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**GEOLOGICAL
PROBLEM SOLVING
WITH LOTUS 1-2-3
FOR EXPLORATION
AND MINING GEOLOGY**

(with programs on diskette)

GEORGE S. KOCH, JR., University of Georgia



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Series Editor's Foreword

This contribution to the Computer Methods in the Geosciences (CMG) series by George Koch, Jr., is his second - the first being, "Exploration-Geochemical Data Analysis with the IBM PC." As with the first book, this one is another practical guide. This work is the direct result of Koch's 35 years experience in the exploration for and exploitation of mineral deposits. Coauthor with R.F. Link of the successful book on "Statistical Analysis of Geological Data" (v. 1, 1970/v. 2, 1971, John Wiley & Sons), it brings a wealth of experience to the material presented in this volume.

The reader is taken through the book step-by-step with detailed instructions on just how to accomplish each step. Subjects range through statistics, such as confidence intervals, fitting a straightline, analysis of variance, probability, and exploration models to exploration geochemical data, from economic considerations such as compound interest, depreciation and depletion, and discounted cash flow to blocking ore from workings and drillhole data, ore concentration and smelter settlement to Peters's model for mineral-property evaluation. The object of the book is to teach the user how Lotus 1-2-3 can help solve the many numerical problems facing the practitioner today.

Spreadsheets are being used more and more as earth-science users learn of the many applications to which they can be made. They are versatile and easy to use. Koch has adeptly put together the 1-2-3 programs as well as the text explaining their use. Examples are taken from his own work as well as from the literature and range from Tennessee zinc to Nevada placer gold deposits. Intended to be used by both the professional and student, this book can serve as a text/lab manual or a reference book. It should be noted too, that the material is so structured that it could be used in other geological fields, including petroleum geology.

This volume is the eighth in the series initiated in 1982. First published by Hutchinson Ross Publishing Company (Stroudsburg, PA) and later Van Nostrand Reinhold (New York), the series is being continued by Pergamon Press. The series (1) promotes the subject of geomathematics by making available material in plain English; (2) introduces the reader to the subject and where to get more information; and (3) keeps the geological public informed of latest developments in a fast-moving field. As stated in the first volume of the series it is ".....designed to be self-contained and open-ended." The same high quality is promised by the publisher and this contribution by George Koch certainly fulfills that promise.

Other titles which are in preparation include:

- Computer Applications for Geologic Exploration: Resource and Hazards, by William Green.
- Simulating Sedimentary Transport by Waves, by Paul A. Martinez and John W. Harbaugh.

DANIEL F. MERRIAM

Preface

Geologists working in the mineral industries are accustomed to solving many numerical problems, some concerned with exploring for deposits, and others with evaluation of deposits once they are located. Required is a knowledge of varied techniques, most of which may be classified as statistical or financial. The software package 1-2-3 (TM) of the Lotus Development Corporation provides a powerful new tool for implementing these techniques. The purpose of this book is to present effective 1-2-3 problem-solving methods for geology as applied in the mineral industries.

This book provides 1-2-3 programs (also named worksheets) together with a brief text that explains them. The statistical and financial principles, not explained in this book, may be presented in books for which this book may be a companion work.

This book is for both professionals and students. Professionals will determine it suitable for self-study. It also can be used as a laboratory manual for students.

Using the floppy diskette supplied, you will be able to solve problems at once, following the brief instructions. For effective learning, I have structured the ideas and worksheets from simple to complex. In order

to make it easy for you to modify the worksheets if you wish, I have made them simple and have defined and labeled the variables clearly. Advanced features of 1-2-3 are used only when absolutely necessary.

1-2-3 is ideally suited for our purpose because it combines a spreadsheet with mathematical and statistical analysis and graphical displays. The term *spreadsheet* implies in accounting “the traditional financial modeling tools: the accountant’s columnar pad, pencil, and calculator” (Ewing, D.P., and others, 1987, p. 12). In geology, the spreadsheet becomes an orthogonal x,y grid on which to drillholes, take soil samples, or lay out a mining plan. And, in mathematical/statistical terms, the spreadsheet becomes a matrix of data or calculated terms.

Therefore, with 1-2-3 your only investment in time (or, overhead in computer language) is learning its rudiments, rather than having to learn several systems, perhaps one for statistics, one for databases, and yet another one for financial calculations. And, because 1-2-3 is used widely, many books are available if you need them. Help also is available from the Lotus Development Corp. Moreover, because of the widespread acceptance of 1-2-3, we can anticipate that the system will be maintained in the future. This book’s programs should be compatible with future releases of updated versions of 1-2-3.

I have learned programming in 1-2-3 is highly rewarding in contrast to programming in a high-level language such as FORTRAN or BASIC. Programming in 1-2-3 is to programming in BASIC or FORTRAN as building with premade window and door units is to building with only dimension lumber and glass. Correcting any bugs that may exist in this book or in other programs that you may write will be less of a problem than in high-level languages because of 1-2-3’s structure and organizational support.

Although the programs in this book are written for problems about metallic ore deposits, they also are pertinent for industrial minerals and are relevant for problem solving in other geological specialties including petroleum geology.

The example problems come from many sources. Some are from the excellent 1987 book “Exploration and Mining Geology,” by W. Peters

(John Wiley & Sons, New York); in fact, I have included most of Peters's numerical examples; others were taught me by H.E. McKinstry. A number are from my work, published and unpublished.

T.J. Bornhorst, J.T. Hanley, D. Papacharalampos, W.C. Peters, and T.B. Thompson reviewed various drafts of this book and made many helpful suggestions. Of course, I am responsible for any mistakes and shortcomings.

GEORGE S. KOCH, JR.

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CHAPTER 1

Introduction

This book contains Lotus 1-2-3 worksheets (on a floppy diskette) and a brief explanatory text. In addition, you need to have 1-2-3 (version 2.01 or later) available.

I assume that you have a basic understanding of 1-2-3. If you are unfamiliar with the system, the "Introduction to 1-2-3" of the Lotus Development Corporation will be helpful. The Corporation's 1-2-3 manuals are useful for reference. An excellent introduction to 1-2-3 in general is by Ewing and others (1987), and a book on statistics with 1-2-3 has been written by Kilpatrick (1987).

I assume that you are familiar with the elementary concepts and standard notation of mathematical statistics, available in a geological framework in books by Koch and Link (1970, 1971), Davis (1986), or in standard statistical textbooks. Also assumed is familiarity with basic methods of mineral-deposit evaluation, as developed by Peters (1987) in a geologically oriented book or by Stermole and Stermole (1987) in a book on economic evaluation.

The floppy diskette contains 1-2-3 worksheets, which have the property that the model, consisting of the formulae, is not separated from the data, which consist of the values that change from one problem to another. This situation is unlike that in higher level languages (for example, FORTRAN or BASIC), in which the data and formulae are

clearly separated. I provide worksheets for representative geological and mining situations. To use the worksheets for your data, simply replace the example data with your data, altering the numbers of rows and columns if necessary. Alternatively, remove all of the data, to create dataless worksheets, named *templates*, in 1-2-3 jargon. (Be careful to remove only data, not formulae; you can identify data cells by inspection on the screen, or by printing out the contents of all cells, using the /Print/Printer/Options /Other/Cell-Formula command.)

The rest of this chapter tells you how to get started using the worksheets, provides some other introductory material, and ends with a general conclusion for this book.

HOW TO START

Many of us prefer to see something on the screen before reading text. If you belong to this group, here is how to start.

You need an IBM Personal Computer, with enough memory to run 1-2-3, and preferably with a graphics card, or a compatible computer made by another vendor. Desirable but not required are a monitor that that will display graphs and a printer that will make hard copies of them.

The diskette in this book contains 1-2-3 worksheets. You will need to copy this diskette (termed the original diskette) to one or more working diskettes or to a hard disk. There are two reasons for doing this. First, it is desirable to save the original diskette so that it will not wear out through repeated use, and, second, you might inadvertently erase or alter a worksheet in a way that you do not intend, so you need to be able to retrieve an unaltered worksheet from an original diskette. (I generally keep two or three working diskettes.) Because the original diskette is write protected, it is difficult to damage while you are copying, but, if you are uncertain about the procedure, you may want to consult your computer manual and practice on other diskettes.

To copy the original diskette to a working diskette, follow the instructions with your computer. For the IBM Personal Computer with two diskette drives, first format a diskette. The command to copy and verify 1-2-3 worksheet files is

COPY *.* B:/V.

Alternatively, you can copy the original diskette to a directory on a hard disk.

The worksheets take up about 310,000 of the some 360,000 characters that can be stored on a floppy diskette.

CONVENTIONS

Long worksheets are programmed for manual recalculation. To recalculate, press the function key (F9).

All references to particular keys are to the keyboard of the IBM Personal Computer. Cell references are in capital letters and numbers, for example, D7. For ease in reference, I have numbered the rows of each worksheet in column A and have left column B empty. Columns are labeled wherever necessary.

Most formulae follow the notation in Peters's (1987) book, except for compound interest factors, for which I have adopted Stermole and Stermole's (1987) notation. In general, metric units and U.S. dollars are used.

For clarity, I have labeled most of the variables used in the worksheets, using 1-2-3's `/Range Name Label` command. In the individual worksheets, the variables are listed in 1-2-3's alphabetical order, as provided by the command `/Range Name Table`.

Lotus 1-2-3 limits labels to 15 upper-case alphabetic and certain other characters; within this limitation my intent is to make the labels easy to read. Toward this end, I have used these conventions:

- *Words, letters, and numbers within a label are separated by periods (.) rather than by symbols like slashes (/) and dashes (-) that can be confused with mathematical operators.
- *Compound interest factors are preceded by a poundsign (#).
- *Labels for dollar values are preceded by an exclamation mark (!), rather than a dollar sign, because 1-2-3 uses a dollar sign for a particular purpose (to set cell references).

*Percentage factors are preceded by a percent sign (%).

*Labels for metals start with the chemical symbol, for example AG.

Macros, which are programs written in 1-2-3's command language, are used in Chapters 6 and 7. They are listed in the worksheets and named in alphabetical order.

CONCLUSIONS

As you practice with the worksheets in this book, you will doubtless learn many ways to adopt them for your particular purposes. They will also suggest other worksheets that you devise. 1-2-3 worksheets and templates are being published in increasing numbers in geology and mining engineering.

What precautions need you take? For at least two reasons, it is difficult to locate mistakes in worksheets: (1) worksheets do not separate formulae and data and (2) macros are, at present, difficult to read and correct. This situation, we can hope is temporary. Moreover, one always need to be aware of the choices possible to solve a particular problem, one or another spreadsheet, perhaps a statistical-analysis package, or a high-level language.

In the future, I anticipate that it will be easier to interface the various types of computational aids with one another than at present. And more functions will be developed for 1-2-3 that will make scientific, statistical, and engineering computing more routine than at present.

CHAPTER 2

Confidence Intervals

In this chapter, we will use 1-2-3 to calculate confidence intervals for a population mean. The principles and methods are explained by Koch and Link (1970, p. 79-104), Peters (1987), p. 476), and by standard statistical texts.

2.1 CONFIDENCE INTERVALS FOR A FICTITIOUS DATA SET

Suppose that you have drilled five exploration diamond-drillholes in the Middle Tennessee zinc district. Suppose further that the zinc assays are 2, 4, 6, 6, and 7%, as recorded in Part 1 of worksheet C2A.WK1.

In Part 2, we calculate a confidence limit at a 10-percent risk level for the data in Part 1. Line 26 records the number of drillholes; line 28 the arithmetic average (sample mean); line 30 the sample variance; and line 32 the sample standard deviation. The value for Student's *t* is entered in line 34 from a standard table present in statistical textbooks (also available in Koch and Link, 1970, p. 346). Line 36 is *D*, the distance that is one-half the width of the confidence interval; line 38 is *UCL*, the upper confidence limit; and line 40 is *LCL*, the lower confidence limit. For this

data set, the sample variance is exactly 4, and the sample standard deviation is exactly 2. Therefore, it is easy to check the 1-2-3 arithmetic that calculates these statistics. Because 1-2-3's function @VAR calculates a population variance rather than a sample variance, the formula in line 30 is corrected to obtain the required sample variance. Part 3 of worksheet C2A.WK1 is the table of labels used to make the formulae easy to read.

2.2 CONFIDENCE INTERVALS FOR DATA FROM THE ELMWOOD, TENNESSEE ZINC DEPOSIT

The data in the previous section are fictitious, selected for their simplicity. In Part 4, we repeat the calculations for data from six surface boreholes drilled to evaluate the Elmwood, Tennessee zinc deposit. These holes are six of 121 holes drilled by the New Jersey Zinc company to evaluate the deposit (Callahan, 1977); the data are from Chapter 6 (worksheet C6B.WK1), where the geological situation is discussed. The confidence interval, from 4.10 to 5.60 percent zinc, includes the mined grade of 4.2 percent for the first year of production (Koch and Schuene-meyer, 1982, p. 660).

2.3 CONFIDENCE INTERVALS FOR THE DON TOMAS VEIN, FRISCO MINE

The Don Tomas vein of the Frisco mine, Chihuahua, Mexico contained a large and rich ore shoot some 600 m long and 40 m high (Koch and Link, 1964); Figure 2.1 is a vertical longitudinal section. The vein was discovered and explored by diamond drilling.

Part 1 of worksheet C2B.WK1 tabulates data for 18 diamond-drill-holes bored through the vein. (The units, meter-grams per tonne, and meter-percent per tonne are appropriate for evaluating narrow tabular ore bodies.) Part 2 gives the calculations and confidence intervals for the five metals that were produced from this vein. As explained by Koch and Link (1970, p. 100), the confidence limits include the grade of ore mined

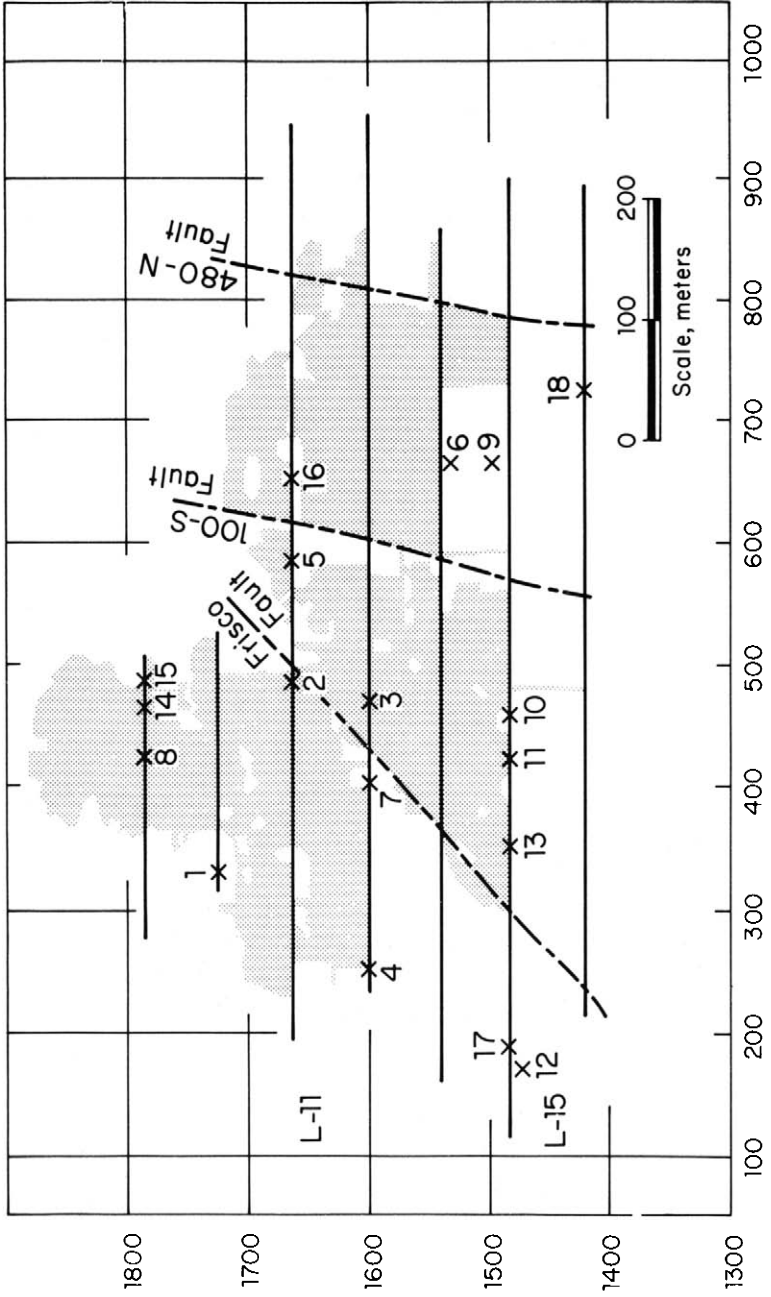


Figure 2.1 Don Tomas vien, Frisco mine, vertical longitudinal section

as established from 1,829 drift samples taken after the vein was developed for mining.

Usually, confidence intervals narrow as more data (observations, in statistical nomenclature) are obtained, corresponding in this example to more holes being drilled. They narrow for two reasons: (1) in the equation used to calculate D in line 51, the number of observations, N, in the denominator, increases, and (2) the value of t, in the numerator, decreases. Figure 2.2 graphs this change for lead. The interval width narrows from 2 to 13 holes as expected. It then widens at hole 14, because the grade of lead in this hole is so high, an illustration of statistical fluctuation. The interval width for holes 15 to 18 continues to narrow progressively.

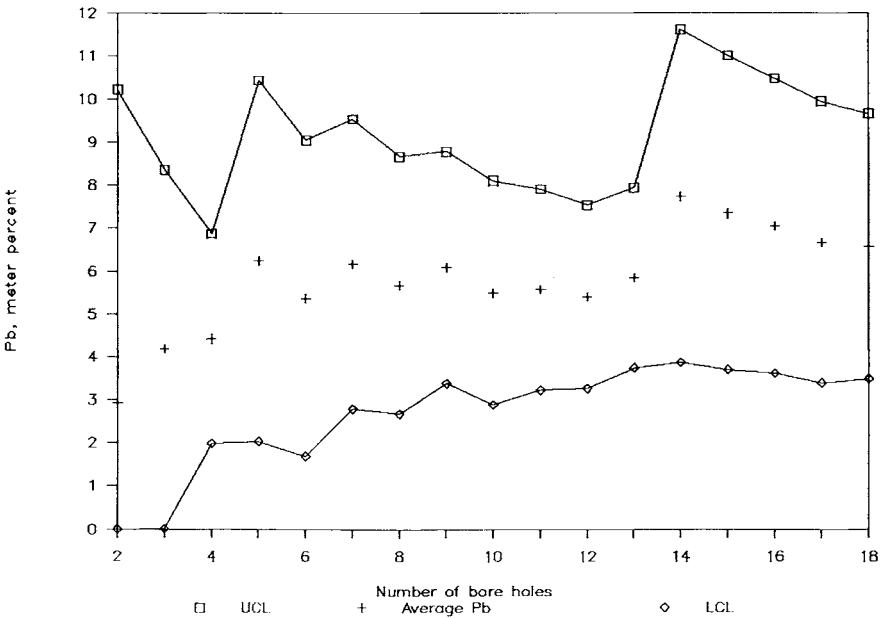


Figure 2.2 Confidence intervals for lead, Don Tomas vein, Frisco mine.

Worksheet C2A Confidence intervals for Elmwood mine

```

1  1 Dec 87, file C2A.WK1 on disks GSK 059-061
2
3  Worksheet C2A.WK1. Confidence intervals for the Elmwood
4  mine
5  1. Fictitious assay data from drilling in Middle
6  Tennessee
7  -----
8      Zn,
9      %
10 -----
11         2
12         4
13         6
14         6
15         7
16 -----
17
18
19 2. Calculation of a confidence interval for the popula-
20 tion mean
21         C           D
22 -----
23 Item           Zn,
24                %
25 -----
26 N                5
27
28 AVG                5
29
30 VAR                4
31
32 STD                2
33
34 T(5%)            2.132
35
36 D                1.91

```

Worksheet C2A (continued)

37		
38	UCL	6.91
39		
40	LCL	3.09
41		
42	-----	
43		
44		
45	3. Table of labels	
46		
47	AVERAGE	D28
48	D	D36
49	HOLES	C11..C15
50	N	D26
51	STD.DEVIATION	D32
52	T	D34
53	VARIANCE	D30
54		
55		
56	4. Elmwood mine data (from worksheet C6B.WK1).	
57		
58	-----	
59	Zn,	
60	%	
61	-----	
62	3.6	
63	6.0	
64	5.2	
65	5.9	
66	5.5	
67	3.4	
68	3.6	
69	5.6	
70	-----	
71		
72		

Worksheet C2A (continued)

73 2. Calculation of a confidence interval for the
 74 population mean

75

	C	D
76	-----	-----
77	Item	Zn,
78		%
79	-----	-----
80	N	8
81		
82	AVG	4.85
83		
84	VAR	1.25
85		
86	STD	1.12
87		
88	T (5%)	1.895
89		
90	D	0.75
91		
92	UCL	5.60
93		
94	LCL	4.10
95		
96	-----	-----

Worksheet C2B Confidence intervals for Frisco mine

1 11 Dec 87, file C2B.WK1 on disks GSK 059-061

2

3 Worksheet C2B.WK1. Confidence intervals for the Frisco mine

4

5 1. Assay data from diamond-drillholes bored through the
6 Don Tomas vein (from Koch and Link, 1964, p.24).

