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} + E(\frac{Z_i^2}{h_i}) \frac{\partial h_i}{\partial \omega} \frac{\partial h_i}{\partial \omega^T}\right)$$

$$E\left(\frac{\partial^2 L_i(\theta)}{\partial \omega \partial \omega^T}\right) = -E(\{ Z_i^2 - 1 \}) \cdot \frac{1}{2h_i} \frac{\partial h_i}{\partial \omega} + \frac{1}{2h_i^2} \frac{\partial h_i}{\partial \omega} \frac{\partial h_i}{\partial \omega^T}$$
which gives

\[ E \left( \frac{\partial^2 L_i(\theta)}{\partial \omega \partial \omega^T} \right) = \frac{1}{2h_i^2} \frac{\partial h_i}{\partial \omega} \frac{\partial h_i}{\partial \omega^T} \]

The sample diagonal block is therefore:

\[ E \left( \frac{\partial^2 L(\theta)}{\partial \omega \partial \omega^T} \right) = \sum_{i=1}^{n} \frac{1}{2h_i^2} \frac{\partial h_i}{\partial \omega} \frac{\partial h_i}{\partial \omega^T} \]  \hspace{1cm} (19.20)

**Calculation of the diagonal block \( \frac{\partial^2 L_i(\theta)}{\partial b \partial b^T} \)**

Recalling from Section 19.1 that the first derivative is:

\[ \frac{\partial L_i(\theta)}{\partial b} = -\epsilon_i X_i + \frac{1}{h_i} \frac{\partial h_i}{\partial b} \left\{ \frac{\epsilon_i^2}{h_i} - 1 \right\} \]

Taking second derivatives w.r.t. to \( b \) we have:

\[ \frac{\partial^2 L_i(\theta)}{\partial b \partial b^T} = -\frac{X_i \frac{\partial \epsilon_i}{\partial b}}{h_i} - \epsilon_i X_i \frac{\partial (1/h_i)}{\partial b} \frac{\partial h_i}{\partial b^T} + \frac{1}{2h_i} \frac{\partial h_i}{\partial b} \frac{\partial h_i}{\partial b^T} \frac{\partial h_i}{\partial b} \left\{ \frac{\epsilon_i^2}{h_i} - 1 \right\} \]

\[ - \frac{1}{2h_i} \frac{\partial h_i}{\partial b} \frac{\partial h_i}{\partial b^T} \left\{ \frac{\epsilon_i^2}{h_i} - 1 \right\} \]

But since \( \epsilon_i = y_i - X_i^T b - b_0 \)

\[ \frac{\partial \epsilon_i}{\partial b^T} = \frac{\partial (y_i - X_i^T b - b_0)}{\partial b^T} = X_i^T \quad \text{and} \quad \frac{\partial \epsilon_i^2}{\partial b^T} = -2 \epsilon_i X_i \]

We have:

\[ \frac{\partial^2 L_i(\theta)}{\partial b \partial b^T} = \frac{X_i X_i^T}{h_i} + \frac{\epsilon_i X_i \frac{\partial h_i}{\partial b}}{h_i} + \frac{1}{2} \frac{\partial (1/h_i)}{\partial b} \frac{\partial h_i}{\partial b^T} \frac{\partial h_i}{\partial b} \left\{ \frac{\epsilon_i^2}{h_i} - 1 \right\} + \frac{\epsilon_i X_i \frac{\partial h_i}{\partial b}}{h_i} + \frac{\epsilon_i^2}{2h_i^2} \frac{\partial h_i}{\partial b} \frac{\partial h_i}{\partial b^T} \]

Therefore using standardized residuals and taking conditional expectations of the block at time instant \( i \) we have:

\[ \left( \frac{\partial^2 L_i(\theta)}{\partial b \partial b^T} \right) = \frac{X_i X_i^T}{h_i} + \frac{E(\epsilon_i) X_i \frac{\partial h_i}{\partial b}}{h_i} + \frac{1}{2} \frac{\partial (1/h_i)}{\partial b} \frac{\partial h_i}{\partial b^T} \frac{\partial h_i}{\partial b} \frac{E(\epsilon_i^2)}{\partial b} \frac{\partial h_i}{\partial b^T} \]

\[ + \frac{E(\epsilon_i) X_i \frac{\partial h_i}{\partial b}}{h_i} + \frac{E(\epsilon_i^2)}{2h_i^2} \frac{\partial h_i}{\partial b} \frac{\partial h_i}{\partial b^T} \]

which gives:

\[ E \left( \frac{\partial^2 L_i(\theta)}{\partial b \partial b^T} \right) = \frac{X_i X_i^T}{h_i} + \frac{1}{2h_i^2} \frac{\partial h_i}{\partial b} \frac{\partial h_i}{\partial b^T} \]
The sample diagonal block is therefore:

\[ E \left( \frac{\partial^2 L(\theta)}{\partial b_0 \partial b_0^T} \right) = \sum_{i=1}^{n} \left( \frac{X_i X_i^T}{h_i} + \frac{1}{2h_i^2} \frac{\partial h_i}{\partial b} \frac{\partial h_i}{\partial b^T} \right) \]  \hspace{1cm} (19.21)

**Calculation of the diagonal block \( \frac{\partial^2 L_i(\theta)}{\partial b_0 \partial b_0^T} \)**

Recalling from Section 19.1 that the first derivative is:

\[ \frac{\partial L_i(\theta)}{\partial b_0} = -\frac{\epsilon_i}{h_i} + \frac{1}{2h_i} \frac{\partial h_i}{\partial b_0} \left\{ \frac{\epsilon_i^2}{h_i} - 1 \right\} \]

Taking second derivatives w.r.t. to \( b_0 \) we have:

\[ \frac{\partial^2 L_i(\theta)}{\partial b_0 \partial b_0^T} = -\frac{1}{h_i} \frac{\partial \epsilon_i}{\partial b_0} - \frac{\epsilon_i}{h_i} - \frac{1}{h_i} \frac{\partial (1/h_i)}{\partial b_0} \frac{\partial h_i}{\partial b_0} - \frac{1}{2h_i} \frac{\partial h_i}{\partial b_0} \frac{\partial h_i}{\partial b_0} \left\{ \frac{\epsilon_i^2}{h_i} - 1 \right\} \]

\[ -\frac{1}{2h_i^2} \frac{\partial h_i}{\partial b_0} \left\{ \frac{\epsilon_i^2}{h_i} - 1 \right\} \]

But since \( \epsilon_i = y_i - X_i^T b - b_0 \)

\[ \frac{\partial \epsilon_i}{\partial b_0} = \frac{\partial (y_i - X_i^T b - b_0)}{\partial b_0} = -\frac{\partial b_0}{\partial b_0} = -1 \quad \text{and} \quad \frac{\partial^2 \epsilon_i}{\partial b_0} = -2 \epsilon_i \]

We have:

\[ \frac{\partial^2 L_i(\theta)}{\partial b_0 \partial b_0^T} = \frac{1}{h_i} + \frac{\epsilon_i h_i}{h_i^2} \frac{\partial h_i}{\partial b_0} - \frac{1}{2h_i} \frac{\partial h_i}{\partial b_0} \frac{\partial h_i}{\partial b_0} \left\{ \frac{\epsilon_i^2}{h_i} - 1 \right\} + \frac{\epsilon_i h_i}{h_i^2} \frac{\partial h_i}{\partial b_0} + \frac{\epsilon_i^2}{2h_i^2} \frac{\partial h_i}{\partial b_0} \frac{\partial h_i}{\partial b_0} \]

Therefore using standardized residuals and taking conditional expectations of the block at time instant \( i \) we have:

\[ E \left( \frac{\partial^2 L_i(\theta)}{\partial b_0 \partial b_0^T} \right) = \frac{1}{h_i} + \frac{E(\epsilon_i) h_i}{h_i^2} \frac{\partial h_i}{\partial b_0} - \frac{1}{2h_i} \frac{\partial h_i}{\partial b_0} \frac{\partial h_i}{\partial b_0} E\left( \{ \epsilon_i^2 \} \right) \]

\[ + \frac{E(\epsilon_i) h_i}{h_i^2} \frac{\partial h_i}{\partial b_0} + \frac{E(\epsilon_i^2)}{2h_i^2} \frac{\partial h_i}{\partial b_0} \frac{\partial h_i}{\partial b_0} \]

which gives:

\[ E \left( \frac{\partial^2 L_i(\theta)}{\partial b_0 \partial b_0^T} \right) = \frac{1}{h_i} + \frac{1}{2h_i^2} \frac{\partial h_i}{\partial b_0} \frac{\partial h_i}{\partial b_0} \]

The sample diagonal block is therefore:

\[ E \left( \frac{\partial^2 L(\theta)}{\partial b_0 \partial b_0^T} \right) = \sum_{i=1}^{n} \left( \frac{1}{h_i} + \frac{1}{2h_i^2} \frac{\partial h_i}{\partial b_0} \frac{\partial h_i}{\partial b_0} \right) \]  \hspace{1cm} (19.22)
In summary we obtain:

### The blocks of the Fisher information matrix

\[
E \left( \frac{\partial^2 \mathcal{L}(\theta)}{\partial \omega \partial \omega^T} \right) = \sum_{i=1}^{n} \frac{1}{2h_i^2} \frac{\partial h_i}{\partial \omega} \frac{\partial h_i}{\partial \omega^T} 
\]

(19.23)

\[
E \left( \frac{\partial^2 \mathcal{L}(\theta)}{\partial b \partial b_0^T} \right) = \sum_{i=1}^{n} \left( \frac{1}{h_i} + \frac{1}{2h_i^2} \frac{\partial h_i}{\partial b} \frac{\partial h_i}{\partial b_0^T} \right) 
\]

(19.24)

\[
E \left( \frac{\partial^2 \mathcal{L}(\theta)}{\partial b \partial b^T} \right) = \sum_{i=1}^{n} \left( \frac{X_i X_i^T}{h_i} + \frac{1}{2h_i^2} \frac{\partial h_i}{\partial b} \frac{\partial h_i}{\partial b^T} \right) 
\]

(19.25)

It can be seen that these diagonal blocks of the information matrix involve the outer product of the following first derivative vectors \( \sum_{i=1}^{n} \frac{\partial h_i}{\partial \omega}, \sum_{i=1}^{n} \frac{\partial h_i}{\partial b} \), and also the square of the scalar derivative \( \sum_{i=1}^{n} \frac{\partial h_i}{\partial b_0} \). Once these terms have been computed it is easy to calculate the information matrix. Chapter 20 provides details on how this can be accomplished.
Chapter 20

GJR–GARCH algorithms

We will now use the information in Chapter 19 to show how the partial derivatives of the log likelihood can be computed. Practical details concerning initial estimates and pre-observed values are discussed. Pseudocode is also provided to facilitate computer implementations of the regression-GJR–GARCH model.

The notation used in this section is as follows:

- \( \text{num} \) the number of terms in the GARCH sequence, \( \text{num} \) is synonymous with the mathematical symbol \( n \).
- \( \text{mn} \) indicates whether the mean term \( b_0 \) is included in the model. If \( \text{mn} = 1 \) then \( b_0 \) is included, otherwise it is not.
- \( \text{nreg} \) the number of regression terms in the model, \( \text{nreg} \) is synonymous with the mathematical symbol \( k \).
- \( \text{npar} \) the number of heteroskedastic parameters in a standard symmetric Gaussian GARCH model, that is \( 1 + p + q \).
- \( \hat{b} \) the initial estimate for \( b \), the \( k \) element vector of regression coefficients.
- \( \hat{b}_0 \) the initial estimate for \( b_0 \), the mean term.
- \( \theta_k \) the \( k \)th element of the regression-GARCH parameter vector \( \theta \). The order of the elements is the same as given in Chapter 17. That is:
  \[
  \theta_1 = \alpha_0, \theta_{k+1} = \alpha_k, k = 1, \ldots, q, \theta_{1+q+k} = \beta_k, k = 1, \ldots, p,
  \theta_{npar+1} = \gamma, \text{ etc.}
  \]
- \( \mathcal{H}(i-k) \) a function which has the value 1 when \( i > k \) and zero otherwise.
- \( N_p \) the total number of parameters to estimate. In the Gaussian regression-GJR–GARCH \( N_p = 2 + p + q + \text{mn} + \text{nreg} \), and in the Student’s \( t \) distribution \( N_p = 3 + p + q + \text{mn} + \text{nreg} \).

All other symbols have been previously defined in Chapters 14 and 15.

20.1 INITIAL ESTIMATES AND PRE-OBSERVED VALUES

In this section we consider how to estimate the initial values that are required for computing both the log likelihood and its partial derivatives.

The initial estimates of the regression coefficients, \( \hat{b}_i, i = 1 - \text{mn}, \ldots, k \), can be obtained using linear regression.
If \( mn = 1 \) then the residuals are calculated as:

\[
e_i = y_i - X_i \hat{b} - \hat{b}_0, \quad i = 1, \ldots, n
\]  

otherwise they are:

\[
e_i = y_i - X_i \hat{b}, \quad i = 1, \ldots, n
\]

In all GARCH processes the conditional variance \( h_i \) satisfies a recursive equation. For instance the basic linear GARCH model has:

\[
h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j e_{t-j}^2 + \sum_{j=1}^{p} \beta_j h_{t-j}, \quad t = 1, \ldots, n
\]

This means that the conditional variance for the term \( h_1 \) is given by:

\[
h_1 = \alpha_0 + \sum_{j=1}^{q} \alpha_j e_{1-j}^2 + \sum_{j=1}^{p} \beta_j h_{1-j}
\]

which relies on the terms \( e_0^2, e_1^2, \ldots, e_{1-j}^2 \), and \( h_0, h_{-1}, \ldots, h_{1-j} \), that refer to times before the sequence started. We will call this terms pre-observed values. There are various methods of providing estimates for these values.

One simple approach is to model an alternative time series which starts at the data point \( i = \max(p, q) \) and has the reduced length \( n - \max(p, q) \). The first \( \max(p, q) \) terms are then used to calculate the pre-observed values.

The pre-observed values of \( e_i \) can now use the actual values, and \( e_i^2 \) can be used as an estimate for \( h_i \).

However, this method is not entirely satisfactory as we are not modelling the true data and also the single value \( e_1^2 \) is unlikely to be a good estimate of the conditional variance \( h_1 \).

Here we use a different technique. The initial value for the variance, \( \sigma_0^2 \), is taken as the average value of \( e_1^2 \) using the first \( \tau \) terms of the sequence:

\[
\sigma_0^2 = \frac{1}{\tau} \sum_{i=1}^{\tau} e_i^2
\]  

The optimal value of \( \tau \) to use will depend on the nature of the data. If the sequence has high initial volatility then \( \tau \) should be short enough to capture this. For sequences with less initial variation the estimate \( \sigma_0^2 \) will benefit from an increased value of \( \tau \).

Here we used, the compromise value \( \tau = N_p \). This value is used in Equation 20.3 to calculate the pre-observed conditional variance and residuals squared, i.e.:

\[
e_i^2 = h_i = \sigma_0^2, \quad i \leq 0
\]

The pre-observed values for the residuals, \( e_i \), are taken as:

\[
e_i = E[e_j] = 0, \quad \text{where } i \leq 0 \text{ and } j = 1, \ldots, n
\]
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Since \( \partial h_i / \partial \theta_k \) is calculated recursively from previous terms such as \( \Sigma_{j=1}^{p} \beta_j (\partial h_{i-j} / \partial \theta_k) \) and \( \Sigma_{j=1}^{q} \alpha_j (\partial \varepsilon_{i-j}^2) / (\partial \theta_k) \) we can make use of the fact that \( \partial \varepsilon_i^2 / \partial \theta_k = \partial h_i / \partial \theta_k = 0 \), \( k = 1, \ldots, N_p, i \leq 0 \), i.e.:

\[
\frac{\partial \varepsilon_i^2}{\partial \omega} = \frac{\partial \varepsilon_i^2}{\partial b_0} = \frac{\partial \varepsilon_i^2}{\partial b} = 0, \quad i \leq 0
\]  

(20.4)

and

\[
\frac{\partial h_i}{\partial \omega} = \frac{\partial h_i}{\partial b_0} = \frac{\partial h_i}{\partial b} = 0, \quad i \leq 0
\]

(20.5)

Note: Although this is correct for a Gaussian distribution it is not strictly true for a Student’s \( t \) distribution, since the derivative term \( \partial h_i / \partial \nu \) does not depend on its previous value.

Using the above results, for \( i \leq q \) we now have:

\[
\sum_{j=1}^{q} \alpha_j \frac{\partial \varepsilon_{i-j}}{\partial \theta_k} = \sum_{j=1}^{i-1} \alpha_j \frac{\partial \varepsilon_{i-j}}{\partial \theta_k}
\]

(20.6)

and

\[
\sum_{j=1}^{p} \beta_j \frac{\partial h_{i-j}}{\partial \theta_k} = \sum_{j=1}^{i-1} \beta_j \frac{\partial h_{i-j}}{\partial \theta_k} \mathcal{H}(i-j)
\]

(20.7)

Further details are provided in the pseudocode provided in the following section.

20.2 GAUSSIAN DISTRIBUTION

20.2.1 The log likelihood

Deal with the first \( q \) terms of the sequence:

\( \gamma = \hat{\gamma} \)

\( \mathcal{L}(\theta) = 0 \)

For \( i = 1 \) To num

If (\( mn == 1 \)) \( \epsilon_i = y_i - X_i^T \hat{b} \)

If (\( mn == 0 \)) \( \epsilon_i = y_i - \hat{b}_0 - X_i^T \hat{b} \)

Next i

For \( i = 1 \) To \( q \)

\( h_i = \alpha_0 + \sum_{j=1}^{i-1} (\alpha_j + \gamma S_{i-j}) \varepsilon_{i-j}^2 + \sum_{j=1}^{q} \alpha_j \sigma_0^2 + \sum_{k=1}^{i-1} h_{i-k} \beta_k \)

Store the current value of \( h_i \) and keep all the previous values of \( h_i \).

\( \mathcal{L}(\theta) = \mathcal{L}(\theta) + \frac{1}{2} \left( \log(h_i) + \frac{\varepsilon_i^2}{h_i} \right) \)

Next i
Deal with the remaining terms of the sequence:

For $i = q + 1$ to $num$

$$h_i = \alpha_0 + \sum_{j=1}^{q} (\alpha_j + \gamma S_{t-j}) \epsilon_{i-j} + \sum_{k=1}^{p} h_{i-k} \beta_k$$

Store the current value of $h_i$ and keep $N_p$ previous values of $h_i$.

$$\mathcal{L}(\theta) = \mathcal{L}(\theta) + \frac{1}{2} \left( \log(h_i) + \frac{\epsilon_i^2}{h_i} \right)$$

Next $i$

### 20.2.2 The first derivatives of the log likelihood

Algorithm for the first $q$ terms of the sequence:

$$\frac{\partial \mathcal{L}(\theta)}{\partial \theta_k} = 0, \quad k = 1, \ldots, N_p$$

For $i = 1$ to $q$

$$\frac{\partial \epsilon_i}{\partial \alpha_0} = 1 + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \alpha_0}$$

For $j = 1$ to $i - 1$

$$\frac{\partial \epsilon_i}{\partial \alpha_j} = \epsilon_{i-j}^2$$

Next $j$

For $j = i$ to $q$

$$\frac{\partial \epsilon_i}{\partial \alpha_j} = \sigma_0^2$$

Next $j$

For $j = 1$ to $q$

$$\frac{\partial \epsilon_i}{\partial \gamma_j} = \frac{\partial \epsilon_i}{\partial \gamma_j} + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \gamma_j}$$

Next $j$

For $j = 1$ to $p$

$$\frac{\partial \epsilon_i}{\partial \beta_j} = h_{i-j} + \sum_{k=1}^{p} \beta_j \frac{\partial h_{i-k}}{\partial \beta_k}$$

Next $j$

$$\frac{\partial \epsilon_i}{\partial \gamma} = \sum_{j=1}^{i-1} \epsilon_{i-j}^2 + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \gamma}$$
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\[ h_i = \alpha_0 + \sum_{k=1}^{p} h_{i-k} \beta_k + \sum_{j=1}^{i-1} (\alpha_j + \gamma S_{i-j}) \varepsilon_{i-j}^2 + \sum_{j=i}^{q} \alpha_j \sigma_0^2 \]

if \( mn == 1 \) then

\[ \frac{\partial h_i}{\partial b_0} = -2 \sum_{k=1}^{i-1} (\alpha_k + \gamma S_{i-k}) \varepsilon_{i-k} + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial b_0} \mathcal{H}(i-k) \]

end if

For \( j = 1 \) to \( n_{\text{reg}} \)

\[ \frac{\partial h_i}{\partial b_j} = -2 \sum_{k=1}^{i-1} (\alpha_k + \gamma S_{i-k}) \varepsilon_{i-k} X_{i-k}^j + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial b_j} \mathcal{H}(i-k) \]

Next \( j \)

Store the current values of \( h_i \) and \( \partial h_i / \partial \theta \) and keep all the previous values of \( h_i \) and \( \partial h_i / \partial \theta \).

For \( k = 1 \) to \( n_{\text{par}} + 1 \)

\[ \frac{\partial \mathcal{L}(\theta)}{\partial \theta_k} = \frac{\partial \mathcal{L}(\theta)}{\partial \theta_k} - \frac{1}{2h_i} \left( \varepsilon_i^2 - 1 \right) \frac{\partial h_i}{\partial \theta_k} \]

Next \( k \)

if \( mn == 1 \) then

\[ \frac{\partial \mathcal{L}(\theta)}{\partial b_0} = \frac{\partial \mathcal{L}(\theta)}{\partial b_0} - \frac{1}{h_i} \frac{1}{2h_i} \left( \frac{\varepsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial b_0} \]

end if

For \( k = 1 \) to \( n_{\text{reg}} \)

\[ \frac{\partial \mathcal{L}(\theta)}{\partial b_k} = \frac{\partial \mathcal{L}(\theta)}{\partial b_k} - \frac{1}{h_i} \left( \frac{\varepsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial b_k} \]

Next \( k \)

Next \( i \)

Algorithm for the remaining terms of the sequence:

For \( i = q+1 \) to \( \text{num} \)

\[ \frac{\partial h_i}{\partial \alpha_0} = 1 + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \alpha_0} \]

For \( j = 1 \) to \( q \)

\[ \frac{\partial h_i}{\partial \alpha_j} = \varepsilon_{i-j}^2 \]
Next j
For \( j = 1 \) to \( q \)
\[
\frac{\partial h_i}{\partial \alpha_j} = \frac{\partial h_i}{\partial \alpha_j} + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \alpha_j}
\]
Next j
For \( j = 1 \) to \( p \)
\[
\frac{\partial h_i}{\partial \beta_j} = h_{i-j} + \sum_{k=1}^{p} \beta_j \frac{\partial h_{i-k}}{\partial \beta_k}
\]
Next j
\[
\frac{\partial h_i}{\partial \gamma} = \sum_{j=1}^{q} \epsilon_{i-j}^2 + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \gamma}
\]
\[
h_i = \alpha_0 + \sum_{k=1}^{p} h_{i-k} \beta_k + \sum_{j=1}^{q} (\alpha_j + \gamma S_{i-j}) \epsilon_{i-j}^2
\]
if \((mn == 1)\) then
\[
\frac{\partial h_i}{\partial b_0} = -2 \sum_{k=1}^{q} (\alpha_k + \gamma S_{i-k}) \epsilon_{i-k} + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial b_0} \mathcal{H}(i-k)
\]
end if
For \( j = 1 \) to \( n_{\text{reg}} \)
\[
\frac{\partial h_i}{\partial b_j} = -2 \sum_{k=1}^{q} (\alpha_k + \gamma S_{i-k}) \epsilon_{i-k} X_{i-k}^j + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial b_j} \mathcal{H}(i-k)
\]
Next j

Store the current values of \( h_i \) and \( \partial h_i / \partial \theta \) and keep \( N_p \) previous values of \( h_i \) and \( \partial h_i / \partial \theta \).