7

8

9 C	D	E	F	G	H	I	J
10	-----						
11 Hole	Au,	Ag,	Pb,	Cu,	Zn,	x,	y,
12 No.	m-g/T	m-g/T	m-%	m-%	m-%	metrs	metrs
13	-----						
14 1	0.15	16.8	1.77	0.090	1.44	328	1,728
15 2	0.24	115.2	4.08	0.072	13.20	496	1,665
16 3	1.20	384.0	6.72	0.120	5.88	460	1,603
17 4	0.00	172.8	5.16	0.240	4.92	250	1,603
18 5	0.00	893.2	13.44	0.700	5.11	583	1,665
19 6	0.00	80.0	1.04	0.080	1.04	665	1,518
20 7	1.30	239.2	10.92	0.416	5.98	400	1,603
21 8	0.07	37.8	2.24	0.238	4.76	422	1,787
22 9	0.00	708.0	9.36	0.360	8.64	665	1,499
23 10	0.00	4.0	0.24	0.060	0.24	458	1,483
24 11	0.00	88.0	6.25	0.190	12.50	413	1,483
25 12	1.08	252.0	3.60	0.594	9.36	166	1,473
26 13	0.00	261.0	11.10	0.250	25.40	346	1,483
27 14	0.00	294.8	32.34	1.210	36.08	465	1,787
28 15	0.00	45.0	1.98	0.315	4.59	483	1,787
29 16	0.49	256.2	2.52	0.140	3.22	652	1,665
30 17	0.00	170.1	0.63	0.903	10.71	183	1,483
31 18	0.35	987.0	4.90	0.455	1.75	727	1,424

32

33 Note: m-g/T = meter-grams per tonne; m-% = meter-percent per tonne

Worksheet C2B (continued)

34

35

36 2. Calculation of a confidence interval for the
 population mean

37 -----

38 Item	Au,	Ag,	Pb,	Cu,	Zn,
39	m-g/T	m-g/T	m-%	m-%	m-%
40 -----					
41 N	18	18	18	18	18
42					
43 AVG	0.27	278	6.57	0.36	8.60
44					
45 VAR	0.20	85,901	56.68	0.10	82.58
46					
47 STD	0.45	293	7.53	0.32	9.09
48					
49 T(5%)	1.74	2	1.74	1.74	1.74
50					
51 D	0.18	120	3.09	0.13	3.73
52					
53 UCL	0.46	398	9.66	0.49	12.33
54					
55 LCL	0.09	158	3.48	0.23	4.87

56

57 -----

58

59

60 3. Table of labels

61

62 AVERAGE	D43
63 D	D51
64 HOLES	D14..D31
65 N	D41
66 STD.DEVIATION	D47
67 T	D49
68 VARIANCE	D45

CHAPTER 3

Fitting a Straightline To Exploration Data

In this chapter, we will fit a straightline to exploration data through the statistical method of linear regression, which is explained by Koch and Link (1971, p. 7-15) and by standard statistical textbooks. This method determines the equation of the straightline with the property that the sum of the squared vertical deviations of the points from the line is a minimum.

Linear regression is also useful for problems of ore-deposit evaluation, for instance to compare different methods of sampling or assaying.

3.1 GOLD PLACER DATA FROM MANHATTAN, NEVADA

Worksheet C3A.WK1 contains data for gold nuggets collected in a stream near Manhattan, Nevada (Ferguson, 1916). The following hypothesis will be tested: that gold percentage rises with increasing distance from the place where the stream crosses a lode gold deposit that is the source of the nuggets. The supposition is that, as the nuggets travel downstream, silver and base metals are leached from them.

Columns C and D of the data table give the paired original values for distance and gold percentage. After the data table, I did a linear regression using the /Data Regression command. The output, which is presented in lines 49 to 57, is explained in the references cited in the previous paragraph and in 1-2-3 manuals.

In lines 63 to 66, we set up a table of labels. CONSTANT, the y-intercept, is in cell F50, and X.COEFFICIENT, the slope of the line, is in cell E56. Remembering the equation for a straightline,

$$y\text{-hat} = ax + b,$$

where y-hat is the estimated gold percentage, x is the distance, and a and b are constants named the slope and intercept, respectively, we obtain for cell E12 the formula,

$$(\$X.COEFFICIENT)*(C12)+(\$CONSTANT).$$

By copying this formula into cells E13..E43, we obtain the fitted y-hat values corresponding to the original data points. (The y-hat values are the gold values predicted by the linear equation for each of the x-values of distance.)

Figure 3.1 shows the original data points and the plotted straightline.

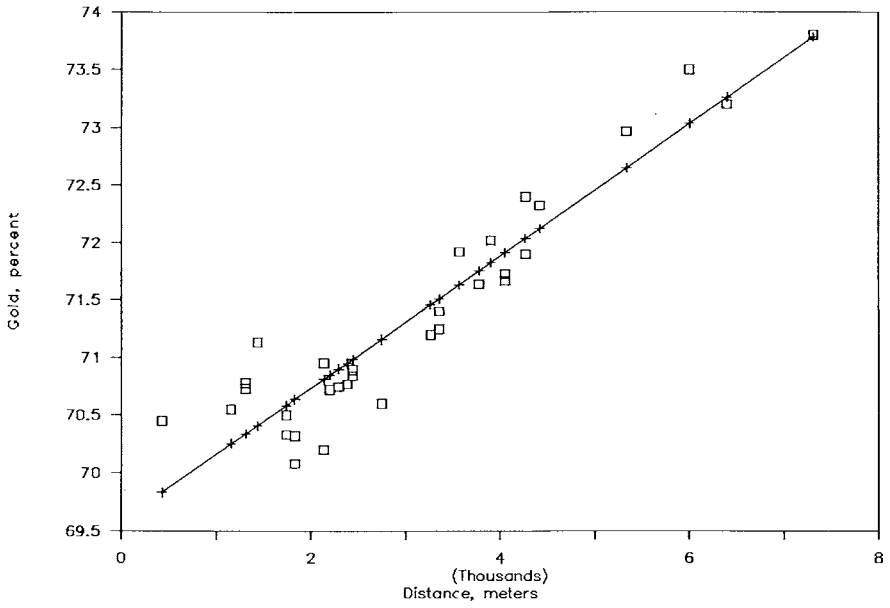


Figure 3.1 Linear regression, gold nuggets, Manhattan, Nevada.

Worksheet C3A Linear regression

1 15 Sep 87, file C3A.WK1 on disks GSK 056-0

2

3 Worksheet C3A.WK1. Linear regression

4

5 Part 1. Table of data

6

7 C D E F

8

9 x, y, y-hat,

10 m % %

11

12 427 70.45 69.83

13 1158 70.55 70.25

14 1311 70.73 70.34

15 1311 70.78 70.34

16 1433 71.13 70.41

17 1737 70.33 70.58

18 1737 70.50 70.58

19 1829 70.08 70.64

20 1829 70.32 70.64

21 2134 70.95 70.81

22 2134 70.20 70.81

23 2195 70.72 70.85

24 2286 70.75 70.90

25 2377 70.77 70.95

26 2438 70.84 70.99

27 2438 70.90 70.99

28 2743 70.60 71.16

29 3261 71.20 71.46

30 3353 71.40 71.51

31 3353 71.25 71.51

32 3566 71.92 71.63

33 3780 71.64 71.76

34 3901 72.02 71.83

35 4054 71.73 71.91

36 4054 71.67 71.91

37 4267 72.40 72.04

38 4267 71.90 72.04

Worksheet C3A (continued)

39	4420	72.33	72.12
40	5334	72.96	72.65
41	6005	73.50	73.03
42	6401	73.20	73.26
43	7315	73.80	73.78

44
45
46

47 Part 2. 1-2-3's linear regression

48

49 Regression Output:

50	Constant	69.58880
51	Std Err of Y Est	0.342396
52	R Squared	0.884272
53	No. of Observations	32
54	Degrees of Freedom	30
55		
56	X Coefficient(s)	0.000573
57	Std Err of Coef.	0.000037

58
59

60
61
62

63 3. Table of labels

64

65	CONSTANT	F50
66	X.COEFFICIENT	E56

CHAPTER 4

Analysis Of Variance

If we have several sets of exploration data, we may want to compare them to learn whether they come from bodies with similar mineralization, as measured in quantitative terms that may be grades of ore, geochemical values, tonnages, or one of many other variables. In statistical nomenclature, we are comparing sample data to determine whether populations are evidently the same or different. For example, we may have sampled several Nevada prospects for gold. Do the data support the hypothesis that the gold content of the orebodies is the same, recognizing that the sampling provides only an estimate of the true grades?

We can answer this question by making a one-way analysis of variance, as explained by Koch and Link (1970, p. 132-201) and by standard statistical textbooks.

The one-way analysis of variance is only the simplest form of this powerful method of statistical analysis. Using 1-2-3, other forms of the analysis of variance can be programmed. In this book, I provide a worksheet for one other form, the randomized block analysis of variance. This analysis is particularly useful in mining geology for comparing various methods of sampling, sample preparation, or chemical analysis. Li (1964) provides many other easy-to-program models.

Sections 4.1 and 4.2 explain the one-way analysis of variance; sections 4.3 and 4.4 explain the randomized-block analysis of variance.

4.1 ONE-WAY ANALYSIS OF VARIANCE

Worksheet C4A.WK1 for the one-way analysis of variance follows the form of Li (1964, p. 194-196).

Part 1 is a table of data; there are three groups, numbered 1 to 3, and eight values, of which two are in the first sample, two in the second, and four in the third. Pretend that we drilled holes in three areas in the Middle Tennessee zinc district and obtained these zinc analyses in the eight holes.

Part 2 is the one-way analysis of variance. 1-2-3 calculates all of the entries, except for the tabled value 3.78 in cell I44, which is the F-value at the 10% level with 2 and 5 degrees of freedom from Koch and Link (1970, p. 348) or from any statistical textbook. Because the calculated F-value is larger than the tabled F-value, we conclude that the three groups of samples represent groups with significantly different average grades. In terms of our example, this small amount of drilling suggests that the three areas in Middle Tennessee have significantly different grades of zinc mineralization.

Part 3 is the table of labels.

Part 4 details the calculations. Part 4.A lists the sums, counts, and means; Part 4.B is the table to compute the elements, I, II, and III, which are used to calculate the analysis of variance. Part 4.C gives the squared matrix of the table of data. The table provides for as many as six data groups (samples, in statistical terminology) and as many as 20 data points (observations in statistical terminology) per group. If the data table has too few rows, you can increase the number with the command /Worksheet Insert Row, in order to avoid changing the formulae. Increasing the number of columns will require minor modifications of the formulae.

4.2 COMPARING VEIN SETS IN THE FRISCO MINE, CHIHUAHUA, MEXICO

In the Frisco mine, Chihuahua, Mexico, veins fall into one of four sets according to their strike and dip (Koch, 1956, p. 14-15). Part 1 of worksheet C4B.WK1 gives the percentage of the vein that is ore for the four sets. Part 2 gives the one-way analysis of variance. Because the calculated F-value of 3.2 is larger than the tabled F-value of 2.31, we conclude that evidently the percentage of ore in the vein sets differs significantly. As discussed in my original paper, this difference presumably stems from the tectonic history of the ore deposit.

4.3 RANDOMIZED-BLOCK ANALYSIS OF VARIANCE

Worksheet C4C.WK1 for the randomized-block analysis of variance follows the form of Li (1964, p. 244).

Part 1 is a table of data, consisting of readings of pH from the top, middle, and bottom of six core samples of soil. There are three groups, numbered 1 to 3 (corresponding to top, middle, and bottom), and six values in each group (corresponding to the six samples). The left-hand row contains the sample totals $T(r)$, used to calculate element II in cell F86.

The purpose of this analysis of variance is to test whether the pH in the top, middle, and bottom soil samples is the same; in statistical terms, we test the hypothesis $H(0)$ that there is no treatment effect against the alternative hypothesis $H(1)$ that there is a treatment effect.

Part 2 is the randomized-block analysis. As for the one-way analysis of variance, 1-2-3 calculates all of the entries, except for the tabled values of F in cells I44 and I45. Because the calculated F-value, 31.79, is larger than the tabled F-value, 2.92, we conclude that the pH in the top, middle, and bottom soil samples are significantly different, or, in statistical terms, that there is a treatment effect.

This analysis of variance also can be used to investigate whether the six places sampled, on the average, have significantly different soil pH values. In statistical terms, the hypothesis $H(0)$ that the soil is the same

at the six places, that is, no replication effect exists, is compared with the alternative hypothesis $H(1)$ that the soil is significantly different. Because the calculated F-value to test this hypothesis is 6.43, which is larger than the tabled F-value of 2.52, we conclude that the soil is significantly different.

The labels in Parts 1 and 2 are general ones. To present this example analysis for publication, I would change the group numbers in line 12 to indicate the soil type, top, middle, and bottom. And, in cell C44, the appropriate label would be "Among soil depths," and in cell C45, "Among core samples." This relabeling, easily done in 1-2-3, will help you and your users understand the geological significance of an analysis of variance.

Parts 3 and 4 correspond to those of worksheet C4A.WK1 for the one-way analysis of variance, with a few changes in detail.

4.4 COMPARING METHODS OF SAMPLE PREPARATION FOR GOLD ORE

Given geological samples of gold ore to be assayed, what methods of sample preparation yield reliable results at the lowest cost? That is, how best can time and money be allocated among crushing, grinding, pulverizing, splitting, and assaying? Statistical methods to answer this general question include the randomized-block analysis of variance. In this section, I use data from the Homestake Mine, Lead, South Dakota (Koch and Link, 1972, p.13).

Worksheet C4D.WK1 analyzes data from three methods of sample preparation: crushing, grinding, and pulverizing, which are listed as groups 1, 2, and 3, respectively in Part 1 of the worksheet.

Part 2 gives the analysis of variance. Because the F-value of 0.58 for replication variability is smaller than the tabled value of 2.92, we conclude that there is no significant difference among the three methods of sample preparation for these data.

The randomized-block analysis of variance is only one of several statistical methods that we used in the original paper.

Worksheet C4A One-way analysis of variance

1 25 Nov 87, file C4A.WK1 on disks GSK 059-061

2
3 Worksheet C4A.WK1 for one-way analysis of variance

4 Example problem from Li (1964, p. 194-196)

5
6 Part 1. Table of data

	D	E	F	G	H	I
Sample						
totals	1	2	3	4	5	6
	7	7	3	4	5	6
	9	5	2	3	3	

Worksheet C4A (continued)

30									
31									
32									
33									
34									
35									
36									
37	Part 2.	Analysis of variance							
38									
39			E	F	G	H	I		
40									
41			Sum of	Degrees of	Mean				
42	Source of variation	squares	freedom	square	F	F	F(10%)		
43									
44	Among-samples	36	2	18.0	15.0	3.78			
45	Within-samples	6	5	1.2					
46	Total	42	7						
47									
48									
49									
50	Part 3.	Table of labels							
51									
52	DF.AMONG	F44							
53	DF.TOTAL	F46							
54	DF.WITHIN	F45							
55	G	H35							
56	G*G	F88							
57	I	F88							
58	II	F89							

Worksheet C4A (continued)

59 III F90
 60 K D79
 61 MS.AMONG G44
 62 MS.WITHIN G45
 63 N E88
 64 SS.AMONG E44
 65 SS.TOTAL E46
 66 SS.WITHIN E45
 67
 68
 69
 70

71 Part 4. Computation

72
 73 4.A. Sums, counts, and means

74
 75 SUM(W) 16 12 12
 76 COUNT(W) 2 2 4
 77 AVG(W) 8 6 3
 78 SUM(W)^2/N 128 72 36
 79 K
 80 3

Worksheet C4A (continued)

81 4.B. Computing table for one-way analysis of variance
 82 (from Koch and Link, 1970, p. 141)

83	84	85	86	87	88	89	90	91	92
Type	Total of	Number of	Squares/	Element					
of sum	squares	observ.	observ.						
Grand	1600	8	200	(I)					
Sample			236	(II)					
Observ.	242	1	242	(III)					

Worksheet C4A (continued)

	4-C. Squares of observations								
	C	D	E	F	G	H	I		
93									
94									
95									
96									
97	Squared								
98	sample								
99	1	2	3	4	5	6			
100									
101	49	49	49	9					
102	81	25	16						
103			4						
104			9						
105									
106									
107									
108									
109									
110									
111									
112									
113									
114									
115									
116									
117									
118									
119									
120									
121									

Worksheet C4B One-way analysis of variance, Frisco mine

1 25 Nov 87, file C4B.WK1 on disks GSK 059-061
 2
 3 Worksheet C4B.WK1. One-way analysis of variance, Frisco mine
 4 Comparison of four vein sets in the Frisco mine
 5
 6 Part 1. Table of data

	D	E	F	G	H	I
10 Sample						
11 totals						
12	1	2	3	4	5	6
13						
14	83	10	53	76		
15	61	43	58	50		
16	56	26	35	60		
17	57	15	19	50		
18	25	40		64		
19	67	67		33		
20	67			67		
21	33			80		
22	20			83		
23				90		
24				40		
25						
26						
27						
28						
29						

Worksheet C4B (continued)

59	III	F90
60	K	D79
61	MS.AMONG	G44
62	MS.WITHIN	G45
63	N	E88
64	SS.AMONG	E44
65	SS.TOTAL	E46
66	SS.WITHIN	E45
67		
68		
69		
70		
71	Part 4. Computation	
72		
73	4.A. Sums, counts, and means	
74		
75	SUM(W)	469
76	COUNT(W)	201
77	AVG(W)	6
78	SUM(W)^2/N24440.111	33.5
79	K	41.25
80		6806.25
		693
		11
		63
		43659

Worksheet C4B (continued)

81 4.B. Computing table for one-way analysis of variance
 82 (from Koch and Link, 1970, p. 141)

83	84	85	86	87
Type	Total of squares	Number of observ.	Squares/observ.	Element
88 Grand	2334784	30	77826.133	(I)
89 Sample			81638.861	(II)
90 Observ.	91824	1	91824	(III)

91
 92

Worksheet C4B (continued)

	4-C. Squares of observations						
	C	D	E	F	G	H	I
93							
94							
95							
96							
97	Squared						
98	sample						
99	1	2	3	4	5	6	
100							
101	6889	100	2809	5776			
102	3721	1849	3364	2500			
103	3136	676	1225	3600			
104	3249	225	361	2500			
105	625	1600		4096			
106	4489	4489		1089			
107	4489			4489			
108	1089			6400			
109	400			6889			
110				8100			
111				1600			
112							
113							
114							
115							
116							
117							
118							
119							
120							
121							

Worksheet C4C Randomized-block analysis of variance

1 1 Dec 87, file C4C.WK1 on disks GSK 059-061
 2
 3 Worksheet C4C.WK1 for randomized-block analysis of variance
 4 Example problem from Li (1964, p. 244)
 5

6 Part 1. Table of data
 7

	C	D	E	F	G	H	I
10 Sample							
11 totals	1	2	3	4	5	6	
14	22.3	7.5	7.6	7.2	7.6	7.2	
15	21.0	7.2	7.1	6.7	7.1	6.7	
16	21.5	7.3	7.2	7.0	7.2	7.0	
17	21.9	7.5	7.4	7.0	7.4	7.0	
18	22.4	7.7	7.7	7.0	7.7	7.0	
19	22.2	7.6	7.7	6.9	7.7	6.9	

Worksheet C4C (continued)

Part 2. Analysis of variance

	E	F	G	H	I	
	Sum of Degrees of					
	squares	freedom	square	F	F (10%)	
44	Replication	0.4894	5	0.09789	6.43	2.52
45	Treatment	0.9678	2	0.48389	31.79	2.92
46	Residual	0.1522	10	0.01522		
47	Total	1.6094	17			

Part 3. Table of labels

52	DF.REPLICATION	F44
53	DF.RESIDUAL	F46
54	DF.TOTAL	F47
55	DF.TREATMENT	F45
56	G*G	D85
57	I	F85
58	II	F86

Worksheet C4C (continued)

59	III	F87
60	IV	F88
61	K	E86
62	MS.REPLICATION	G44
63	MS.RESIDUAL	G46
64	MS.TREATMENT	G45
65	N	D76
66	SS.REPLICATION	E44
67	SS.RESIDUAL	E46
68	SS.TOTAL	E47
69	SS.TREATMENT	E45
70		
71		
72	4.A.	Sums and counts
73		
74	SUM(W)	44.8 44.7 41.8
75	SUM(W)^2	2007.04 1998.09 1747.24
76	COUNT(W)	6 6 6
77		
78	4.B.	Computing table for randomized block
79		(from Koch and Link, 1970, p. 159)
80		

Worksheet C4C (continued)

81	-----						
82	Type	Total of	Number of	Squares/	Element		
83	of sum	squares	observ.	observ.			
84	-----						
85	Grand	17239.69	18	957.7606	(I)		
86	Replicat.	2874.75	3	958.2500	(II)		
87	Treat.	5752.37	6	958.7283	(III)		
88	Observ.	959.37	1	959.3700	(IV)		
89	-----						
90							
91							
92							

Worksheet C4C (continued)

	4-C. Squares of observations					
	C	D	E	F	G	I
93						
94						
95						
96						
97	Squared					
98	sample					
99	1	2	3	4	5	6
100						
101	497.29	56.25	57.76	51.84		
102	441.00	51.84	50.41	44.89		
103	462.25	53.29	51.84	49.00		
104	479.61	56.25	54.76	49.00		
105	501.76	59.29	59.29	49.00		
106	492.84	57.76	59.29	47.61		
107						
108						
109						
110						
111						
112						
113						
114						
115						
116						
117						
118						
119						
120						
121						

Worksheet C4D Randomized-block analysis of variance, Homestake mine

1 1 Dec 87, file C4D.WK1 on disks GSK 059-061

2

3 Worksheet C4D.WK1. Randomized-block analysis of variance for

4 Homestake mine data. Data from Koch and Link (1972, p. 13).

5

6 Part 1. Table of data

7

8	C	D	E	F	G	H	I
9	-----						
10	Sample	Group Number					
11	totals	-----					
12		1	2	3	4	5	6
13		-----					
14	23.797	5.198	6.254	12.3445			
15	5.405	0.3255	2.1327	2.9472			
16	3.965	0.0872	0.001	3.8765			
17	0.367	0.2345	0.001	0.1315			
18	3.698	0.001	0.001	3.6955			
19	39.803	23.956	8.8585	6.9887			
20	3.549	2.4835	0.3767	0.689			
21	0.969	0.429	0.334	0.2057			
22	0.879	0.001	0.2147	0.6633			
23	11.449	8.3112	0.894	2.244			
24	2.236	0.1947	0.385	1.6565			
25	18.930	4.299	6.4242	8.207			
26	4.751	0.645	1.3372	2.769			
27	45.712	1.3077	17.1602	27.244			
28	37.878	22.9445	7.779	7.1545			

Worksheet C4D (continued)

29	25.885	18.9772	3.184	3.724					
30	58.338	11.3167	25.5457	21.4755					
31	2.447	0.832	0.734	0.881					
32	0.003	0.001	0.001	0.001					
33	39.373	22.4892	7.0732	9.8102					
34									
35									
36									
37	Part 2. Analysis of variance								
38									
39									
40									
41									
42	Source of variation	Sum of squares	Degrees of freedom	Mean square	F	F (10%)			
43									
44	Replication	2156.0567	19	113.47667	3.78	2.52			
45	Treatment	34.7959	2	17.39795	0.58	2.92			
46	Residual	1141.2244	38	30.03222					
47	Total	3332.0770	59						
48									
49									
50	Part 3. Table of labels								
51									
52	DF.REPLICATION			F44					
53	DF.RESIDUAL			F46					
54	DF.TOTAL			F47					
55	DF.TREATMENT			F45					
56	G*G			D85					
57	I			F85					

Worksheet C4D (continued)

58	II	F86
59	III	F87
60	IV	F88
61	K	E86
62	MS.REPLICATION	G44
63	MS.RESIDUAL	G46
64	MS.TREATMENT	G45
65	N	D76
66	SS.REPLICATION	E44
67	SS.RESIDUAL	E46
68	SS.TOTAL	E47
69	SS.TREATMENT	E45
70		
71		
72	4.A.	Sums and counts
73		
74	SUM(W)	124.0339 88.6911 116.7086
75	SUM(W)^2	15384.408 7866.1112 13620.897
76	COUNT(W)	20 20 20
77		
78	4.B.	Computing table for randomized block
79		(from Koch and Link, 1970, p. 159)
80		

Worksheet C4D (continued)

	Type of sum squares	Total of squares	Number of observ.	Squares/ observ.	Element
81					
82					
83					
84					
85	Grand	108526.49	60	1808.7749	(I)
86	Replicat.	11894.494	3	3964.8316	(II)
87	Treat.	36871.416	20	1843.5708	(III)
88	Observ.	5140.8519	1	5140.8519	(IV)
89					
90					
91					
92					

Worksheet C4D (continued)

	4-C. Squares of observations						
	C	D	E	F	G	H	I
93							
94							
95							
96							
97							
98	Squared						
99	sample						
100	totals	1	2	3	4	5	6
101		566.273	27.019	39.113	152.387		
102		29.218	0.106	4.548	8.686		
103		15.719	0.008	0.000	15.027		
104		0.135	0.055	0.000	0.017		
105		13.672	0.000	0.000	13.657		
106		1584.295	573.890	78.473	48.842		
107		12.597	6.168	0.142	0.475		
108		0.938	0.184	0.112	0.042		
109		0.773	0.000	0.046	0.440		
110		131.084	69.076	0.799	5.036		
111		5.001	0.038	0.148	2.744		
112		358.352	18.481	41.270	67.355		
113		22.574	0.416	1.788	7.667		
114		2089.578	1.710	294.472	742.236		
115		1434.743	526.450	60.513	51.187		
116		670.044	360.134	10.138	13.868		
117		3403.311	128.068	652.583	461.197		
118		5.988	0.692	0.539	0.776		
119		0.000	0.000	0.000	0.000		
120		1550.202	505.764	50.030	96.240		
121							

CHAPTER 5

Exploration Geochemical Data

1-2-3 provides useful procedures for analyzing sets of exploration geochemical data that are not too large. (The size will depend on your computer memory; a maximum typical size may be a few hundred sites with ten to twenty variables per site.) This chapter applies some of these procedures to a small set of geochemical data.

For larger data sets, it is more effective to use a system of BASIC programs (Koch, 1987), or one of the general purpose programs for scientific data analysis, for example, MINITAB (Ryan, Joiner, and Ryan, 1985). Fortunately, it is not difficult to transfer files from 1-2-3 to other programs (Koch, 1987, p. 16) or from other programs to 1-2-3.

5.1 SUMMARY STATISTICS

Worksheet C5A.WK1 calculates summary statistics for a Norwegian data set, which Howarth and Sinding-Larsen (1983) used to demonstrate multivariate statistical analysis. Part 1 is their data set; Part 2 gives basic statistics, calculated using 1-2-3 functions (except for CV, the coefficient of variation, which is the ratio of the standard deviation, STD,

to the arithmetic average or sample mean, AVG). As with worksheet C4A.WK1, in order to avoid resetting the formulae in Part 2, you can increase the number of rows using the 1-2-3 command /Worksheet Insert Row, and the number of columns using the command /Copy.

5.2 FREQUENCY DISTRIBUTIONS AND HISTOGRAMS

Worksheet C5B.WK1 calculates frequency distributions for the values of Zn and Fe in worksheet C5A.WK1. Part 1 is the same as Part 1 of the previous worksheet. Part 2 provides the frequency distributions, obtained using 1-2-3's /Data Distribution command. Here are the step-by-step directions for calculating the frequency distribution for zinc.

- (1) In range C54..C78, establish the class intervals (named "bins" by 1-2-3) using the /Data Fill command. I used a starting value of 0 and a step value of 50.
- (2) Using the /Data Distribution command, calculate the frequencies. 1-2-3 first asks for the range of values, which is D15..D39, and then for the range of bins, established in the previous step as C54..C78.
- (3) Make a histogram for zinc (Figure 5.1) by using the graphics commands, as follows:
 - (a) Select /Graph Type and enter B for a bar graph.
 - (b) Select X and specify the range C54..C75 for the range of values on the x-axis.
 - (c) Select A and specify the range D54..D75 for the range of values on the y-axis.
 - (d) Select Options Titles and specify titles for the two title lines and the x- and y-axes.

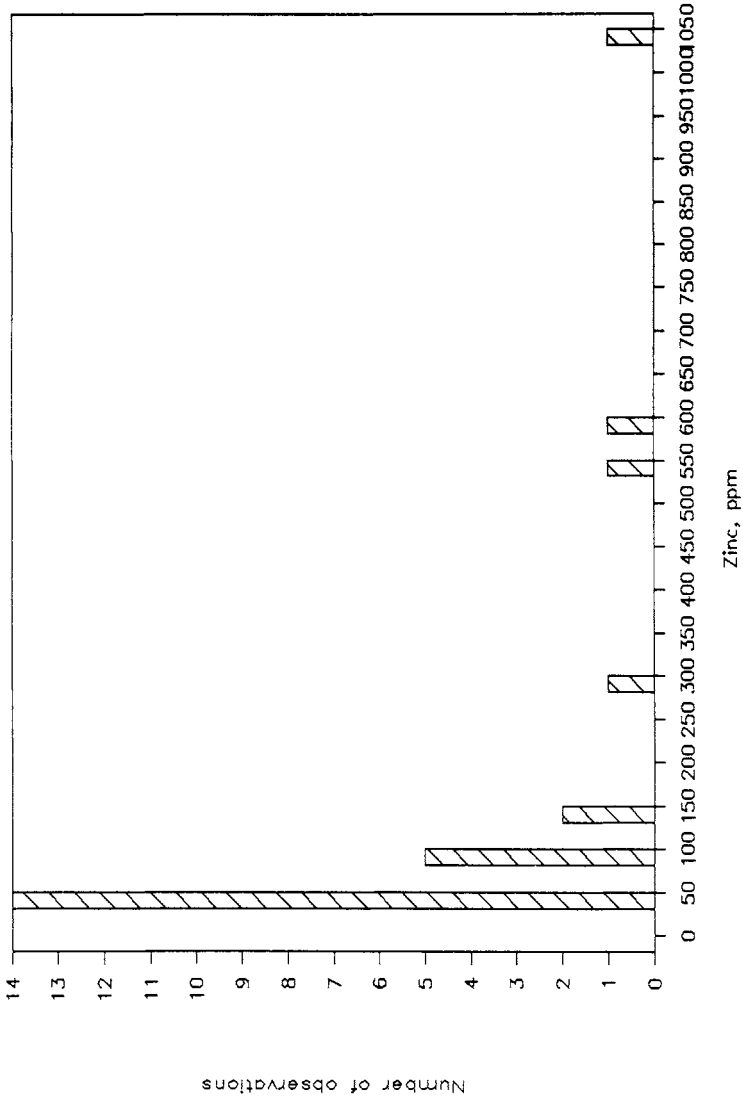


Figure 5.1 Histogram for Zn, Norwegian data set

(e) Select View to display the graph on the screen.

Repeating the analysis for Fe gives the results in the worksheet and in Figure 5.2.

1-2-3's /Data Distribution command does not provide some information that is conventional for frequency distributions, and the graphics commands display bar graphs rather than usual histograms with the bars adjacent to one another. In Part 3 of the worksheet, I have recast the frequency distribution for Zn in a conventional form, through a few additional steps. You can make a conventional histogram with 1-2-3's / Graph Type XY command by specifying the coordinates of the corners of the bars for the X and A ranges. Doing this is simple but tedious, and seems to me to defeat 1-2-3's purpose of providing graphs that are easy to both make and read. Although 1-2-3 macros could be written to handle the tasks explained in this paragraph, I would prefer to use another computing method altogether.

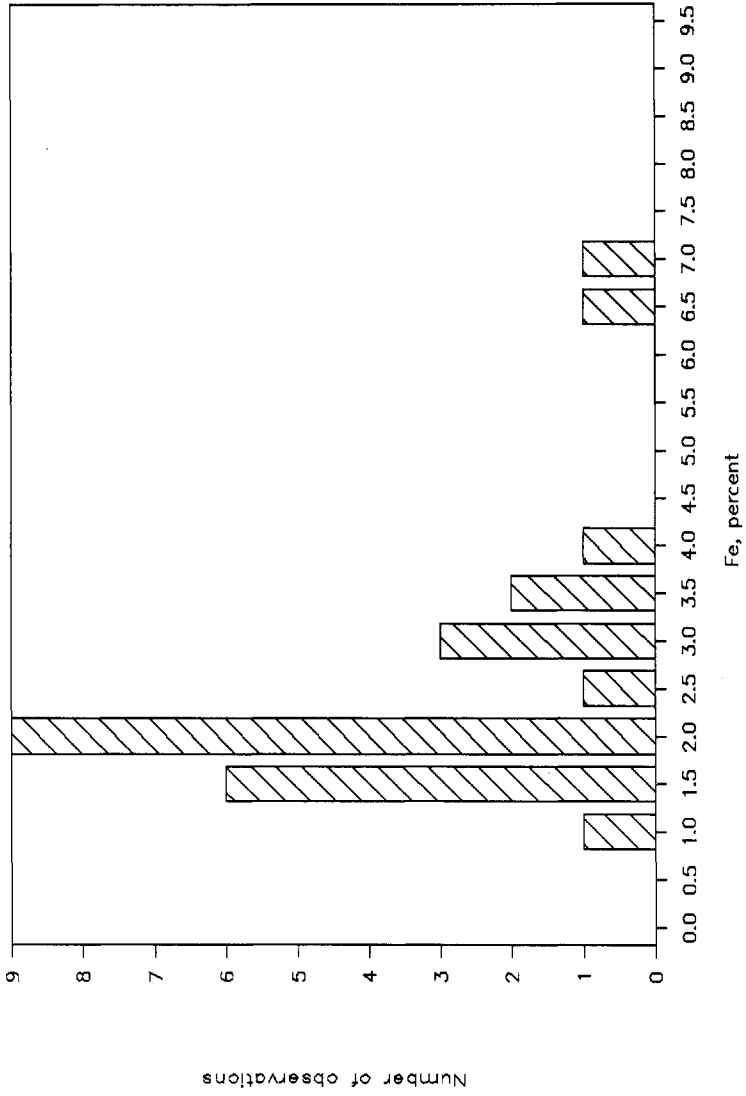


Figure 5.2 Histogram for Fe, Norwegian data

Worksheet C5A Basic statistics for exploration geochemical data

1 28 Aug 87, file C5A.WK1 on disks GSK 059-061

2
3 Worksheet C5A.WK1. Basic statistics for
4 exploration geochemical data
5
6

7 Part 1. Table of stream-sediment data from Norway
8 (from Howarth and Sinding-Larsen, 1983, p. 210) 9

	C	D	E	F	G	H	I
Sample No.	Zn ppm	Fe pct	Mn ppm	Cd ppm	Cu ppm	Pb ppm	
1	24	1.08	330	0.4	7	5	
2	25	1.18	420	0.3	9	7	
3	42	2.06	910	0.6	12	6	
4	50	1.73	700	0.5	15	9	
5	52	1.74	690	0.5	18	10	
6	29	1.06	510	0.6	10	8	
7	26	1.08	530	0.5	10	7	
8	23	0.93	260	0.4	7	7	
9	89	1.84	670	0.9	32	6	
10	72	3.35	530	0.5	11	8	
11	31	1.51	350	0.4	11	4	
12	115	1.59	650	1.2	37	6	
13	535	2.59	960	3.5	350	9	
14	48	1.71	570	0.3	17	9	
15	1010	2.71	1070	5.6	590	9	

Worksheet C5A (continued)

30	16	560	3.05	450	3.1	490	8
31	17	48	1.43	590	0.4	8	8
32	18	44	1.70	710	0.4	11	16
33	19	36	1.28	410	0.2	7	10
34	20	33	1.94	490	1.0	20	12
35	21	45	1.79	260	0.8	27	12
36	22	118	6.70	1930	1.2	33	17
37	23	274	6.10	5920	3.1	63	27
38	24	81	2.62	970	0.9	22	13
39	25	80	3.61	900	1.0	23	10
40							
41							
42							

Worksheet C5A (continued)

43 Part 2. Basic statistics for all samples

44	C	D	E	F	G	H	I
45	Statistic	Zn	Fe	Mn	Cd	Cu	Pb
46		ppm	pct	ppm	ppm	ppm	ppm
47	Statistic	Zn	Fe	Mn	Cd	Cu	Pb
48		ppm	pct	ppm	ppm	ppm	ppm
49							
50	N	25	25	25	25	25	25
51	AVG	140	2.26	871	1.1	74	10
52	VAR	53464	2.09	1226344	1.7	24446	23
53	STD	231	1.44	1107	1.3	156	5
54	CV	1.66	0.64	1.27	1.15	2.12	0.49
55	MAX	1010	6.70	5920	5.6	590	27
56	MIN	23	0.93	260	0.2	7	4
57							

58 N, number of observations; AVG, average (mean); VAR, variance;

59 STD, standard deviation; CV, coefficient of variation;

60 MAX, maximum value; MIN, minimum value.

61

62

63 Part 3. Table of labels

64

65 AVG D51

66 CV D54

67 MAX D55

68 MIN D56

69 N D50

70 STD D53

71 VAR D52

Worksheet C5B Frequency distributions and histograms for exploration geochemical data

1 10 Sep 87, file C5B.WK1 on disks GSK 059-061

2
3 Worksheet C5B.WK1. Frequency distributions and histograms
4 for exploration geochemical data
5
6

7 Part 1. Table of stream-sediment data from Norway
8 (from Howarth and Sinding-Larsen, 1983, p. 210)
9

C	D	E	F	G	H	I
Sample No.	Zn ppm	Fe pct	Mn ppm	Cd ppm	Cu ppm	Pb ppm
1	24	1.08	330	0.4	7	5
2	25	1.18	420	0.3	9	7
3	42	2.06	910	0.6	12	6
4	50	1.73	700	0.5	15	9
5	52	1.74	690	0.5	18	10
6	29	1.06	510	0.6	10	8
7	26	1.08	530	0.5	10	7
8	23	0.93	260	0.4	7	7
9	89	1.84	670	0.9	32	6
10	72	3.35	530	0.5	11	8
11	31	1.51	350	0.4	11	4
12	115	1.59	650	1.2	37	6
13	535	2.59	960	3.5	350	9
14	48	1.71	570	0.3	17	9
15	1010	2.71	1070	5.6	590	9

Worksheet C5B (continued)

30	16	560	3.05	450	3.1	490	8
31	17	48	1.43	590	0.4	8	8
32	18	44	1.70	710	0.4	11	16
33	19	36	1.28	410	0.2	7	10
34	20	33	1.94	490	1.0	20	12
35	21	45	1.79	260	0.8	27	12
36	22	118	6.70	1930	1.2	33	17
37	23	274	6.10	5920	3.1	63	27
38	24	81	2.62	970	0.9	22	13
39	25	80	3.61	900	1.0	23	10
40							
41							
42							

Part 2. Frequency distributions for Zn and Fe

	C	D	E	F	G	H	I
43							
44							
45							
46							
47							
48							
49							
50							
51							
52							
53							
54	0	0	0	0.0	0		
55	50	14		0.5	0		
56	100	5		1.0	1		
57	150	2		1.5	6		
58	200	0		2.0	9		

Worksheet C5B (continued)

59	250	0	2.5	1
60	300	1	3.0	3
61	350	0	3.5	2
62	400	0	4.0	1
63	450	0	4.5	0
64	500	0	5.0	0
65	550	1	5.5	0
66	600	1	6.0	0
67	650	0	6.5	1
68	700	0	7.0	1
69	750	0	7.5	0
70	800	0	8.0	0
71	850	0	8.5	0
72	900	0	9.0	0
73	950	0	9.5	0
74	1000	0		0
75	1050	1		
76	1100	0		
77	1150	0		
78	1200	0		
79		0		
80				
81				
82				

Worksheet C5B (continued)

83 Part 3. Frequency distribution for Zn

84	C	D	E	F	G	H	I
85	Class Interval	Frequency	Relative Freq. (%)	Cumulative Freq.	Relative Cumulative Freq. (%)		
86	0	50	0	0	0	0	0
87	50	100	14	56	14	56	56
88	100	150	5	20	19	76	76
89	150	200	2	8	21	84	84
90	200	250	0	0	21	84	84
91	250	300	0	0	21	84	84
92	300	350	1	4	22	88	88
93	350	400	0	0	22	88	88
94	400	450	0	0	22	88	88
95	450	500	0	0	22	88	88
96	500	550	0	0	22	88	88
97	550	600	1	4	23	92	92
98	600	650	1	4	24	96	96
99	650	700	0	0	24	96	96
100	700	750	0	0	24	96	96
101	750	800	0	0	24	96	96
102	800	850	0	0	24	96	96
103	850	900	0	0	24	96	96
104	900	950	0	0	24	96	96
105	950	1000	0	0	24	96	96

Worksheet C5B (continued)

112	1000	1050	0	0	24	96
113	1050	1100	1	4	25	100
114	-----	-----	-----	-----	-----	-----

CHAPTER 6

Exploration Models

This chapter will discuss two simulation models for drilling on grids; for simplicity we will consider only square grids. After practicing with the worksheets, you will be able to modify them for other situations or write additional ones. The worksheets in this chapter and the next contain “macros”, which are short programs written in 1-2-3’s command language. Macros make it easy to perform certain operations.