For \( k = 1 \) to \( n_{\text{par}} + 1 \)
\[
\frac{\partial \mathcal{L}(\theta)}{\partial \theta_k} = \frac{\partial \mathcal{L}(\theta)}{\partial \theta_k} - \frac{1}{2h_i} \left( \frac{\epsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial \theta_k}
\]
Next k
if \((mn == 1)\) then
\[
\frac{\partial \mathcal{L}(\theta)}{\partial b_0} = \frac{\partial \mathcal{L}(\theta)}{\partial b_0} - \epsilon_i \mathcal{H}(i) - \frac{1}{2h_i} \left( \frac{\epsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial b_0}
\]
end if
For \( k = 1 \) to \( n_{\text{reg}} \)
\[
\frac{\partial \mathcal{L}(\theta)}{\partial b_k} = \frac{\partial \mathcal{L}(\theta)}{\partial b_k} - \frac{X_i^k \epsilon_i}{h_i} - \frac{1}{2h_i} \left( \frac{\epsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial b_k}
\]
Next k
Next i
20.3 STUDENT’S $t$ DISTRIBUTION

20.3.1 The log likelihood

Deal with the first $q$ terms of the sequence:

$$\gamma = \hat{\gamma}$$

$$L(\theta) = 0$$

$$M_\nu = \log(\Gamma((\nu + 1)/2)) - \log(\Gamma(\nu/2)) - \frac{1}{2}\log(\nu - 2)$$

For $i = 1$ to $\text{num}$

If $mn == 1$  $\epsilon_i = y_i - X_i^T \hat{\beta}$

If $mn == 0$  $\epsilon_i = y_i - \hat{b}_0 - X_i^T \hat{\beta}$

Next $i$

For $i = 1$ to $q$

$$h_i = \alpha_0 + \sum_{j=1}^{i-1} (\alpha_j + \gamma S_{i-j}) \epsilon_{i-j}^2 + \sum_{j=i}^{q} \alpha_j \sigma_0^2 + \sum_{k=1}^{p} \beta_k h_{i-k}$$

Store the current value of $h_i$ and keep all the previous values of $h_i$.

$$L(\theta) = L(\theta) - M_\nu + \frac{1}{2} \log(h_i) + \frac{\nu + 1}{2} \log\left(1 + \frac{\epsilon_i^2}{(\nu - 2)h_i}\right)$$

Next $i$

Deal with the remaining terms of the sequence:

For $i = q+1$ to $\text{num}$

$$h_i = \alpha_0 + \sum_{j=1}^{q} (\alpha_j + \gamma S_{i-j}) \epsilon_{i-j}^2 + \sum_{k=1}^{p} \beta_k h_{i-k}$$

Store the current value of $h_i$ and keep $N_p$ previous values of $h_i$.

$$L(\theta) = L(\theta) - M_\nu + \frac{1}{2} \log(h_i) + \frac{\nu + 1}{2} \log\left(1 + \frac{\epsilon_i^2}{(\nu - 2)h_i}\right)$$

Next $i$

20.3.2 The first derivatives of the log likelihood

Algorithm for the first $q$ terms of the sequence:

$$\frac{\partial L(\theta)}{\partial \theta_k} = 0, \quad k = 1, \ldots, N_p$$

For $i = 1$ to $q$
Compute $h_i$ as described in Section 20.2.1. Also calculate the derivatives $\partial h_i / \partial \theta_j$, $j = 1, \ldots, N_p$, as described for a Gaussian distribution in Section 20.1.2. Store $h_i$ and $\partial h_i / \partial \theta_j$, $j = 1, \ldots, N_p$, and keep all the previous values of $h_i$ and $\partial h_i / \partial \theta_i$.

Set $G = \frac{(\nu + 1)}{(\nu - 2) + \epsilon_i^2 / h_i}$

For $k = 1$ to $npar + 1$

$$\frac{\partial L(\theta)}{\partial \theta_k} = \frac{\partial L(\theta)}{\partial \theta_k} - \frac{1}{2h_i} \left(1 - \frac{\epsilon_i^2}{h_i} G\right) \frac{\partial h_i}{\partial \theta_k}$$

Next $k$

$$\frac{\partial L(\theta)}{\partial \nu} = \frac{\partial L(\theta)}{\partial \nu} - \frac{1}{2} \psi\left(\frac{\nu + 1}{2}\right) + \frac{1}{2} \psi\left(\frac{\nu}{2}\right) + \frac{1}{2(\nu - 2)}$$

$$+ \frac{1}{2} \log\left(1 + \frac{\epsilon_i^2}{(\nu - 2)h_i}\right) - \frac{\epsilon_i^2}{2(\nu - 2)h_i} G$$

if (mn == 1) then

$$\frac{\partial L(\theta)}{\partial b_0} = \frac{\partial L(\theta)}{\partial b_0} - \frac{\epsilon_i}{h_i} G - \frac{1}{2h_i} \left(\frac{\epsilon_i^2}{h_i} G - 1\right) \frac{\partial h_i}{\partial b_0}$$

end if

For $k = 1$ to $nreg$

$$\frac{\partial L(\theta)}{\partial b_k} = \frac{\partial L(\theta)}{\partial b_k} - X_i^k \frac{\epsilon_i}{h_i} G - \frac{1}{2h_i} \left(\frac{\epsilon_i^2}{h_i} G - 1\right) \frac{\partial h_i}{\partial b_k}$$

Next $k$

Next $i$

Algorithm for the remaining terms of the sequence:

For $i = q + 1$ to $num$

Compute $h_i$ as described in Section 20.2.1. Also calculate the derivatives $\partial h_i / \partial \theta_j$, $j = 1, \ldots, N_p$, as described for a Gaussian distribution in Section 20.1.2. Store $h_i$ and $\partial h_i / \partial \theta_j$, $j = 1, \ldots, N_p$, and keep $N_p$ previous values of $h_i$ and $\partial h_i / \partial \theta_i$.

Set $G = \frac{(\nu + 1)}{(\nu - 2) + \epsilon_i^2 / h_i}$

For $k = 1$ to $npar + 1$

$$\frac{\partial L(\theta)}{\partial \theta_k} = \frac{\partial L(\theta)}{\partial \theta_k} - \frac{1}{2h_i} \left(1 - \frac{\epsilon_i^2}{h_i} G\right) \frac{\partial h_i}{\partial \theta_k}$$

Next $k$
\[
\frac{\partial \mathcal{L}(\theta)}{\partial \nu} = \frac{\partial \mathcal{L}(\theta)}{\partial \nu} - \frac{1}{2} \psi\left(\frac{\nu + 1}{2}\right) + \frac{1}{2} \psi\left(\frac{\nu}{2}\right) + \frac{1}{2(\nu - 2)} \\
+ \frac{1}{2} \log\left(1 + \frac{\epsilon_i^2}{(\nu - 2)h_i}\right) - \frac{\epsilon_i^2}{2(\nu - 2)h_i} G
\]

if \((mn == 1)\) then
\[
\frac{\partial \mathcal{L}(\theta)}{\partial b_0} = \frac{\partial \mathcal{L}(\theta)}{\partial b_0} - \epsilon_i G - \frac{1}{2h_i} \left(\frac{\epsilon_i^2}{h_i} G - 1\right) \frac{\partial h_i}{\partial b_0}
\]
end if

For \(k = 1\) to \(n_{\text{reg}}\)
\[
\frac{\partial \mathcal{L}(\theta)}{\partial b_k} = \frac{\partial \mathcal{L}(\theta)}{\partial b_k} - \frac{X_k^i \epsilon_i}{h_i} G - \frac{1}{2h_i} \left(\frac{\epsilon_i^2}{h_i} G - 1\right) \frac{\partial h_i}{\partial b_k}
\]
Next \(k\)
Next \(i\)
Chapter 21

GARCH software

In this chapter we will describe some of the expected capabilities of practical GARCH software, and also how to test whether the software performs as expected.

21.1 EXPECTED SOFTWARE CAPABILITIES

To illustrate we will consider the requirements for the regression-GJR–GARCH model discussed in Chapter 20. We will assume a GARCH modelling component is to be developed and will list some important input and output properties that should be considered in its design.

**Inputs**

- The conditional probability distribution to use, i.e. Gaussian distribution, Student’s $t$ distribution, etc.
- The required initial estimates for the model parameters.
- The input data, $y_i, i = 1, \ldots, n$, and $X_i, i = 1, \ldots, n$.
- The number of GARCH model parameters, $a_j, j = 0, \ldots, q, \beta_j = 1, \ldots, p$, regression coefficients $b_j, j = 1, \ldots, k$, and mean term $b_0$.
- A flag to indicate whether the GARCH stationary constraint is to be enforced.
- A flag to indicate whether the user wants to provide initial estimates for the regression coefficients $b_j, j = 1, \ldots, k$, mean term $b_0$, and pre-observed conditional variance $\sigma_0^2$, or let the component calculate these.

**Outputs**

- The estimated GARCH model parameters, $\hat{\theta}$.
- The value of the minimized log likelihood, $\mathcal{L}(\hat{\theta})$.
- The estimated conditional variances, $h_i, i = 1, \ldots, n$.
- The estimated residuals, $\epsilon_i, i = 1, \ldots, n$.
- The standard errors associated with each estimated parameter.
- The covariance matrix of the estimated GARCH model parameters $\hat{\theta}$.

It is also useful to have information concerning the underlying numerical optimization of the log likelihood, such as the maximum number of iterations allowed for convergence, and also the tolerance used during the optimization process.
The software could also provide the scores for each estimated GARCH model parameter, \( \partial \mathcal{L}(\theta)/\partial \theta_i \neq \theta_i \). In the absence of constraints all these partial derivatives, at \( \mathcal{L}(\hat{\theta}) \) should be nearly zero. However, when the stationary constraint is imposed, a high value for the \( k \)th score indicates that the feasible boundary for this parameter has been reached. This means that it has not been possible to optimize \( \mathcal{L}(\theta) \) any further through variation of the parameter \( \theta_k \).

21.2 TESTING GARCH SOFTWARE

We will now give details concerning the implementation and testing, see Levy (2000), of regression-GJR–GARCH estimation software, developed using the algorithms outlined in Chapter 20. The log likelihood was minimized using a general purpose quasi-Newton type numerical optimizer, see Gill et al. (1981), and Murtagh and Saunders (1983), which employed either analytic or numeric derivatives (calculated using finite differences). The optimizer also had the capability of returning a finite-difference approximation to the Hessian at the solution point, \( \hat{\theta} \), which was used as the second-derivative estimate of the Fisher information matrix, \( \mathcal{F} \). All the results presented here are based on Monte Carlo simulations involving the generation and parameter estimation of 200 regression-GJR–GARCH sequences. Each sequence was created using the NAG routine G05HMF and estimated with the following optimization settings:

- The maximum number of iterations required for convergence to a solution set to 100.
- GARCH stationary condition enforced.
- The optimality tolerance set to \( 10^{-8} \), that is, the optimal value of the log likelihood has eight figure accuracy.

The simulation results are shown in Tables 21.1 to 21.10. The first column labelled ‘Estimated Value’ refers to the average parameter estimate using 200 simulations. The second column labelled ‘Estimated Standard Error’ refers to the average of the standard errors computed by the GARCH software. The third column labelled ‘Standard Error of Estimates’ refers to the actual standard error of the parameter estimates. The parameters are output in the order in which they occur in \( \theta \), i.e.:

- A Gaussian process has \( \theta = (\omega^T, b_0, b^T) \).
- A Student’s \( t \) distribution has \( \theta = (\omega^T, \nu, b_0, b^T) \).

Here \( \omega^T = (\omega_0, \omega_1, \ldots, \omega_q, \beta_1, \ldots, \beta_p, \gamma) \) and \( b^T = (b_1, \ldots, b_k) \).

Each table also reports the total CPU time in seconds required to estimate the model parameters for the 200 GARCH sequences. The tables labelled Numeric Derivatives refer to results obtained using a finite-difference approximation to both the gradient and the Hessian. Those labelled Analytic Derivatives refer to results obtained using the algorithms of Chapter 20 and an approximation to the Fisher
information matrix based on the conditional expectation of the Hessian at the solution point, $\hat{\theta}$.

### 21.2.1 Gaussian distribution

In Tables 21.1 to 21.6 we present the results of Monte Carlo simulations to check the parameter estimation software for the following Gaussian regression-GJR-GARCH(1,1) process:

$$k = 2 \quad \alpha_0 = 0.01 \quad \alpha_1 = 0.1 \quad \beta_1 = 0.8$$

$$\gamma = 0.2 \quad b_0 = 1.1 \quad b_1 = -1.5 \quad b_2 = 2.5$$

$$X_i^1 = \frac{1}{100} + 0.7 \times \sin\left(\frac{i}{100}\right), \quad X_i^2 = \frac{1}{2} + \left(\frac{i}{1000}\right), \text{ for } i = 1, \ldots, n$$

where the value of $\beta_1$ was taken as *realistically high* for a financial time series.

The initial values for the regression coefficients, $b_i$, $i = 0, \ldots, k$, and the pre-observed conditional variance $\sigma_0^2$ were estimated using OLS regression, as outlined in Section 20.1. The initial estimates for the elements of the parameter vector $\omega$ were all set to 0.1.

#### Numeric derivatives

Table 21.1  Sequence length 300, CPU time = 111.1 s

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0167</td>
<td>0.0155</td>
<td>0.0139</td>
<td>0.01</td>
</tr>
<tr>
<td>0.0869</td>
<td>0.0679</td>
<td>0.04217</td>
<td>0.10</td>
</tr>
<tr>
<td>0.7911</td>
<td>0.0975</td>
<td>0.03019</td>
<td>0.80</td>
</tr>
<tr>
<td>0.2075</td>
<td>0.0975</td>
<td>0.02788</td>
<td>0.20</td>
</tr>
<tr>
<td>1.1225</td>
<td>0.3737</td>
<td>0.1738</td>
<td>1.10</td>
</tr>
<tr>
<td>-1.5211</td>
<td>0.1910</td>
<td>0.6548</td>
<td>-1.50</td>
</tr>
<tr>
<td>2.4646</td>
<td>0.6003</td>
<td>1.4521</td>
<td>2.50</td>
</tr>
</tbody>
</table>

Table 21.2  Sequence length 1000, CPU time = 380.7 s

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0111</td>
<td>0.0037</td>
<td>0.0038</td>
<td>0.01</td>
</tr>
<tr>
<td>0.0956</td>
<td>0.0336</td>
<td>0.0304</td>
<td>0.10</td>
</tr>
<tr>
<td>0.8001</td>
<td>0.0301</td>
<td>0.0269</td>
<td>0.80</td>
</tr>
<tr>
<td>0.2014</td>
<td>0.0531</td>
<td>0.0511</td>
<td>0.20</td>
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<tr>
<td>1.0976</td>
<td>0.0639</td>
<td>0.0623</td>
<td>1.10</td>
</tr>
<tr>
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<td>0.0356</td>
<td>0.0361</td>
<td>-1.50</td>
</tr>
<tr>
<td>2.5004</td>
<td>0.0613</td>
<td>0.0585</td>
<td>2.50</td>
</tr>
</tbody>
</table>
### Analytic derivatives

#### Table 21.3 Sequence length 3000, CPU time = 1246.0 s

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0105</td>
<td>0.0020</td>
<td>0.0020</td>
<td>0.01</td>
</tr>
<tr>
<td>0.1000</td>
<td>0.0173</td>
<td>0.0174</td>
<td>0.10</td>
</tr>
<tr>
<td>0.7982</td>
<td>0.0163</td>
<td>0.0152</td>
<td>0.80</td>
</tr>
<tr>
<td>0.2015</td>
<td>0.0290</td>
<td>0.0296</td>
<td>0.20</td>
</tr>
<tr>
<td>1.0990</td>
<td>0.0210</td>
<td>0.0240</td>
<td>1.10</td>
</tr>
<tr>
<td>−1.5000</td>
<td>0.0180</td>
<td>0.0190</td>
<td>−1.50</td>
</tr>
<tr>
<td>2.5004</td>
<td>0.0096</td>
<td>0.0180</td>
<td>2.50</td>
</tr>
</tbody>
</table>

#### Table 21.4 Sequence length 300, CPU time = 141.8 s

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0167</td>
<td>0.0155</td>
<td>0.0097</td>
<td>0.01</td>
</tr>
<tr>
<td>0.0870</td>
<td>0.0677</td>
<td>0.0538</td>
<td>0.10</td>
</tr>
<tr>
<td>0.7970</td>
<td>0.0799</td>
<td>0.0559</td>
<td>0.80</td>
</tr>
<tr>
<td>0.2074</td>
<td>0.0976</td>
<td>0.0899</td>
<td>0.20</td>
</tr>
<tr>
<td>1.1228</td>
<td>0.3734</td>
<td>0.2577</td>
<td>1.10</td>
</tr>
<tr>
<td>−1.5209</td>
<td>0.1911</td>
<td>0.1627</td>
<td>−1.50</td>
</tr>
<tr>
<td>2.4639</td>
<td>0.5997</td>
<td>0.3922</td>
<td>2.50</td>
</tr>
</tbody>
</table>

#### Table 21.5 Sequence length 1000, CPU time = 520.2 s

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0111</td>
<td>0.0037</td>
<td>0.0036</td>
<td>0.01</td>
</tr>
<tr>
<td>0.0956</td>
<td>0.0336</td>
<td>0.0286</td>
<td>0.10</td>
</tr>
<tr>
<td>0.8001</td>
<td>0.0301</td>
<td>0.0260</td>
<td>0.80</td>
</tr>
<tr>
<td>0.2014</td>
<td>0.0531</td>
<td>0.0470</td>
<td>0.20</td>
</tr>
<tr>
<td>1.0976</td>
<td>0.0639</td>
<td>0.0586</td>
<td>1.10</td>
</tr>
<tr>
<td>−1.5008</td>
<td>0.0356</td>
<td>0.0339</td>
<td>−1.50</td>
</tr>
<tr>
<td>2.5004</td>
<td>0.0613</td>
<td>0.0554</td>
<td>2.50</td>
</tr>
</tbody>
</table>

#### Table 21.6 Sequence length 3000, CPU time = 1597.3 s

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0105</td>
<td>0.0020</td>
<td>0.0019</td>
<td>0.01</td>
</tr>
<tr>
<td>0.1000</td>
<td>0.0173</td>
<td>0.0165</td>
<td>0.10</td>
</tr>
<tr>
<td>0.7982</td>
<td>0.0163</td>
<td>0.0147</td>
<td>0.80</td>
</tr>
<tr>
<td>0.2015</td>
<td>0.0290</td>
<td>0.0270</td>
<td>0.20</td>
</tr>
<tr>
<td>1.0990</td>
<td>0.0210</td>
<td>0.0225</td>
<td>1.10</td>
</tr>
<tr>
<td>−1.5000</td>
<td>0.0180</td>
<td>0.0179</td>
<td>−1.50</td>
</tr>
<tr>
<td>2.5004</td>
<td>0.0096</td>
<td>0.0103</td>
<td>2.50</td>
</tr>
</tbody>
</table>
It can be seen that the estimated values are in agreement with the actual values, and as expected, the standard error of the parameter estimates decreases as the GARCH sequence length increases.

It is also evident that, for small sample sizes, the use of analytic derivatives leads to significantly better estimates of the standard errors. As the sequence length is increased both the numeric and analytic results become very similar. Here the numeric approach has the advantage of being considerably faster. This is because finite-difference approximations to the derivatives can be achieved by merely evaluating the log likelihood at several points in the neighbourhood of the current estimate for the parameter vector $\theta$. This is in contrast to analytic derivatives, which are computed recursively using all the terms in the GARCH sequence.

### 21.2.2 Student’s $t$ distribution

In Tables 21.7 to 21.10 we present the results of Monte Carlo simulations to check the parameter estimation software for the following Student’s $t$ regression-GJR–GARCH(1,2) process:

$$
egin{align*}
  k &= 2, 
  \alpha_0 &= 0.08, 
  \alpha_1 &= 0.05, 
  \alpha_2 &= 0.1, 
  \beta_1 &= 0.4, \\
  \gamma &= 0.2, 
  \nu &= 4.2, 
  b_0 &= 1.1, 
  b_1 &= -1.5, 
  b_2 &= 2.5, \\
  X_1^i &= \frac{1}{100} + 0.7 \times \sin\left(\frac{i}{100}\right), 
  X_2^i &= \frac{1}{2} + \left(\frac{i}{1000}\right), 
  \text{for } i = 1, \ldots, n
\end{align*}
$$

The initial values for the regression coefficients, $b_i$, $i = 0, \ldots, k$ and the pre-observed conditional variance $\sigma_0^2$ were estimated using OLS regression, as outlined in Section 20.1. The initial estimates for all the elements of the parameter vector $\omega$ were all set to 0.1. In addition the initial value for the number of degrees of freedom for the Student’s $t$ distribution, $\nu$, was taken as 100.0, which effectively assumes a Gaussian distribution as the starting approximation.

#### Numeric derivatives

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0820</td>
<td>0.0241</td>
<td>0.0354</td>
<td>0.08</td>
</tr>
<tr>
<td>0.0921</td>
<td>0.0726</td>
<td>0.1683</td>
<td>0.10</td>
</tr>
<tr>
<td>0.1064</td>
<td>0.0716</td>
<td>0.1941</td>
<td>0.10</td>
</tr>
<tr>
<td>0.3863</td>
<td>0.1138</td>
<td>0.2033</td>
<td>0.40</td>
</tr>
<tr>
<td>0.2176</td>
<td>0.0841</td>
<td>0.1451</td>
<td>0.20</td>
</tr>
<tr>
<td>4.4595</td>
<td>0.8290</td>
<td>0.8964</td>
<td>4.20</td>
</tr>
<tr>
<td>1.1016</td>
<td>0.0581</td>
<td>0.0726</td>
<td>1.10</td>
</tr>
<tr>
<td>-1.4973</td>
<td>0.0311</td>
<td>0.0354</td>
<td>-1.50</td>
</tr>
<tr>
<td>2.4970</td>
<td>0.0619</td>
<td>0.0767</td>
<td>2.50</td>
</tr>
</tbody>
</table>
Analytic Derivatives

The characteristics of these tables are similar to those of Section 21.2.1. However, here nine GARCH model parameters are estimated in contrast to the seven model parameters in Section 21.2.1. It is also interesting to note the high standard error associated with the Student’s $t$ distribution parameter $\nu$. 

### Table 21.8 Sequence length 3000, CPU time $= 1933.2$ s

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0802</td>
<td>0.0124</td>
<td>0.0108</td>
<td>0.08</td>
</tr>
<tr>
<td>0.0969</td>
<td>0.0371</td>
<td>0.0328</td>
<td>0.10</td>
</tr>
<tr>
<td>0.1007</td>
<td>0.0416</td>
<td>0.0394</td>
<td>0.10</td>
</tr>
<tr>
<td>0.3974</td>
<td>0.0645</td>
<td>0.0565</td>
<td>0.40</td>
</tr>
<tr>
<td>0.2059</td>
<td>0.0441</td>
<td>0.0404</td>
<td>0.20</td>
</tr>
<tr>
<td>4.2642</td>
<td>0.3563</td>
<td>0.3307</td>
<td>4.20</td>
</tr>
<tr>
<td>1.0972</td>
<td>0.0176</td>
<td>0.0190</td>
<td>1.10</td>
</tr>
<tr>
<td>−1.4985</td>
<td>0.0158</td>
<td>0.0152</td>
<td>−1.50</td>
</tr>
<tr>
<td>2.5011</td>
<td>0.0081</td>
<td>0.0087</td>
<td>2.50</td>
</tr>
</tbody>
</table>

### Table 21.9 Sequence length 800, CPU time $= 770.9$ s

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0820</td>
<td>0.0241</td>
<td>0.0257</td>
<td>0.08</td>
</tr>
<tr>
<td>0.0922</td>
<td>0.0724</td>
<td>0.1189</td>
<td>0.10</td>
</tr>
<tr>
<td>0.1063</td>
<td>0.0716</td>
<td>0.1116</td>
<td>0.10</td>
</tr>
<tr>
<td>0.3863</td>
<td>0.1138</td>
<td>0.1376</td>
<td>0.40</td>
</tr>
<tr>
<td>0.2176</td>
<td>0.0841</td>
<td>0.1056</td>
<td>0.20</td>
</tr>
<tr>
<td>4.4596</td>
<td>0.8289</td>
<td>0.8229</td>
<td>4.20</td>
</tr>
<tr>
<td>1.1016</td>
<td>0.0581</td>
<td>0.0634</td>
<td>1.10</td>
</tr>
<tr>
<td>−1.4973</td>
<td>0.0311</td>
<td>0.0314</td>
<td>−1.50</td>
</tr>
<tr>
<td>2.4970</td>
<td>0.0619</td>
<td>0.0678</td>
<td>2.50</td>
</tr>
</tbody>
</table>

### Table 21.10 Sequence length 3000, CPU time $= 2987.6$ s

<table>
<thead>
<tr>
<th>Estimated value</th>
<th>Estimated standard error</th>
<th>Standard error of estimates</th>
<th>Correct values</th>
</tr>
</thead>
<tbody>
<tr>
<td>0.0802</td>
<td>0.0124</td>
<td>0.0123</td>
<td>0.08</td>
</tr>
<tr>
<td>0.0969</td>
<td>0.0371</td>
<td>0.0390</td>
<td>0.10</td>
</tr>
<tr>
<td>0.1007</td>
<td>0.0416</td>
<td>0.0531</td>
<td>0.10</td>
</tr>
<tr>
<td>0.3974</td>
<td>0.0645</td>
<td>0.0706</td>
<td>0.40</td>
</tr>
<tr>
<td>0.2059</td>
<td>0.0441</td>
<td>0.0425</td>
<td>0.20</td>
</tr>
<tr>
<td>4.2642</td>
<td>0.3563</td>
<td>0.3287</td>
<td>4.20</td>
</tr>
<tr>
<td>1.0972</td>
<td>0.0176</td>
<td>0.0189</td>
<td>1.10</td>
</tr>
<tr>
<td>−1.4985</td>
<td>0.0158</td>
<td>0.0150</td>
<td>−1.50</td>
</tr>
<tr>
<td>2.5011</td>
<td>0.0081</td>
<td>0.0086</td>
<td>2.50</td>
</tr>
</tbody>
</table>
It has been shown that analytic derivatives provide more accurate results than numeric derivatives for short GARCH sequences. However, numeric derivatives are considerably faster. This suggests that practical GARCH software should have the ability to switch from analytic to numeric derivatives when appropriate. The precise benefits to be gained from using analytic derivatives will depend on the numerical optimization software used and the accuracy of the finite-difference approximations to the derivatives.

The results demonstrate that good GARCH model estimates can be obtained even when the initial parameter estimates are simple, see Section 22.5. This suggests the construction of easy to use software packages in which initial estimates (for $\omega$ and $\nu$) are not required from the user.
Chapter 22

GARCH process identification

In this chapter we consider the practical aspects of GARCH modelling. We deal with the statistical tests that can be performed on the modelled data in order to identify the best GARCH process. The results of using a GJR–GARCH model on S&P 500 index data are presented. Also two GARCH windows demonstrations are discussed which illustrate the practical use of the mathematics given earlier in this chapter. Each demonstration either uses a standard linear GARCH model or an AGARCH-I model, and the conditional probability distribution of the residuals is assumed to be Gaussian. Detailed information concerning the construction of these applications is provided elsewhere in the book.

22.1 LIKELIHOOD RATIO TEST

A popular approach for testing the significance of parameters estimated using maximum likelihood techniques is the **likelihood ratio test**. It is useful in the following situation.

Suppose we have modelled data using a GARCH process $N_p$ parameters, $\theta_k$, $k = 1, \ldots, N_p$, and have obtained a maximized log likelihood $\mathcal{L}(\theta)$. We now want to know if by increasing the number of model parameters to $N_p + m$ we can obtain a significantly better model to the data. If we let the (improved) maximized log likelihood using the increased number of parameters be $\mathcal{L}(\hat{\theta})$, then we can use the result that:

$$2 \left[ \mathcal{L}(\hat{\theta}) - \mathcal{L}(\theta) \right] \approx \chi^2(m)$$

22.2 SIGNIFICANCE OF THE ESTIMATED PARAMETERS

As described in Section 18.2.1 the significance of an estimated parameter can be determined by the value of its $t$ statistic. For the $i$th parameter estimate $\hat{\theta}_i$, the $t$ statistic is $\frac{\hat{\theta}_i}{\sigma_i}$, where $\sigma_i$ is the standard error.

It is common practice to reject the null hypothesis of no significance at the 0.5 per cent level, in which case estimates with $t$ statistic values greater than 2.57 are considered significant.

22.3 THE INDEPENDENCE OF THE STANDARDIZED RESIDUALS

In Section 15.2 we showed that the GARCH(p,q) process:

$$h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j \epsilon_{i-j}^2 + \sum_{j=1}^{p} \beta_j h_{i-j}, \quad i = 1, \ldots, n, \quad \epsilon_i \sim R(0, h_i)$$

would be satisfied when the standardized residuals

$$\frac{\epsilon_i}{\sqrt{h_i}} \sim R(0, 1)$$

are independent.
Gives rise to an ARMA($\kappa$, $p$) process in $\epsilon_i^2$ of the form:

$$
\epsilon_i^2 = \alpha_0 + \sum_{j=1}^{p} (\alpha_j + \beta_j) \epsilon_{i-j}^2 - \sum_{j=1}^{p} \beta_j \nu_{i-j} + \nu_i
$$

where $\kappa = \max(p, q)$. If the model is correctly specified then the standardized sequence $Z_i^2 = \epsilon_i^2 / \nu_i$, $i = 1, n$ should constitute white noise. This can be checked by computing the sample autocorrelations.

The statistical independence of the elements $Z_i^2$ can be checked by computing the values of the sample autocorrelations.

The $k$th sample autocorrelation is defined as:

$$
r_k = \sum_{i=1}^{n} Z_i^2 Z_{i-k}^2, \quad i = k + 1, \ldots, n \tag{22.1}
$$

The Box–Pierce $Q$–statistic is defined as:

$$
Q_{\text{stat}} = \sum_{k=1}^{p} r_k \tag{22.2}
$$

If the model is correctly specified then $Q_{\text{stat}}$ has a $\chi^2_2$ distribution with $P - \kappa - q$ degrees of freedom. High values of $Q_{\text{stat}}$ lead to reject of the hypothesis that the standardized residuals are independently distributed.

### 22.4 THE DISTRIBUTION OF THE STANDARDIZED RESIDUALS

The standardized residuals $Z_i = \epsilon_i / \sqrt{\nu_i}$, $i = 1, \ldots, n$ should have the distribution $\mathcal{R}(0, 1)$.

In the case of $\mathcal{R}(0, 1)$ being a Gaussian distribution $N(0, 1)$ we can check for non-normality in the following manner.

Remembering that the kurtosis is defined as:

$$
\kappa = \frac{E[\epsilon_i^4]}{\sigma_i^4}
$$

and that the skewness is defined as:

$$
\mathcal{S} = \frac{E[\epsilon_i^3]}{\sigma_i^3}
$$

where $E[\epsilon_i^4] = \frac{1}{n} \sum_{i=1}^{n} \epsilon_i^4$, $E[\epsilon_i^3] = \frac{1}{n} \sum_{i=1}^{n} \epsilon_i^3$, and $\sigma^2 = \frac{1}{n} \sum_{i=1}^{n} \epsilon_i^2$

For large samples we have:

$$
\mathcal{S} \sim N(0, 6/n) \quad \text{and} \quad \kappa \sim N(3, 24/n)
$$

A test statistic for non-normality of the residuals is given by:

$$
N_{\text{stat}} = \frac{n}{6} \mathcal{S}^2 + \frac{n}{24} (\kappa - 3) \tag{22.3}
$$
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Under the null hypothesis $N_{\text{stat}}$ has a $\chi^2_2$ distribution, Harvey (1990). In practical terms this means that if $N_{\text{stat}} < 3$ then we can reject the alternative hypothesis of non-normality in favour of normality.

22.5 MODELLING THE S&P 500 INDEX

This section concerns the use of GJR–GARCH to model 3000 daily returns from the S&P 500 index, for the years 1960–1972; see Levy (2003) for more details.

Analytic derivatives were used in the numerical optimization, and the shocks were either from a Gaussian distribution or a Student’s $t$ distribution. Tables 22.1 and 22.2 show the maximized log likelihood, $LGF(\theta)$, and parameter estimates for GJR–GARCH(1,1), AR(1)–GJR–GARCH(1,1), and AR(2)–GJR–GARCH(1,1) models. The format of these results is $\hat{\theta} (\Delta \hat{\theta}) t$, where $\hat{\theta}$ is the vector of estimated model parameters, $\Delta \hat{\theta}$ is the vector of estimated standard errors, and $t$ is the vector of significance statistics, $t = \hat{\theta} / \Delta \hat{\theta}$. Here we consider parameters with $t > 1.96$ (i.e. 2.5 per cent probability level) as significant.

It is evident from Table 22.1 that the preferred Gaussian model for the S&P 500 index data is AR(1)–GJR–GARCH(1,1), with $\alpha_0 = 0.0196$, $\alpha_1 = 0.0716$, $\beta_1 = 0.7938$, $\gamma = 0.1851$, $c = 0.0267$, and $\phi_1 = 0.2280$.

The results for the Student’s $t$ distribution in Table 22.2 show that the log likelihood surface is very flat, and the parameters $\phi_1$, and $\phi_2$ are not significant at the 2.5 per cent probability level. Here the preferred model for the S&P 500 index data is GJR–GARCH(1,1), with $\alpha_0 = 0.0128$, $\alpha_1 = 0.0545$, $\beta_1 = 0.8373$, $\gamma = 0.1568$, $\nu = 8.1160$, and $c = 0.0483$.

It was found that the optimized parameter estimates did not depend strongly on the initial estimates. This observation was investigated by studying the shape of the log likelihood surface for both the Gaussian distribution AR(1)–GJR–GARCH(1,1) model, and the Student’s $t$ distribution GJR–GARCH(1,1) model. Figures 22.1 and 22.2 show the results for analytic derivatives. Here the value of $LGF(\theta)$ is plotted as each GARCH parameter is individually incremented in steps of 0.04, while all

Table 22.1 Gaussian distribution, estimated model parameters. $\hat{\theta}$ is the vector of estimated model parameters, $\Delta \hat{\theta}$ is the vector of estimated standard errors, $t = \hat{\theta} / \Delta \hat{\theta}$ is the vector of significance statistics, and $LGF(\hat{\theta})$ is the value of the maximized log likelihood at $\hat{\theta}$

<table>
<thead>
<tr>
<th>Parameters</th>
<th>GJR–GARCH(1,1)</th>
<th>AR(1)–GJR–GARCH(1,1)</th>
<th>AR(2)–GJR–GARCH(1,1)</th>
</tr>
</thead>
<tbody>
<tr>
<td>$\theta$</td>
<td>$LGF(\hat{\theta}) = 132.716$</td>
<td>$LGF(\hat{\theta}) = 198.32$</td>
<td>$LGF(\hat{\theta}) = 134.23$</td>
</tr>
<tr>
<td>$\phi_1$</td>
<td>0.0189 (0.0028) [6.78]</td>
<td>0.0196 (0.0028) [6.83]</td>
<td>0.0184 (0.0027) [6.71]</td>
</tr>
<tr>
<td>$\phi_2$</td>
<td>0.0680 (0.0131) [5.19]</td>
<td>0.0716 (0.0132) [5.41]</td>
<td>0.0663 (0.0129) [5.12]</td>
</tr>
<tr>
<td>$\phi_2$</td>
<td>0.8106 (0.0162) [50.01]</td>
<td>0.7938 (0.0167) [47.35]</td>
<td>0.8135 (0.0160) [50.07]</td>
</tr>
<tr>
<td>$\gamma$</td>
<td>0.1524 (0.0203) [7.48]</td>
<td>0.1851 (0.0243) [7.60]</td>
<td>0.1533 (0.0205) [7.46]</td>
</tr>
<tr>
<td>$\phi_3$</td>
<td>0.0458 (0.0087) [5.23]</td>
<td>0.0267 (0.0085) [3.11]</td>
<td>0.0420 (0.0089) [4.72]</td>
</tr>
<tr>
<td>$\phi_4$</td>
<td>0.2280 (0.0191) [11.94]</td>
<td>0.0180 (0.0201) [0.89]</td>
<td>0.0279 (0.0197) [1.41]</td>
</tr>
</tbody>
</table>
Table 22.2 Student's $t$ distribution, estimated model parameters. $\hat{\theta}$ is the vector of estimated model parameters, $\Delta \hat{\theta}$ is the vector of estimated standard errors, $t = \hat{\theta} / \Delta \hat{\theta}$ is the vector of significance statistics, and $LGF(\hat{\theta})$ is the value of the maximized log likelihood at $\hat{\theta}$.

<table>
<thead>
<tr>
<th>Parameters</th>
<th>GJR–GARCH(1,1)</th>
<th>AR(1)–GJR–GARCH(1,1)</th>
<th>AR(2)–GJR–GARCH(1,1)</th>
</tr>
</thead>
<tbody>
<tr>
<td>$\theta$</td>
<td>$LGF(\hat{\theta}) = -2554.96$</td>
<td>$LGF(\hat{\theta}) = -2554.95$</td>
<td>$LGF(\hat{\theta}) = -2553.17$</td>
</tr>
<tr>
<td>$\alpha_0$</td>
<td>0.0128 (0.0029) [4.35]</td>
<td>0.0128 (0.0030) [4.21]</td>
<td>0.0125 (0.0032) [3.92]</td>
</tr>
<tr>
<td>$\alpha_1$</td>
<td>0.0545 (0.0149) [3.65]</td>
<td>0.0544 (0.0153) [3.55]</td>
<td>0.0521 (0.0147) [3.54]</td>
</tr>
<tr>
<td>$\beta_1$</td>
<td>0.8373 (0.0208) [40.21]</td>
<td>0.8373 (0.0216) [38.81]</td>
<td>0.8402 (0.0224) [37.43]</td>
</tr>
<tr>
<td>$\gamma$</td>
<td>0.1568 (0.0236) [6.64]</td>
<td>0.1570 (0.0236) [6.65]</td>
<td>0.1573 (0.0705) [2.23]</td>
</tr>
<tr>
<td>$\nu$</td>
<td>8.1160 (1.0360) [7.83]</td>
<td>8.1260 (1.0910) [7.44]</td>
<td>8.1080 (3.0980) [2.61]</td>
</tr>
<tr>
<td>$c$</td>
<td>0.0483 (0.0092) [5.27]</td>
<td>0.0481 (0.0095) [5.04]</td>
<td>0.0451 (0.0264) [1.71]</td>
</tr>
<tr>
<td>$\phi_1$</td>
<td>0.0021 (0.0194) [0.11]</td>
<td>0.0021 (0.0194) [0.11]</td>
<td>0.0010 (0.1001) [0.01]</td>
</tr>
<tr>
<td>$\phi_2$</td>
<td>0.0351 (0.1191) [0.29]</td>
<td>0.0351 (0.1191) [0.29]</td>
<td>0.0351 (0.1191) [0.29]</td>
</tr>
</tbody>
</table>

Figure 22.1 The partial log likelihood surface for the Gaussian AR(1)–GJR–GARCH(1,1) model presented in Table 22.1. In order to display the results on a single graph symbols have been used which incorporate various scale factors. The symbol definitions are as follows: $LF(\alpha_0) = LGF(\alpha_0)\hat{\theta}$, $LF(\alpha_1) = 0.5 \times LGF(\alpha_1)\hat{\theta}$, $LF(\beta_1) = 0.2 \times LGF(\beta_1)\hat{\theta}$, $LF(\gamma) = 3 \times LGF(\gamma)\hat{\theta}$, $LF(c) = LGF(c)\hat{\theta}$, $LF(\phi_1) = 4 \times LGF(\phi_1)\hat{\theta}$. The parameter values range from 0 to 0.76, with an increment of 0.04.
the other GARCH parameters are held fixed at the values which maximize the log likelihood function. For the $k$th GARCH model parameter, $\theta_k$, we will denote this partial log likelihood function by $LGF(\theta_k)$. Inspection of Figures 22.1 and 22.2 shows that, to within the step tolerance, the location of the partial log likelihood maxima are in agreement with the parameter estimates given in Tables 22.1 and 22.2. It can also be seen that the partial log likelihood surface is both smooth and convex. This explains why a Newton-type numerical optimizer can converge to a global maximum even when poor initial estimates are supplied.

22.6 EXCEL DEMONSTRATION

This demonstration is concerned with modelling currency exchange rate returns data. It illustrates how to identify the best GARCH model to suit a particular time series, and is composed of the following components:

- Two ActiveX plot controls, to display the original data and also the modelled standardized residuals $Z_i = \epsilon_i / \sqrt{h_i}$, $i = 1, \ldots, n$.
- Microsoft TextBox controls to allow the user to select the order of the GARCH model.
Microsoft Radio controls to allow the user to select the type of GARCH model.

Microsoft Button controls to perform actions such as: Clear, Calculate, and Show Data.

A Microsoft Grid control to display the modelled values and parameter estimates.

It can be seen that the user has the choice of selecting either an AGARCH-I or an AGARCH-II model, and there is also an asymmetry option. It should be noted that when AGARCH-I is used without asymmetry this is equivalent to the standard linear GARCH model. All the results presented here are for an AGARCH-I model either with or without asymmetry.

We found the Microsoft Grid control to be a very versatile means of showing information. In this example it was used to display the following results:

- The initial parameter estimates. Although they were all automatically set to 0.1, they could if required be edited to different values.
- The computed parameter estimates, \( \hat{\theta}_k, k = 1, \ldots, N_p \).
- The standard errors, \( \sigma_k, k = 1, \ldots, N_p \).
- The significance statistics for the estimated parameters, \( T_k = \hat{\theta}_k / \sigma_k, k = 1, \ldots, N_p \).
- The value of the log likelihood for the estimated parameter values, \( -L(\theta) \).
- The partial derivatives (also termed scores) \( \partial L(\theta) / \partial \theta_k, k = 1, \ldots, N_p \) for the estimated parameters.
- The normality test statistic \( N_{stat} \) described in Section 22.4.

The top graph in Figures 22.3 to 22.7 plots the exchange rate returns data, and is identical in each figure. It can be seen that the data clearly exhibits volatility clustering. The bottom graph in Figures 22.3 to 22.7 plots the standardized residuals \( Z_i = \epsilon_i / \sqrt{h_i}, i = 1, \ldots, n \). If the data has been modelled well then these values should have the distribution \( \text{NID}(0,1) \). This means that the closer the lower graph corresponds to white noise the better the GARCH model. Of course it may be difficult to perform this appraisal visually, so that is why we also report the normality test statistic. We could also look at the autocorrelation structure, but we have not done that here.

Figure 22.3 shows the results of using a simple GARCH(0,1), that is ARCH(1), on the data. The log likelihood is 4122.57 and the normality test statistic, \( N_{stat} \), is 1499.48. This value of \( N_{stat} \) is very large, and we can clearly see clustering in the lower graph. In Figures 22.4 and 22.5 we experiment with using high order ARCH models to describe the data.

Figure 22.4 shows the results of using a GARCH(0,10). Here the log likelihood is 4337.0 and the normality test statistic, \( N_{stat} \), is 2.922. It can be seen that only the parameters \( \alpha_0, \alpha_1, \alpha_2, \) and \( \alpha_5 \) have \( t \) statistic values above 3.0. This model is certainly better than the simple GARCH(0,1) model since the log likelihood is higher and also \( N_{stat} \) is considerably lower.

Figure 22.5 shows the results of using an AGARCH-I(0,10). The inclusion of asymmetry has improved on the GARCH(0,10) model, since the log likelihood has now increased to 4353.97 and \( N_{stat} \) has reduced to 1.562. It can be seen that the parameters \( \alpha_0, \alpha_1, \alpha_2, \alpha_4, \alpha_5, \) and the asymmetry parameter \( \gamma \) have \( t \) statistic values above 3.
Figure 22.3  Modelling the data with GARCH(0,1)

Figure 22.4  Modelling the data with GARCH(0, 2)
**GARCH process identification**

Figure 22.5 Modelling the data with GARCH(0,10)

Figure 22.6 Modelling the data with GARCH(1,1)
In Figures 22.6 and 22.7 we see if a GARCH(1,1) model will do better than the ARCH(10) models.

Figure 22.6 shows the results of using a GARCH(1,1). Here the likelihood is 4340.95 and the normality test statistic, $N_{\text{stat}}$, is 12.39. All the parameters have $t$ statistic values above 3, and the value $\beta_1$ is 56.922. However, these results are not as good as those we obtained from the AGARCH-I(0,10) model.

Figure 22.7 shows the results of using an AGARCH-I(1,1). Here the likelihood is 4357.67 and the normality test statistic, $N_{\text{stat}}$, is 0.575. All the parameters have $t$ statistic values above 3, and the value $\beta_1$ is 59.886. Clearly this is the preferred model is an AGARCH-I(1,1) model with:

\[
\begin{align*}
\alpha_0 &= 5.55 \times 10^{-5}, & \alpha_1 &= 0.128, & \beta_1 &= 0.852, & \gamma &= 0.0175
\end{align*}
\]

It is interesting to note that the asymmetry parameter $\gamma$ is positive. This is because we are modelling currency exchange rate returns data and (since it is not clear that negative exchange rate returns indicate bad news) there is no preference in the sign of the asymmetry. However, if we were modelling stock market returns data we would expect $\gamma$ to be negative.

22.7 INTERNET EXPLORER DEMONSTRATION

This demonstration illustrates how GARCH modelling could be carried out with a Web Browser on an Internet Web page.
It shows how ActiveX components on a Web page can be used to model the volatility of the share price for a particular company. This Web page contains the following ActiveX components:

- Three Microsoft ActiveX button controls
- Two ActiveX graphical controls, called Plot1 and Plot2
- The GARCH modelling ActiveX control, GARCH1.

All the data input and computation is performed by the aggregated ActiveX component GARCH1. This control was created using Visual Basic and uses textboxes to input the model parameters and company name. It also extracts company share returns from a Microsoft Access database and models this data by calling computational GARCH routines contained within a Visual C++ DLL.

Figures 22.8 to 22.10 show how this demonstration is used. Appropriate values of p, q and the company name are entered into the textboxes of GARCH1. The data corresponding to the selected company (in this case it is merely called ASSET) is displayed on the lower plot region by clicking the button labelled ‘Show Data’. This runs the subroutine Show_Data_Click() which calls the EXTRACT_DATA property of GARCH1 to retrieve data from the Access database and then display it using the ActiveX component Plot2. When the button labelled ‘Calculate’ is clicked Calculate_GARCH_Click() is run and a GARCH(p,q) process used to model the data. If the user wishes to try alternative value of p and q, then clicking the ‘Clear’ button deletes the previous results and the data can be remodelled.

**Figure 22.8** The original data displayed on the Web page
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**Figure 22.9** Modelling the data with ARCH(1)

**Figure 22.10** Modelling the data with ARCH(4)
Chapter 23

Multivariate time series

So far we have been concerned with modelling the volatility of single assets. However, most practical financial applications consist of portfolios containing many assets, and it is therefore necessary to model multivariate time series and their associated time varying covariance matrices.

One of the main problems connected with doing this is the number of parameters that may require estimating. For instance a typical portfolio consisting of 100 assets has a $100 \times 100$ covariance matrix which is described by 5050 terms. One way round this problem is to use principal component analysis to reduce the number of parameters which describe the multivariate process.

23.1 PRINCIPAL COMPONENT GARCH

Principal component or orthogonal GARCH provides a parsimonious way in which to model multivariate time series, see Ding (1994) and Alexander (2000).

Consider $T$ observations of $m$ variables and let these be represented by the $T \times m$ matrix $Y$, so that the $i$th column of $Y$, $Y^i$, contains the $T$ observations of the $i$th variable. We will also denote the row vector containing the values of the $m$ variables at instant $t$ by $Y_t$. If the unconditional mean of the $Y^i$ is $\mu_i$ then we can construct the matrix $X$, where the $i$th column of $X$ is $X^i = Y^i - \mu_i$, and the $m \times m$ covariance matrix, $C$, of the data matrix $Y$ is given by:

$$ C = X^T X $$  \hspace{1cm} (23.1)

Since the covariance matrix is symmetric and positive definite we can obtain the following spectral decomposition

$$ C = W \Sigma W^T $$  \hspace{1cm} (23.2)

where the columns of the $m \times m$ matrix $W$ contain the eigenvectors of $C$ and the elements of the $m \times m$ diagonal matrix $\Sigma$ contain the corresponding eigenvalues, $\lambda_i$, $i = 1, \ldots, m$.

If $k$ of the eigenvalues are much greater than the other $m - k$ eigenvalues, then a good approximation to the covariance matrix is:

$$ C \approx W_k \Sigma_k W_k^T $$
where \( \Sigma_k \) is the \( k \times k \) diagonal matrix containing the \( k \)th largest eigenvalues, and \( W_k \) is the \( m \times k \) matrix of eigenvectors formed using the appropriate \( k \) columns of \( W \).

The time-dependent scores, \( S_i^t, \ t = 1, \ldots, T \), corresponding to the \( i \)th eigenvector (principal component) are calculated as:

\[
S_i^t = X_i W^i, \quad t = 1, \ldots, T
\]

where \( S_i^t \) is the \((T \times 1)\) score vector at time instant \( t \) corresponding to the \( i \)th eigenvalue, \( X_i \) is the the row of the \((T \times m)\) matrix \( X \) in Equation 23.1 corresponding to time \( t \), and \( W^i \) is the corresponding \((m \times 1)\) eigenvector.

In matrix notation, if the \( k \) largest eigenvalues are used, then we obtain the following scores matrix:

\[
S = X W_k
\]

where \( S \) is now a \( T \times k \) matrix, \( X \) is a \( T \times m \) matrix, and \( W_k \) is a \( m \times k \) matrix.

All the columns of \( S \) (that is the score vector corresponding to each principal component) are orthogonal, and so \( S^T S \) results in the diagonal matrix \( \Sigma \). This can easily be shown as follows:

\[
S^T S = (XW)^T X W
\]

\[
= W^T (X^T X) W
\]

\[
= W^T C W
\]

\[
= \Sigma
\]

The variance of the score vector for the \( i \)th principal component is thus equal to the \( i \)th eigenvalue, \( \lambda_i \). That is:

\[
E[S_i^T S_i^t] = \sum_{i=1}^{T} S_i^t S_i^t = \lambda_i, \quad i = 1, \ldots, k
\]

The instantaneous covariance matrix at time \( t \), \( V_t \), can thus be approximated as follows:

\[
V_t = X_t^T X_t
\]

where \( X_t \) is the row of matrix \( X \) corresponding to time \( t \).