6.1 A SIMULATION OF GRID DRILLING

In mineral exploration, grid drilling is used widely, particularly for the discovery and evaluation of bedded deposits including uranium, coal, kaolin, and many others. In an area to be explored, if we can estimate the percentage of total area that is mineralized and assume that the mineralized areas are “randomly distributed,” we can then simulate the effect of drilling holes to discover ore deposits. “Randomly distributed” indicates that any part of the area has an equal chance of containing an orebody as any other. Simulation is “a class of techniques that involve setting up a model of a real situation and then performing experiments on the model,” to quote from the introduction to a comprehensive and classic book on simulation in geology (Harbaugh and Bonham-Carter,

1970, p. 1-2). The ideas in this chapter are drawn from a paper by Koch and Schuenemeyer (1982). First, I will explain the worksheet and then suggest some modifications that you may want to make.

Worksheet C6A.WK1 simulates the drilling of vertical holes on a square grid of 100 potential locations in a 10 by 10 array. After you specify drillhole locations in these 100 potential locations, the worksheet indicates which of the “drillholes” encountered ore. To run the model, execute the four steps summarized at the beginning of the worksheet in any order that you select.

Part 1 of the worksheet is an abbreviated set of instructions for the worksheet’s four steps. Step 1 allows you to specify the percent area that is mineralized. To do this, press ALT-a (this notation indicates to press the ALT key and then the a key, while holding down the ALT key); this sequence of keys calls a 1-2-3 command named a “Macro,” used in 1-2-3 for many purposes). On the second line of the screen, the following request appears:

Enter percent area mineralized:

In response, enter any appropriate percentage. For the worksheet on the diskette, I entered 10. (The worksheet stores this percentage in cell C105 and the corresponding decimal fraction in cell C106.

In Step 2, first place cell A21 at the upper-left-hand corner of your screen, and then enter drillhole locations in the array MATRIX.HOLES (Part 2) in cells C21..L30. These locations can be identified by any numerical values; for illustration, I entered the numbers 1 to 10. The hits then appear as asterisks in the array MATRIX.HITS in cells C31..L40. Matrices ARRAY.HOLES and ARRAY.HITS are formatted so that they appear on one screen through your placement of cell A21.

How does the program determine whether a drillhole hits ore? This is done by comparing the matrix of hole locations, MATRIX.HOLES, with MATRIX.DATA (Part 3), which is a matrix (in cells C51..L60), of random numbers between 0 and 1. If the random number is low enough,

relative to the percent area selected as mineralized in Step 1, a hit is recorded. For the example, only hole 5 in cell F25 is a hit, because only the random number 0.010 in cell F55 is less than 0.1.

Step 3 allows you to erase one set of hole locations and start over by using the macro ALT-b.

Step 4 allows you to select a different set of random numbers for array MATRIX.DATA by using the macro ALT-c, which copies the random numbers in array MATRIX.RAND (Part 4) into array MATRIX.DATA. The array MATRIX.RAND is reset each time you press function key 9.

Here are some notes about this model. In order to save room on the diskette, I used a 10 by 10 grid. 1-2-3 will allow you to make the grid larger if you first copy the worksheet to a diskette used for only this model. Although the model is for holes on a square grid, the computer monitor display is not square in order to simplify programming. Using the command /Worksheet Global Column-width, you can reformat the cell size if you wish. Also, the model can readily be modified for non-square grids. And it would not be difficult to introduce a gradient into the random numbers, corresponding to a change in mineralization favorability in some direction. For example, this change would be appropriate for a mine in the Coeur d'Alene district (Koch, Schuene-meyer, and Link, 1974). Taking account of the probabilities of recognition given a hit also would be interesting (Koch and Schuenemeyer, 1982, p. 654).

6.2 THE ELMWOOD, TENNESSEE ZINC DEPOSIT

Between 1964 and about 1980, more than 20 mining companies explored the Middle Tennessee district near Nashville for zinc deposits. These deposits are located at depths of between 300 and 600 meters in gently dipping rocks of Ordovician age. Because the ore deposits occur below a major unconformity which is not exposed in the district, surface geology provided little or no guide to ore; nor were geophysics or geochemistry of much help. Therefore, nearly all exploration was by diamond drilling.

Although active exploration was essentially stopped in the 1980's for economic reasons, the district contains a major zinc resource for the future.

In 1967, The New Jersey Zinc Company discovered the Elmwood mine near Carthage, Tennessee after drilling 79 holes; Callahan (1977) details the exploration campaign. After the discovery, the deposit was evaluated through drilling another 89 holes on a square grid with a hole spacing of 1000 feet.

Worksheet C6B.WK1 maps the zinc grades located by drilling at Elmwood. We can use these data for two simulation exercises: (1) to simulate the discovery process, and (2) to define a mining area, based on the entire data set. These two exercises are taken up in turn.

Part 1 is instructions for the four macros that run the model. Part 2 is array MATRIX.DATA, which is an 11 by 11 array, consisting of Callahan's (1977, p. 1390) 89 holes, plus another 17 holes drilled by the New Jersey Zinc Company, plus 15 holes simulated by me to complete the grid. (Callahan's original data are given in array ORIGINAL.DATA, displayed in Part 7.) When you retrieve the worksheet, the values in array MATRIX.DATA are not displayed (because macro \0 runs automatically), so that you can "discover" them by "drilling" holes. To see the values, press Alt-c; to hide them once more, press Alt-d.

To "drillholes," press Alt-a, and follow the instructions for entering holes according to the rows and columns of array MATRIX.HOLES, in Part 3. The rows and columns are numbered starting in the upper-left-hand corner according to 1-2-3's system. For the worksheet provided on the diskette, two holes are drilled, in row 2, column 3, and in row 4, column 5. The results of the drilling simulation appear in Part 4 in array MATRIX.HITS. For the example, the values are 2 and 5 percent zinc.

You then can continue the simulation to get an average grade for all the holes drilled or for selected holes by pressing Alt-b. This macro selects holes by cell addresses. Selecting addresses F70 and H72 gives the average grade of 3.5 percent zinc.

For the second simulation, you can display the entire array `MATRIX.DATA` by pressing `Alt-c`, and then proceed as noted. The assumption is that the entire grid has been drilled.

Part 5 lists the macros, and Part 6 is the table of labels. Part 7 is Callahan's original data matrix. To use this original matrix in your simulations, press `Alt-e`, which copies it to array `MATRIX.DATA`. Then, if you drill a hole where there is an empty cell, the worksheet records an o (lower-case "oh").

Here are some notes on this model. To scale the arrays for printing, I reduced the column-widths to 2, with the result that their labels are abbreviated. In the array `MATRIX.DATA`, row and column labels 0 indicates 10, and in the second set of labels, 1 indicates 11. In this matrix, the rows and columns start in the upper-left-hand corner, following 1-2-3's convention; in the convention usual in geology and engineering, the array is in the southeast or fourth quadrant. The IF statement in cell D69 is:

```
@IF(@ISSTRING(D49=0,"",@IF(D31)>0,D31,"o")).
```

The `@ISSTRING` function tests for a blank cell in array `MATRIX.HOLES`; if blank, a blank is placed in the corresponding cell in `MATRIX.HITS`. Otherwise, the `@IF` function tests for a value greater than 0 in array `MATRIX.DATA`; if greater, the value is recorded in the corresponding cell in `MATRIX.HITS`; if smaller, which happens if you use array `ORIGINAL.MATRIX`, an o is recorded to indicate no zinc. The greater-than-0 value can be readily changed for other simulations.

Worksheet C6A Simulation of drilling exploration holes

- 1 02 Feb 89, file C6A.WK1 on disks GSK 059-061 2
- 3 Worksheet C6A.WK1. Simulation of drilling exploration holes
- 4
- 5
- 6 Part 1. Abbreviated running instructions (Details in text.)
- 7
- 8 (1) To specify the percent area mineralized, press Alt-a.
- 9
- 10 (2) Specify drill-hole locations by entering numbers in array
- 11 MATRIX.HOLES, C21..L30. The hit locations will appear in array
- 12 MATRIX.HITS, C31..L40.
- 13
- 14 (3) To erase a set of drill-hole locations, press Alt-b.
- 15
- 16 (4) To reset the mineralized cells, press Alt-c.
- 17
- 18

Worksheet C6A (continued)

```
19 Part 2. Array MATRIX.HOLES, C21..L30 and MATRIX.HITS, C31..L40.
20 C D E F G H I J K L
21 1
22
23 2
24
25 3 4 5 6 7 8 9 10
26
27
28
29
30
31
32
33
34
35
36
37
38
39
40
41
42
43
44
45
46
47
```

Worksheet C6A (continued)

48
 49 Part 3. Array MATRIX.DATA, C51..L60, of values between 0 and 1.
 50
 51 0.572 0.603 0.051 0.814 0.835 0.943 0.709 0.486 0.209 0.327
 52 0.403 0.532 0.019 0.443 0.849 0.576 0.621 0.575 0.676 0.005
 53 0.909 0.287 0.084 0.558 0.855 0.778 0.181 0.080 0.459 0.197
 54 0.996 0.858 0.699 0.483 0.403 0.209 0.357 0.499 0.432 0.249
 55 0.811 0.536 0.704 0.010 0.179 0.459 0.987 0.324 0.396 0.065
 56 0.082 0.155 0.106 0.427 0.552 0.287 0.462 0.289 0.002 0.559
 57 0.204 0.800 0.918 0.048 0.624 0.132 0.281 0.889 0.265 0.790
 58 0.448 0.639 0.482 0.647 0.041 0.378 0.209 0.010 0.810 0.937
 59 0.552 0.261 0.725 0.961 0.057 0.868 0.768 0.700 0.771 0.823
 60 0.166 0.716 0.454 0.781 0.044 0.919 0.549 0.475 0.735 0.616
 61
 62
 63 Part 4. Array MATRIX.RAND, C66..L75, of random numbers
 64 between 0 and 1.
 65
 66 0.765 0.188 0.235 0.477 0.397 0.325 0.533 0.606 0.725 0.395
 67 0.798 0.129 0.113 0.939 0.663 0.976 0.661 0.211 0.091 0.781
 68 0.528 0.644 0.550 0.058 0.269 0.414 0.486 0.098 0.817 0.641
 69 0.332 0.232 0.952 0.020 0.532 0.410 0.373 0.937 0.313 0.241
 70 0.732 0.365 0.195 0.940 0.241 0.081 0.472 0.828 0.710 0.007
 71 0.146 0.340 0.119 0.720 0.045 0.673 0.099 0.428 0.122 0.540
 72 0.913 0.961 0.157 0.653 0.813 0.563 0.347 0.911 0.867 0.551
 73 0.530 0.409 0.626 0.125 0.012 0.740 0.243 0.869 0.243 0.225
 74 0.884 0.761 0.821 0.015 0.571 0.722 0.557 0.225 0.786 0.004
 75 0.596 0.528 0.462 0.969 0.807 0.878 0.225 0.049 0.952 0.284
 76

Worksheet C6A (continued)

```

77
78 Part 5. Table of labels
79
80 % .AREA.MINERAL C102
81 FRACT.AREA.MIN. D102
82 MATRIX.DATA C51..L60
83 MATRIX.HITS C31..L40
84 MATRIX.HOLES C21..L30
85 MATRIX.RAND C66..L75
86 \A D93..D94
87 \B D96..D97
88 \C D99
89
90
91 Part 6. Macros
92
93 \a {GETNUMBER "Enter percent area mineralized:",C102}
94 {CALC}
95
96 \b /reMATRIX.HOLES~
97 {CALC}
98
99 \c /rvMATRIX.RAND~MATRIX.DATA~
100
101
102 10 0.1 C105 = % area mineralized; D105 = decimal fraction

```

Worksheet C6B Simulation of drilling holes at Elmwood, Tennessee

- 1 02 Feb 89, file C6B.WK1 on disks GSK 059-061 2
- 3 Worksheet C6B.WK1. Simulation of drilling holes at
- 4 Elmwood, Tennessee
- 5
- 6 Part 1. Abbreviated running instructions (Details in text.)
- 7
- 8 (1) To "drill" holes, press Alt-a.
- 9
- 10 (2) To calculate the table of summary statistics in Part 4,
- 11 press Alt-b.
- 12
- 13 (3) To hide the array MATRIX.DATA, press Alt-d.
- 14
- 15 (4) To display the array MATRIX.DATA, press Alt-c.
- 16
- 17
- 18
- 19
- 20
- 21
- 22
- 23
- 24

Worksheet C6B (continued)

```
25 Part 2. Array MATRIX.DATA (To display, press Alt-c; to hide,  
26 press Alt-d.  
27  
28  
29 C D E F G H I J K L M N O P Q R S  
30 . . . . .  
31 . . . . .  
32 . . . . .  
33 . . . . .  
34 . . . . .  
35 . . . . .  
36 . . . . .  
37 . . . . .  
38 . . . . .  
39 . . . . .  
40 . . . . .  
41 . . . . .  
42 . . . . .  
43  
44
```

Worksheet C6B (continued)

```

45 Part 3. Array MATRIX.HOLES. Press Alt-a to "drill" holes.
46
47
48 . 1 2 3 4 5 6 7 8 9 0 1 .
49 1
50 2 *
51 3
52 4 *
53 5
54 6
55 7
56 8
57 9
58 0
59 1
60 . . . . .
61
62

```

Worksheet C6B (continued)

63 Part 4. Array MATRIX.HITS and Table of Summary Statistics
 64 Press Alt-b to calculate the Table.
 65

	D	E	F	G	H	I	J	K	L	M	N	
66	
67	
68	
69	
70	.	2	
71	
72	.	.	.	5	
73	
74	
75	
76	
77	
78	
79	
80	
81	
82	
83	
84	
85	
86	
87	
88	

	Hole No.	Zn, %
	1	2
	2	5
	3	
	4	
	5	
	6	
	7	
	8	
	9	
	10	

No. of holes	2
Average grade, %	3.5

Worksheet C6B (continued)

```

89 Part 5. Macros
90
91      /RED49..N59~
92      {GOTO}B45~
93      {GETNUMBER "Enter the numbe
94      {FOR COUNT1,1,NUM1,1,SUB1}
95
96      SUB1
97      {GETNUMBER "Enter the hole
98      {GETNUMBER "Enter the hole
99      {PUT MATRIX.HOLES, COL, ROW, "
100      {CALC}
101
102      COL          5
103      COUNT1      3
104      NUM1        2
105      ROW         4
106
107      \b
108
109      /RES72..S81~
110      {GOTO}B63~
111      {GETNUMBER "Enter the numbe
112      {DOWN 9}
113      {RIGHT 17}
114      {FOR COUNT,1,NUMBER,1,ROUTI
115      {CALC}
116      {GOTO}S81~
117      {DOWN 5}
118
119      ROUTINE /XNEnter a drill-hole locat

```

Worksheet C6B (continued)

```

118                                     {DOWN}
119
120
121                                     NUMBER      2
122                                     COUNT        3
123
124
125                                     \c          /RFGD31..N41~
126                                                 {GOTO}B25~
127
128                                     \d          /RFHD31..N41~
129                                     \o
130
131                                     \e          /CD156..N166~D31..N41~
132                                                 {GOTO}B25~
133
134 Part 6. Table of labels
135
136 COL                                Q101
137 COUNT                              Q122
138 COUNT1                             Q102
139 MATRIX.HOLES C48..N59
140 NUM1                               Q103
141 NUMBER                             Q121
142 ROUTINE                            Q116
143 ROW                                Q104
144 SUB1                               Q95

```


CHAPTER 7

Probability and Related Calculations

This chapter explains some probability calculations used in exploration and also ways to plot tonnage-grade curves. The final section tells how to plot logarithmic graphs using 1-2-3.

7.1 TONNAGE-GRADE RELATIONS

Lasky (1950) proposed that the tonnage and grade of ore deposits follow an exponential relationship that plots as a straightline on semi-logarithmic graph paper. Koch and Link (1971, p. 255-262) summarize Lasky's work, and Harris (1984, p. 59-92) gives a comprehensive account of Lasky's and later investigations of this complex and controversial subject.

Worksheet C7A.WK1 provides the data to graph Lasky's law (Figure 7.1) with his constants for a hypothetical porphyry copper deposit. Lasky's formula appears in cells G32..G34, and the scaled logarithmic values in cells H32..H34. The previously cited references explain the mathematics, and the plotting procedure is explained in Section 7.3. You can readily modify this worksheet to plot graphs for other deposits using data from Lasky, Harris, or other workers.

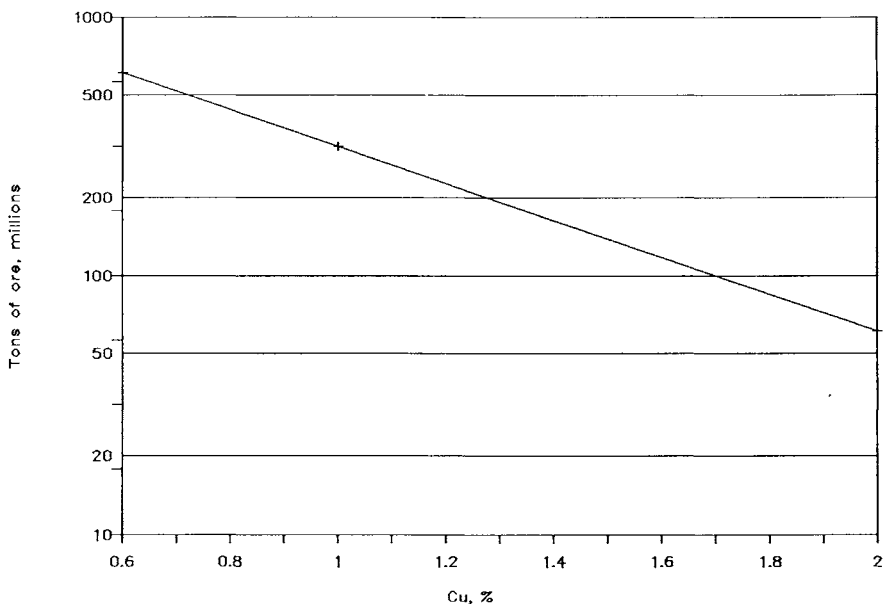


Figure 7.1 Lasky's law, hypothetical porphyry copper deposit

Worksheet C7B.WK1 provides the data to graph later work on copper deposits by Cox, Wright, and Coakley, (1981), who tabulate (Table 1, p. F3-F6) published reserves and resources of copper in the United States. In the worksheet, the tonnage data are listed in range C21..C94, and the grade data in range E21..E94.

Figure 7.2 plots these data for the entire United States, and Figure 7.3 plots the subset for operating mines and announced developments in Arizona. Using 1-2-3, it is easy to edit the data list to obtain a graph for any interesting subset, for instance, for all deposits in other states, or all deposits of a certain geologic type.