From Equation 23.3 we have \( S = X W \) and therefore:

\[
SW^T = X W W^T = X, \quad \text{since} \quad WW^T = I
\]

Thus \( X = SW^T \) and \( X^T X = WS^T SW^T \).

Since \( S^T S \) is diagonal we can write \( V_t \) as:

\[
V_t = W \Omega_t W^T, \quad t = 1, \ldots, T
\]

where each time dependent variance, \( \Omega_t \), \( i = 1, \ldots, k \), in the diagonal matrix \( \Omega_t \) is modelled using a \textit{univariate} GARCH process. Note: The eigenvectors \( W \) are assumed to be time independent.
For example, suppose we want to model the time dependent covariance matrix of three exchange rate return series \( r^i_j, \) \( i = 1, \ldots, 3, \) \( j = 1, \ldots, T, \) then we would employ the steps explained below.

**Construct the covariance matrix, \( C \)**

The value of the element in the \( i \)th row and \( j \)th column of matrix \( C \) is:

\[
C_{i,j} = \sum_{k=1}^{T} (r^i_k - \bar{r}^i)(r^j_k - \bar{r}^j), \quad i = 1, \ldots, 3, \quad j = 1, \ldots, 3
\]

where \( \bar{r}^i \) is the mean value of the \( i \)th return series.

**Form the eigenvalue decomposition**

From Equation 23.2 we have:

\[
C = W\Sigma W^T
\]

where the \( i \)th eigenvector is the column vector \( W^i = (W^i_1, W^i_2, W^i_3)^T \). We will assume that \( \lambda_1 \sim \lambda_2, \) and \( \lambda_1 \gg \lambda_3, \) so \( k = 2. \)

**Use the eigenvectors to form the scores**

From Equation 23.1 the scores are given by \( S = XW_k, \) where \( k = 2 \)

**The scores for the first principal component are:**

\[
S^1_1 = X^1_1 W^1_1 + X^2_2 W^2_2 + X^3_3 W^3_3 \\
S^1_2 = X^1_2 W^1_1 + X^2_2 W^2_2 + X^3_3 W^3_3 \\
S^1_3 = X^1_3 W^1_1 + X^2_2 W^2_2 + X^3_3 W^3_3 \\
\vdots
\vdots
\vdots
S^1_T = X^T_1 W^1_1 + X^T_2 W^2_2 + X^T_3 W^3_3
\]

**The scores for the second principal component are:**

\[
S^2_1 = X^2_1 W^2_1 + X^2_2 W^2_2 + X^3_3 W^3_3 \\
S^2_2 = X^2_1 W^2_1 + X^2_2 W^2_2 + X^3_3 W^3_3 \\
S^2_3 = X^2_1 W^2_1 + X^2_2 W^2_2 + X^3_3 W^3_3 \\
\vdots
\vdots
\vdots
S^2_T = X^T_1 W^2_1 + X^T_2 W^2_2 + X^T_3 W^3_3
\]
From Equation 23.5 we use:

Use an appropriate univariate GARCH process to model the variance of the scores, 

\[ h_t^i = (S_t^i)^2 = \Omega_t^i, \quad i = 1, \ldots, k, \quad t = 1, \ldots, T \]

for each sequence \( S_t^i, \ t = 1, \ldots, T \), where \( i \leq k \). So, using \( k = 2 \) and assuming a basic GARCH(\( p,q \)) process, then the two diagonal elements (\( \Omega_t^1 \) and \( \Omega_t^2 \)) of \( \Omega_t \) are modelled as follows:

The GARCH(\( p_1,q_1 \)) process for the first principal component score vector is:

\[ \Omega_t^1 = (S_t^1)^2 = h_t^1 = \alpha_0^1 + \sum_{j=1}^{p_1} \alpha_j^1 \varepsilon_{t-j}^1 + \sum_{j=1}^{q_1} \beta_j^1 h_{t-j}, \quad t = 1, \ldots, T \]

and the GARCH(\( p_2,q_2 \)) process for the second principal component score vector is:

\[ \Omega_t^2 = (S_t^2)^2 = h_t^2 = \alpha_0^2 + \sum_{j=1}^{p_2} \alpha_j^2 \varepsilon_{t-j}^2 + \sum_{j=1}^{q_2} \beta_j^2 h_{t-j}, \quad t = 1, \ldots, T \]

### Model the univariate scores using GARCH

From Equation 23.5 we use:

\[ V_t = W \Omega_t W^T, \quad t = 1, \ldots, T \]

For this example it is easy to perform the matrix multiplications as follows:

\[
V_t = \begin{pmatrix} V_{11} & V_{12} & V_{13} \\ V_{21} & V_{22} & V_{23} \\ V_{31} & V_{32} & V_{33} \end{pmatrix} = \begin{pmatrix} W_{11}^1 & W_{11}^2 \\ W_{21}^1 & W_{21}^2 \\ W_{31}^1 & W_{31}^2 \end{pmatrix} \begin{pmatrix} h_t^1 & h_t^2 & h_t^3 \\ 0 & h_t^1 & h_t^2 \\ 0 & 0 & h_t^1 \end{pmatrix} \begin{pmatrix} W_{11}^1 & W_{12}^1 & W_{13}^1 \\ W_{22}^2 & W_{22}^2 & W_{22}^2 \\ W_{33}^3 & W_{33}^3 & W_{33}^3 \end{pmatrix}
\]

This yields the orthogonal GARCH estimate of the time dependent covariance matrix as:

\[
V_t = \begin{pmatrix} h_t^1 W_{11}^1 W_{11}^1 + h_t^2 W_{12}^2 W_{11}^1 & h_t^1 W_{11}^1 W_{12}^2 + h_t^2 W_{12}^2 W_{12}^2 & h_t^1 W_{11}^1 W_{13}^1 + h_t^2 W_{12}^2 W_{13}^2 \\ h_t^1 W_{21}^1 W_{11}^1 + h_t^2 W_{21}^2 W_{12}^2 & h_t^1 W_{21}^1 W_{12}^2 + h_t^2 W_{21}^2 W_{12}^2 & h_t^1 W_{21}^1 W_{13}^1 + h_t^2 W_{21}^2 W_{13}^2 \\ h_t^1 W_{31}^1 W_{11}^1 + h_t^2 W_{31}^2 W_{12}^2 & h_t^1 W_{31}^1 W_{12}^2 + h_t^2 W_{31}^2 W_{12}^2 & h_t^1 W_{31}^1 W_{13}^1 + h_t^2 W_{31}^2 W_{13}^2 \end{pmatrix}
\]

More details concerning orthogonal GARCH models can be found in Van der Weide (2002).
Appendix A

Computer code for Part I

This Appendix contains complete code for the examples referenced in the main text.

A.1 THE ODL FILE FOR THE DERIVATIVE PRICING CONTROL

The complete ODL file for the derivative pricing control used in Section 3.3 and Chapter 4 is given below.

```c
// NAGDBS.odl : type library source for ActiveX Control project.
// This file will be processed by the Make Type Library (mktyplib) tool to
// produce the type library (NAGDBS.tlb) that will become a resource in
// NAGDBS.ocx.
#include <olectl.h>
#include <idispids.h>
[uuid(8B7F2A94—E828—11D2—AD08—0060087ED9F1),version(1.0),
helpfile(''NAGDBS.hlp''),
helpstring(''NAGDBSActiveX Control module''),
control] library NAGDBSLib
{
#include <slectl.h>
#include <idispid.h>
[uuid(8B7F2A94—E828—11D2—AD08—0060087ED9F1),version(1.0),
helpfile(''NAGDBS.hlp''),
helpstring(''NAGDBSActiveX Control module''),
control] library NAGDBSLib
{
importlib(STDOLE_TLB);
importlib(STDTYPE_TLB);
typedef enum
{
[helpstring(''Call'')] Call = 0,
[helpstring(''Put'')] Put = 1
} PUTCALLTYPE;
typedef enum
{
[helpstring(''European'')] European = 0,
[helpstring(''American'')] American = 1,
[helpstring(''Cntrl_American'')] Cntrl_American = 2
} EXTTYPE;
typedef enum
{
[helpstring(''Lattice u = 1/d'')] Lattice_standard = 0,
[helpstring(''Lattice p = 1/2'')] Lattice_prob_half = 1,
[helpstring(''Analytic'')] Analytic = 2
} METHODTYPE;
// Primary dispatch interface for CNAGDBSCtrl
[uuid(8B7F2A95—E828—11D2—AD08—0060087ED9F1),
helpstring("Dispatch interface for NAGDBS Control"), hidden] disinterface_DNAGDBS
{
properties:
// NOTE — ClassWizard will maintain property
// information here.
// Use extreme caution when editing this section.
// {AFX_ODL_PROP(CNAGDBSCtrl)
[id(1)] METHODTYPE method;
[id(2)] EXTTYPE extype;
[id(3)] double sigma;
[id(4)] long numsteps;
```
Appendix A

```c++
[id(5)] double intrate;
[id(6)] double dividends;
[id(7)] double curval;
[id(8)] double optval;
[id(9)] double strike;
[id(10)] FUTALLTYPE putcall;
[id(11)] double maturity;
[id(DISPID_CAPTION), bindable, requestedit] BSTR Caption;
[id(DISPID_BACKCOLOR), bindable, requestedit] OLE_COLOR BackColor;
[id(DISPID_FORECOLOR), bindable, requestedit] OLE_COLOR ForeColor;
//}}AFX_ODL_PROP

methods:
// NOTE — ClassWizard will maintain method information here.
// Use extreme caution when editing this section.
//{{AFX_ODL_METHOD(CNAGDBSCtrl)
[id(12)] void Calculate();
[id(13)] void greeks(double* greekvals);
//}}AFX_ODL_METHOD

[id(DISPID_ABOUTBOX)] void AboutBox();
//}}AFX_ODL_PROP

//}}AFX_ODL_PROP

//}}AFX_APPEND_ODL

//}}AFX_APPEND_ODL
```
Appendix B

Some more option pricing formulae

In this section we list some more derivative pricing formulae; use is made of the following symbols:

\[
\begin{align*}
  d_1 &= \frac{\log(S/E) + (r - q + \sigma^2/2)\tau}{\sigma\sqrt{\tau}} \\
  d_2 &= \frac{\log(S/E) + (r - q - \sigma^2/2)\tau}{\sigma\sqrt{\tau}} = d_1 - \sigma\sqrt{\tau}
\end{align*}
\]  

(B.1) (B.2)

B.1 BINARY OPTIONS

A binary cash or nothing call option pays nothing if the stock price ends up below the strike and an amount \(Q\) if it ends up above the strike price. The value is:

\[V_c = Q \exp(-r\tau)N_1(d_2)\]  

(B.3)

A binary asset or nothing call option pays nothing if the stock price ends up below the strike and the stock price itself if it ends up above the strike price. The value is:

\[V_c = S \exp(-r\tau)N_1(d_1)\]  

(B.4)

B.2 OPTION TO EXCHANGE ONE ASSET FOR ANOTHER

\[V = S_2 \exp(-q_2\tau)N_1(d_1 - 1) - S_1 \exp(-q_1\tau)N_1(d_2)\]  

(B.5)

where

\[
\begin{align*}
  d_1 &= \frac{\log(S_2/S_1) + (q_1 - q_2 + \sigma^2/2)\tau}{\sigma\sqrt{\tau}} \\
  d_2 &= d_1 - \sigma\sqrt{\tau}
\end{align*}
\]

and

\[\sigma = \sqrt{\sigma_1^2 + \sigma_2^2 - 2\rho\sigma_1\sigma_2}\]

It is interesting to note that this formula is independent of the interest rate \(r\).
B.3 LOOKBACK OPTIONS

The value of a European lookback call at time zero is:

\[
V_c = S \exp(-q\tau) \left\{ N_1(a_1) - \frac{\sigma^2}{2(r-q)} N_1(-a_1) \right\} \\
- S_{min} \exp(-r\tau) \left\{ N_1(a_2) - \frac{\sigma^2}{2(r-q)} \exp(Y_1) N_1(-a_3) \right\}
\]

where

\[
a_1 = \frac{\log(S/S_{min}) + (r-q + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}, \quad a_2 = a_1 - \sigma\sqrt{\tau} \\
a_3 = \frac{\log(S/S_{min}) + (-r+q + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}, \quad Y_1 = -\frac{\log(S/S_{min})2(-r-q-\sigma^2/2)}{\sigma^2}
\]

and \( S_{min} \) is the minimum stock price achieved to date. If the lookback option has just been originated then \( S_{min} = S \) and the valuation simplifies to:

\[
V_c = S \exp(-q\tau) \{ N_1(g_1) - WN_1(-g_1) \} - S \exp(-r\tau) \{ N_1(g_2) - WN_1(g_2) \}
\]

where

\[
W = \frac{\sigma^2}{2(r-q)}, \quad g_1 = \frac{(r-q + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}, \quad g_2 = g_1 - \sigma\sqrt{\tau}
\]

The value of a European lookback put is:

\[
V_p = S \exp(-q\tau) \left\{ -N_1(b_2) + \frac{\sigma^2}{2(r-q)} N_1(-b_2) \right\} \\
+ S_{max} \exp(-r\tau) \left\{ N_1(b_1) - \frac{\sigma^2}{2(r-q)} \exp(Y_2) N_1(-b_3) \right\}
\]

where

\[
b_1 = \frac{\log(S_{max}/S) + (-r+q + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}, \quad b_2 = b_1 - \sigma\sqrt{\tau} \\
b_3 = \frac{\log(S_{max}/S) + (-r-q - \sigma^2/2)\tau}{\sigma\sqrt{\tau}}, \quad Y_2 = -\frac{\log(S_{max}/S)2(r-q-\sigma^2/2)}{\sigma^2}
\]

and \( S_{max} \) is the maximum stock price achieved to date. If the lookback option has just been originated then \( S_{max} = S \) and the valuation simplifies to:

\[
V_p = S \exp(-q\tau) \{ -N_1(b_2) + WN_1(-b_2) \} + S \exp(-r\tau) \{ N_1(b_1) - WN_1(b_1) \}
\]

where

\[
W = \frac{\sigma^2}{2(r-q)}, \quad b_1 = \frac{(-r+q + \sigma^2/2)\tau}{\sigma\sqrt{\tau}}, \quad b_2 = b_1 - \sigma\sqrt{\tau}
\]
Appendix C

Derivation of the Greeks for vanilla European options

C.1 INTRODUCTION

In this section we will present some useful results which will be used later on to derive expressions for the Greeks.

A fundamental result of calculus is that:

$$\frac{\partial}{\partial x} \int f(x)dx = f(x)$$  \hspace{1cm} (C.1)

Also the indefinite integral, $\int f(x)dx$, can be expressed as a definite integral with variable upper bound as follows:

$$\int f(x)dx = \int_{a}^{x} f(x)dx + c$$

so

$$\frac{\partial}{\partial x} \int_{a}^{x} f(x)dx = f(x)$$ \hspace{1cm} (C.2)

We can now use this result to obtain the derivative of the cumulative distribution function:

$$N_1(x) = \frac{1}{\sqrt{2\pi}} \int_{-\infty}^{x} \exp(-x^2/2)dx$$

which gives

$$\frac{\partial N_1(x)}{\partial x} = n(x)$$ \hspace{1cm} (C.3)

where

$$n(x) = \frac{1}{\sqrt{2\pi}} \exp(-x^2/2)$$

We now derive various results for the parameters $d_1$ and $d_2$ which appear in the Black–Scholes equation, see Part I Section 2.3.3.

$$d_1 = \frac{\log(S/E) + (r - q + \sigma^2/2)(T-t)}{\sigma \sqrt{T-t}}$$ \hspace{1cm} (C.4)
Appendix C

and

\[ d_2 = \frac{\log(S/E) + (r - q - \sigma^2/2)(T - t)}{\sigma \sqrt{T - t}} = d_1 - \sigma \sqrt{T - t} \]  \hspace{1cm} (C.5)

We have:

\[ \frac{\partial d_2}{\partial S} = \frac{1}{S \sigma \sqrt{T - t}} \]  \hspace{1cm} (C.6)

\[ \frac{\partial d_2}{\partial \sigma} = \frac{d_1}{\sigma} - \sqrt{T - t} \]  \hspace{1cm} (C.7)

\[ \frac{\partial \Delta_1}{\partial r} = \frac{\sqrt{T - t}}{\sigma} \]  \hspace{1cm} (C.8)

\[ \frac{\partial \Delta_2}{\partial t} = \frac{\sigma}{2(T - t)} \]  \hspace{1cm} (C.9)

Also:

\[ n(d_2) = \frac{1}{\sqrt{2\pi}} \exp(-d_2^2/2) \]

\[ = \frac{1}{\sqrt{2\pi}} \exp(-d_1^2/2) \exp\left\{ \sigma d_1 \sqrt{T - t - \sigma^2(T - t)/2} \right\} \]

\[ = n(d_1) \exp\left\{ \log(S/E) + (r - q + \sigma^2/2)(T - t) - \sigma^2(T - t)/2 \right\} \]

so

\[ n(d_2) = \frac{S}{E} n(d_1) \exp(r(T - t)) \exp(-q(T - t)) \]  \hspace{1cm} (C.10)

C.2 Gamma

Gamma is defined as the second derivative of the option value with respect to the underlying stock price. This means, see Section C.3, it is the rate of change of Delta with the underlying stock price.

For a European call the value of Gamma is:

\[ \Gamma_c = \frac{\partial^2 c}{\partial S^2} = \frac{\partial \Delta_c}{\partial S} = \frac{\partial}{\partial S} \left\{ N_1(d_1) \exp(-q(T - t)) \right\} \]

where the value of \( \Delta_c \) is given in Section C.3. So

\[ \Gamma_c = \exp(-q(T - t)) \frac{\partial N_1(d_1)}{\partial S} = \exp(-q(T - t)) n(d_1) \frac{\partial d_1}{\partial S} \]

Therefore:

\[ \Gamma_c = \frac{n(d_1)}{S \sigma \sqrt{T - t}} \exp(-q(T - t)) \]  \hspace{1cm} (C.11)

The value of Gamma for a European put can be calculated similarly:

\[ \Gamma_p = \frac{\partial^2 p}{\partial S^2} = \frac{\partial \Delta_p}{\partial S} = \frac{\partial}{\partial S} \left\{ (N_1(d_1) - 1) \exp(-q(T - t)) \right\} \]
where we have used the value of $\Delta_p$, derived in Section C.3. Therefore:

$$
\Gamma_p = \exp(-q(T-t)) \frac{\partial(N_1(d_1) - 1)}{\partial S} = \exp(-q(T-t))n(d_1) \frac{\partial d_1}{\partial S}
$$

So

$$
\Gamma_p = \Gamma_c = \frac{n(d_1)}{S\sigma\sqrt{T-t}} \exp(-q(T-t)) \tag{C.12}
$$

So the value of $\Gamma$ for both a put and a call is the same.

### C.3 DELTA

Delta is defined as the rate of change of option value with the underlying stock price. For a European call we have:

$$
\Delta_c = \frac{\partial c}{\partial S} = \frac{\partial}{\partial S} \{S\exp(-q(T-t))N_1(d_1) - \exp(-r(T-t))N_1(d_2)\}
$$

So

$$
\Delta_c = \exp(-q(T-t)) \left\{ N_1(d_1) + Sn(d_1) \frac{\partial d_1}{\partial S} \right\} - \exp(-r(T-t))n(d_2) \frac{\partial d_2}{\partial S} \tag{C.13}
$$

Substituting for $n(d_2)$, and $\partial d_2/\partial S$ we obtain:

$$
\Delta_c = \exp(-q(T-t))N_1(d_1) \tag{C.14}
$$

In similar manner we have for a European put:

$$
\Delta_p = \frac{\partial p}{\partial S} = \frac{\partial}{\partial S} \{\exp(-r(T-t))(1 - N_1(d_2)) + S\exp(-q(T-t))(1 - N_1(d_1))\}
$$

So

$$
\Delta_p = -\exp(-r(T-t))n(d_2) \frac{\partial d_2}{\partial S}
$$

$$
- \exp(-q(T-t)) \left\{ (1 - N_1(d_1)) + Sn(d_1) \frac{\partial d_1}{\partial S} \right\} \tag{C.15}
$$

substituting for $n(d_2)$, and $\partial d_2/\partial S$ we obtain:

$$
\Delta_p = \exp(-q(T-t))\{N_1(d_1) - 1\} \tag{C.16}
$$

### C.4 THETA

Theta is defined as the rate of change of the option value with time. For a European call option we have:

$$
\Theta_c = \frac{\partial c}{\partial t} = \frac{\partial}{\partial t} \{S\exp(-q(T-t))N_1(d_1) - \exp(-r(T-t))N_1(d_2)\}
$$

$$
= q\exp(-q(T-t))SN_1(d_1) + \exp(-q(T-t))Sn(d_1) \frac{\partial d_1}{\partial t}
$$

$$
- r\exp(-r(T-t))N_1(d_2) - \exp(-r(T-t))n(d_2) \frac{\partial d_2}{\partial t}
$$
substituting for \( n(d_2) \) and \( \partial d_2 / \partial t \) we obtain:

\[
\Theta_c = q \exp(-q(T - t))S N_1(d_1) - r E \exp(-r(T - t))N_1(d_2)
+ \exp(-q(T - t))S n(d_1) \frac{\partial d_1}{\partial t} - E \exp(-r(T - t))n(d_1)
\]

\[
\times \frac{S}{E} \exp(r(T - t)) \exp(-q(T - t)) \left\{ \frac{\partial d_1}{\partial t} + \frac{\sigma}{2(T - t)} \right\}
\]

\[
= q \exp(-q(T - t))S N_1(d_1) - r E \exp(-r(T - t))N_1(d_2)
- \frac{Sn(d_1) \sigma \exp(-q(T - t))}{2\sqrt{T - t}}
\]

Therefore the value of theta is:

\[
\Theta_c = \exp(-q(T - t)) \left\{ q - S N_1(d_1) \frac{Sn(d_1) \sigma}{2\sqrt{T - t}} \right\} - r E \exp(-r(T - t))N_1(d_2) \quad \text{(C.17)}
\]

For a put we can similarly show that

\[
\Theta_p = \frac{\partial \rho}{\partial t} = \frac{\partial}{\partial t} \left\{ E \exp(-r(T - t))(1 - N_1(d_2)) - S \exp(-q(T - t))(1 - N_1(d_1)) \right\}
\]

\[
\Theta_p = r E \exp(-r(T - t))(1 - N_1(d_2)) - E \exp(-r(T - t))n(d_2) \frac{\partial d_2}{\partial t}
- qS \exp(-q(T - t))(1 - N_1(d_1)) + S \exp(-q(T - t))n(d_1) \frac{\partial d_1}{\partial t}
\]

substituting for \( n(d_2) \) and \( \partial d_2 / \partial t \) we obtain:

\[
\Theta_p = r E \exp(-r(T - t))N_1(-d_2) - q S \exp(-q(T - t))N_1(-d_1)
- E \exp(-r(T - t)) \exp(r(T - t)) \exp(-q(T - t))
\]

\[
\times n(d_1) \frac{S}{E} \left\{ \frac{\partial d_1}{\partial t} + \frac{\sigma}{2(T - t)} \right\} + S \exp(-q(T - t))n(d_1) \frac{\partial d_1}{\partial t}
\]

So we have:

\[
\Theta_p = - \exp(-q(T - t)) \left\{ q S N_1(-d_1) + \frac{Sn(d_1) \sigma}{2\sqrt{T - t}} \right\}
+ r E \exp(-r(T - t))N_1(-d_2) \quad \text{(C.18)}
\]

### C.5 RHO

\( \text{Rho} \) is the rate of change of the option value with interest rate.

For a call we have:

\[
\rho_c = \frac{\partial c}{\partial r} = \frac{\partial}{\partial r} \left\{ S \exp(-q(T - t))N_1(d_1) - E \exp(-r(T - t))N_1(d_2) \right\}
\]

\[
= S \exp(-q(T - t))n(d_1) \frac{\partial d_1}{\partial r} + E(T - t)N_1(d_2) - E \exp(-r(T - t))n(d_2) \frac{\partial d_2}{\partial r}
\]

substituting for \( n(d_2) \) and \( \partial d_2 / \partial r \) we obtain:

\[
\rho_c = E(T - t)N_1(d_2) \quad \text{(C.19)}
\]
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For a European put we have:
\[ \rho_p = \frac{\partial P}{\partial r} - \frac{\partial P}{\partial \sigma} \{ E \exp(-r(T-t))(1 - N_1(d_2)) - S \exp(-q(T-t))(1 - N_1(d_2)) \} \]
\[ = - E(T-t)(1 - N_1(d_2)) - E \exp(-r(T-t))n(d_2) \frac{\partial d_2}{\partial r} \]
\[ + S \exp(-q(T-t))n(d_1) \frac{\partial d_1}{\partial r} \]
\[ = - E(T-t)N_1(-d_2) - E \exp(-r(T-t))n(d_2) \frac{\partial d_2}{\partial r} \]
\[ + S \exp(-q(T-t))n(d_1) \frac{\partial d_1}{\partial r} \]

substituting for \( n(d_2) \) and \( \partial d_2/\partial r \) we obtain:
\[ \rho_p = - E(T-t)N_1(-d_2) \quad (C.20) \]

C.6 VEGA

Vega is the rate of change of option value with volatility. For a call we have:
\[ \mathcal{V}_c = \frac{\partial c}{\partial \sigma} \]
\[ = \frac{\partial}{\partial \sigma} \{ S \exp(-q(T-t))N_1(d_1) - E \exp(-r(T-t))N_1(d_2) \} \]
\[ = S \exp(-q(T-t))n(d_1) \frac{\partial d_1}{\partial \sigma} - E \exp(-r(T-t))n(d_2) \frac{\partial d_2}{\partial \sigma} \quad (C.21) \]

substituting for \( n(d_2) \) and \( \partial d_2/\partial \sigma \) we obtain:
\[ \mathcal{V}_c = S \exp(-q(T-t))n(d_1) \sqrt{T-t} - Sn(d_1) \exp(-q(T-t)) \{ \frac{\partial d_1}{\partial \sigma} - \sqrt{T-t} \} \]

Therefore
\[ \mathcal{V}_c = S \exp(-q(T-t))n(d_1) \sqrt{T-t} \quad (C.22) \]

For a European put we have:
\[ \mathcal{V}_p = \frac{\partial p}{\partial \sigma} \]
\[ = \frac{\partial}{\partial \sigma} \{ E \exp(-r(T-t))(1 - N_1(d_2)) - S \exp(-q(T-t))(1 - N_1(d_1)) \} \]
\[ = - E \exp(-r(T-t))n(d_2) \frac{\partial d_2}{\partial \sigma} + S \exp(-q(T-t))n(d_1) \frac{\partial d_1}{\partial \sigma} \quad (C.23) \]

substituting for \( n(d_2) \) and \( \partial d_2/\partial \sigma \) we obtain:
\[ \mathcal{V}_p = S \exp(-q(T-t))n(d_1) \sqrt{T-t} \quad (C.24) \]

which is the same as for a call.
Appendix D

Multiasset binomial lattices

D.1 TRUNCATED TWO ASSET BINOMIAL LATTICE

```c
void truncated_2D_binomial(double *value, double tol, double S1, double S2, 
    double X, double sigma1, double sigma2, double rho, double T, double r, double q1, 
    double q2, Integer put, Integer M, Integer opt_type, Integer is_american, Integer *iflag)
{
    /* Input parameters:
       tol — the parameter which controls when lattice truncation occurs
       S1 — the current price of the underlying asset 1
       S2 — the current price of the underlying asset 2
       X — the strike price
       sigma1 — the volatility of asset 1
       sigma2 — the volatility of asset 2
       rho — the correlation coefficient between asset 1 and asset 2
       T — the time to maturity
       r — the interest rate
       q1 — the continuous dividend yield for asset 1
       q2 — the continuous dividend yield for asset 2
       put — if put is 0 then a call option, otherwise a put option
       M — the number of time steps, the zeroth time step is the root node of the lattice
       opt_type — if opt_type is 0 then an option on the maximum of two assets, otherwise an option on the
                  minimum of two assets,
       is_american — if is_american is 0 then a European option, otherwise an American option
    */
    #define UU 0
    #define UD 1
    #define DD 2
    #define DU 3
    dt = T/(double)M;
    sqrt_dt = sqrt(dt);
    jp1 = sigma1*sqrt_dt;
    jp2 = sigma2*sqrt_dt;
    mu1 = r - q1 - sigma1*sigma1*half;
    mu2 = r - q2 - sigma2*sigma2*half;
    u1 = exp(jp1); /* assign the jump sizes */
    u2 = exp(jp2);
```
\[
\begin{align*}
d_1 &= \exp(-\sqrt{dt}); \\
d_2 &= \exp(-\sqrt{dt}); \\
p[\text{UU}] &= \text{quarter} \left( 1 + \sqrt{dt} \left( \frac{\mu_1}{\sigma_1} + \frac{\mu_2}{\sigma_2} \right) \right); \\
p[\text{UD}] &= \text{quarter} \left( 1 + \sqrt{dt} \left( \frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} \right) \right); \\
p[\text{DD}] &= \text{quarter} \left( 1 + \sqrt{dt} \left( -\frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} \right) \right); \\
\text{if } (p[i] < 0) \text{ or } (p[i] > 1.0) \text{ printf } (\text{"ERROR p out of range"}); \\
t_{\text{vd}} &= \text{max_index} + 1; \\
\text{discount} &= \exp(-r \times dt); \\
\text{for } (i = 0; i < 4; i++) \\
\quad p[i] &= p[i] \times \text{discount}; \\
\text{V} \left( i, j \right) &= \text{v[i * t_{\text{vd}} + s]} \\
\text{P1} &= 0; \\
\text{for } (i = 0; i < \text{max_index}; i++) \\
\quad \text{P2} &= 0; \\
\text{for } (j = 0; j < \text{max_index}; j++) \\
\quad &\quad \text{V} \left( i, j \right) &= \text{V} \left( i, j \right) + \text{V} \left( i + 1, j \right) + \text{V} \left( i, j + 1 \right); \\
\quad &\quad \text{P2} &= \text{P2} + 2; \\
\quad \text{P1} &= \text{P1} + 2; \\
\text{if } (\text{is_american}) \\
\text{\quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad \quad 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\[
P_1 = P_1 + 2;
\]

else{/* using a restricted lattice */
    if (odd_step){/* compute the option values in reverse order */
        ((only to index 1) so that don't overwrite storage */
            array_size = max_index;
            P_1 = 2 \cdot max_index - 1;
            for (i = array_size; i > 0; i--){
                hold = p[UD] \cdot V(i-1, j) + p[UU] \cdot V(i-1, j) + p[DU] \cdot V(i, j-1) + p[DD] \cdot V(i, j-1);
                if (is_american){
                    "Insert code fragment 4.1 to deal with option type: put/call, max/min"
                }
                else{
                    V(i, j) = hold;
                }
            } // Reverse order
        } // odd_step
    } else{/* even time step, grow extra nodes at the top and bottom. Compute the */
        /* option values in forward order making sure that don't overwrite storage */
            array_size = max_index + 1;
            P_0 = 0;
            for (i = 0; i < array_size; i++){
                P_1 = 0;
                for (j = 0; j < array_size; j++){
                    hold = p[UU] \cdot V(1, j);
                    if (i == 0){
                        hold = p[UU] \cdot V(1, j);
                    }
                    else if (j == 0){
                        hold = p[UU] \cdot V(1, j);
                    }
                    else if (i == array_size - 1){
                        hold = p[DD] \cdot V(array_size - 1, j);
                    }
                    else if (j == array_size - 1){
                        hold = p[DD] \cdot V(i, array_size - 1);
                    }
                    else{
                        hold = p[UD] \cdot V(i+1, j) + p[UU] \cdot V(i+1, j) + p[DU] \cdot V(i+1, j) + p[DD] \cdot V(i, j);
                    }
                    if (is_american){
                        "Insert code fragment 4.1 to deal with option type: put/call, max/min"
                    }
                    else{
                        V(i, j) = hold;
                    }
                } // j loop
            } // i loop
    } // even_step
}

tmp = V(0, 0);
*value = tmp;

Code excerpt D.1  Function that uses a truncated binomial lattice to compute options on the maximum and minimum of two assets

D.2 RECURSIVE TWO ASSET BINOMIAL LATTICE

Here, in Code excerpt D.2, we show the code for a recursive binomial lattice to price options on the maximum or minimum of two assets.
The parameter $M$ is the number of time intervals used, and the lattice is constructed under the assumption that $M$ is even. The parameter $rec$ control whether or not a recursive call to the function is made.

```c
void RECURSIVE_2D_binomial ( double *value, double S1, double S2, double X,
    double sigma1, double sigma2, double rho, double T, double r, double q1, double q2, 
    Integer put, Integer M, Integer opt_type, Integer is_american, Integer recc, Integer *iflag)
{
    /* Input parameters:
        S1 — the current price of the underlying asset 1
        S2 — the current price of the underlying asset 2
        X — the strike price
        sigma1 — the volatility of asset 1
        sigma2 — the volatility of asset 2
        rho — the correlation coefficient between asset 1 and asset 2
        T — the time to maturity
        r — the interest rate
        q1 — the continuous dividend yield for asset 1
        q2 — the continuous dividend yield for asset 2
        put — if put is 0 then a call option, otherwise a put option
        M — the number of time steps, the zeroth node is the root node of the lattice
        opt_type — if opt_type is 0 then an option on the maximum of two asset, otherwise an option on the
                    minimum of two assets
        is_american — if is_american is 0 then a European option, otherwise an American option
        recc — if recc is 0 then not a recursive call, otherwise a recursive call
    
    Output parameters:
        value — the value of the option,
        iflag — an error indicator.
    */
    double discount,t1,dt,d1,d2,u1,u2;
    Integer i,j,m1,n,iflagx,jj,ind;
    double zero = 0.0,hold;
    double temp,ds1,ds2,dv1,dv2,h,tmp;
    double *s1, *s2, *v;
    double p[4];
    Integer P1,P2,tdv;
    double loc_T, sqrt_dt, t, mu1, mu2, jp1, jp2;
    double one = 1.0, half = 0.5, quarter = 0.25;
    Integer v1;
    Integer loc_M, loc_recc, loc_iflag;
    Integer loc_is_american;
    if (!((M+1)/2 == M/2)){
        printf (" ERROR THE NUMBER OF TIME STEPS IS NOT EVEN \n");
        return;
    }
    if (!((recc == 0) || (recc == 1))){
        printf ("ERROR IN THE VALUE OF RECC recc == %ld \n", recc);
        return;
    }
    tdv = M+1;
    #define V(I,J) v[(I) * tdv + (J)]
    #define U0 0
    #define UD 1
    #define DD 2
    #define DU 3
    dt = T/(double)M;
    sqrt_dt = sqrt(dt);
    jp1 = sigma1*sqrt_dt;
    jp2 = sigma2*sqrt_dt;
    mu1 = r - q1 - sigma1*sigma1*half;
    mu2 = r - q2 - sigma2*sigma2*half;
    u1 = exp(jp1);
    u2 = exp(jp2);
    d1 = exp(-jp1);
    d2 = exp(-jp2);
    p[U0] = quarter*(one + sqrt_dt * ( (mu1/sigma1) + (mu2/sigma2) ) + rho);
    p[UD] = quarter*(one + sqrt_dt * ( (mu1/sigma1) - (mu2/sigma2) ) - rho);
    p[DD] = quarter*(one + sqrt_dt * ( -(mu1/sigma1) - (mu2/sigma2) ) + rho);
    p[DU] = quarter*(one + sqrt_dt * ( -(mu1/sigma1) + (mu2/sigma2) ) - rho);
    for (i=0; i<4; ++i){
```
if ((p[i] < zero) || (p[i] > 1.0)) printf("ERROR p out of range\n");
} 

discount = exp(-r*dt);
for (i = 0; i < 4; ++i) 
  p[i] = p[i]*discount;
/* Allocate the arrays v[(M+1)*(M+1)], s1[2*M+1] and s2[2*M+1] */
 for (i = 0; i < M; ++i) /* assign the 2*M+1 asset values for s1 */
  s1[M+i] = u1*s1[M+i-1];
  s2[M+i] = d1*s2[M+i-1];
 for (i = 1; i <= M; ++i) /* assign the 2*M+1 asset values for s2 */
  s2[M+i] = u2*s2[M+i-1];
  s2[M+i] = d2*s2[M+i-1];
/* Calculate the option values at maturity */
if (recc == 0) /* called without recursion */
P1 = 0;
 for (i = 0; i <= M; ++i) 
  P2 = 0;
  for (j = 0; j <= M; ++j) {
    if (opt_type == 0) /* Maximum of two assets */
      if (put) {
        V(i,j) = MAX(X-MAX(s1[P1],s2[P2]),zero);
      }
      else {
        V(i,j) = MAX(MAX(s1[P1],s2[P2])-X,zero);
      }
    else/* Minimum of two assets */
      if (put) {
        V(i,j) = MAX(X-MIN(s1[P1],s2[P2]),zero);
      }
      else {
        V(i,j) = MAX(MIN(s1[P1],s2[P2])-X,zero);
      }
    P2 = P2 + 2;
  }
  P1 = P1 + 2;
}
else/* called with recursive last step */
P1 = 1;
 for (i = 0; i <= M-1; ++i) 
  P2 = 1;
  for (j = 0; j <= M-1; ++j) {
    loc_T = dt;
    loc_M = 10;
    loc_recc = 0;
    loc_iflag = 0;
    loc_is_american = is_american;
    recursive_2D_binomial(hold, s1[P1], s2[P2], X, sigma1, sigma2, rho, 
      loc_T, r, q1, q2, put, loc_M, opt_type, loc_is_american, loc_recc, &loc_iflag);
    if (is_american){
      if (opt_type == 0) /* Maximum of two assets */
        if (put) {
          V(i,j) = MAX(hold,X-MAX(s1[P1],s2[P2]));
        }
        else {
          V(i,j) = MAX(hold,MAX(s1[P1],s2[P2])-X);
        }
      else/* Minimum of two assets */
        if (put) {
          V(i,j) = MAX(hold,X-MIN(s1[P1],s2[P2]));
        }
        else {
          V(i,j) = MAX(hold,MIN(s1[P1],s2[P2])-X);
        }
    }
    else{
      V(i,j) = hold;
    }
    P2 = P2 + 2;
  }
\[ P_1 = P_1 + 2; \]
}\)
for (m1 = M−1−rec; m1 > 0; --m1) /* work backwards through the lattice to calculate the option value */
\[ P_1 = M−m1; \]
/* Identical code to the equivalent loop of the standard 2 dimensional binomial lattice
see code excerpt 5.7 */
\}
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Appendix D

\[ p_{dd\ell d} = \frac{1}{16} \left\{ 1 + \sqrt{\Delta I} \left( -\frac{\mu_1}{\sigma_1} + \frac{\mu_2}{\sigma_2} - \frac{\mu_3}{\sigma_3} - \frac{\mu_4}{\sigma_4} \right) - \rho_{12} + \rho_{13} + \rho_{14} - \rho_{23} - \rho_{24} + \rho_{34} \right\} \]

\[ p_{dd\ell u} = \frac{1}{16} \left\{ 1 + \sqrt{\Delta I} \left( -\frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} + \frac{\mu_3}{\sigma_3} + \frac{\mu_4}{\sigma_4} \right) + \rho_{12} - \rho_{13} - \rho_{14} - \rho_{23} - \rho_{24} + \rho_{34} \right\} \]

\[ p_{dd\ell ud} = \frac{1}{16} \left\{ 1 + \sqrt{\Delta I} \left( -\frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} + \frac{\mu_3}{\sigma_3} - \frac{\mu_4}{\sigma_4} \right) + \rho_{12} - \rho_{13} + \rho_{14} - \rho_{23} + \rho_{24} - \rho_{34} \right\} \]

\[ p_{dd\ell uu} = \frac{1}{16} \left\{ 1 + \sqrt{\Delta I} \left( -\frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} - \frac{\mu_3}{\sigma_3} + \frac{\mu_4}{\sigma_4} \right) + \rho_{12} + \rho_{13} - \rho_{14} + \rho_{23} - \rho_{24} - \rho_{34} \right\} \]

\[ p_{dd\ell d\ell} = \frac{1}{16} \left\{ 1 + \sqrt{\Delta I} \left( -\frac{\mu_1}{\sigma_1} - \frac{\mu_2}{\sigma_2} - \frac{\mu_3}{\sigma_3} - \frac{\mu_4}{\sigma_4} \right) + \rho_{12} + \rho_{13} - \rho_{14} + \rho_{23} + \rho_{24} + \rho_{34} \right\} \]
Appendix E

Derivation of the conditional mean and covariance for a multivariate normal distribution

Let \( X = [X_1/X_2] \) be distributed as \( N_p(\mu, \Sigma) \) with \( \mu = [\mu_1/\mu_2] \) and \( \Sigma = [(\Sigma_{11}\Sigma_{12})/(\Sigma_{21}\Sigma_{22})] \) and \( |\Sigma_{22}| > 0 \).

We will prove that the conditional distribution of \( X_1 \), given that \( X_2 = x_2 \), is normal and has:

Mean\( = \mu_1 + \Sigma_{11}\Sigma_{22}^{-1}(x_2 - \mu_2) \), and covariance\( = \Sigma_{11} - \Sigma_{12}\Sigma_{22}^{-1}\Sigma_{21} \). Let the inverse of \( \Sigma \) be \( \Sigma^{-1} \), where:

\[
\Sigma^{-1} = \begin{pmatrix}
\Sigma_{11} & \Sigma_{12} \\
\Sigma_{21} & \Sigma_{22}
\end{pmatrix}
\]

So \( \Sigma^{-1}\Sigma = I_p \), where \( I_p \) represents the \( p \times p \) unit matrix, and:

\[
\begin{pmatrix}
\Sigma_{11} & \Sigma_{12} \\
\Sigma_{21} & \Sigma_{22}
\end{pmatrix}
\cdot
\begin{pmatrix}
\Sigma_{11} & \Sigma_{12} \\
\Sigma_{21} & \Sigma_{22}
\end{pmatrix}
= \begin{pmatrix}
I_q & 0 \\
0 & I_{p-q}
\end{pmatrix}
\]

Multiplying out these matrices yields the following equations:

\[
\Sigma_{11}\Sigma_{11} + \Sigma_{21}\Sigma_{21} = I_q \quad (E.3)
\]

\[
\Sigma_{21}\Sigma_{11} + \Sigma_{22}\Sigma_{22} = 0 \quad (E.4)
\]

\[
\Sigma_{11}\Sigma_{12} + \Sigma_{12}\Sigma_{22} = 0 \quad (E.5)
\]

\[
\Sigma_{21}\Sigma_{12} + \Sigma_{22}\Sigma_{22} = I_{p-q} \quad (E.6)
\]

Multiplying Equation E.5 on the left by \((\Sigma_{11})^{-1}\) and on the right by \(\Sigma_{22}^{-1}\) gives:

\[
(\Sigma_{11})^{-1}\Sigma_{12} = -\Sigma_{12}\Sigma_{22}^{-1} \quad (E.7)
\]

Multiplying Equation E.3 on the left by \((\Sigma_{11})^{-1}\) yields

\[
\Sigma_{11} + (\Sigma_{11})^{-1}\Sigma_{12}\Sigma_{21} = (\Sigma_{11})^{-1} \quad (E.8)
\]

and substituting for \((\Sigma_{11})^{-1}\Sigma_{12}\) from Equation E.7 into Equation E.8 gives

\[
(\Sigma_{11})^{-1} = \Sigma_{11} - \Sigma_{12}\Sigma_{22}^{-1}\Sigma_{21} \quad (E.9)
\]

The joint probability density function of \( x \) is:

\[
f(x) = (2\pi)^{-p/2}|\Sigma|^{-1/2}\exp\left\{-\frac{1}{2}(x - \mu)^T\Sigma^{-1}(x - \mu)\right\}
\]
writing $x, \mu$ and $\Sigma^{-1}$ in their partitioned form and expanding gives:

$$f(x) = (2\pi)^{-p/2} |\Sigma|^{-1/2} \exp \left[ -\frac{1}{2} \left\{ (x_1 - \mu_1)^T \Sigma^{11} (x_1 - \mu_1) + 2(x_1 - \mu_1)^T \Sigma^{12} (x_2 - \mu_2) \right\} \right]$$

(E.10)

The conditional distribution of $x_1$ given the value of $x_2$ is thus obtained by dividing this density by the marginal density of $x_2$, and treating $x_2$ as constant in the resulting expression. The only portion of the resultant that is not constant is the portion involving terms in $x_1$. It can easily be shown that:

$$f(x_1|x_2) \propto \exp \left[ -\frac{1}{2} \left\{ (x_1 - \mu_1)^T \Sigma^{11} (x_1 - \mu_1) + 2(x_1 - \mu_1)^T \Sigma^{12} (x_2 - \mu_2) \right\} \right]$$

where the constant of proportionality is obtained using $\int f(x_1|x_2)dx_1 = 1$.

If we let

$$G = (x_1 - \mu_1)^T \Sigma^{11} (x_1 - \mu_1) + 2(x_1 - \mu_1)^T \Sigma^{12} (x_2 - \mu_2)$$

we then obtain:

$$G = \left\{ x_1 - \mu_1 + (\Sigma^{11})^{-1} \Sigma^{12} (x_2 - \mu_2) \right\}^T \Sigma^{11} \left\{ x_1 - \mu_1 + (\Sigma^{11})^{-1} \Sigma^{12} (x_2 - \mu_2) \right\}$$

$$- (x_2 - \mu_2)^T \Sigma^{21} (\Sigma^{12})^{-1} (x_2 - \mu_2)$$

(E.11)

where, for instance we have used, the fact that the scalar quantity

$$\left\{ (x_1 - \mu_1)^T \Sigma^{12} (x_2 - \mu_2) \right\}^T = (x_2 - \mu_2)^T \Sigma^{21} (x_1 - \mu_1)$$

Since the last term in Equation E.11 only involves constants (as far as $f(x_1|x_2)$ is concerned), it follows that:

$$f(x_1|x_2) \propto \exp \left[ -\frac{1}{2} \left\{ x_1 - \mu_1 + (\Sigma^{11})^{-1} \Sigma^{12} (x_2 - \mu_2) \right\}^T \right.$$

$$\times \Sigma^{11} \left\{ x_1 - \mu_1 + (\Sigma^{11})^{-1} \Sigma^{12} (x_2 - \mu_2) \right\}$$

which is the density of a multivariate normal distribution that has a mean of $\mu_1 - (\Sigma^{11})^{-1} \Sigma^{12} (x_2 - \mu_2)$, which from Equation E.7 can be expressed as $\mu_1 + \Sigma_{12}\Sigma_{22}^{-1} (x_2 - \mu_2)$. The covariance matrix is $(\Sigma^{11})^{-1}$, which from Equation E.9 can be written as $\Sigma_{11} - \Sigma_{12}\Sigma_{22}^{-1}\Sigma_{21}$.
Appendix F

Standard statistical results

F.1 THE LAW OF LARGE NUMBERS

Let $X_1, X_2, \ldots$ be a sequence of independent, identically distributed random variables (IID), each with expected value $\mu$ and variance $\sigma^2$. Define the sequence of averages

$$Y_n = \frac{X_1 + X_2 + \cdots + X_n}{n}, \quad n = 1, 2, \ldots$$

Then $Y_n$ converges to $\mu$ as $n \to \infty$.

We will not rigorously prove this theorem but show that it is plausible. For the mean of $Y_n$ we have:

$$E[Y_n] = \frac{1}{n} (E[X_1] + E[X_2] + \cdots + E[X_n]) = \frac{1}{n} n \mu = \mu$$

For the variance of $Y_n$ we have:

$$Var(Y_n) = \sum_{i=1}^{n} Var\left(\frac{X_i}{n}\right) = \sum_{i=1}^{n} \frac{\sigma^2}{n^2} = \frac{\sigma^2}{n}$$

where we have used the fact that the variance of the sum of independent random variables is the sum of their variances, see Section F.2.

So as $n \to \infty$, we have $Var(Y_n) \to 0$.

F.2 THE CENTRAL LIMIT THEOREM

This is similar to the Law of Large numbers. In this case we divide by $\sqrt{n}$ instead of $n$, which prevents the variance of $Y_n$ converging to zero as $n \to \infty$.

Let $X_1, X_2, \ldots$ be a sequence of independent, identically distributed random variables (IID), each with expected value $\mu$ and variance $\sigma^2$. Define:

$$Z_n = \frac{(X_1 - \mu) + (X_2 - \mu) + \cdots + (X_n - \mu)}{\sqrt{n}}, \quad n = 1, 2, \ldots$$

so that each $Z_n$ has expected value zero and variance

$$Var(Z_n) = \sum_{i=1}^{n} Var\left(\frac{X_i - \mu}{\sqrt{n}}\right) = \sum_{i=1}^{n} \frac{\sigma^2}{n} = \sigma^2$$
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The central Limit Theorem states that as \( n \to \infty \) the distribution of \( Z_n \) approaches that of a normal random variable (say \( x \)) with mean zero and variance \( \sigma^2 \). In other words the probability density function of \( Z_n \) is:

\[
P(Z_n) \to \frac{1}{\sigma \sqrt{2\pi}} \exp \left( -\frac{x^2}{2\sigma^2} \right) \quad \text{as} \quad n \to \infty
\]

F.3 THE MEAN AND VARIANCE OF LINEAR FUNCTIONS OF RANDOM VARIABLES

Let \( X \) be a variate from a given distribution, and \( Z \) be the following linear function of this variate:

\[
Z = a + bX
\]

where \( a \) and \( b \) are constants. Then

\[
E[Z] = E[a] + E[bX] = a + bE[X]
\]

and

\[
Var[Z] = E[(Z - E[Z])^2] = E[(a + bX - a - bE[X])^2]
\]

\[
= E[(bX - bE[X])^2] = E[b^2(X - E[X])^2] = b^2E[(X - E[X])^2]
\]

Therefore the mean is \( bE[X] \), and the variance is \( b^2Var[X] \).