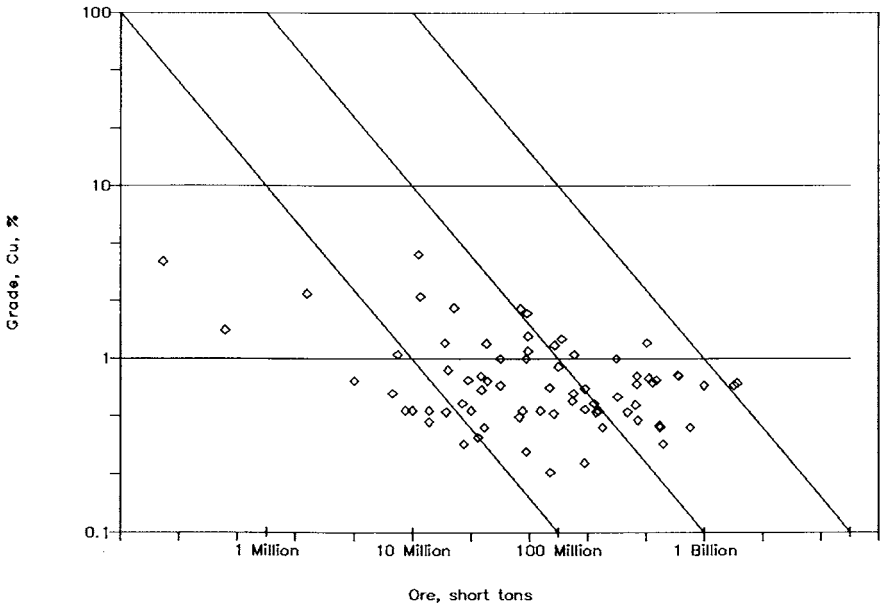


Figure 7.2 U.S. copper reserves, tonnages, and grades
(from Cox, 1981)

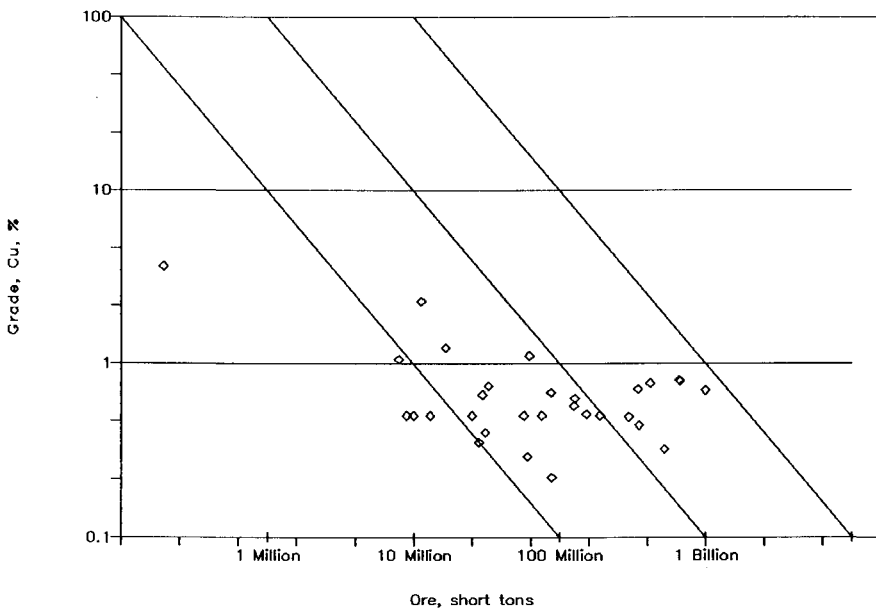


Figure 7.3 Arizona copper reserves , tonnages, and grades (from Cox, 1981)

7.2 PROBABILITY CALCULATIONS

Worksheet C7C.WK1 provides 1-2-3 macros to facilitate some of the probability calculations used in exploration. These calculations are explained for mineral deposits by Koch and Link (1971, p. 187-228) and Peters (1987, p. 444-448) and for petroleum exploration by Harbaugh, Doveton, and Davis (1977). There are too many approaches and formulae to include in this book. The worksheet provides macros for two simple situations, Gambler's ruin, and calculation of a factorial, needed for many probability formulae, but not furnished as a function by 1-2-3.

7.3 PLOTTING LOGARITHMIC GRAPHS

These directions for plotting logarithmic graphs are simplified from those of Fine (1987); they are given here because of the importance of logarithmic graphs in geology, and because Fine's article may not be readily available to you.

Because 1-2-3 does not provide for logarithmic plots, it is necessary to select the range of values to plot, calculate the logarithms of this range, plot the range, and relabel the range with the original values rather than the logarithms. None of these steps is complicated in itself, but combining them together in 1-2-3 takes some planning and patience. The principal point to remember is that 1-2-3's basic concept is that of the spreadsheet, or array, so that the data to be plotted need to be expressed as an array.

To explain logarithmic plotting, I will demonstrate how to plot Figure 7.1. In worksheet C7A, the X values are listed in range E11..E34, the A values in range F11..F30, and the B values in range H11..H34. 1-2-3 plots points only where X values are paired with A or B values. The A values define the boundaries of the graph and the horizontal lines; the B values define the data points.

This is the sequence of commands with their purposes:

- (1) /Graph Type XY, to select the appropriate display.
- (2) /Graph Options Scale Y Scale Format Hidden Quit Quit, to suppress automatic scaling on the Y axis.
- (3) /Graph, select X, and specify range E11..E34.
- (4) /Graph, select A, and specify range F11..F30.
- (5) /Graph Options Format A Lines Quit Quit, to plot lines only.

- (6) /Graph Options Data-Labels A (Specify range I11..I29) Left Quit Quit, to plot logarithmic values for the labels.
- (7) /Graph, select B, and specify range H11..H34 to plot the data points. Even though the data points appear only in the range H32..H34, the entire range must be specified.
- (8) /Graph Options Titles, First, and specify the first title line. Repeat for the second title line, and for the X- and Y-axis titles.
- (9) Select View to display the graph on the screen.

Worksheet C5B Graph of Figure 7.1

1 15 Sep 87, file C7A.WK1 on disks GSK 059-061

3 Worksheet C7A.WK1. Graph of Figure 7.1

	C	D	E	F	G	H	I
	Y	Scaled LN	X	A	Tonnage, millions	B	A Labels
10	10	0.00	0.60	0.00			10
11	20	0.06	2.00	0.00			
12	50	0.14					
13	100	0.20	0.60	0.06			20
14	200	0.26	2.00	0.06			
15	500	0.34					
16	1000	0.40	0.60	0.14			50
17			2.00	0.14			
18							
19							
20			0.60	0.20			100
21			2.00	0.20			
22							
23			0.60	0.26			200
24			2.00	0.26			
25							
26			0.60	0.34			500
27			2.00	0.34			
28							
29			0.60	0.40			1000

Worksheet C7A (continued)

30				
31	2.00	0.40		
32	0.60		610.54	0.3571
33	1.00		316.23	0.3000
34	2.00		61.05	0.1571
35	-----			

Worksheet C7B (continued)

30	350.0	1.77	0.44	0.19	2.50	0.0		
31	11.3	1.03	2.28	0.41				
32	297.0	1.74	0.49	0.21	0.00		0	0 1 Million
33	28.0	1.22	0.35	0.16	0.50		0	0 10 Million
34	88.0	1.47	0.22	0.10	1.00		0	0 100 Million
35	0.2	0.15	3.65	0.47	1.50		0	0 1 Billion
36	13.0	1.06	0.50	0.21	2.00		0	
37	31.0	1.25	0.40	0.18	2.50		0	
38	523.0	1.86	0.32	0.15				
39	60.0	1.39	0.29	0.14	1.77			0.26
40	189.0	1.64	0.50	0.21	1.40			0.31
41	25.2	1.20	0.50	0.21	1.47			0.25
42	56.8	1.38	0.50	0.21	1.26			0.26
43	9.0	0.98	0.50	0.21	1.11			0.33
44	667.0	1.91	0.79	0.27	1.59			0.21
45	1000.0	2.00	0.70	0.25	0.95			0.31
46	10.0	1.00	0.50	0.21	1.24			0.25
47	662.5	1.91	0.80	0.27	1.55			0.23
48	416.0	1.81	0.77	0.27	1.77			0.19
49	126.6	1.55	0.63	0.24	1.03			0.41
50	75.0	1.44	0.50	0.21	1.74			0.21
51	405.0	1.80	1.23	0.33	1.22			0.16
52	94.0	1.49	1.20	0.32	1.47			0.10
53	128.0	1.55	1.06	0.31	0.15			0.47
54	152.0	1.59	0.67	0.25	1.06			0.21
55	17.0	1.12	0.49	0.21	1.25			0.18
56	253.0	1.70	0.60	0.23	1.86			0.15
57	7.3	0.93	0.63	0.24	1.39			0.14
58	29.3	1.23	0.79	0.27	1.64			0.21

Worksheet CTB (continued)

59	443.0	1.82	0.73	0.26	1.20	0.21
60	344.4	1.77	0.79	0.27	1.38	0.21
61	19.3	1.14	1.97	0.39	0.98	0.21
62	17.5	1.12	0.86	0.28	1.91	0.27
63	40.0	1.30	1.00	0.30	2.00	0.25
64	61.2	1.39	1.84	0.38	1.00	0.21
65	1602.0	2.10	0.70	0.25	1.91	0.27
66	22.0	1.17	0.55	0.22	1.81	0.27
67	337.0	1.76	0.54	0.22	1.55	0.24
68	24.0	1.19	0.75	0.26	1.44	0.21
69	470.0	1.84	0.75	0.26	1.80	0.33
70	55.0	1.37	1.95	0.39	1.49	0.32
71	181.0	1.63	0.49	0.21	1.55	0.31
72	250.0	1.70	1.00	0.30	1.59	0.25
73	800.0	1.95	0.40	0.18	1.12	0.21
74	200.0	1.65	0.40	0.18	1.70	0.23
75	40.0	1.30	0.70	0.25	0.93	0.24
76	175.0	1.62	0.55	0.22	1.23	0.27
77	4.0	0.80	0.74	0.26	1.82	0.26
78	0.5	0.36	1.46	0.35	1.77	0.27
79	105.0	1.51	1.30	0.33	1.14	0.39
80	62.0	1.40	1.35	0.34	1.12	0.28
81	100.0	1.50	0.90	0.29	1.30	0.30
82	100.0	1.50	0.90	0.29	1.39	0.38
83	1679.0	2.11	0.72	0.26	2.10	0.25
84	93.0	1.48	0.48	0.20	1.17	0.22
85	151.0	1.59	0.25	0.12	1.76	0.22
86	1.9	0.64	2.37	0.41	1.19	0.26
87	22.4	1.18	0.32	0.15	1.84	0.26

Worksheet C7B (continued)

88	13.0	1.06	0.43	0.19	1.37	0.39
89	495.0	1.85	0.41	0.18	1.63	0.21
90	32.0	1.25	1.22	0.33	1.70	0.30
91	500.0	1.85	0.40	0.18	1.95	0.18
92	54.0	1.37	0.46	0.20	1.65	0.18
93	60.0	1.39	1.00	0.30	1.30	0.25
94	11.0	1.02	4.00	0.48	1.62	0.22
95					0.80	0.26
96					0.36	0.35
97					1.51	0.33
98					1.40	0.34
99					1.50	0.29
100					1.50	0.29
101					2.11	0.26
102					1.48	0.20
103					1.59	0.12
104					0.64	0.41
105					1.18	0.15
106					1.06	0.19
107					1.85	0.18
108					1.25	0.33
109					1.85	0.18
110					1.37	0.20
111					1.39	0.30
112					1.02	0.48
113						

Worksheet C7C Probability calculations

- 1 15 Sep 87, file C7C.WK1 on disks GSK 059-061
- 2
- 3 Worksheet C7C.WK1. Probability calculations
- 4
- 5
- 6 Part 1. Abbreviated running instructions (Details in text)
- 7
- 8 (1) To calculate the probability of gambler's ruin through
- 9 a run of bad luck, press Alt-a.
- 10
- 11 Number of prospects to be explored: 10
- 12 Percent probability of success in
- 13 each exploration: 20
- 14 Probability of ruin: 11%
- 15
- 16 (2) To calculate a factorial, press Alt-B.
- 17
- 18 Factorial number, N 5
- 19 Factorial value, N! 120
- 20
- 21
- 22
- 23
- 24
- 25
- 26
- 27
- 28
- 29

Worksheet C7C (continued)

```

30
31
32
33
34
35 Part 2. List of macros
36 \a
37 {GOTO}C8~
38 {DOWN 3}~
39 {RIGHT 5}~
40 {GETNUMBER "Enter the number of prospects to be explor
41 {DOWN}
42 {GETNUMBER "Enter the percent probability of success i
43 {DOWN}
44 {LET RUIN, (1-(P/100)) ^N}~
45
46
47
48
49
50
51
52
53
54 Factorial macro (Ewing and others, 1987)
55
56 \b
57 {GOTO}C16~
58 {DOWN 2}~
59 {RIGHT 5}~

```

Worksheet CTC (continued)

```

59 {LET IT,1}
60 {GETNUMBER "Enter number: ",FACTORIAL}
61 {FOR COUNTER,1,FACTORIAL,1,SUB}
62 {DOWN}~
63
64
65 SUB {LET IT,+IT*COUNTER}
66 {RETURN}
67
68 COUNTER 6
69
70
71 Part 3. Table of labels
72
73 COUNTER E68
74 FACTORIAL H18
75 IT H19
76 N H11
77 P H13
78 RUIN H14
79 SUB D65
80 \A D37
81 \B D56

```

CHAPTER 8

Compound Interest

In this chapter, I will first introduce compound-interest formulae, then provide a worksheet to tabulate discrete compounding interest factors at any selected rate of interest, and finally solve typical problems using these factors. The presentation and notation follows that of Frank Stermole of the Colorado School of Mines, who taught me economic evaluation and investment decision methods in a short course that “has been presented more than 300 times since 1970 to more than 10,000 practicing engineers, scientists, accountants, and managers working in petroleum, mineral, and general industry.” (Quotation from course brochure.)

8.1 COMPOUND INTEREST FORMULAE

Stermole and Stermole (1987, p. 17) designate the factors according to this rule: the first letter in each factor designates the quantity that is calculated, whereas the second letter designates the quantity that is given. The subscripts are the period interest rate, I , followed by the number of interest compounding periods, N .

Table 8.1 Compound-interest factors and formulae. After Stermole and Stermole (1987, p. 22) with notation adjusted to 1-2-3's style.

Name	Notation	Formula
Single Payment Compound Amount Factor	$F/P(I,N)$	$(1+I)^N$
Single Payment Present-Worth Factor	$P/F(I,N)$	$1/((1+I)^N)$
Uniform Series Compound Amount Factor	$F/A(I,N)$	$((1+I)^N-1)/I$
Sinking-Fund Deposit Factor	$A/F(I,N)$	$I/((1+I)^N-1)$
Capital-Recovery Factor	$A/P(I,N)$	$(I*((1+I)^N))/((1+I)^N-1)$
Uniform Series Present-Worth Factor $(I*((1+I)^N))$	$P/A(I,N)$	$((1+I)^N-1)/I$

Notation (from Stermole and Stermole, 1987, p. 16)

- P: Present single sum of money
- F: A future single sum of money
- A: The amount of each payment in a uniform series of equal payments
- N: The number of interest compounding periods
- I: The period compound interest rate

All of these relationships are used in evaluation. Probably most familiar is the single payment compound-amount factor, $F/P(I,N)$. We use this factor to calculate the future worth, F , after N years (or other period), of a present sum of money, P , with interest at I percent compounded per year (or other period). The formula is

$$F = P*(1 + i)^N.$$

We use the single payment present-worth factor to calculate the present value of a future sum, F , by solving the previous equation for P to obtain

$$P = F/((1 + I)^N).$$

We might use this formula to discount back to present time a series of payments to be received after a mine starts operation at some time in the future.

The uniform series compound amount factor, $F/A(I,N)$, gives the future value of a series of equal payments. This formula will calculate the value at retirement of \$2000-per-year IRA investments, assuming that interest rates are constant.

The sinking-fund deposit factor, $A/F(I,N)$, calculates the series of payments required to provide a specified amount of money at some time in the future. For instance, you may wish to determine the payments necessary to accumulate the down-payment for a house at a specified time in the future.

The capital-recovery factor, $A/P(I,N)$, gives the uniform series of payments equal to a present sum, for instance the house payments for a specified number of periods for a loan of a certain amount.

The uniform series present-worth factor, $P/A(I,N)$ is used to determine the present single sum of money, P , that is equivalent to a uniform series of equal payments, A , for N periods at I percent interest per period. For our purposes, these periods are usually equal to years and in the rest

of this book I use the terminology “years” instead of “periods.” The formula is

$$P = A*((1+I)^N-1)/(I(1+I)^N).$$

We could use this formula to determine the present value of a deposit that will produce a certain cash flow every year for a specified number of years.

The arithmetic gradient factor, not discussed in this book, is included to complete Stermole and Stermole’s (1987) tables as explained in the next section.

8.2 TABLE OF DISCRETE COMPOUNDING FACTORS

Worksheet C8A.WK1 is a table of discrete compounding factors for from 1 to 30 years, set up in the format of Stermole and Stermole (1987, p. 431). In the worksheet, the interest rate is set at 10%, but you can change this in cell G8 to any other value. Of course, you may need to adjust the cell widths or formats in order to display your results.

Figures 8.1 to 8.3 graph the interest factors for values of N from 1 to 30.

8.3 SOLVING INTEREST PROBLEMS

1-2-3 can readily be programmed to solve interest problems. Worksheet C8B.WK1 provides the required formulae. Part 1 of this worksheet is a data-entry table. The values entered in cells E14 to E19 are processed in Part 2 to give the six discrete compounding factors in cells E29 to E41. 1-2-3 provides functions for the factors most used in business; the others are programmed. The examples are those in the file as stored on the floppy diskette for 10% interest and six years.

Part 3 of the worksheet solves ten typical problems discussed in the next paragraph. In the worksheet, the solution formulae are given in

column E, and the values in this column are those obtained when the values in the data-entry table are those in the printed table ($I = 0.1$; $N = 6$; all other values = 1). The values in column F are those obtained when the values in Part 1 are set for the particular problem. They are copied from the values computed by formula in column E by using the 1-2-3 function /RV. For example, after setting the data-entry table for problem 8.1, the value 1.611 appears in cell E53. By following the prompts that appear when you key the command /RV, you can copy the value (not the formula) to cell F53.

Part 4 is the table of labels. Because of the way 1-2-3 alphabetizes, label 8.10 follows label 8.1 instead of label 8.9.

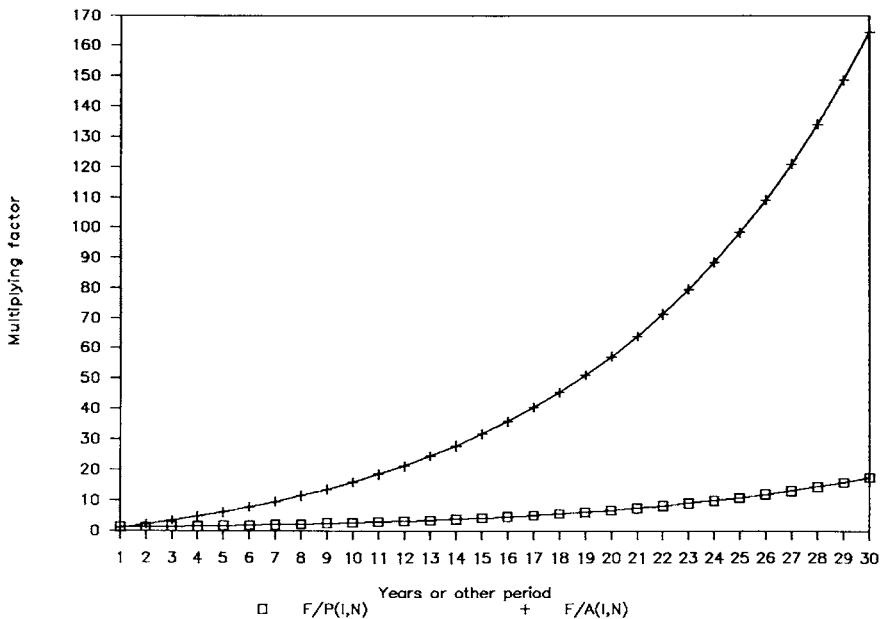


Figure 8.1 Interest factors, $F/P(I,N)$, $F/A(I,N)$

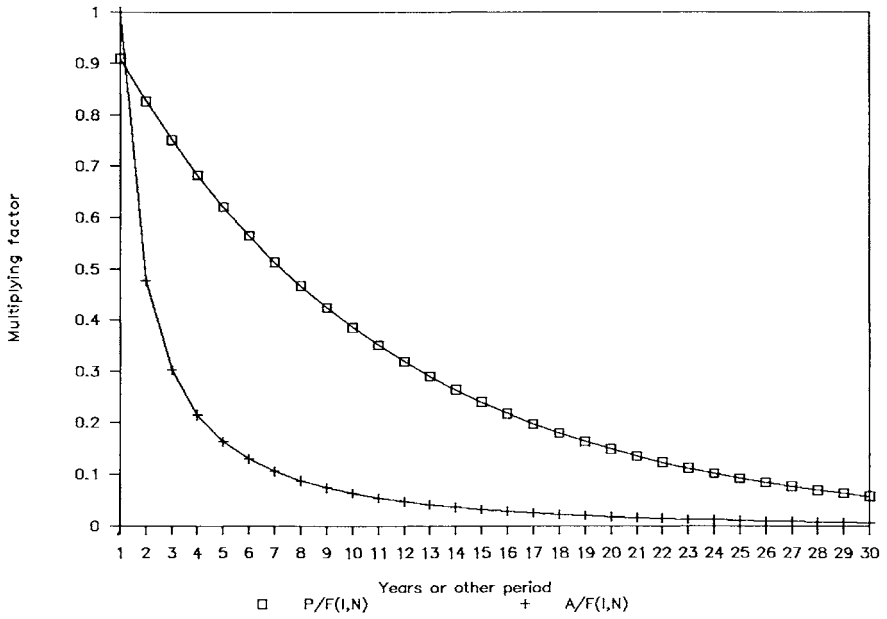


Figure 8.2 Interest factors, P/F(I,N), A/F(I,N)

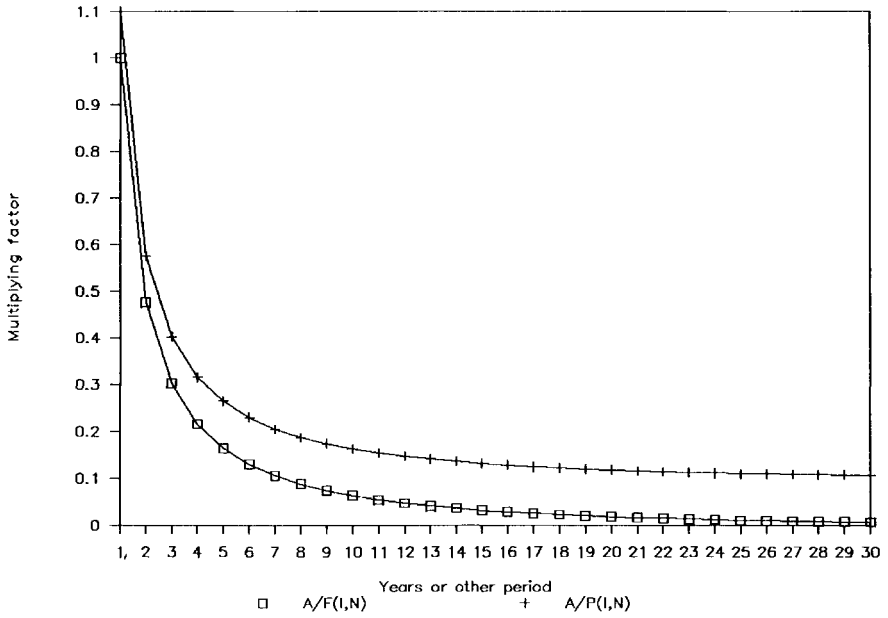


Figure 8.3 Interest factors, $A/F(I,N)$, $A/P(I,N)$

If you practice with these problems, note that I have labeled the ranges and cells so that you can move around the worksheet using the GoTo function key (F5). For instance, if you press function key (F5) followed by 8.1, the cursor will move to problem 8.1.