The variance of the sum of random identical independently distributed variables (IID).

F.3.1 The sum of 2 variables

Let \( Z_2 = X_1 + X_2 \), where \( X_1 \) and \( X_2 \) are IID variables. Then we have:

\[
Var[Z_2] = E[((X_1 + X_2) - E[X_1 + X_2])^2]
\]

\[
= E[((X_1 - E[X_1]) + (X_2 - E[X_2]))^2]
\]

\[
= E[(X_1 - E[X_1])^2 + (X_2 - E[X_2])^2 + 2(X_1 - E[X_1])(X_2 - E[X_2])]
\]

\[
= E[(X_1 - E[X_1])^2] + E[(X_2 - E[X_2])^2]
\]

\[
= Var[X_1] + Var[X_2],
\]

where we have used the fact that, since the variables are independent

\[
E[(X_1 - E[X_1])(X_2 - E[X_2])] = 0.
\]

Therefore:

\[
Var[X_1 + X_2] = Var[X_1] + Var[X_2]
\]
F.3.2 The sum of 3 variables

Let $Z_3 = X_1 + X_2 + X_3$, where $X_1, X_2,$ and $X_3$ are IID variables. Then we have:

$$
\text{Var}[Z_3] = E[(X_1 + X_2 + X_3) - E(X_1 + X_2 + X_3)]^2
$$

$$
= E[((X_1 - E[X_1]) + (X_2 - E[X_2]) + (X_3 - E[X_3]))^2]
$$

$$
= E[(X_1 - E[X_1])^2 + (X_2 - E[X_2])^2 + (X_3 - E[X_3])^2]
$$

$$
+ 2(X_1 - E[X_1])(X_2 - E[X_2]) + 2(X_1 - E[X_1])(X_3 - E[X_3])
$$

$$
+ 2(X_2 - E[X_2])(X_3 - E[X_3])
$$

$$
= E[(X_1 - E[X_1])^2] + E[(X_2 - E[X_2])^2] + E[(X_3 - E[X_3])^2]
$$

$$
= \text{Var}[X_1] + \text{Var}[X_2] + \text{Var}[X_3]
$$

where, as before, we have used the fact that $E[(X_i - E[X_i]) (X_j - E[X_j])] = 0$, $i = 1, \ldots, 3$, $j = 1, \ldots, 3$, $i \neq j$. Therefore:

$$
\text{Var}[X_1 + X_2 + X_3] = \text{Var}[X_1] + \text{Var}[X_2] + \text{Var}[X_3]
$$

F.3.3 The sum of $n$ variables

Let $Z_n = \sum_{i=1}^{n} X_i$

Then we have:

$$
\text{Var}[Z_n] = E \left[ \left( \sum_{i=1}^{n} X_i - E[X_i] \right)^2 \right]
$$

$$
= \sum_{i=1}^{n} E[(X_i - E[X_i])^2] + \sum_{i=1}^{n} \sum_{j=1}^{n} E[(X_i - E[X_i])(X_j - E[X_j])]
$$

$$
= \sum_{i=1}^{n} E[(X_i - E[X_i])^2]
$$

$$
= \sum_{i=1}^{n} \text{Var}[X_i]
$$

Therefore:

$$
\text{Var} \left[ \sum_{i=1}^{n} X_i \right] = \sum_{i=1}^{n} \text{Var}[X_i]
$$

F.4 STANDARD ALGORITHMS FOR THE MEAN AND VARIANCE

In this section we provide standard results concerning the computation of the mean and variance (covariance) of the observations contained in a given data set.

The variance of $X$ is defined as:

$$
\text{Var}[X] = E[(X - E[X])^2]
$$
The simplest way of computing the variance is to use a two pass method. We illustrate this with a simple Monte Carlo program which is designed to stop when the result has attained a given accuracy.

```c
double result[1000000] // need to provide a large array to store the results
tol = 0.1
mean_X = 0.0
i = 1
while (variance > tol){ // keep going until the variance is small enough
    // call a Monte Carlo function, with n parameters, which return the current estimate
    result[i] = my_monte_carlo(param_1, param_2,...,param_n)
    mean_X = (mean_X + result[i])/i // first pass to calculate the mean
    for (j=1, j<i; ++i){ // second pass to calculate the variance
        variance = (result[i] - mean_X)*(result[i] - mean_X)
    }
    variance = variance/(double)i
    i = i + 1
}
```

Although numerical stable (West, 1979) this approach requires the allocation of the very large array `result`, and also contains a nested for loop. We can get round this problem by expanding the terms in the variance as follows:

\[
\text{Var}[X] = E[(X - E[X])^2] = E[X^2] + (E[X])^2 - 2XE[X]
\]

Therefore \( \text{Var}[X] = E[X^2] - (E[X])^2 \). This approach leads to the so-called text-book algorithm which allows the variance to be computed by using only one pass through the data. The program for our original problem then becomes:

```c
tol = 0.1
mean_X = 0.0
mean_X_squared = 0.0
i = 1.0
while (variance > tol){ // keep going until the variance is small enough
    // call a Monte Carlo function, with n parameters, which return the current estimate
    result = my_monte_carlo(param_1, param_2,...,param_n)
    // calculate the running mean
    mean_X = (mean_X + result)/i
    // calculate the running mean value of the square of the result
    mean_X_squared = (mean_X_squared + result*result)/i
    // calculate the running variance
    variance = mean_X_squared - (mean_X*mean_X)
    i = i + 1.0
}
```

Although this method doesn’t require extra memory allocation and doesn’t require a second pass through the data, it is numerically unstable (Chan et al., 1982; West, 1979) and the algorithm given in Section F.5 should be used if accurate results are required.
The textbook algorithm can easily be extended to compute the covariance rather than the variance.

If we consider two random variates \( X \) and \( Y \) then the covariance, \( \text{COV}[X, Y] \), is defined as:

\[
\text{COV}[X, Y] = E[(X - E[X])(Y - E[Y])]
\]

This can be expanded as follows:

\[
\]

Therefore the covariance is given by the following equation:

\[
\text{COV}[X, Y] = E[XY] - E[X]E[Y] \quad (F.2)
\]

### F.5 THE HANSON AND WEST ALGORITHM FOR THE MEAN AND VARIANCE

Here we describe a method of computing the mean and variance (covariance) of a data set that is more numerically stable than the textbook algorithm given in Section F.4, West (1979).

We will consider an \( n \times p \) data matrix of \( n \) observations on \( p \) variates. The observations are represented by the \( p \) element vector \( X_i, i = 1, \ldots, n \).

Let the mean of the first \( i - 1 \) observations be denoted by \( \bar{X}_{[i-1]} = \frac{\sum_{k=1}^{i-1} X_k}{i-1} \). Then we have:

\[
\bar{X}_i = \frac{\sum_{k=1}^{i} X_k}{i} = \frac{\sum_{k=1}^{i-1} X_k + X_i}{i} = \frac{i-1}{i-1} \bar{X}_{[i-1]} + \frac{1}{i-1} X_i
\]

Therefore:

\[
\bar{X}_i = \frac{\bar{X}_{[i-1]} + X_i}{i} = \bar{X}_{[i-1]} + \frac{1}{i} (X_i - \bar{X}_{[i-1]}) \quad (F.3)
\]

Let \( \sigma^2_{[i-1]} \) be the variance of the first \( i - 1 \) observations. This means that the sum of squares about the mean \( S^2_{[i-1]} \) of the first \( i - 1 \) observations is \( S^2_{[i-1]} = (i - 1) \sigma^2_{[i-1]} \). Now from the definition of variance we have:

\[
\frac{S^2_{[i-1]}}{i-1} = \sigma^2_{[i-1]} = \left( \frac{1}{i-1} \sum_{k=1}^{i-1} X^2_k \right) - (\bar{X}_{[i-1]})^2
\]

so

\[
\sum_{k=1}^{i-1} X^2_k = S^2_{[i-1]} + (i - 1)(\bar{X}_{[i-1]})^2
\]
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Now the inclusion of the \(i\)th observation \(X_i\) results in the new sum of squares about the mean \(S_{[i]}^2\):

\[
S_{[i]}^2 = \left( \sum_{k=1}^{i} X_k^2 \right) - i (\bar{X}_{[i]})^2
\]

\[
S_{[i]}^2 = \left( \sum_{k=1}^{i-1} X_k^2 \right) + X_i^2 - i (\bar{X}_{[i]})^2
\]

\[
S_{[i]}^2 = S_{[i-1]}^2 + (i - 1)(\bar{X}_{[i-1]})^2 + X_i^2 - i (\bar{X}_{[i]})^2
\]  \hspace{1cm} (F.4)

But

\[
i (\bar{X}_{[i]})^2 = \frac{i}{i^2} (X_i + (i - 1)\bar{X}_{[i-1]})^2
\]

\[
i (\bar{X}_{[i]})^2 = \frac{1}{i} \left\{ (i - 1)^2 (\bar{X}_{[i-1]})^2 + 2(i - 1)X_i\bar{X}_{[i-1]} + X_i^2 \right\}
\]  \hspace{1cm} (F.5)

So we have:

\[
S_{[i]}^2 = S_{[i-1]}^2 + (i - 1)(\bar{X}_{[i-1]})^2 + X_i^2 - \frac{(i - 1)^2}{i} (\bar{X}_{[i-1]})^2 - \frac{2(i - 1)}{i} X_i\bar{X}_{[i-1]} - \frac{X_i^2}{i}
\]

\[
= S_{[i-1]}^2 + (i - 1) \left( 1 - \frac{i - 1}{i} \right) (\bar{X}_{[i-1]})^2 + \left( 1 - \frac{i - 1}{i} \right) X_i^2 - \frac{2(i - 1)}{i} X_i\bar{X}_{[i-1]}
\]

Therefore

\[
S_{[i]}^2 = S_{[i-1]}^2 + \frac{(i - 1)}{i} (\bar{X}_{[i-1]})^2 + \frac{(i - 1)}{i} X_i^2 - \frac{2(i - 1)}{i} X_i\bar{X}_{[i-1]}
\]  \hspace{1cm} (F.6)

The above equation can be written in more compact form since:

\[
\left( \frac{i - 1}{i} \right) \left( X_i - \bar{X}_{[i-1]} \right) \left( X_i - \bar{X}_{[i-1]} \right) = \frac{i - 1}{i} X_i^2 + \frac{i - 1}{i} (\bar{X}_{[i-1]})^2 - 2 \frac{(i - 1)}{i} X_i\bar{X}_{[i-1]}
\]

which gives the final updating equation for the sum of squares about the mean as:

\[
S_{[i]}^2 = S_{[i-1]}^2 + \left( \frac{i - 1}{i} \right) \left( X_i - \bar{X}_{[i-1]} \right) \left( X_i - \bar{X}_{[i-1]} \right)
\]  \hspace{1cm} (F.7)

This useful equation gives the sum of squares about the mean \(S_{[i]}^2\), given the previous sum of squares about the mean \(S_{[i-1]}^2\), the previous mean \(\bar{X}_{[i-1]}\), and the new data point \(X_i\).

The estimated variance, \(Var[X]\), computed using the data \(X_i, i = 1, \ldots, n\), is therefore given by:

\[
Var[X] = \frac{S_{[n]}^2}{n - 1}
\]

The following code excerpt shows how the algorithm works in practice

```
tol = 0.1
// call a Monte Carlo function, with n parameters, which return the current estimate
result = my_monte_carlo(param_1, param_2, \ldots, param_n)
```
The above method can easily be extended to compute the covariance of two variables $X$ and $Y$. The covariance is defined as follows:

$$ COV(X, Y) = \frac{1}{n-1} \sum_{i=1}^{n} (X_i - \bar{X})(Y_i - \bar{Y}) $$

where

$$ \bar{X} = \frac{1}{n} \sum_{i=1}^{n} X_i, \quad \bar{Y} = \frac{1}{n} \sum_{i=1}^{n} Y_i $$

$X_i$ and $Y_i$ denote the $i$th data values of $X$ and $Y$ respectively, and the expression $(X_i - \bar{X})(Y_i - \bar{Y})$ is termed the $i$th cross product about the means $\bar{X}$ and $\bar{Y}$. As before we will also let $\bar{X}_i$ and $\bar{Y}_i$ denote the running means of the first $i$ observations, of the $X$ and $Y$ variables respectively.

The $i$th sum of the cross products about the means is updated according to the following equation:

$$ P_{[i]} = P_{[i-1]} + \left( \frac{i-1}{i} \right) (X_i - \bar{X}_{[i-1]}) (Y_i - \bar{Y}_{[i-1]}) \quad (F.8) $$

where $P_{[i]}$ denotes the updated sum of the cross products about the mean, $P_{[i-1]}$ denotes the previous sum of the cross products about the mean, $\bar{X}_{[i-1]}$ is the previous mean of the variable $X$, $\bar{Y}_{[i-1]}$ is the previous mean of the variable $Y$, and the new variate values are $X_i$ and $Y_i$.

The estimated covariance, $COV[X, Y]$, computed using the data $X_i, i = 1, \ldots, n$, and $Y_i, i = 1, \ldots, n$, is therefore given by:

$$ COV[X, Y] = \frac{P_{[n]}}{n-1} $$

F.6 JENSEN’S INEQUALITY

This states that if the function $h(X)$ of a random variable $X$ is convex and $E[X] = \mu$, then $E[h(X)] \geq h(\mu)$. 
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F.6.1 Proof

Let $X$ be a random variable with expected value $E[X] = \mu$ and the variable $Y$ be a nonlinear function of $X$, $Y = h(X)$. Then $\partial Y / \partial X = h'(X)$. If $Z$ is the tangent to $h(X)$ at the point $\mu$ then:

$$Z = h(\mu) + h'(\mu)(X - \mu)$$

Since $h(\mu)$ and $h'(\mu)$ are constants we have that:

$$E[Z] = h(\mu) + h'(\mu)E[(X - \mu)] = h(\mu) + h'(\mu)(E[X] - \mu) = h(\mu)$$

If the function $h(X)$ is convex then $Y \geq Z$ everywhere. Then regardless of the distribution of $X$ we have:

$$E[Y] \geq E[Z], \text{ but } E[Z] = h(\mu) \text{ so } E[Y] \geq h(\mu)$$

Therefore for a convex function $h(X)$ we have that:

$$E[h(X)] \geq h(\mu)$$

For a concave function we obviously have:

$$E[h(X)] \leq h(\mu)$$

An example of a convex function is $h(X) = X^2$. So regardless of the distribution of $X$ we have that $E(X^2) \geq (E[X])^2$.

An example of a concave function is $h(X) = \log(X)$. So regardless of the distribution of $X$ we have that $E(\log(X)) \leq \log(E[X])$. 
Appendix G

Derivation of barrier option integrals

G.1 THE DOWN AND OUT CALL

We will now derive the formula for the value, \( c_{do} \), of a down and out call option which was given in Part I Section 2.4.2.

\[
c_{do} = \frac{\exp(-r\tau)}{\sigma\sqrt{\pi}} \int_{X = \log(E/S)}^{\infty} \{S \exp(X) - E\} f(X > B) dX
\]

where

\[
f(X > B) = \frac{1}{\sigma\sqrt{\pi}} \exp\left(-\frac{\{X - (r - \sigma^2/2)\tau\}^2}{2\sigma^2\tau}\right)
\times \left\{1 - \exp\left(\frac{2 \log(B/S)(X - \log(B/S))}{\sigma^2\tau}\right)\right\}
\]

We will represent this integral as:

\[c_{do} = I_A + I_B\]

where

\[I_A = \frac{\exp(-r\tau)}{\sigma\sqrt{\pi}} \int_{X = \log(E/S)}^{\infty} \{S \exp(X) - E\} \exp\left(-\frac{\{X - (r - \sigma^2/2)\tau\}^2}{2\sigma^2\tau}\right) dX\]

and

\[I_B = -\frac{\exp(-r\tau)}{\sigma\sqrt{\pi}} \int_{X = \log(E/S)}^{\infty} \{S \exp(X) - E\} \exp\left(-\frac{\{X - (r - \sigma^2/2)\tau\}^2}{2\sigma^2\tau}\right) \times \exp\left(\frac{2 \log(B/S)(X - \log(B/S))}{\sigma^2\tau}\right) dX\]

G.1.1 Evaluation of integral \(I_A\)

Now comparing \(I_A\) with Equation 2.34 in Part I Section 2.3.3 we can identify \(I_A\) as \( c(S,E,\tau)\), the price of a European call. That is:

\[I_A = SN_1(d_1) - E \exp(-r\tau)N_1(d_2)\]  

(G.2)
where:
\[ d_1 = \frac{\log(S/E) + (r + \sigma^2/2)\tau}{\sigma\sqrt{\tau}} \quad \text{and} \quad d_2 = \frac{\log(S/E) + (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}} \]

G.1.2 Evaluation of integral \( I_B \)

We will now consider the integral \( I_B \), and let \( I_B = I_C + I_D \) where:
\[
I_C = -\frac{S}{\sigma\sqrt{\tau}} \sqrt{2\pi} \int_{X=\log(E/S)}^{\infty} \exp(X) \exp\left(-\frac{\{X - (r - \sigma^2/2)\tau\}^2}{2\sigma^2\tau}\right) \times \exp\left(\frac{2\log(B/S)(X - \log(B/S))}{\sigma^2\tau}\right) dX
\]
and
\[
I_D = \frac{E}{\sigma\sqrt{\tau}} \sqrt{2\pi} \int_{X=\log(E/S)}^{\infty} \exp\left(-\frac{\{X - (r - \sigma^2/2)\tau\}^2}{2\sigma^2\tau}\right) \times \exp\left(\frac{2\log(B/S)(X - \log(B/S))}{\sigma^2\tau}\right) dX
\]

G.1.3 Evaluation of integral \( I_D \)

We will first consider \( I_D \) and factor the integrand as follows:
\[
-\exp\left(-\frac{\{X - (r - \sigma^2/2)\tau\}^2}{2\sigma^2\tau}\right) \exp\left(\frac{2\log(B/S)(X - \log(B/S))}{\sigma^2\tau}\right)
\]
\[= \exp\left(-\frac{\{X - (r - \sigma^2/2)\tau\}^2 - 4\log(B/S)(X - \log(B/S))}{2\sigma^2\tau}\right) \]
\[= \exp\left(-\frac{\{X - (r - \sigma^2/2)\tau - 2\log(B/S)\}^2}{2\sigma^2\tau}\right) \exp\left(\frac{4\log(B/S)(X - \log(B/S))}{2\sigma^2\tau}\right)
\]

This means that \( I_D \) can be expressed as:
\[
I_D = \frac{B}{S}^{2(r-\sigma^2/2)/\sigma^2} \frac{E \exp(-r\tau)}{\sigma\sqrt{\tau}\sqrt{2\pi}} \times \int_{X=\log(E/S)}^{\infty} \exp\left(-\frac{\{X - (r - \sigma^2/2)\tau - 2\log(B/S)\}^2}{2\sigma^2\tau}\right) dX
\]
Letting \( u = (X - (r - \sigma^2/2)\tau - 2\log(B/S))/\sigma\sqrt{\tau} \) we have \( dX = \sigma\sqrt{\tau} du \) and
Appendix G

\[ I_D = \left( \frac{B}{S} \right)^{2(r-\sigma^2/2)/\sigma^2} E \frac{\exp(-rt)}{\sigma \sqrt{2\pi}} \int_{u=k_3}^{\infty} \exp\left(-\frac{u^2}{2}\right) du \]

where

\[ k_3 = \frac{\log(E/S) - (r-\sigma^2/2)\tau - 2\log(B/S)}{\sigma \sqrt{\tau}} = \frac{\log(ES/B^2) - (r-\sigma^2/2)\tau}{\sigma \sqrt{\tau}} \]

So

\[ I_D = \left( \frac{B}{S} \right)^{2r/\sigma^2 - 1} E \exp(-rt) N_1(-k_3) \]  

(G.4)

Letting \( d_3 = -k_3 \) we have:

\[ I_D = \left( \frac{B}{S} \right)^{2r/\sigma^2 - 1} E \exp(-rt) N_1(d_3), \text{ where } \]

\( d_3 = \frac{\log(B^2/SE) + (r-\sigma^2/2)\tau}{\sigma \sqrt{\tau}} \)  

(G.5)

G.1.4 Evaluation of integral \( I_C \)

Now consider the term

\[ I_C = S \frac{\exp(-rt)}{\sigma \sqrt{2\pi}} \int_{X=\log(E/S)}^{\infty} \exp(X) \exp\left(-\frac{\{X - (r-\sigma^2/2)\tau\}^2}{2\sigma^2 \tau}\right) \]

\[ \times \exp\left(\frac{2\log(B/S)(X - \log(B/S))}{2\sigma^2 \tau}\right) dX \]

Now we have:

\[ \exp(X) \exp\left(-\frac{(X - (r-\sigma^2/2)\tau)^2}{2\sigma^2 \tau}\right) \exp\left(\frac{2\log(B/S)(X - \log(B/S))}{\sigma^2 \tau}\right) \]

\[ = \exp\left(-\frac{(X - (r-\sigma^2/2)\tau)^2 - 2a^2 \tau X - 4\log(B/S)X + 4(\log(B/S))^2}{2\sigma^2 \tau}\right) \]

\[ = \exp\left(\frac{(\sigma^2 \tau)^2 + 2(r-\sigma^2/2)\tau^2 \sigma^2 + 4(r-\sigma^2/2)\tau \log(B/S) + 4\sigma^2 \tau \log(B/S)}{2\sigma^2 \tau}\right) \]

\[ \times \exp\left(-\frac{(X - (r-\sigma^2/2)\tau - \sigma^2 \tau - 2\log(B/S))^2}{2\sigma^2 \tau}\right) \]

\[ = \exp(rt) \exp\left(\frac{2r}{\sigma^2} + 1\right) \log(B/S) \exp\left(-\frac{(X - (r-\sigma^2/2)\tau - \sigma^2 \tau - 2\log(B/S))^2}{2\sigma^2 \tau}\right) \]

\[ = \exp(rt) \left( \frac{B}{S} \right)^{2r/\sigma^2 + 1} \exp\left(-\frac{(X - (r-\sigma^2/2)\tau - \sigma^2 \tau - 2\log(B/S))^2}{2\sigma^2 \tau}\right) \]
So we have:

\[ IC = -\left(\frac{B}{S}\right)^{2r/\sigma^2+1} \frac{S}{\sigma \sqrt{\tau}} dX \]

Letting \( u = (X - (r - \sigma^2/2)\tau - \sigma^2\tau - 2 \log(B/S))/(\sigma \sqrt{\tau}) \) we have \( dX = \sigma \sqrt{\tau} du \) and

\[ IC = S\left(\frac{B}{S}\right)^{2r/\sigma^2+1} N_1(-k_4) \] (G.6)

where

\[ k_4 = \frac{\log(E/S) - (r - \sigma^2/2)\tau - \sigma^2\tau - 2 \log(B/S)}{\sigma \sqrt{\tau}} = \frac{\log(ES/B^2) - (r + \sigma^2/2)\tau}{\sigma \sqrt{\tau}} \]

which gives

\[ IC = S\left(\frac{B}{S}\right)^{2r/\sigma^2+1} N_1(-k_4) \]

or letting \( d_4 = -k_4 \) we have

\[ IC = -S\left(\frac{B}{S}\right)^{2r/\sigma^2+1} N_1(d_4), \quad \text{where} \quad d_4 = \frac{\log(B^2/ES) + (r + \sigma^2/2)\tau}{\sigma \sqrt{\tau}} \] (G.7)

Therefore the value for the down and out call option is: \( c_{do} = I_A + I_C + I_D \) which, on collecting all the terms, yields:

\[
\begin{align*}
\text{Value of the down and out call option} \\
\quad c_{do} = S \left( N_1(d_1) - N_1(d_4) \left(\frac{B}{S}\right)^{2r/\sigma^2+1} \right) \\
- E \exp(-r\tau) \left( N_1(d_2) - N_1(d_3) \left(\frac{B}{S}\right)^{2r/\sigma^2-1} \right)
\end{align*}
\] (G.8)

### G.2 THE UP AND OUT CALL

We will now derive the formula for the value, \( c_{uo} \), of an up and out call option which was given in Part I Section 2.4.3.

\[ c_{uo} = \frac{\exp(-r\tau)}{\sigma \sqrt{\tau} \sqrt{2\pi}} \int_{X = \log(E/S)}^{\log(B/S)} \{S \exp(X) - E\} f(X < B) dX \] (G.9)
where

\[ f(X < B) = \frac{1}{\sigma \sqrt{2\pi}} \sqrt{\frac{2}{\pi}} \exp \left( -\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau} \right) \]
\[ \times \left\{ 1 - \exp \left( \frac{2\log(B/S)(X - \log(B/S))}{\sigma^2\tau} \right) \right\} \]  \hspace{1cm} (G.10)

We will represent this integral as:

\[ c_{uo} = I_A + I_B \]

where:

\[ I_A = \frac{\exp(-r\tau)}{\sigma \sqrt{2\pi}} \int_{X = \log(E/S)}^{\log(B/S)} \left\{ S \exp(X) - E \right\} \exp \left( -\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau} \right) dX \]

and

\[ I_B = -\frac{\exp(-r\tau)}{\sigma \sqrt{2\pi}} \int_{X = \log(E/S)}^{\log(B/S)} \left\{ S \exp(X) - E \right\} \exp \left( -\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau} \right) \]
\[ \times \exp \left( \frac{2\log(B/S)(X - \log(B/S))}{\sigma^2\tau} \right) dX \]

### G.2.1 Evaluation of integral \( I_A \)

Letting \( I_A = I_1 + I_2 \) where

\[ I_1 = \frac{S \exp(-r\tau)}{\sigma \sqrt{2\pi}} \int_{X = \log(E/S)}^{\log(B/S)} \exp(X) \exp \left( -\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau} \right) dX \]

and

\[ I_2 = -E \frac{\exp(-r\tau)}{\sigma \sqrt{2\pi}} \int_{X = \log(E/S)}^{\log(B/S)} \exp \left( -\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau} \right) dX \]

From our previous derivation of the Black–Scholes formula in Part I Section 2.3.3 we have:

\[ I_1 = \frac{S \exp(-r\tau)}{\sigma \sqrt{2\pi}} \int_{u = k_1}^{k_2} \exp \left( -\frac{u^2}{2} \right) du = S\{N_1(k_2) - N_1(k_1)\} \]

where

\[ k_1 = \frac{\log(E/S) - (r + \sigma^2/2)\tau}{\sigma \sqrt{\tau}} \quad \text{and} \quad k_2 = \frac{\log(B/S) - (r + \sigma^2/2)\tau}{\sigma \sqrt{\tau}} \]

\[ I_2 = -E \frac{\exp(-r\tau)}{\sigma \sqrt{2\pi}} \int_{u = k_3}^{k_4} \exp \left( -\frac{u^2}{2} \right) du = -E \exp(-r\tau)\{N_1(k_4) - N_1(k_3)\} \]
where \( k_3 = \frac{\log(E/S) - (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}} \) and \( k_4 = \frac{\log(B/S) - (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}} \)

Therefore

\[
I_A = S\{N_1(k_2) - N_1(k_1)\} - E \exp(-r\tau)\{N_1(k_4) - N_1(k_3)\} \quad (G.11)
\]

Letting \( I_B = I_C + I_D \) where:

\[
I_C = -\frac{S \exp(-r\tau)}{\sigma\sqrt{\tau}\sqrt{2\pi}} \int_{X = \log(E/S)}^{\log(B/S)} \exp(X) \exp\left(-\frac{(X - (r - \sigma^2/2)\tau)^2}{2\sigma^2\tau}\right) \\
\times \exp\left(\frac{2\log(B/S)(X - \log(B/S))}{\sigma^2\tau}\right) dX
\]

and

\[
I_D = \frac{E \exp(-r\tau)}{\sigma\sqrt{\tau}\sqrt{2\pi}} \int_{X = \log(E/S)}^{\log(B/S)} \exp\left(-\frac{(X - (r - \sigma^2/2)\tau - 2\log(B/S))^2}{2\sigma^2\tau}\right) \\
\times \exp\left(\frac{2\log(B/S)(X - \log(B/S))}{\sigma^2\tau}\right) dX
\]

G.2.2 Evaluation of integral \( I_D \)

In a similar manner to that in Section G.1 we have:

\[
I_D = \left(\frac{B}{S}\right)^{2(r-\sigma^2/2)/\sigma^2} \frac{E \exp(-r\tau)}{\sigma\sqrt{\tau}\sqrt{2\pi}} \\
\times \int_{X = \log(E/S)}^{\log(B/S)} \exp\left(-\frac{(X - (r - \sigma^2/2)\tau - 2\log(B/S))^2}{2\sigma^2\tau}\right) dX
\]

Letting \( u = \frac{X - (r - \sigma^2/2)\tau - 2\log(B/S)}{\sigma\sqrt{\tau}} \) gives

\[
I_D = \left(\frac{B}{S}\right)^{2(r-\sigma^2/2)/\sigma^2} \frac{E \exp(-r\tau)}{\sigma\sqrt{\tau}\sqrt{2\pi}} \int_{u = k_5}^{k_6} \exp\left(-\frac{u^2}{2}\right) du \quad (G.12)
\]

where

\[
k_5 = \frac{\log(E/S) - (r - \sigma^2/2)\tau - 2\log(B/S)}{\sigma\sqrt{\tau}} = \frac{\log(ES/B^2) - (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}}
\]

and

\[
k_6 = \frac{\log(B/S) - (r - \sigma^2/2)\tau - 2\log(B/S)}{\sigma\sqrt{\tau}} = \frac{\log(S/B) - (r - \sigma^2/2)\tau}{\sigma\sqrt{\tau}}
\]
Therefore:

\[ I_D = \left( \frac{B}{S} \right)^{2r/\sigma^2-1} E\{\exp(-r\tau)N_1(k_6) - N_1(k_5)\} \]  \hspace{1cm} (G.13)

### G.2.3 Evaluation of integral \( I_C \)

Now consider the term

\[ I_C = -\frac{S \exp(-r\tau)}{\sigma \sqrt{\tau} \sqrt{2\pi}} \int_{X = \log(E/S)}^{\log(B/S)} \exp(X) \exp\left(-\frac{(X - (r - \sigma^2/2)\tau - \sigma^2\tau - 2\log(B/S))^2}{2\sigma^2 \tau}\right) \times \exp\left(\frac{2 \log(B/S)(X - \log(B/S))}{\sigma^2 \tau}\right) dX \]

In a similar manner to that in Section G.1 we have:

\[ I_C = -\left( \frac{B}{S} \right)^{2r/\sigma^2+1} \frac{S}{\sigma \sqrt{\tau} \sqrt{2\pi}} \times \int_{X = \log(E/S)}^{\log(B/S)} \exp\left(-\frac{(X - (r - \sigma^2/2)\tau - \sigma^2\tau - 2\log(B/S))^2}{2\sigma^2 \tau}\right) dX \]

Letting \( u = (X - (r - \sigma^2/2)\tau - \sigma^2\tau - 2\log(B/S))/(\sigma \sqrt{\tau}) \) gives

\[ I_C = -S \left( \frac{B}{S} \right)^{2r/\sigma^2+1} \left\{ N_1(k_8) - N_1(k_7) \right\} \]  \hspace{1cm} (G.14)

where

\[ k_7 = \frac{\log(E/S) - (r - \sigma^2/2)\tau - \sigma^2\tau - 2\log(B/S)}{\sigma \sqrt{\tau}} \]

\[ k_8 = \frac{\log(B/S) - (r - \sigma^2/2)\tau - \sigma^2\tau - 2\log(B/S)}{\sigma \sqrt{\tau}} \]

So we have: \( c_{uo} = I_A + I_C + I_D \), which on collecting terms gives:

### Value of the up and out call option

\[ c_{uo} = S \left( \frac{B}{S} \right)^{2r/\sigma^2+1} \left\{ N_1(k_7) - N_1(k_8) \right\} - \left( \frac{B}{S} \right)^{2r/\sigma^2-1} \times E\{\exp(-r\tau) N_1(k_5) - N_1(k_6)\} \]

\[ + S\{N_1(k_2) - N_1(k_1)\} - E\exp(-r\tau)\{N_1(k_4) - N_1(2k_3)\} \]  \hspace{1cm} (G.15)
Appendix H

Algorithms for an AGARCH-I process

Here we provide pseudocode which calculates the log likelihood and is partial derivatives for a regression-AGARCH-I process. We consider residuals which have either a Gaussian distribution or a Student’s t distribution. The notation used is the same as that given in Section 20 of PART III.

H.1 GAUSSIAN DISTRIBUTION

H.1.1 The log likelihood

Deal with the first q terms of the sequence:

\[ \gamma = \gamma \]
\[ \mathcal{L}(\theta) = 0 \]

For i = 1 To num

If (mn == 1) \[ \epsilon_i = y_i - X_i^T \hat{b} \]

If (mn == 0) \[ \epsilon_i = y_i - \hat{b}_0 - X_i^T \hat{b} \]

Next i

For i = 1 To q

\[ h_i = \alpha_0 + \sum_{j=1}^{i-1} \alpha_j (\epsilon_{i-j} + \gamma)^2 + \sum_{j=i}^{q} \alpha_j \sigma_0^2 + \sum_{k=1}^{p} h_{i-k} \beta_k \]

Store the current value of \( h_i \) and keep all the previous values of \( h_i \).

\[ \mathcal{L}(\theta) = \mathcal{L}(\theta) + \frac{1}{2} \left( \log(h_i) + \frac{\epsilon_i^2}{h_i} \right) \]

Next i

Deal with the remaining terms of the sequence:

For i = q + 1 To num

\[ h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j (\epsilon_{i-j} + \gamma)^2 + \sum_{k=1}^{p} \beta_k h_{i-k} \]
Store the current value of $h_i$ and keep $N_p$ previous values of $h_i$.

$\mathcal{L}(\theta) = \mathcal{L}(\theta) + \frac{1}{2} \left( \log(h_i) + \frac{e_i^2}{h_i^2} \right)$

Next $i$

H.1.2 The first derivatives of the log likelihood

Algorithm for the first $q$ terms of the sequence:

$$\frac{\partial \mathcal{L}(\theta)}{\partial \theta_k} = 0, \quad k = 1, \ldots, N_p$$

For $i = 1$ to $q$

$$\frac{\partial h_i}{\partial \alpha_0} = 1 + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \alpha_0}$$

For $j = 1$ to $i - 1$

$$\frac{\partial h_i}{\partial \alpha_j} = (\epsilon_{i-j} + \gamma)^2$$

Next $j$

For $j = i$ to $q$

$$\frac{\partial h_i}{\partial \alpha_j} = \sigma_0^2$$

Next $j$

For $j = 1$ to $p$

$$\frac{\partial h_i}{\partial \beta_j} = h_{i-j} + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \beta_k}$$

Next $j$

$$\frac{\partial h_i}{\partial \gamma} = \sum_{j=1}^{i-1} 2(\epsilon_{i-j} + \gamma)\alpha_j + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \gamma}$$

$h_i = \alpha_0 + \sum_{k=1}^{p} h_{i-k} \beta_k + \sum_{j=1}^{i-1} \alpha_j (\epsilon_{i-j} + \gamma)^2 + \sum_{j=i}^{q} \alpha_j \sigma_0^2$

if $(mn == 1)$ then

$$\frac{\partial h_i}{\partial b_0} = -2 \sum_{k=1}^{i-1} (\epsilon_{i-k} + \gamma)\alpha_k + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial b_0} + \mathcal{H}(i - k)$$
end if
For \( j = 1 \) to \( n_{\text{reg}} \)

\[
\frac{\partial h_i}{\partial b_j} = -2 \sum_{k=1}^{i-1} (\epsilon_{i-k} + \gamma) a_k x_i^{j-k} + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial b_j} H(i-k)
\]

Next \( j \)

Store the current values of \( h_i \) and \( \partial h_i / \partial \theta \) and keep all the previous values of \( h_i \) and \( \partial h_i / \partial \theta \).

For \( k = 1 \) to \( n_{\text{par}} + 1 \)

\[
\frac{\partial \mathcal{L}(\theta)}{\partial \theta_k} = \frac{\partial \mathcal{L}(\theta)}{\partial b_k} - \frac{1}{2 h_i} \left( \frac{\epsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial b_k}
\]

Next \( k \)
if (\( mn == 1 \)) then

\[
\frac{\partial \mathcal{L}(\theta)}{\partial b_0} = \frac{\partial \mathcal{L}(\theta)}{\partial b_0} - \frac{\epsilon_i}{h_i} - \frac{1}{2 h_i} \left( \frac{\epsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial b_0}
\]
end if
For \( k = 1 \) to \( n_{\text{reg}} \)

\[
\frac{\partial \mathcal{L}(\theta)}{\partial b_k} = \frac{\partial \mathcal{L}(\theta)}{\partial b_k} - \frac{X_i^{k} \epsilon_i}{h_i} - \frac{1}{2 h_i} \left( \frac{\epsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial b_k}
\]

Next \( k \)
Next \( i \)

Algorithm for the remaining terms of the sequence:

For \( i = q + 1 \) to \( \text{num} \)

\[
\frac{\partial h_i}{\partial \alpha_0} = 1 + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \alpha_0}
\]

For \( j = 1 \) to \( q \)

\[
\frac{\partial h_i}{\partial \alpha_j} = \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \alpha_j}
\]

Next \( j \)
For \( j = 1 \) to \( p \)

\[
\frac{\partial h_i}{\partial \beta_{i-j}} = h_{i-j} + \sum_{k=1}^{p} \beta_j \frac{\partial h_{i-k}}{\partial \beta_k}
\]

Next \( j \)
\[ \frac{\partial h_i}{\partial \gamma} = \frac{\partial h_i}{\partial \gamma} + \sum_{j=1}^{q} 2(\epsilon_{i-j} + \gamma) \alpha_j + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial \gamma} \]

\[ h_i = \alpha_0 + \sum_{k=1}^{p} h_{i-k} \beta_k + \sum_{j=1}^{q} \alpha_j (\epsilon_{i-j} + \gamma)^2 \]

if (\(mn = 1\)) then

\[ \frac{\partial h_i}{\partial b_0} = -2 \sum_{k=1}^{q} (\epsilon_{i-k} + \gamma) \alpha_k + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial b_0} H(i-k) \]

end if

For \(j = 1\) to \(n_{\text{reg}}\)

\[ \frac{\partial h_i}{\partial b_j} = -2 \sum_{k=1}^{q} (\epsilon_{i-k} + \gamma) \alpha_k X_{i-k}^j + \sum_{k=1}^{p} \beta_k \frac{\partial h_{i-k}}{\partial b_j} H(i-k) \]

Next \(j\)

Store the current values of \(h_i\) and \(\partial h_i/\partial \theta\) and keep \(N_p\) previous values of \(h_i\) and \(\partial h_i/\partial \theta\).

For \(k = 1\) to \(n_{\text{par}} + 1\)

\[ \frac{\partial L(\theta)}{\partial \theta_k} = \frac{\partial L(\theta)}{\partial \theta_k} - \frac{1}{2h_i} \left( \frac{\epsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial \theta_k} \]

Next \(k\)

if (\(mn = 1\)) then

\[ \frac{\partial L(\theta)}{\partial b_0} = \frac{\partial L(\theta)}{\partial b_0} - \frac{\epsilon_i}{h_i} - \frac{1}{2h_i} \left( \frac{\epsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial b_0} \]

end if

For \(k = 1\) to \(n_{\text{reg}}\)

\[ \frac{\partial L(\theta)}{\partial b_k} = \frac{\partial L(\theta)}{\partial b_k} - \frac{X_k^j \epsilon_i}{h_i} - \frac{1}{2h_i} \left( \frac{\epsilon_i^2}{h_i} - 1 \right) \frac{\partial h_i}{\partial b_k} \]

Next \(k\)

Next \(i\)

H.2 STUDENT’S \(t\) DISTRIBUTION

H.2.1 The log likelihood

Deal with the first \(q\) terms of the sequence:

\[ \gamma = \hat{\gamma} \]

\[ L(\theta) = 0 \]

\[ M_\nu = \log(\Gamma((\nu + 1)/2)) - \log(\Gamma(\nu/2)) - \frac{1}{2} \log(\nu - 2) \]
Appendix H

For $i = 1$ to $\text{num}$

\[ \begin{align*}
    \text{If } (m_n = 1) & \quad \epsilon_i = y_i - \mathbf{x}_i^T \hat{\mathbf{b}} \\
    \text{If } (m_n = 0) & \quad \epsilon_i = y_i - \hat{b}_0 - \mathbf{x}_i^T \hat{\mathbf{b}}
\end{align*} \]

Next $i$

For $i = 1$ to $q$

\[ h_i = \alpha_0 + \sum_{j=1}^{i-1} \alpha_j (\epsilon_{i-j} + \gamma)^2 + \sum_{j=i}^{q} \alpha_j \sigma_0^2 + \sum_{k=1}^{p} h_{i-k} / \beta_k \]

Store the current value of $h_i$ and keep all the previous values of $h_i$.

\[ \mathcal{L}(\theta) = \mathcal{L}(\theta) - \mathcal{M}_\nu + \frac{1}{2} \log(h_i) + \frac{\nu + 1}{2} \log \left( 1 + \frac{\epsilon_i^2}{(\nu - 2)h_i} \right) \]

Next $i$

Deal with the remaining terms of the sequence:

For $i = q + 1$ to $\text{num}$

\[ h_i = \alpha_0 + \sum_{j=1}^{q} \alpha_j (\epsilon_{i-j} + \gamma)^2 + \sum_{k=1}^{p} \beta_k h_{i-k} \]

Store the current value of $h_i$ and keep $N_p$ previous values of $h_i$.

\[ \mathcal{L}(\theta) = \mathcal{L}(\theta) - \mathcal{M}_\nu + \frac{1}{2} \log(h_i) + \frac{\nu + 1}{2} \log \left( 1 + \frac{\epsilon_i^2}{(\nu - 2)h_i} \right) \]

Next $i$

H.2.2 The first derivatives of the log likelihood

Algorithm for the first $q$ terms of the sequence:

\[ \frac{\partial \mathcal{L}(\theta)}{\partial \theta_k} = 0, \quad k = 1, \ldots, N_p \]

For $i = 1$ to $q$

Compute $h_i$ as described in Section H.2.1. Also calculate the derivatives $\partial \epsilon_i / \partial \theta_j$, $j = 1, \ldots, N_p$ as described for a Gaussian distribution in Section H.1.2.

Store $h_i$ and $\partial \epsilon_i / \partial \theta_j$, $j = 1, \ldots, N_p$, and keep all the previous values of $h_i$ and $\partial \epsilon_i / \partial \theta$.

Set

\[ G = \left( \frac{(\nu + 1)}{(\nu - 2) + (\epsilon_i^2 / h_i)} \right) \]

For $k = 1$ to $\text{npar} + 1$
\[
\frac{\partial L(\theta)}{\partial \theta_k} = \frac{\partial L(\theta)}{\partial \theta_k} - \frac{1}{2h_i} \left( 1 - \frac{c_i^2}{h_i} \right) \frac{\partial h_i}{\partial \theta_k}
\]

Next \(k\)

\[
\frac{\partial L(\theta)}{\partial \nu} = \frac{\partial L(\theta)}{\partial \nu} - \frac{1}{2} \psi \left( \frac{\nu + 1}{2} \right) + \frac{1}{2} \psi \left( \frac{\nu}{2} \right) + \frac{1}{2(\nu - 2)} 
+ \frac{1}{2} \log \left( 1 + \frac{c_i^2}{(\nu - 2)h_i} \right) - \frac{c_i^2}{2(\nu - 2)h_i} G
\]

if \((mn = 1)\) then

\[
\frac{\partial L(\theta)}{\partial b_0} = \frac{\partial L(\theta)}{\partial b_0} - \frac{c_i}{h_i} G - \frac{1}{2h_i} \left( \frac{c_i^2}{h_i} G - 1 \right) \frac{\partial h_i}{\partial b_0}
\]

end if

For \(k = 1\) to \(\text{npreg}\)

\[
\frac{\partial L(\theta)}{\partial b_k} = \frac{\partial L(\theta)}{\partial b_k} - X^k_j c_i G - \frac{1}{2h_i} \left( \frac{c_i^2}{h_i} G - 1 \right) \frac{\partial h_i}{\partial b_k}
\]

Next \(k\)

Next \(i\)

Algorithm for the remaining terms of the sequence:

For \(i = q+1\) to \(\text{num}\)

Compute \(h_i\) as described in Section H.2.1. Also calculate the derivatives \(\partial h_i/\partial \theta_j\), \(j = 1, \ldots, N_p\) as described for a Gaussian distribution in Section H.1.2.

Store \(h_i\) and \(\partial h_i/\partial \theta_j\), \(j = 1, \ldots, N_p\), and keep \(N_p\) previous values of \(h_i\) and \(\partial h_i/\partial \theta\).

Set

\[
G = \left( \frac{(\nu + 1)}{(\nu - 2) + (c_i^2/h_i)} \right)
\]

For \(k = 1\) to \(\text{npreg} + 1\)

\[
\frac{\partial L(\theta)}{\partial \theta_k} = \frac{\partial L(\theta)}{\partial \theta_k} - \frac{1}{2h_i} \left( 1 - \frac{c_i^2}{h_i} \right) \frac{\partial h_i}{\partial \theta_k}
\]

Next \(k\)

\[
\frac{\partial L(\theta)}{\partial \nu} = \frac{\partial L(\theta)}{\partial \nu} - \frac{1}{2} \psi \left( \frac{\nu + 1}{2} \right) + \frac{1}{2} \psi \left( \frac{\nu}{2} \right) + \frac{1}{2(\nu - 2)} 
+ \frac{1}{2} \log \left( 1 + \frac{c_i^2}{(\nu - 2)h_i} \right) - \frac{c_i^2}{2(\nu - 2)h_i} G
\]
Appendix H

if (mn == 1) then

\[ \frac{\partial \mathcal{L}(\theta)}{\partial b_0} = \frac{\partial \mathcal{L}(\theta)}{\partial b_0} - \frac{\epsilon_i}{h_i} \frac{G - 1}{2h_i} \left( \frac{\epsilon_i^2}{h_i} G - 1 \right) \frac{\partial h_i}{\partial b_0} \]

end if

For k = 1 to nreg

\[ \frac{\partial \mathcal{L}(\theta)}{\partial b_k} = \frac{\partial \mathcal{L}(\theta)}{\partial b_k} - \frac{X_k^i \epsilon_i}{h_i} \frac{G - 1}{2h_i} \left( \frac{\epsilon_i^2}{h_i} G - 1 \right) \frac{\partial h_i}{\partial b_k} \]

Next k

Next i
Appendix I

The general error distribution

This section proves various relations for the general error distribution.
The density function for the general error distribution is:

\[ f(\varepsilon_i) = \mathcal{K} \exp\left(-\frac{1}{2} \left| \frac{\varepsilon_i}{\lambda} \right|^a\right), \quad \text{where} \quad \mathcal{K} = \frac{a}{\lambda^2(1+1/a)\Gamma(1/a)} \]  \hfill (I.1)

### I.1 VALUE OF \( \lambda \) FOR VARIANCE \( h_i \)

Calculation of the scale factor \( \lambda \) required for a general error distribution with mean zero and variance \( h_i \).

The variance of the distribution, \( E(\varepsilon_i^2) \), is given by:

\[ E(\varepsilon_i^2) = \mathcal{K} \int_{-\infty}^{\infty} \varepsilon_i^2 \exp\left(-\frac{1}{2} \left| \frac{\varepsilon_i}{\lambda} \right|^a\right) d\varepsilon_i = 2\mathcal{K} \int_{0}^{\infty} \varepsilon_i^2 \exp\left(-\frac{1}{2} \left( \frac{\varepsilon_i}{\lambda} \right)^a\right) d\varepsilon_i \]

Using the standard integrals in Appendix K.1 with \( n = 2 \), \( p = a \) and \( b = 1/2(1/\lambda)^a \) gives:

\[ h_i = \frac{2\mathcal{K}}{a} \Gamma\left(\frac{3}{a}\right) \left\{ \frac{1}{2} \left( \frac{1}{\lambda} \right) \right\}^{a-3/a} \]

Which after some simplification yields:

\[ h_i = \frac{2\mathcal{K}2^{3/a}a^3}{a} \Gamma\left(\frac{3}{a}\right) \]

Substituting for \( \mathcal{K} \) and simplifying then gives:

\[ h_i = \frac{\lambda^22^{2/a} \Gamma(3/a)}{\Gamma(1/a)} \]

The required value of \( \lambda \) is therefore:

\[ \lambda = \left\{ h_i 2^{-2/a} \frac{\Gamma(1/a)}{\Gamma(3/a)} \right\}^{1/2} \]

### I.2 THE KURTOSIS

\[ E(\varepsilon_i^4) = \mathcal{K} \int_{-\infty}^{\infty} \varepsilon_i^4 \exp\left(-\frac{1}{2} \left| \frac{\varepsilon_i}{\lambda} \right|^a\right) d\varepsilon_i \]

\[ = 2\mathcal{K} \int_{0}^{\infty} \varepsilon_i^4 \exp\left(-\frac{1}{2} \left( \frac{\varepsilon_i}{\lambda} \right)^a\right) d\varepsilon_i \]
However from standard mathematical tables:
\[ \int_{0}^{\infty} \xi_{i}^{a} \exp(-b\xi_{i}^{p}) = \frac{\Gamma(k)}{pb^{k}} \]

where \( p = a, \) \( b = (1/2)(1/\lambda)^{a} \) and \( k = 5/a \) which gives:
\[ E[\xi_{i}^{a}] = 2K2^{5/a}\lambda^{5} \frac{\Gamma(5/a)}{a} = 2^{2/a}\lambda^{2}h_{i}\frac{\Gamma(5/a)}{\Gamma(3/a)} \]

From Appendix I.1 we have:
\[ E[\xi_{i}^{2}] = h_{i} = \frac{2K2^{3/a}\lambda^{3}}{a} \Gamma\left(\frac{3}{a}\right) \quad \text{and} \quad \lambda^{2} = \frac{h_{i}2^{-2/a}\Gamma(1/a)}{\Gamma(3/a)} \]

Therefore:
\[ E[\xi_{i}^{4}] = h_{i}^{2}\frac{\Gamma(5/a)\Gamma(1/a)}{\Gamma(3/a)\Gamma(3/a)} \]

Which gives the kurtosis as:
\[ K = \frac{E[\xi_{i}^{4}]}{(E[\xi_{i}^{2}])^{2}} = h_{i}^{2}\frac{\Gamma(5/a)\Gamma(1/a)}{\Gamma(3/a)\Gamma(3/a)} = \frac{\Gamma(5/a)\Gamma(1/a)}{\Gamma(3/a)\Gamma(3/a)} \]

### I.3 THE DISTRIBUTION WHEN THE SHAPE PARAMETER, \( a \) IS VERY LARGE

If the distribution has variance \( h_{i} \) then, from Appendix I.1:
\[ \lambda = \left( \frac{2^{-2/a}\Gamma(1/a)h_{i}}{\Gamma(3/a)} \right)^{1/2} \]

Now for \( 0 < x < 1 \) we have \( \Gamma(1 + x) = 1 + a_{1}x + a_{2}x^{2} + a_{3}x^{3} + \cdots, \) where the coefficients are \( |a_{i}| < 1, \) see Abramowitz and Stegun (1968).