The following is an explanation of the ten problems presented in Part 3:

- 8.1 Calculate the future value of \$1. invested at 10% interest compounded annually for 5 years (Peters, 1987, p. 268). (Set E17 to 5.)
- 8.2 Suppose that you borrow \$1 from your friendly loan shark with the agreement that you will repay him doubling the amount of the loan for every day you keep the money. That is, you will owe him \$2 tomorrow, \$4 the day after tomorrow, etc. Suppose further that although you fully intend to repay him on the day after tomorrow, you let the matter slide for 20 days. How much do you owe him? (Set E17 to 20 and E18 to 1.)
- 8.3 How much would you need to invest at 10 % interest compounded annually to yield \$1 in five years (Peters, 1987, p. 269)? (Set E17 to 5 and E18 to 0.1.)
- 8.4 Calculate the uniform series of equal payments made at the end of each year for 15 years that are equivalent to a \$10,000 payment 15 years from now if interest is 9 percent per year compounded annually. (Set E17 to 10000, E17 to 15, and E18 to 0.09.)
- 8.5 Calculate the present value of a series of \$1000 payments to be made at the end of each year for 6 years if interest is 20 percent per year compounded annually. Repeat for 12 years. Note that doubling the number of years does not double the present value. (Set C16 to 1000, C17 to 6, and C18 to 0.2; then set C7 to 12.)
- 8.6 How much money (after taxes) do you need to win in the New York State Lottery in order to stop working for 20 years. Assume that you have no money now, that you will settle for an income of \$30,000 a year, and that your winnings can be invested to yield 12% interest. Set E16 to 30000, E17 to 20, and E18 to 0.12.)

- 8.7 A mine is expected to pay dividends of \$200,000 a year for ten years. A certain investor wishes 15% a year on his investment. What is the present value of the anticipated income at this rate of interest? (Set E16 to 200000, E17 to 10, and E18 to 0.15.)
- 8.8 A second investor wishes a 20% return on the investment of question 8.7. What is the present value in his situation? (Set E18 to 0.2.)
- 8.9 This problem is from Peters (1987, p. 270). "Take a capital investment with a starting date value of \$4 million and consider that the investment is expected to return \$2 million in the first and second years, \$1 million in the third year, \$0.5 million in the fourth and fifth years, and \$1 million in the sixth year. The interest rate — hurdle rate — is 15% compounded annually. (Key the returns into range D76..D81, and calculate the discount factors in range E76. E81.)
- 8.10 Suppose that you, at age 30, decide to put \$2000 a year into an Individual Retirement Account (IRA) to provide for your retirement at age 65. Suppose that you have two choices: (1) To start now, contribute \$2000 a year for 8 years, and then stop, or, (2) To wait 8 years, and then contribute \$2000 a year for the rest of your life until you retire? At 10% interest, which choice will give you a larger retirement fund? (In line 98, calculate the future value of your \$2000 annual payments for 8 years; then, in line 99, calculate the future value of this amount (cell E98) after 27 years. In line 101, calculate the future value of \$2000 annual payments for 27 years.)

Worksheet C8A Discrete compounding interest factors

1 12 Jan 88, file C8A.WK1 on disks GSK 059-061
 2
 3 Worksheet C8A.WK1 for discrete compounding interest factors
 4
 5 Part 1. Interest rate
 6
 7 Set the interest rate in cell
 8 G8 as a decimal fraction 0.1
 9

10
 11 Part 2. Discrete compounding factors for -----> 10.0%
 12

	F/P(I,N)	P/F(I,N)	F/A(I,N)	A/F(I,N)	A/P(I,N)	P/A(I,N)	A/G(I,N)
13							
14							
15							
16							
17	1.100	0.9091	1.000	1.000000	1.10000	0.909	0.909
18	1.210	0.8264	2.100	0.47619	0.57619	1.736	0.476
19	1.331	0.7513	3.310	0.30211	0.40211	2.487	0.937
20	1.464	0.6830	4.641	0.21547	0.31547	3.170	1.381
21	1.611	0.6209	6.105	0.16380	0.26380	3.791	1.810
22	1.772	0.5645	7.716	0.12961	0.22961	4.355	2.224
23	1.949	0.5132	9.487	0.10541	0.20541	4.868	2.622
24	2.144	0.4665	11.436	0.08744	0.18744	5.335	3.004
25	2.358	0.4241	13.579	0.07364	0.17364	5.759	3.372
26	2.594	0.3855	15.937	0.06275	0.16275	6.145	3.725
27	2.853	0.3505	18.531	0.05396	0.15396	6.495	4.064
28	3.138	0.3186	21.384	0.04676	0.14676	6.814	4.388
29	3.452	0.2897	24.523	0.04078	0.14078	7.103	4.699

Worksheet C8A (continued)

30	14	3.797	0.2633	27.975	0.03575	0.13575	7.367	4.996
31	15	4.177	0.2394	31.772	0.03147	0.13147	7.606	5.279
32	16	4.595	0.2176	35.950	0.02782	0.12782	7.824	5.549
33	17	5.054	0.1978	40.545	0.02466	0.12466	8.022	5.807
34	18	5.560	0.1799	45.599	0.02193	0.12193	8.201	6.053
35	19	6.116	0.1635	51.159	0.01955	0.11955	8.365	6.286
36	20	6.727	0.1486	57.275	0.01746	0.11746	8.514	6.508
37	21	7.400	0.1351	64.002	0.01562	0.11562	8.649	6.719
38	22	8.140	0.1228	71.403	0.01401	0.11401	8.772	6.919
39	23	8.954	0.1117	79.543	0.01257	0.11257	8.883	7.108
40	24	9.850	0.1015	88.497	0.01130	0.11130	8.985	7.288
41	25	10.835	0.0923	98.347	0.01017	0.11017	9.077	7.458
42	26	11.918	0.0839	109.182	0.00916	0.10916	9.161	7.619
43	27	13.110	0.0763	121.100	0.00826	0.10826	9.237	7.770
44	28	14.421	0.0693	134.210	0.00745	0.10745	9.307	7.914
45	29	15.863	0.0630	148.631	0.00673	0.10673	9.370	8.049
46	30	17.449	0.0573	164.494	0.00608	0.10608	9.427	8.176
47								
48								
49								
50								
51	#A.F.I.N							
52	#A.G.I.N							
53	#A.P.I.N							
54	#F.A.I.N							
55	#F.P.I.N							
56	#P.A.I.N							
57	#P.F.I.N							
58	I							

Part 3. Table of range names

Worksheet C8B Solving interest problems

1 29 May 89, file C8B.WK1 on disks GSK 059-061

2

3 Worksheet C8B.WK1. Solving interest problems

4

5 Part 1. Data-entry table.

6 Enter the required values in column E.

7

8 C D E

9 -----

10 Range Description Value

11 Name -----

12 -----

13

14 P Present value 1

15 F Future value 1

16 A Annuity 1

17 N Number of periods (years) 6

18 I Interest rate 0.1

19 G Arithmetic gradient series 1

20 -----

21

22

Worksheet C8B (continued)

Part 2. Discrete compounding factors.

	Factor	Formula	Example (I=0.1, N=6)
23			
24			
25			
26			
27			
28			
29	$F/P(I, N)$	$(P) * ((1+I)^N)$	1.772
30			
31	$P/F(I, N)$	$(F) * (1 / (1+I)^N)$	0.5645
32			
33	$F/A(I, N)$	$@FV(A, I, N)$	7.716
34			
35	$A/F(I, N)$	$(F) * ((I) / (((1+I)^N) - 1))$	0.12961
36			
37	$A/P(I, N)$	$@PMT(P, I, N)$	0.22961
38			
39	$P/A(I, N)$	$@PV(A, I, N)$	4.355
40			
41	$A/G(I, N)$	$(G) * ((1/I) - (N/I) * (#A.F.I.N))$	10.000
42			
43			
44			

Worksheet C8B (continued)

45 Part 3. Solutions to problems

46	C	D	E	F
47	-----	-----	-----	-----
48	Problem	Factor	Solution	Solution
49	Number		Formula	Value
50	-----	-----	-----	-----
51				
52				
53	8.1	$(P) * ((1+I)^N)$	\$1.772	\$1.611
54				
55	8.2	$(P) * ((1+I)^N)$	\$2	\$1,048,576
56				
57	8.3	$(F) * (1 / (1+I)^N)$	\$0.5645	\$0.6209
58				
59	8.4	$(F) * ((I) / (((1+I)^N) - 1))$	\$0.13	\$340.59
60				
61	8.5	@PV(A, I, N)	\$4	\$3,326
62			\$4	\$4,439
63				
64	8.6	@PV(A, I, N)	\$4	\$224,083
65				
66	8.7	@PV(A, I, N)	\$4	\$1,003,754
67				
68	8.8	@PV(A, I, N)	\$4	\$838,494
69				

Worksheet C8B (continued)

70	8.9				
71					
72					
73	Year				
74		Income,		15% Discount	Present
75		\$M		Factor	Value
76	1		2.0	0.870	\$1,739,130
77	2		2.0	0.756	\$1,512,287
78	3		1.0	0.658	\$657,516
79	4		0.5	0.572	\$285,877
80	5		0.5	0.497	\$248,588
81	6		1.0	0.432	\$432,328
82					
83	Totals		7.0		\$4,875,727
84	Less capital investment				(\$4,000,000)
85					
86	Net present value (acquisition value)				\$875,727
87					
88					
89					
90					

Worksheet C8B (continued)

91	8.10		
92			
93			
94	Age		30
95	Years until retirement		35
96	Years until retirement - 8		27
97			
98		@FV(2000,0.1,8)	\$22,872
99		(E98)*((1.1)^E96)	\$299,849
100			
101		@FV(2000,0.1,E96)	\$242,200
102			
103			
104			
105	Part 4.	Table of labels	
106			
107	#A.F.I.N	C55	
108	#A.P.I.N	E37	
109	#F.A.I.N	E33	
110	#F.P.I.N	E29	
111	#P.A.I.N	E39	
112	#P.F.I.N	E31	
113	8.1	C53	
114	8.10	C91	
115	8.2	C55	
116	8.3	C57	
117	8.4	C59	
118	8.5	C61	
119	8.6	C64	

Worksheet C8B (continued)

120	8.7	C66
121	8.8	C68
122	8.9	C70
123	A	E16
124	F	E15
125	G	E19
126	I	E18
127	N	E17
128	P	E14
129	PART.1	C14..E19
130	PART.2	C29..E41
131	PART.3	C45
132	PART.4	C93

CHAPTER 9

Depreciation and Depletion

In the United States, “depreciation” and “depletion” are accounting terms used to calculate income tax. They are needed to determine cash flow, which is the basis for any economic evaluation.

Depreciation, as used in economic evaluation, “is a tax allowance that is a reasonable allowance for the exhaustion, wear and tear and obsolescence of property used in a trade or business, or of property held by a tax payer for the production of income” (Stermole and Stermole, 1987, p. 229).

Depletion is the “recovery of an owner’s economic interest in mineral (including oil and gas) reserves through federal tax deductions related to removal of the mineral over the economic life of the property” (Stermole and Stermole, 1987, p. 455).

Neither depreciation or depletion involve a corresponding flow of money from the enterprise, any more than personal income tax deductions for individuals, blindness, age, etc. exactly represent corresponding expenditures of money.

Depreciation and depletion are explained briefly by Peters (1987, p. 279-280) and in detail by Stermole and Stermole (1987, p.222-256).

9.1 YEARLY NET INCOME AND CASH FLOW CALCULATION

Worksheet C9A.WK1 is a yearly income and cash-flow calculation based on the format of Peters (1987, p. 279), with additional lines to calculate cost depletion. Part 1 gives fixed data for the project; row 11 assumes a property cost of \$1 million, which is used to calculate cost depletion.

Part 2 includes all of Peters's (1987, p. 279) entries, as well as additional ones to calculate cost depletion. In line 26, depreciation is treated as straightline, the method usually used in preliminary work. Stermole and Stermole (1987, p. 229-241) explain several alternative methods, which may be necessary for detailed work. 1-2-3 provides special functions for the *straightline*, *double-declining-balance*, and *sum-of-the-years'-digits* methods.

Depletion is calculated in rows 30 to 36. Allowable percentage depletion is calculated in rows 30 to 32. Cell D30 calculates percentage depletion by multiplying the percentage factor from D14 by the revenue in D22, and cell D31 calculates the 50% allowable limit. Cell D32 contains the smaller of the values in the two previous cells. Cost depletion is calculated in cell D34. Cell D36 contains the allowable depletion, which is the larger of the values in cells D32 and D34.

Subtracting selected depletion (D36) from net income before depletion (D28) gives taxable income (D38), and subtracting income tax (D39) gives net income in cell D41.

Finally, we obtain cash flow (D45) by adding back depreciation and depletion in rows 42 and 43, remembering that these two items were subtracted only to calculate income tax and do not in themselves represent actual expenditures of cash.

9.2 THE RIDGEWAY, SOUTH CAROLINA, GOLD MINE

A note in *Engineering and Mining Journal* (E&MJ) for May, 1987 (p. 78) states that “A feasibility report issued by the Ridgeway Mining Co. to Amselco Minerals Inc. and Galactic Resources Ltd. projects commercial gold production at the Ridgeway, South Carolina, project starting by mid-1988 at a rate of 158,000 tr oz/yr of gold over the first four years (1989-92), and 133,000 tr oz/yr over the 11-year life of the mine. Amselco and Galactic are joint venture partners in the project, 15 miles northeast of Columbia. Capital costs to bring the Ridgeway project into production are estimated at \$76 million.”

Using these data (supplemented by information from company reports), we can construct worksheet C9B.WK1. To the items in Part 1 of worksheet C9A.WK1, we add production, selling price, and operating cost. The worksheet formulation of this problem illustrates the flexibility of the approach; the assumptions can readily be changed — now by knowledgeable people — or later as more data are published.

Worksheet C9A Yearly net income and cash flow

1	12 Jan 88, file C9A.WK1 on disks GSK 059-61
2	
3	Worksheet C9A.WK1. Yearly net income and cash flow
4	(after Peters, 1987, p. 279)
5	
6	
7	Part 1. Fixed data
8	
9	C
10	-----
11	Property cost \$1,000,000
12	Capital investment \$8,500,000
13	Years life 10
14	Depletion percentage 15
15	Income tax percentage 55
16	-----
17	
18	

Worksheet C9A (continued)

19	Part 2.	Yearly net income and cash flow	
20			
21		-----	
22	Revenue		\$15,420,000
23	Operating costs		7,340,000
24			
25	Operating income		8,080,000
26	Depreciation		850,000
27			
28	Net income before depletion		7,230,000
29			
30	Percentage depletion		2,313,000
31	50% limit		3,615,000
32	Selected percentage depletion		2,313,000
33			
34	Cost depletion		100,000
35			
36	Selected depletion		2,313,000
37			
38	Taxable income		4,917,000
39	Income tax @ 55 %		2,704,350
40			
41	Net income		2,212,650
42	Depreciation		850,000
43	Selected depletion		2,313,000
44			
45	Cash flow		5,375,650
46		-----	
47			

Worksheet C9A (continued)

	Part 3.	Table of labels
48		
49		
50		
51	!CASH.FLOW	D45
52	!COST.DPL	D34
53	!DEPRE.&.AMORT.	D26
54	!DPL.&.SELECTED	D32
55	!DPL.50%.LIMIT	D31
56	!DPL.SELECTED	D36
57	!INCOME.TAX	D39
58	!NET.BEFORE.DPL	D28
59	!NET.INCOME	D41
60	!OP.INCOME	D25
61	!OPERATING.COST	D23
62	!PCT.DPL	D30
63	!REVENUE	D22
64	!TAX.INCOME	D38
65	%DEPLETION	D14
66	%INCOME.TAX	D15
67	CAP.INVESTMENT	D12
68	LIFE	D13
69	PROPERTY.COST	D11

Worksheet C9B Yearly net income and cash flow. Ridgeway, South Carolina, gold mine

1	19 Feb 88, file C9B.WK1 on disks GSK 059-061	
2		
3	Worksheet C9B.WK1. Yearly net income and cash flow,	
4	Ridgeway, South Carolina, gold mine	
5		
6		
7	Part 1. Fixed data	
8		
9		C
10		D
11	Property cost	\$18,000,000
12	Capital investment	\$76,000,000
13	Years life	12
14	Depletion percentage	15
15	Income tax percentage	55
16	Production, tr oz/year	135,000
17	Selling price, \$/tr oz	420
18	Operating costs, \$/tr oz	210
19		
20		
21		
22		
23		
24		
25		

Worksheet C9B (continued)

Part 2. Yearly net income and cash flow		
26		
27		
28		
29	Revenue	\$56,700,000
30	Operating costs	28,350,000
31		
32	Operating income	28,350,000
33	Depreciation	6,333,333
34		
35	Net income before depletion	22,016,667
36		
37	Percentage depletion	8,505,000
38	50% limit	11,008,333
39	Selected percentage depletion	8,505,000
40		
41	Cost depletion	1,500,000
42		
43	Selected depletion	8,505,000
44		
45	Taxable income	13,511,667
46	Income tax @ 55 %	7,431,417
47		
48	Net income	6,080,250
49	Depreciation	6,333,333
50	Selected depletion	8,505,000
51		
52	Cash flow	20,918,583
53		
54		

Worksheet C9B (continued)

55		
56	Part 3. Table of labels	
57		
58	!CASH.FLOW	D52
59	!COST.DPL	D41
60	!DEPRE.&.AMORT.	D33
61	!DPL.%.SELECTED	D39
62	!DPL.50%.LIMIT	D38
63	!DPL.SELECTED	D43
64	!INCOME.TAX	D46
65	!NET.BEFORE.DPL	D35
66	!NET.INCOME	D48
67	!OP.INCOME	D32
68	!OPERATING.COST	D30
69	!PCT.DPL	D37
70	!REVENUE	D29
71	!TAX.INCOME	D45
72	%DEPLETION	D14
73	%INCOME.TAX	D15
74	CAP.INVESTMENT	D12
75	LIFE	D13
76	OP.COST.PER.OZ	D18
77	PRICE.PER.OZ	D17
78	PRODUCTION	D16
79	PROPERTY.COST	D11

CHAPTER 10

Discounted Cash Flow Rate of Return

The Discounted cash flow rate of return (DCFROR), also named the discounted cash flow return on investment (DCFROI), or the internal rate of return (IRR), is the method most widely used to compare investment opportunities. We may want to compare one mining venture with another or a mining venture with some other investment.

The discounted cash flow rate of return is simply the interest rate, I , in the formula

$$P = A(1)/(1+I)^1 + A(2)/(1+I)^2 + \dots + A(N)/(1+I)^N$$

where P is the principal (capital investment), $A(N)$ is the yearly cash flow (which may be positive, zero, or negative), I is the interest rate, and N is the number of years.

1-2-3 makes calculation of DCFROR easy by providing a function, which is explained in the next section. Peters (1987, p. 276-279) gives a brief account of DCFROR; Stermole and Stermole (1987, p. 257-283) provide a chapter that thoroughly explains the principle and details; Koch and Link (1971, p. 312-314) work a simple example in detail.

10.1 A SIMPLE EXAMPLE

Worksheet C10A.WK1 solves a simple example problem to illustrate the method. Assume that a truck with an expected useful life of 4 years costs \$10,000. Annual income each year of the life of this truck is expected to be \$8,000, and annual operating costs are expected to be \$3,500. Straightline depreciation is used, and the overall tax rate is 50 percent.

Part 1 of the worksheet lists the fixed data for the problem; part 2 gives the yearly net income and cash flow. Part 3 is the calculation of DCFROR through 1-2-3's function

@IRR(estimate,range).

The range, D39..D43, is labeled CASH.FLOWS.0-4; the estimate is required, because 1-2-3 calculates the value of the function interactively. After 20 iterations, the entry ERR appears if a result to within 0.0000001 is not located. If, while using the @IRR function, you get the message ERR, simply substitute another, closer estimate of DCFROR. For precautions, see a 1-2-3 reference manual. From my initial estimate of 30%, recorded in the function as 0.3, 1-2-3 calculates the correct value of 15. This is exactly the value that makes the net present value equal to 0 at a hurdle rate of 15%.

10.2 A SECOND EXAMPLE OF DCFROR

A second example of DCFROR is provided in worksheet C10B.WK1 which presents a problem from Peters (1987, p. 278). Except for roundoff errors, the values in the worksheet match those of Peters.

In Part 1, we first determined net present value (NPV) at two interest rates that bracket DCFROR. For the data given, these rates are 16% and 17%.

DCFRORE is calculated in Part 2, first by linear interpolation, and then by 1-2-3's formula. The results are carried out to several decimal places

in order to demonstrate that linear interpolation does not give a precisely correct result, although it is certainly close enough for a preliminary evaluation. For one thing, none of the cash flows will be known exactly! Of course, calculation by 1-2-3's formula is faster, particularly if you cannot guess the interest rates that bracket DCFROR.

The worksheet is a convenient one to use to gain an understanding of DCFROR by changing the flows in the range D14..D21, recalculating, and seeing how DCFROR changes.

Worksheet C10A Illustration of DCFROR

```
1 12 Jan 88, file C10A.WK1 on disks GSK 059-061
2
3 Worksheet C10A.WK1. Illustration of DCFROR
4
5
6 Part 1. Fixed data
7
8           C           D
9 -----
10 Cost of the truck           $10,000
11 Life, years                 4
12 Income tax rate, %         50
13 Estimate of DCFROR, %      20
14 -----
15
16
```


Worksheet C10A (continued)

17	Part 2. Yearly net income and cash flow	
18	-----	
19		
20	Revenue	\$8,000
21	Operating costs	3,500
22		
23	Operating income	4,500
24	Depreciation	2,500
25		
26	Taxable income	2,000
27	Income tax	1000
28		
29	Net income	1,000
30	Depreciation	2,500
31		
32	Cash flow	\$3,500
33	-----	
34		
35		
36	Part 3. Calculation of DCFROR	
37	-----	
38		
39	Cash flow (Year 0)	(10,000)
40	Cash flow (Year 1)	3,500
41	Cash flow (Year 2)	3,500
42	Cash flow (Year 3)	3,500
43	Cash flow (Year 4)	3,500
44		
45	DCFROR	15%

Worksheet C10A (continued)

46	-----		
47			
48			
49	Part 4.	Table of labels	
50			
51	%EST.DCFROR		D13
52	CASH.FLOW		D32
53	CASH.FLOW.0		D39
54	CASH.FLOW.1		D40
55	CASH.FLOW.2		D41
56	CASH.FLOW.3		D42
57	CASH.FLOW.4		D43
58	CASH.FLOWS.0-4		D39.,D43
59	DEPRECIATION		D24
60	INCOME.TAX		D27
61	LIFE		D11
62	NET.INCOME		D29
63	OP.COST		D21
64	OP.INCOME		D23
65	REVENUE		D20
66	TAXABLE.INCOME		D26
67	TRUCK.COST		D10

Worksheet C10B Estimating DCFROR

1 12 Jan 88, file C10B.WK1 on disks GSK 059-061

2

3 Worksheet C10B.WK1. Estimating DCFROR

4 Problem from Peters (1987, p. 278)

5

6

7 Part 1. Calculation of present value at 16% and 17%

8

9 C D E F G H

10	Year	Cash Flow	P/F(16,N)	Present Value,16%	P/F(17,N)	Present Value,17%
11	0	\$-230,000				
12	1	50,000	0.862	\$43,103	0.855	\$42,735
13	2	70,000	0.743	52,021	0.731	51,136
14	3	70,000	0.641	44,846	0.624	43,706
15	4	60,000	0.552	33,137	0.534	32,019
16	5	50,000	0.476	23,806	0.456	22,806
17	6	40,000	0.410	16,418	0.390	15,594
18	7	60,000	0.354	21,230	0.333	19,992

22

23 Total present value \$234,561

24 Capital investment -230,000

\$227,987

-230,000

25 Net present value \$4,561

\$-2,013

26

27

28

Worksheet C10B (continued)

29			
30			
31	Part 2.	Calculation of DCFROR	
32			
33			
34	a.	Calculation by linear interpolation	
35			
36		Interpolation	$16\% + (4,561 / (4,561 + 2,013))$
37			
38		Discounted cash flow rate of return (DCFROR) =	16.743%
39			
40			
41	b.	Calculation by 1-2-3's function @IRR	
42			
43		@IRR(.3,D14..D21)	16.738%
44			
45			
46			
47	Part 3.	Table of labels	
48			
49		CASH.FLOW.0	D14
50		CASH.FLOW.1	D15
51		CASH.FLOW.2	D16
52		CASH.FLOW.3	D17
53		CASH.FLOW.4	D18
54		CASH.FLOW.5	D19
55		CASH.FLOW.6	D20
56		CASH.FLOW.7	D21
57		NPV.16	F27

Worksheet C10B (continued)

58	NPV.17	H27
59	TOT.PRES.VAL.16	F24
60	TOT.PRES.VAL.17	H24

CHAPTER 11

Blocking Ore From Data in Development Workings

In this chapter, we will first calculate grade and tonnage of ore from data obtained in development workings, and then calculate specific gravity of ore from its mineralogical composition.

11. 1 GRADE AND TONNAGE OF ORE FROM DEVELOPMENT DATA

The following sample averages represent four blocks of silver ore in the Carlos Francisco Mine at Casapalca, Peru. The development workings in this part of the mine consist of three levels, numbered from the top downward, 6, 7, and 8; and three raises, lettered from north to south, A, B, and C. Table 11.1 gives the length and width dimensions of these workings, together with the silver assays.

Table 11.1 Development data for 4 blocks of ore in the Carlos Francisco Mine.

Working		Length, m	Width, m	Silver, g/T
Level 6	Raise A to Raise B	110	0.9	300
Level 6	Raise B to Raise C	75	1.1	480
Level 7	Raise A to Raise B	110	1.0	540
Level 7	Raise B to Raise C	75	1.1	660
Level 8	Raise A to Raise B	110	1.0	750
Level 8	Raise B to Raise C	75	1.2	600
Raise A Level 6-7		40	0.9	240
Raise B Level 6-7		40	1.5	600
Raise C Level 6-7		40	1.2	540
Raise A Level 7-8		35	1.0	690
Raise B Level 7-8		35	1.2	660
Raise C Level 7-8		35	1.2	720

Sketching a longitudinal section of the four blocks will show you the geological situation. Additional information is this: (1) assays already have been adjusted to the minimum mining width, 0.9 m; (2) the specific gravity is 3.2; (3) the cut-off grade of ore is 450 grams silver per tonne.

From these data, we will calculate:

1. Tonnage of ore and its grade in each of the four blocks.
2. Aggregate tonnage and average grade of the four blocks.
3. Aggregate tonnage and grade after allowing for 10 percent dilution of barren wall rock.

Peters (1987, p. 482-491) explains the basic procedure, which is a weighted average. Using 1-2-3, we can set up the calculations (worksheet C11A.WK1). Part 1 gives the calculations for the block between levels 6 and 7 and raises A and B. Column C lists the arbitrary numbers designating the four sides of block 1; column D the lengths, column E the widths, and column F the assays. Column G lists the products of lengths times widths, and column H the products of lengths times widths times assays. The column sums, in line 17, then are used to calculate the estimates of assay and tonnes in columns I and J.

Because the grade for this block is lower than the cutoff grade, we need to reduce its size as shown in Part 2. The conventional procedure of defining a triangular block is followed.

Once we have set up the worksheet for one block, we can copy the format for the remaining four blocks using the /COPY command.

Part 6 summarizes the calculations for the four blocks. The sums in row 94 are used to calculate the average assay of 612 grams per tonne in row 97. Line 98 gives the 10 percent additional tonnage resulting from dilution of wall rock, which is assumed to contain no silver. Finally, row 100 gives the mined grade.

The table of labels, Part 7, applies only to Block 1.

11.2 CALCULATION OF SPECIFIC GRAVITY

Starting with the ore mineralogy and the grade expressed as a percentage of metal, we can calculate the specific gravity of the ore. The method is explained by Peters (1987, p. 481) and by Koch and Link, (1971, p. 249); the method in worksheet C11B.WK1 is that of the second reference.

Although the calculations are not complicated, formulating the problem seems to puzzle many people. I think it is helpful to present the problem in the concrete terms of determining the number of grams of each mineral in 100 grams of ore (rows 16 and 17 of worksheet C11B.WK1.)

Once the specific gravity is determined (row 10), you can determine other factors, such as the number of cubic feet per short ton, if required, using tables in Peters (1987) or other standard books.

The function relating grade and specific gravity is not linear. Figure 11.1 graphs the quartz/galena curve for various grades of galena.

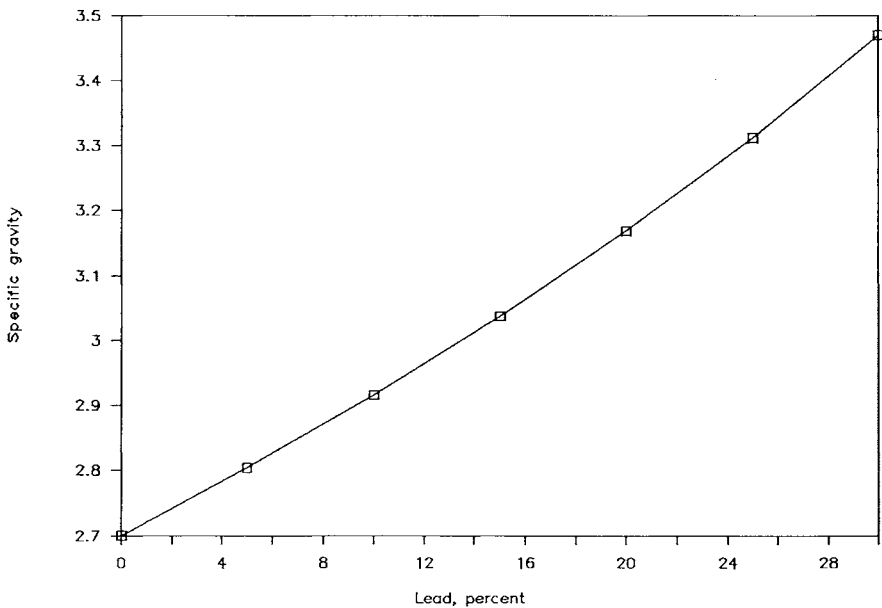


Figure 11.1 S.G. - grade relationship, example of a lead/quartz ore

Worksheet C11A Calculating grade and tonnage of ore

1 02 Feb 89, file C11A.WK1 on disks GSK 059-061

2

3 Worksheet C11.A.WK1. Calculating grade and tonnage of ore

4

5 Part 1. Calculations for block 1 (Levels 6 and 7, Raises A and B)

6

7	C	D	E	F	G	H	I	J
9	Side	Length,	Width,	Assay,	L x W	L x W x A	Block	Block
10	No.	m	m	g/T			assay	tonnes
11								
12	1	110	0.9	300	99.0	29,700		
13	2	40	1.5	600	60.0	36,000		
14	3	110	1.0	540	110.0	59,400		
15	4	40	0.9	240	36.0	8,640		
16								
17		300.0	1.0		305.0	133,740	438	14,315
18								
19								
20								

Worksheet CIAA (continued)

21 Part 2. Size of block 1 reduced to raise grade above cutoff 22
 23 -----

24 Side No.	Length, m	Width, m	Assay, g/T	L x W	L x W x A	Block assay tonnes	Block tonnes
27 2	40	1.5	600	60.0	36,000		
28 3	110	1.0	540	110.0	59,400		
30	150.0	1.1	1140.0	170.0	95,400	561	7,979

31 -----
 32
 33

Part 3. Calculations for block 2 (Levels 6 and 7, Raises B and C)

34
 35
 36 C D E F G H I J
 37 -----

38 Side No.	Length, m	Width, m	Assay, g/T	L x W	L x W x A	Block assay tonnes	Block tonnes
41 1	75	1.1	480	82.5	39,600		
42 2	40	1.2	540	48.0	25,920		
43 3	75	1.1	660	82.5	54,450		
44 4	40	1.5	600	60.0	36,000		
45							
46	230.0	1.2		273.0	155,970	571	11,395

47 -----
 48
 49

Worksheet C11A (continued)

Part 4. Calculations for block 3 (Levels 7 and 8, Raises B and C)										
	C	D	E	F	G	H	I	J		
50										
51										
52										
53										
54	Side	Length,	Width,	Assay,	L x W	L x W x A	Block	Block		
55	No.	m	m	g/T			assay	Block		
56							tonnes	tonnes		
57	1	75	1.1	660	82.5	54,450				
58	2	35	1.2	720	42.0	30,240				
59	3	75	1.2	600	90.0	54,000				
60	4	35	1.2	660	42.0	27,720				
61										
62		220.0	1.2		256.5	166,410	649	9,794		
63										
64										
65										
66	Part 5. Calculations for block 4 (Levels 7 and 8, Raises A and B)									
67										
68	C	D	E	F	G	H	I	J		
69										
70	Side	Length,	Width,	Assay,	L x W	L x W x A	Block	Block		
71	No.	m	m	g/T			assay	Block		
72							tonnes	tonnes		
73	1	110	1.0	540	110.0	59,400				
74	2	35	1.2	660	42.0	27,720				
75	3	110	1.0	750	110.0	82,500				
76	4	35	1.0	690	35.0	24,150				
77										
78		290.0	1.0		297.0	193,770	652	12,617		

Worksheet C11A (continued)

104	Part 7.	Table of labels
105		
106	AG.1	F12
107	AG.2	F27
108	AG.3	F28
109	AG.4	F15
110	BLOCK.L.X.W.X.A	H30
111	L.X.W.1	G12
112	L.X.W.2	G27
113	L.X.W.3	G28
114	L.X.W.4	G15
115	L.X.W.BLOCK	G30
116	L.X.W.X.A.1	H12
117	L.X.W.X.A.2	H27
118	L.X.W.X.A.3	H28
119	L.X.W.X.A.4	H15
120	LENGTH.1	D12
121	LENGTH.2	D27
122	LENGTH.3	D28
123	LENGTH.4	D15
124	LENGTH.BLOCK	D30
125	WIDTH.1	E12
126	WIDTH.2	E27
127	WIDTH.3	E28
128	WIDTH.4	E15
129	WIDTH.BLOCK	E30

Worksheet C11B Calculating specific gravity of ore

```

1 18 Feb 88, file C11B on disks GSK 059-061
2
3 Worksheet C11B.WK1. Calculating specific gravity of ore
4
5 Part 1. Calculation of specific gravity for a galena,
6 quartz ore (from Koch and Link, 1971, p. 249)
7
8 -----
9 Grade of ore, % 5
10 Specific gravity of ore 2.8036
11 -----
12 Atomic weight of Pb 207.2
13 Atomic weight of sulfur 32.1
14 Molecular weight galena 239.3
15 Percent of Pb in galena 86.586
16 Grams of galena in 100 grams of ore 5.775
17 Grams of quartz in 100 grams of ore 94.225
18 Specific gravity of galena 7.5
19 Specific gravity of quartz 2.7
20 Volume of galena 0.770
21 Volume of quartz 34.898
22 Volume of ore 35.668
23 -----
24
25

```

Worksheet C11B (continued)

26 Part 2. Table of labels

27		
28	GN.G.IN.ORE	D16
29	GN.MOL.WT	D14
30	GN.S.G.	D18
31	GN.VOLUME	D20
32	ORE.VOLUME	D22
33	PB.%	D9
34	PB.%.IN.GN	D15
35	PB.ATOMIC.WT	D12
36	QTZ.G.IN.ORE	D17
37	QTZ.S.G.	D19
38	QTZ.VOLUME	D21
39	S.ATOMIC.WT	D13

CHAPTER 12

Blocking Ore from Drillhole Data

This chapter provides worksheets to block ore from drillhole data by making linear and quadratic regressions. I will introduce the method in section 12.1 with a small, simple example of fitting a regression model to structural geologic data.

The example of blocking ore is based on a data set from the Chambishi copper mine in Zambia (Koch, 1975, 1978; Mendelsohn, 1980). Figure 12.1 is a vertical longitudinal section that shows the open pit, blocks of ore below the open pit, and the intersections of 38 diamond-drill holes with the ore-bearing sedimentary bed. Next to each intersection is plotted the grade in copper at that point. We will calculate a quadratic regression and solve the resulting equation for the grade of each block. Then, we will compare this estimate with estimates made in other ways.

12.1 QUADRATIC REGRESSION — RANGELY, COLORADO OIL-AND-GAS FIELD

As a first example of quadratic regression, I use a small data set from the Rangely, Colorado oil-and-gas field, from an earlier analysis (Koch

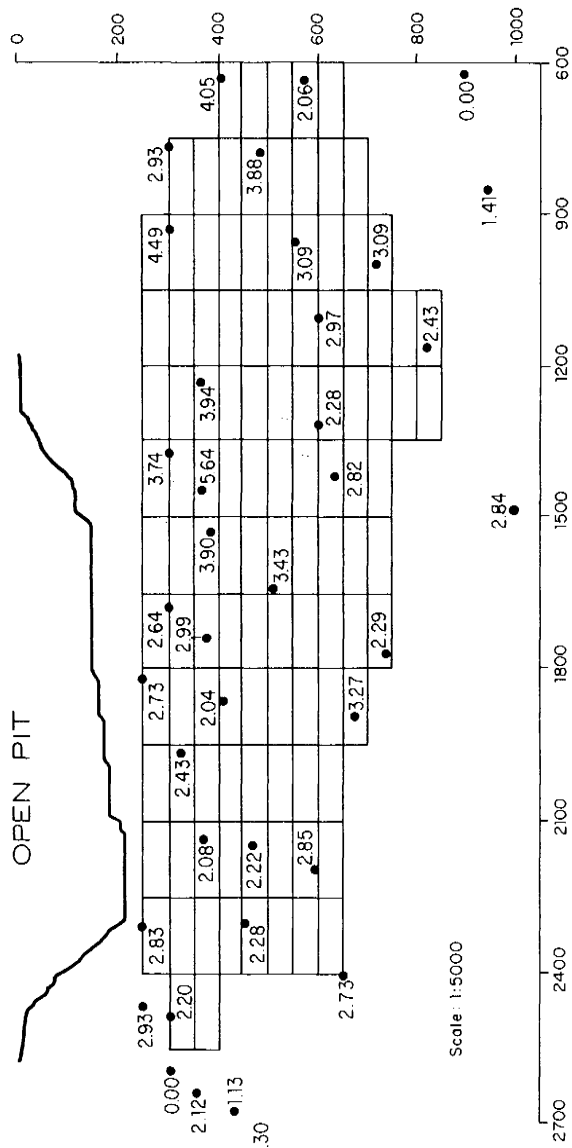


Figure 12.1 Chambishi mine, Zambia, vertical longitudinal section

and Link, 1971, p. 4). The data are elevations in 26 wells of the top of the Weber Sandstone (Pennsylvanian/Permian age) above a datum plane. In worksheet C12A.WK1, Part 1 is a data table listing (in columns C, D, and E) the elevation (w), the east coordinate (x), and the north coordinate (y) for the data plotted in Figure 12.2.