Since \( x\Gamma(x) = \Gamma(1 + x) \) we have, so to third order in \( x: \)
\[ x\Gamma(x) = 1 + a_{1}x + a_{2}x^{2} + a_{3}x^{3} \]

This gives \( \Gamma(x) = (1/x) + a_{1} + a_{2}x + a_{3}x^{2}, \) and \( \Gamma(x) \approx 1/x \) as \( x \to 0. \)

So as \( a \to \infty \) we have the following:
\[ 2^{(1+1/a)} \approx 2, \quad 2^{-2/a} \approx 1, \quad \frac{1}{\Gamma(1/a)} \approx \frac{1}{a}, \quad \frac{\Gamma(1/a)}{\Gamma(3/a)} \approx \frac{3a}{a} = 3, \quad \text{and} \quad \frac{\Gamma(5/a)}{\Gamma(3/a)} \approx \frac{3a}{5a} = \frac{3}{5} \]

The kurtosis is then:
\[ K = \frac{\Gamma(5/a)\Gamma(1/a)}{\Gamma(3/a)\Gamma(3/a)} = \frac{9}{5} \]
Appendix I

Also as \( a \to \infty \lambda \approx (3h_i)^{1/2} \), and for the range \(-(3h_i)^{1/2} < \epsilon_i < (3h_i)^{1/2}\), we have:

\[
\frac{|\epsilon_i|^a}{\lambda} \approx \left| \frac{\epsilon_i}{(3h_i)^{1/2}} \right| \approx 0 \quad \text{and therefore} \quad \exp \left( -\frac{1}{2} \left| \frac{\epsilon_i}{\lambda} \right| \right) \approx 1
\]

Substituting the above results into Equation I.1 the probability density function reduces to:

\[
f(\epsilon_i) \approx \frac{1}{2(3h_i)^{1/2}}
\]

which is a uniform distribution \( \mathcal{U}(-(3h_i)^{1/2}, (3h_i)^{1/2}) \), with lower limit \(-(3h_i)^{1/2}\) and upper limit \(-(3h_i)^{1/2}\). 
Appendix J

The Student’s $t$ distribution

J.1 THE KURTOSIS

This section derives an expression for the kurtosis of the Student’s $t$ distribution. Since the Student’s $t$ distribution density function is:

$$f(x) = K \left[ 1 + \frac{e_i}{h_i(\nu - 2)} \right]^{-(\nu + 1)/2}$$

where

$$K = \frac{\Gamma((\nu + 1)/2)(\nu - 2)^{-1/2}h_i^{-1/2}}{\pi^{1/2}\Gamma(\nu/2)}$$

we have:

$$E[e_i^2] = 2K \int_0^\infty \frac{e_i^2 \, de_i}{(1 + e_i^2/(h_i(\nu - 2)))^{(\nu + 1)/2}}$$

$$= 2K(h_i(\nu - 2))^{(\nu + 1)/2} \int_0^\infty \frac{e_i^2 \, de_i}{(h_i(\nu - 2) + e_i^2)^{\nu/2}}$$

Using the standard integrals in Appendix K with $a = 2$, $b = 2$, $c = (\nu + 1)/2$ and $m = (\nu - 2)h_i$ gives:

$$m^{(a+1-bc)/b} \cdot b = (h_i(\nu - 2))^{(2-\nu)/2} \cdot \Gamma \left( \frac{a+1}{b} \right) = \Gamma(3/2),$$

$$\Gamma \left( c - \frac{a+1}{b} \right) = \Gamma \left( \frac{\nu - 2}{2} \right), \quad \Gamma(c) = \Gamma \left( \frac{\nu + 1}{2} \right)$$

This gives

$$E[e_i^2] = 2K(h_i(\nu - 2))^{(\nu + 1)/2} \left\{ \frac{(h_i(\nu - 2))^{(2-\nu)/2} \sqrt{\pi} \Gamma((\nu - 2)/2)}{4\Gamma((\nu + 1)/2)} \right\}$$

Substituting for $K$ and simplifying we obtain:

$$E[e_i^2] = \frac{h_i(\nu - 2)\Gamma((\nu - 1)/2)}{\Gamma(\nu/2)}$$
Appendix J

But
\[ \left( \frac{\nu - 2}{2} \right) \Gamma \left( \frac{\nu - 2}{2} \right) = \Gamma \left( \frac{\nu - 1}{2} + 1 \right) = \Gamma \left( \frac{\nu}{2} \right) \]

So
\[ E[e_i^2] = h_i(\nu - 2)\Gamma(\nu/2) \quad \frac{2(\nu - 2)\Gamma(\nu/2)}{\nu - 2} = h_i \]

Similarly we have:
\[ E[e_i^4] = 2\mathcal{K} \int_0^\infty \frac{e_i^4 de_i}{(1 + e_i^2/(h_i(\nu - 2)))^{(\nu+1)/2}} = 2\mathcal{K}(h_i(\nu - 2))^{(\nu+1)/2} \int_0^\infty \frac{e_i^4 de_i}{(h_i(\nu - 2) + e_i^2)^{\nu+1/2}} \]

Using the standard integrals in Appendix K with: \( a = 4, \ b = 2, \ c = (\nu + 1)/2 \) and \( m = (\nu - 2)h_i \) gives:
\[
\frac{m^{(a+1-bc)/b}}{b} = \frac{(h_i(\nu - 2))^{(4-\nu)/2}}{2} \quad \Gamma \left( \frac{a + 1}{b} \right) = \Gamma(5/2), \quad \Gamma \left( c - \frac{a + 1}{b} \right) = \Gamma \left( \frac{\nu - 4}{2} \right), \quad \Gamma(c) = \Gamma \left( \frac{\nu + 1}{2} \right)
\]

and
\[ E[e_i^4] = 2\mathcal{K}(h_i(\nu - 2))^{(\nu+1)/2} \left\{ \frac{(h_i(\nu - 2))^{(4-\nu)/2} 3\pi \Gamma((\nu - 4)/2)}{8\Gamma((\nu + 1)/2)} \right\} \]

Substituting for \( \mathcal{K} \) and simplifying we obtain:
\[ E[e_i^4] = \frac{3h_i(\nu - 2)^3\Gamma((\nu - 4)/2)}{4\Gamma(\nu/2)} h_i^2 \]

But
\[ \left( \frac{\nu - 4}{2} \right) \Gamma \left( \frac{\nu - 4}{2} \right) = \Gamma \left( \frac{\nu - 2}{2} \right) \]

and
\[ \left( \frac{\nu - 2}{2} \right) \Gamma \left( \frac{\nu - 2}{2} \right) = \Gamma \left( \frac{\nu}{2} \right) \]

Therefore:
\[ \Gamma \left( \frac{\nu - 4}{2} \right) = \frac{4\Gamma(\nu/2)}{(\nu - 4)(\nu - 2)} \]
So

$$E[e_i^4] = \frac{3(\nu - 2)^2 4\Gamma(\nu/2)h_i^2}{4\Gamma(\nu/2)(\nu - 4)(\nu - 2)} = \frac{3(\nu - 2)h_i^2}{(\nu - 4)}$$

The kurtosis is then:

$$\kappa = \frac{E[e_i^4]}{(E[e_i^2])^2} = \frac{3(\nu - 2)h_i^2}{(\nu - 4)h_i^2} = \frac{3(\nu - 2)}{(\nu - 4)}$$

(J.1)
Appendix K

Mathematical reference

K.1 STANDARD INTEGRALS

Here we quote some useful standard integrals, see for example Beyer (1982).

\[
\begin{align*}
\int_0^\infty y \exp(-ay^2) \, dy &= \frac{1}{2} \\
\int_0^\infty y^2 \exp(-ay^2) \, dy &= \frac{1}{4a} \sqrt{\pi} \\
\int_0^\infty y^4 \exp(-ay^2) \, dy &= \frac{3}{8a^2} \sqrt{\pi} \\
\int_0^\infty y^{2n} \exp(-ay^2) \, dy &= \frac{1 \times 3 \times 5 \cdots (2n - 1)}{2^{n+1}a^n} \sqrt{\pi} \\
\int_0^\infty \epsilon_i^d \exp(-b \epsilon_i^p) \, d\epsilon_i &= \frac{\Gamma(k)}{p b^k}, \quad \text{where } n > -1, \ b > 0, \ m > 0 \ \text{and} \ k = \frac{(n + 1)}{p} \\
\int_0^\infty \frac{\epsilon_i^d d\epsilon_i}{(m + \epsilon_i^p)^c} &= \frac{m(a + 1 - bc)/b}{b} \frac{\Gamma((a + 1)/b)\Gamma(c - (a + 1)/b)}{\Gamma(c)}
\end{align*}
\]

where \( a > -1, \ b > 0, \ m > 0, \) and \( c > (a + 1)/b. \)

K.2 GAMMA FUNCTION

\[
\begin{align*}
\Gamma(1 + x) &= x! \\
x \Gamma(x) &= \Gamma(x + 1) \\
\Gamma\left(\frac{1}{2}\right) &= \sqrt{\pi} \\
\Gamma\left(\frac{3}{2}\right) &= \frac{\sqrt{\pi}}{2} \\
\Gamma\left(\frac{5}{2}\right) &= \frac{3\sqrt{\pi}}{4} \\
\frac{d\Gamma(x)}{dx} &= \psi(x)
\end{align*}
\]
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For $0 \leq x \leq 1$ we have

$$\Gamma(1 + x) = 1 + a_1x + a_2x^2 + a_3x^3 + a_4x^4 + a_5x^5$$

where $a_1 = -0.5748$, $a_2 = 0.9512$, $a_3 = -0.6998$, $a_4 = 0.4245$, and $a_5 = -0.1010$.

K.3 THE CUMULATIVE NORMAL DISTRIBUTION FUNCTION

In this section we show that the cumulative normal distribution function, $N_1(x)$, is related to the complementary error function, $\text{erfc}(x)$, by the following equation:

$$N_1(x) = \frac{1}{2} \text{erfc}(-x/\sqrt{2})$$

(K.1)

If we let the error function be represented by $\text{erf}(x)$ then we have:

$$\text{erf}(x) = \frac{2}{\sqrt{\pi}} \int_0^x \exp(-t^2)dt$$

Now we have the following:

$$\text{erfc}(x) = 1 - \text{erf}(x), \quad \text{erf}(-x) = -\text{erf}(x),$$

$$\text{erf}(\infty) = 1 \quad \text{and} \quad \text{erfc}(-x) = 2 - \text{erfc}(x)$$

We will consider the integral

$$I(x) = \frac{2}{\sqrt{\pi}} \int_0^x \exp(-t^2)dt = \frac{2}{\sqrt{\pi}} \int_0^x \exp(-t^2)dt + \frac{2}{\sqrt{\pi}} \int_0^\infty \exp(-t^2)dt$$

Since

$$\frac{2}{\sqrt{\pi}} \int_0^\infty \exp(-t^2)dt = 1$$

We therefore have

$$I(x) = 1 + \text{erf}(x) = 1 + \{1 - \text{erfc}(x)\} = 2 - \text{erfc}(x)$$

Substituting for $\text{erfc}(x)$ we obtain:

$$I(x) = 2 - \{2 - \text{erfc}(-x)\} = \text{erfc}(-x)$$

So we have

$$\text{erfc}(-x) = \frac{2}{\sqrt{\pi}} \int_{-\infty}^x \exp(-t^2)dt$$

(K.2)

Now the cumulative normal distribution is defined as

$$N_1(x) = \frac{1}{\sqrt{2\pi}} \int_{-\infty}^x \exp(-t^2)dt$$

Letting $u = t\sqrt{2}$, we have $du = \sqrt{2}dt$ and for the upper limit we have $x = t\sqrt{2}$ or $t = x/\sqrt{2}$. 
This integral becomes

\[ N_1(x) = \frac{1}{\sqrt{2\pi}} \int_{-\infty}^{t=x/\sqrt{2}} \exp(-t^2)\sqrt{2}dt \]  

(K.3)

So from Equation K.2 we have

\[ N_1(x) = \frac{1}{2} \text{erfc}(-x/\sqrt{2}) \quad QED \]

**K.4 ARITHMETIC AND GEOMETRIC PROGRESSIONS**

**K.4.1 Arithmetic progression**

The sum of the first \(n\) terms of an arithmetic progression is:

\[ s_n = \frac{n}{2} \left( 2a_1 + (n - 1)d \right) \]  

(K.4)

where \(a_1\) is the first term, and \(d\) is the common difference; that is the terms in the sequence are: \(a_1, a_1 + d, a_1 + 2d, a_1 + 3d, \ldots\)

**K.4.2 Geometric progression**

The sum of the first \(n\) terms of geometric progression is:

\[ s_n = \frac{a_1(1 - r^n)}{1 - r} \]  

(K.5)

where \(a_1\) is the first term, and \(r\) is the common ratio; that is the terms in sequence are: \(a_1, a_1r, a_1r^2, a_1r^3, \ldots\)
Appendix L

The stability of the Black–Scholes finite-difference schemes

L.1 THE GENERAL CASE

In this section we consider the stability of the finite-difference schemes described in Part II Section 10.6.4. It is assumed that the grid contains \( n_s \) asset points, and we will denote the time dependent option values at the \( i \)th and \((i + 1) \)th time instants by the \( n_s \times 2 \) element vectors \( X^i \) and \( X^{i+1} \) respectively. We can therefore write:

\[
T_1 X^i = T_2 X^{i+1} \tag{L.1}
\]

where \( T_1 \) and \( T_2 \) are \( n_s \times n_s \) tridiagonal matrices, and \( x_k^i, k = 1, \ldots, n_s \) will be used to denote the elements of the vector \( X^i \).

The option values at the \( i \)th time instant are computed from those at the \((i + 1) \)th time instant by using

\[
X^i = T_1^{-1} T_2 X^{i+1} \tag{L.2}
\]

However Equation L.2 is only stable if the eigenvalues of the \( n_s \times n_s \) matrix \( T_1^{-1} T_2 \) all have modulus less than one, see Smith (1985).

L.2 THE LOG TRANSFORMATION AND A UNIFORM GRID

We will now prove that the implicit finite-difference method, \( \theta_m = 0 \), when used on the log transformed Black–Scholes equation with a uniform grid is unconditionally stable; which means that the stability does not depend on the values of \( \sigma, \Delta t, \Delta Z \), etc.

From Part II Section 10.6.4 the finite-difference scheme is described by the following tridiagonal system:

\[
\begin{pmatrix}
B & C & 0 & 0 & 0 \\
A & B & C & 0 & 0 \\
0 & 0 & . & . & 0 \\
0 & 0 & 0 & . & 0 \\
0 & 0 & 0 & A & B \\
0 & 0 & 0 & 0 & A & B \\
\end{pmatrix}
\begin{pmatrix}
x_1^i \\
x_2^i \\
. \\
. \\
x_{s-1}^i \\
x_{s-2}^i \\
\end{pmatrix}
=
\begin{pmatrix}
B & \tilde{C} & 0 & 0 & 0 \\
\tilde{A} & \tilde{B} & \tilde{C} & 0 & 0 \\
0 & 0 & . & . & 0 \\
0 & 0 & 0 & \tilde{A} & \tilde{B} \\
0 & 0 & 0 & 0 & \tilde{A} & \tilde{B} \\
\end{pmatrix}
\begin{pmatrix}
x_1^{i+1} \\
x_2^{i+1} \\
. \\
. \\
x_{s-1}^{i+1} \\
x_{s-2}^{i+1} \\
\end{pmatrix}
\]
where

\[ A = \frac{(1 - \Theta_m) \Delta t}{2 \Delta Z^2} \{b \Delta Z - \sigma^2\} \]  \hspace{1cm} (L.3)

\[ B = 1 + (1 - \Theta_m) \Delta t \left\{ r + \frac{\sigma^2}{\Delta Z^2} \right\} \]  \hspace{1cm} (L.4)

\[ C = -\frac{(1 - \Theta_m) \Delta t}{2 \Delta Z^2} \{b \Delta Z + \sigma^2\} \]  \hspace{1cm} (L.5)

\[ \bar{A} = -\frac{\Theta_m \Delta t}{2 \Delta Z^2} \{b \Delta Z - \sigma^2\} \]  \hspace{1cm} (L.6)

\[ \bar{B} = 1 - \Theta_m \Delta t \left\{ r + \frac{\sigma^2}{\Delta Z^2} \right\} \]  \hspace{1cm} (L.7)

\[ \bar{C} = \frac{\Theta_m \Delta t}{2 \Delta Z^2} \{b \Delta Z + \sigma^2\} \]  \hspace{1cm} (L.8)

As in Part II Section 10.6, \( b = r - q - (\sigma^2/2) \) and \( r > 0 \).

Substituting \( \theta_m = 0 \) into Equations L.3 to L.8 we have \( \bar{A} = \bar{C} = 0 \), \( \bar{B} = 1 \) and

\[ A = \frac{\Delta t}{2 \Delta Z^2} \{b \Delta Z - \sigma^2\}, \quad B = 1 + \Delta t \left\{ r + \frac{\sigma^2}{\Delta Z^2} \right\}, \]

\[ C = -\frac{\Delta t}{2 \Delta Z^2} \{b \Delta Z + \sigma^2\} \]

The finite-difference scheme is thus represented by the equations

\[
\begin{pmatrix}
B & C & 0 & 0 & 0 & 0 \\
A & B & C & 0 & 0 & 0 \\
0 & 0 & . & . & 0 & 0 \\
0 & 0 & 0 & . & . & 0 \\
0 & 0 & 0 & A & B & C \\
0 & 0 & 0 & 0 & A & B
\end{pmatrix}
\begin{pmatrix}
x_1^j \\
x_2^j \\
\vdots \\
x_{s+1}^j \\
x_{s+2}^j
\end{pmatrix}
= 
\begin{pmatrix}
1 & 0 & 0 & 0 & 0 \\
0 & 1 & 0 & 0 & 0 \\
0 & 0 & . & . & 0 \\
0 & 0 & . & . & 0 \\
0 & 0 & 0 & 0 & 1 \\
0 & 0 & 0 & 0 & 0
\end{pmatrix}
\begin{pmatrix}
x_1^{j+1} \\
x_2^{j+1} \\
\vdots \\
x_{s-1}^{j+1} \\
x_{s-2}^{j+1}
\end{pmatrix}
\]

or in matrix notation

\[ X^t = T_1^{-1} X^{t+1} \quad (L.9) \]

where \( T_2 = I \) in Equation L.2.

As mentioned in Section L.1, Equation L.9 is stable if the modulus of all the eigenvalues of \( T_1^{-1} \) are less than one. We will now show that this is in fact the case.

If the eigenvalues of \( T_1 \) are \( \lambda_k, k = 1, \ldots, n_{s-2} \), then the eigenvalues of \( T_1^{-1} \) are \( \lambda_k^{-1}, k = 1, \ldots, n_{s-2} \). This means that the system is stable if all the eigenvalues of \( T_1 \) have a modulus greater than one. This result can be proved by considering the eigenvalue with the smallest modulus, \( \lambda_{\min} \). If \( |\lambda_{\min}| > 1 \) then the result is proved.
Appendix L

Now the eigenvalues of $T_1$, see Smith (1985), are given by:

$$
\lambda_k = 1 + \Delta t \left( r + \frac{\sigma^2}{\Delta Z^2} \right) + 2\sqrt{AC} \cos \left( \frac{k\pi}{n_{z-2} + 1} \right), \quad k = 1, \ldots, n_{z-2} \tag{L.10}
$$

where the term

$$
2\sqrt{AC} = \sqrt{\frac{\Delta t^2 (\sigma^4 - b^2 \Delta Z^2)}{\Delta Z^4}} \tag{L.11}
$$

It can be seen that if $b^2 \Delta Z^2 > \sigma^4$ then the eigenvalues are complex and if $\sigma^4 \geq b^2 \Delta Z^2$ then eigenvalues are real. We will consider each of these cases in turn.

L.2.1 Complex eigenvalues: $b^2 \Delta Z^2 > \sigma^4$

We will represent the $k$th complex eigenvalue as:

$$
\lambda_k = R + iY
$$

where the real part is $R = 1 + \Delta t \left( r + \frac{\sigma^2}{\Delta Z^2} \right)$ and the imaginary part is $Y = 2\sqrt{AC} \cos \left( \frac{k\pi}{n_{z-2} + 1} \right)$

Since $|\lambda_k| > |R| + |Y|$ and $|R| > 1$

we conclude that $|\lambda_{\text{min}}| > 1$

L.2.2 Real eigenvalues: $\sigma^4 \geq b^2 \Delta Z^2$

In this case the $k$th eigenvalue is real, and from Equation L.10 we have:

$$
\lambda_k > 1 + \Delta t \left( r + \frac{\sigma^2}{\Delta Z^2} \right) - 2\sqrt{AC}
$$

Since $b^2 \Delta^2 > 0$ from Equation L.11 we have

$$
2\sqrt{AC} < \sqrt{\frac{\sigma^4 \Delta t^2}{\Delta Z^4}} \quad \text{or} \quad \left| 2\sqrt{AC} \right| < \frac{\sigma^4 \Delta t}{\Delta Z^2}
$$

So $\lambda_{\text{min}} > 1 + \Delta t \left( r + \frac{\sigma^2}{\Delta Z^2} \right) - \frac{\sigma^2 \Delta t}{\Delta Z^2}$

Therefore we have $|\lambda_{\text{min}}| > 1 + r \Delta t$

and since $r > 0$, we have: $|\lambda_{\text{min}}| > 1$
Glossary of terms

The notation used is as follows:

\[ \psi(x) \]  The psi function, also called the digamma function, \( (\partial(\log \Gamma(x)))/\partial x = \psi(x) \)

\[ \Gamma(x) \]  The gamma function. If \( x \) is an integer then \( \Gamma(x) = (n - 1)! \)

\[ \log(x) \]  The natural logarithm of \( x \).

\[ E(x) \]  The conditional expectation value of \( x \).

\[ E[x] \]  The unconditional expectation value of \( x \).

\[ NID(a, b) \]  Normally and independently distributed variates, with mean \( a \) and variance \( b \).

\[ R(a, b) \]  An arbitrary distribution, with mean \( a \) and variance \( b \).

\[ IID(a, b) \]  Independently and identically distributed, with lower limit \( a \) and upper limit \( b \).

\[ U(a, b) \]  The uniform distribution, with lower limit \( a \) and upper limit \( b \).

\[ OLS \]  Ordinary least squares.

\[ |x| \]  The absolute value of the variable \( x \).

\[ PDF \]  The probability density function of a given distribution.

\[ D\mathcal{L}(\theta) \]  \( (\partial \mathcal{L}(\theta))/\partial \theta \)

\[ D^2\mathcal{L}(\theta) \]  \( (\partial^2 \mathcal{L}(\theta))/\partial \theta^2 \)

Leptokurtic  The distribution has a kurtosis greater than 3. This implies that the tails of the distribution are thicker than those of a Gaussian.

Platykurtic  The distribution has a kurtosis less than 3. This implies that the tails of the distribution are thinner than those of a Gaussian.

\( \hat{\theta} \)  The vector of estimated GARCH model parameters.

\( \hat{\theta}_i \)  The estimated value of the \( i \)th GARCH model parameter.
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