The quadratic equation for a surface in two dimensions is

$$\hat{w} = A + (B.1)*X + (B.2)*Y + (B.3)*(X^2) + (B.4)*(Y^2) + (B.5)*(XY),$$

where A is a constant and B.1 to B.5 are coefficients. Columns F, G, and H in Part 1 list the squared and cross-product terms calculated from columns D and E by entering the appropriate formulas in cells F13, G13, and H13, and then copying downward to the bottom of the table.

Part 2 gives the regression output, which is similar to that of linear regression (Chapter 3) except that there are five rather than two independent variables.

Part 3 applies the coefficients to x,y data points to calculate the estimated values of W-HAT. Plotting the data points, will give you enough to construct the elevation contours of the dome graphed in Figure 12.3.

12.2 QUADRATIC REGRESSION — CHAMBISHI MINE

Worksheet C12B.WK1 is constructed similar to worksheet C12A.WK1. The regression output is similar. The smaller value of r-squared, 0.60, indicates that less of the variability (about 36 percent) is explained by the model; however, I believe that this moderate smoothing may provide appropriate block estimates. Part 3 gives the predicted value of Cu for each block; I arranged the table for ease in copying the center coordinates of each block and the formula.

Figure 12.4 is a 1-2-3 plot of the blocks, scaled approximately to the scale of Figure 12.1 by adjusting the column widths using the /Work-

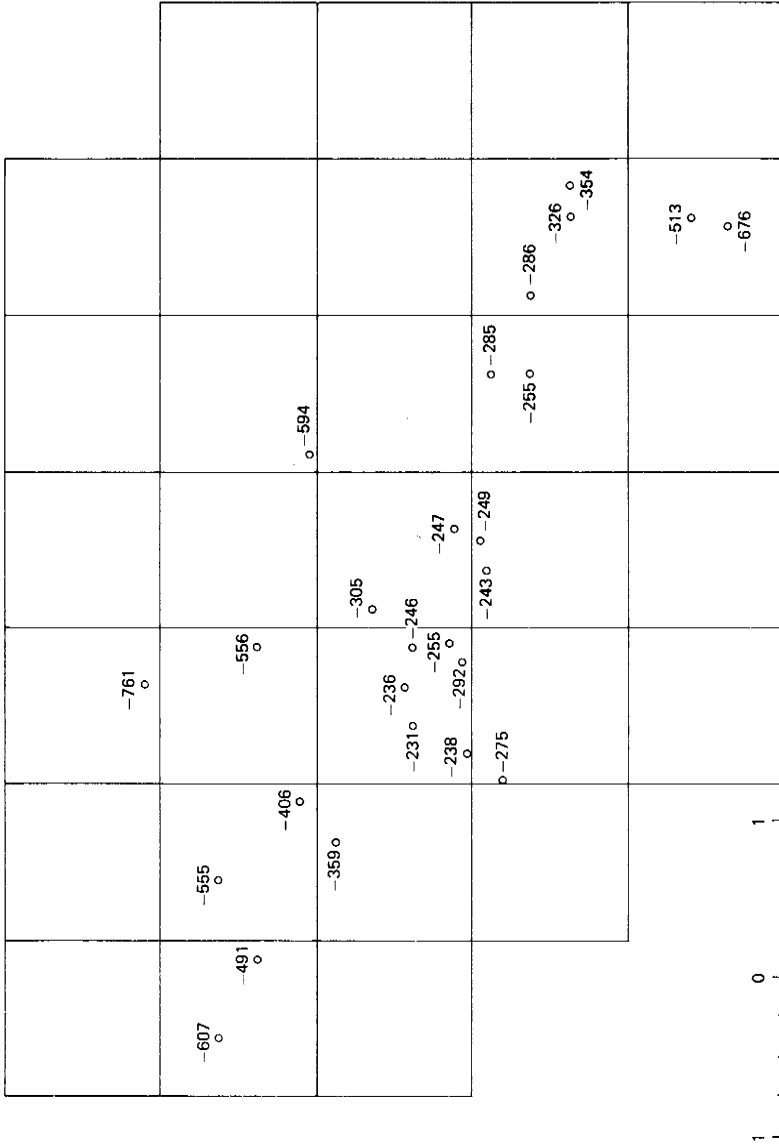


Figure 12.2 Elevations of Weber Sandstone in 29 wells in Rangely oil field, Colorado

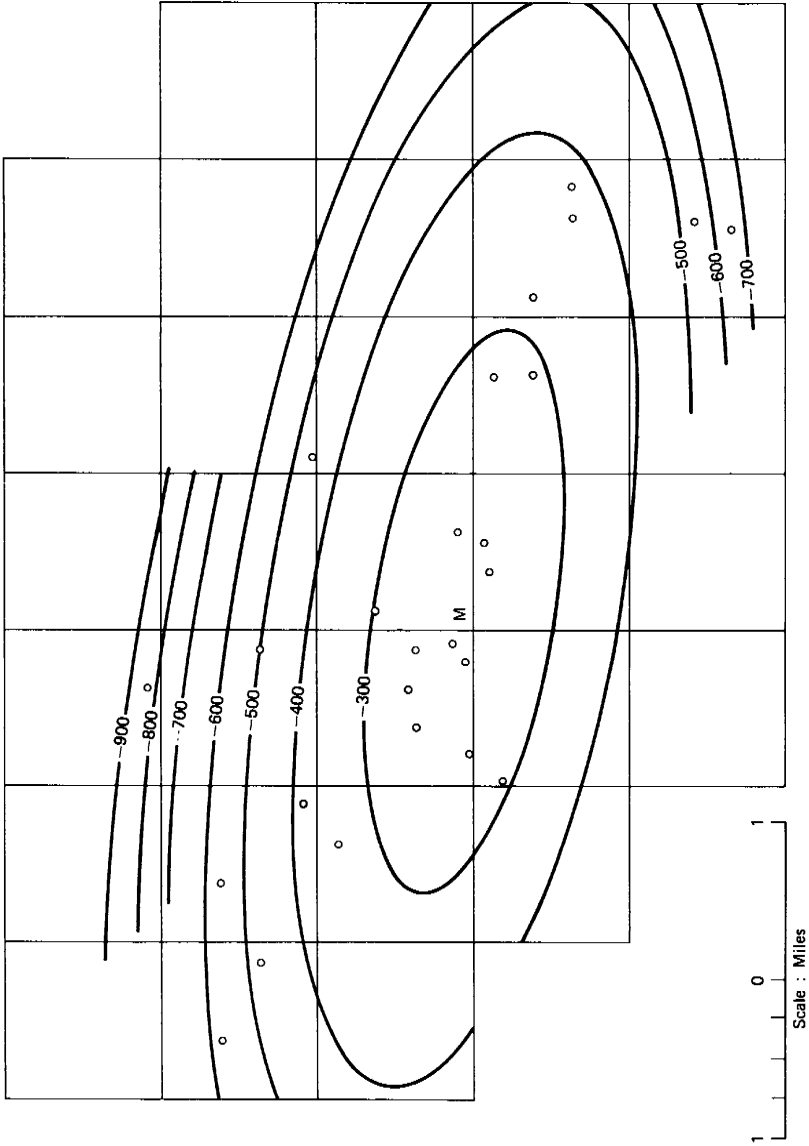


Figure 12.3 Structure contour map on top of Weber Sandstone prepared by quadratic regression analysis.

1.8	2.1	2.5	2.9	3.2	3.5	3.7	3.9	4.1	4.2	4.3	4.0	4.0	3.8	3.8	3.4	3.4	3.1	3.1	3.0	2.7	2.5	2.3	3.4	3.1	2.8	2.6	2.3
1.9	2.2	2.5	2.8	3.1	3.4	3.6	3.7	3.9	4.0	4.0	3.8	3.8	3.8	3.5	3.5	3.4	3.2	3.1	3.0	2.7	2.5	2.3	3.5	3.3	3.0	2.7	2.3
	2.3	2.6	2.8	3.0	3.2	3.3	3.5	3.5	3.5	3.5	3.5	3.5	3.5	3.4	3.4	3.2	3.1	2.9	2.8	2.6	2.5	2.3	3.2	3.1	2.9	2.7	2.3
	2.3	2.6	2.8	3.0	3.1	3.2	3.3	3.4	3.4	3.4	3.4	3.4	3.4	3.4	3.2	3.0	2.9	2.8	2.6	2.5	2.3	3.3	3.1	2.9	2.7	2.3	
	2.4	2.6	2.8	3.0	3.1	3.2	3.2	3.2	3.2	3.2	3.2	3.2	3.2	3.2	3.2	3.0	2.9	2.8	2.6	2.5	2.3	3.3	3.1	2.9	2.7	2.3	
	2.5	2.7	2.8	2.9	3.0	3.0	3.0	3.1	3.1	3.1	3.1	3.1	3.1	3.1	3.0	2.9	2.8	2.7	2.6	2.5	2.3	3.3	3.1	2.9	2.7	2.3	
	2.6	2.7	2.8	2.9	2.9	2.9	2.9	2.9	2.9	2.9	2.9	2.9	2.9	2.9	2.9	2.9	2.8	2.7	2.6	2.5	2.3	3.3	3.1	2.9	2.7	2.3	

Figure 12.4 Plot for Chambishi problem

sheet-Global-Width command. To make this plot, I first copied Part 3 to arrange the columns side by side, and then used the command / Transpose to interchange rows and columns.

Part 4 calculates a confidence interval for the population mean, and Part 5 is the table of labels.

12.3 OTHER WAYS TO ESTIMATE BLOCK GRADES

Many other ways have been devised to estimate block grades. Peters (1987, p. 489) explains weighting by the inverse distance squared, and Hayes and Koch (1984) provide a computer program for this method. Table 12.1 gives several grade estimates made in different ways for the Chambishi mine. In a classic paper, Tukey (1948) showed that any reasonable weighting system will yield similar results overall, although values for individual blocks may differ. (Technically, Tukey showed that weights can change much without changing the standard error of an estimate.) Not until the ore is mined out, and then only if reliable mining records have been kept, do we know for any deposit whether a particular evaluation scheme is "best."

You may want to experiment with estimating block values in various ways and entering your results in column E of Part 3 of worksheet C12B.WK1. If you then compute the mean, and confidence intervals, you can compare them with those in Table 12.1. I think that you will be surprised with how close they are, at least this is the experience of students at the University of Georgia through several years.

Table 12.1 Chambishi mine, Zambia. Grade estimates from 38 boreholes made in various ways. Values are in percent copper.

Source of Estimate	Mean	Standard Deviation	90% Confidence Interval	
			Lower	Upper
Arithmetic average of borehole	2.69	1.09	2.39	2.98
Geologist's estimate	2.94	0.38	2.84	3.03
An early computer contouring program	3.03	0.82	2.90	3.15
Quadratic regression	3.00	0.74	2.89	3.12
Inverse distance				
3 neighbors	3.05	0.66	2.94	3.15
4 neighbors	3.03	0.64	2.93	3.13
Inverse distance squared				
3 neighbors	3.05	0.70	2.94	3.16
4 neighbors	3.04	0.69	2.93	3.15

Worksheet C12A Quadratic regression of Rangely data

1 18 Feb 88, file C12A.WK1 on disks GSK 059-061

2
3 Worksheet C12A.WK1. Quadratic regression of the Rangely data

4
5 Part 1. Table of data for the 26 wells. W, X, and Y are from
6 the original data file (Koch and Link, 1971, p. 34); X**2,
7 Y**2, and XY are computed as explained in the text.

8	9	10	11	12	13	14	15	16	17	18	19	20	21	22	23	24	25	26	27	28	29
C	D	E	F	G	H	W	X	Y	X**2	Y**2	XY										
						607	4.35	3.64	18.9225	13.2496	15.8340										
						491	4.87	3.38	23.7169	11.4244	16.4606										
						555	5.38	3.64	28.9444	13.2496	19.5832										
						406	5.88	3.10	34.5744	9.6100	18.2280										
						556	6.86	3.36	47.0596	11.2896	23.0496										
						761	6.62	4.10	43.8244	16.8100	27.1420										
						594	8.10	3.02	65.6100	9.1204	24.4620										
						359	5.61	2.86	31.4721	8.1796	16.0446										
						231	6.36	2.38	40.4496	5.6644	15.1368										
						236	6.61	2.43	43.6921	5.9049	16.0623										
						246	6.87	2.38	47.1969	5.6644	16.3506										
						238	6.20	2.01	38.4400	4.0401	12.4620										
						292	6.80	2.05	46.2400	4.2025	13.9400										
						255	6.90	2.13	47.6100	4.5369	14.6970										
						305	7.16	2.62	51.2656	6.8644	18.7592										
						247	7.64	2.10	58.3696	4.4100	16.0440										
						275	6.02	1.81	36.2404	3.2761	10.8962										

Worksheet C12A (continued)

30	243	7.38	1.90	54.4644	3.6100	14.0220
31	249	7.56	1.94	57.1536	3.7636	14.6664
32	255	8.63	1.63	74.4769	2.6569	14.0669
33	285	8.63	1.88	74.4769	3.5344	16.2244
34	286	9.13	1.62	83.3569	2.6244	14.7906
35	326	9.63	1.37	92.7369	1.8769	13.1931
36	354	9.84	1.37	96.8256	1.8769	13.4808
37	676	9.59	0.35	91.9681	0.1225	3.3565
38	513	9.62	0.60	92.5444	0.3600	5.7720
39						
40						
41	Part 2.	Regression output				
42						
43						
44	Constant		2628.752			
45	Std Err of Y Est		47.59915			
46	R Squared		0.928588			
47	No. of Observations		26			
48	Degrees of Freedom		20			
49						
50	X Coefficient (s)	-392.922	-968.661	20.72697	153.8660	48.16129
51	Std Err of Coef.	204.040	270.2769	10.29091	19.93190	25.20694
52						
53						

Worksheet C12A (continued)

Part 3. List of W-hat values for X,Y points

	W	X	Y
54			
55			
56			
57			
58	1389	4	0
59	767	4	1
60	452	4	2
61	445	4	3
62	746	4	4
63	1355	4	5
64	1182	5	0
65	608	5	1
66	342	5	2
67	384	5	3
68	733	5	4
69	1390	5	5
70	1017	6	0
71	492	6	1
72	273	6	2
73	363	6	3
74	760	6	4
75	1466	6	5
76	894	7	0
77	416	7	1
78	246	7	2
79	384	7	3
80	830	7	4
81	1583	7	5
82	812	8	0

Worksheet C12A (continued)

83	382	8	1
84	261	8	2
85	447	8	3
86	940	8	4
87	1742	8	5
88	771	9	0
89	390	9	1
90	316	9	2
91	551	9	3
92	1092	9	4
93	1942	9	5
94	772	10	0
95	439	10	1
96	414	10	2
97	696	10	3
98	1286	10	4
99	2184	10	5
100	815	11	0
101	530	11	1
102	552	11	2
103	883	11	3
104	1521	11	4
105	2467	11	5
106	-----		
107			
108			

Worksheet C12A *(continued)*

	Part 4.	Table of labels
109		
110		
111	A	F44
112	B.1	E50
113	B.2	F50
114	B.3	G50
115	B.4	H50
116	B.5	I50

Worksheet C12B Quadratic regression of Chambishi data

1 22 Feb 88, file C12B.WK1 on disks GSK 059-061
 2
 3 Worksheet C12B.WK1. Quadratic regression of the Chambishi data
 4

5 Part 1. Table of data for the 38 drillholes. Cu, X, and Y
 6 values from the original records; X**2, Y**2, and XY are
 7 computed as explained in the text.
 8

9	C	D	E	F	G	H
10	-----					
11	Cu	X	Y	X**2	Y**2	XY
12	-----					
13	4.05	632	412	399,424	169,744	260,384
14	2.06	636	577	404,496	332,929	366,972
15	2.93	763	296	582,169	87,616	225,848
16	3.88	777	485	603,729	235,225	376,845
17	4.49	928	302	861,184	91,204	280,256
18	3.09	955	555	912,025	308,025	530,025
19	3.09	1000	720	1,000,000	518,400	720,000
20	0.00	625	900	390,625	810,000	562,500
21	1.41	855	945	731,025	893,025	807,975
22	2.97	1105	600	1,221,025	360,000	663,000
23	2.43	1165	820	1,357,225	672,400	955,300
24	3.94	1230	368	1,512,900	135,424	452,640
25	2.28	1310	600	1,716,100	360,000	786,000
26	3.74	1375	298	1,890,625	88,804	409,750
27	2.82	1427	635	2,036,329	403,225	906,145
28	5.64	1450	365	2,102,500	133,225	529,250
29	3.90	1530	385	2,340,900	148,225	589,050

Worksheet C12B (continued)

30	2.84	1490	995 2,220,100	990,025	1,482,550
31	3.43	1645	510 2,706,025	260,100	838,950
32	2.64	1675	297 2,805,625	88,209	497,475
33	2.99	1735	375 3,010,225	140,625	650,625
34	2.29	1772	740 3,139,984	547,600	1,311,280
35	2.73	1822	250 3,319,684	62,500	455,500
36	2.04	1865	410 3,478,225	168,100	764,650
37	3.27	1895	675 3,591,025	455,625	1,279,125
38	2.43	1975	327 3,900,625	106,929	645,825
39	2.08	2140	370 4,579,600	136,900	791,800
40	2.22	2147	465 4,609,609	216,225	998,355
41	2.85	2195	595 4,818,025	354,025	1,306,025
42	2.28	2303	450 5,303,809	202,500	1,036,350
43	2.731	2400	650 5,760,000	422,500	1,560,000
44	2.83	2305	250 5,313,025	62,500	576,250
45	2.93	2475	240 6,125,625	57,600	594,000
46	2.20	2475	305 6,125,625	93,025	754,875
47	0.00	2592	300 6,718,464	90,000	777,600
48	2.12	2635	357 6,943,225	127,449	940,695
49	1.13	2670	430 7,128,900	184,900	1,148,100
50	1.30	2775	470 7,700,625	220,900	1,304,250
51					
52					
53					

Worksheet C12B (continued)

```

54 Part 2. Regression Output
55 -----
56 Constant                6.6027
57 Std Err of Y Est        0.7406
58 R Squared                0.60
59 No. of Observations     38
60 Degrees of Freedom      32
61
62 X Coefficient (s)       -3.48E-05 -1.01E-02 -8.27E-07  1.52E-06  4.28E-06
63 Std Err of Coef.       1.50E-03  5.04E-03  3.43E-07  3.25E-06  1.36E-06
64 -----
65
66

```

Part 3. List of Cu values for each block

```

68 -----
69 Cu      x      y
70 -----
71 2.13    2325    275
72 2.51    2175    275
73 2.86    2025    275
74 3.18    1875    275
75 3.45    1725    275
76 3.69    1575    275
77 3.89    1425    275
78 4.05    1275    275
79 4.18    1125    275
80 4.27     975    275
81 1.77    2475    325
82 2.16    2325    325

```


Worksheet C12B (continued)

83	2.52	2175	325
84	2.84	2025	325
85	3.12	1875	325
86	3.36	1725	325
87	3.57	1575	325
88	3.74	1425	325
89	3.87	1275	325
90	3.96	1125	325
91	4.02	975	325
92	4.04	825	325
93	1.85	2475	375
94	2.21	2325	375
95	2.53	2175	375
96	2.82	2025	375
97	3.07	1875	375
98	3.28	1725	375
99	3.45	1575	375
100	3.59	1425	375
101	3.69	1275	375
102	3.75	1125	375
103	3.78	975	375
104	3.76	825	375
105	2.26	2325	425
106	2.56	2175	425
107	2.81	2025	425
108	3.03	1875	425
109	3.20	1725	425
110	3.35	1575	425
111	3.45	1425	425

Worksheet C12B (continued)

112	3.52	1275	425
113	3.55	1125	425
114	3.54	975	425
115	3.50	825	425
116	3.42	675	425
117	2.33	2325	475
118	2.58	2175	475
119	2.81	2025	475
120	2.99	1875	475
121	3.14	1725	475
122	3.25	1575	475
123	3.32	1425	475
124	3.36	1275	475
125	3.35	1125	475
126	3.31	975	475
127	3.24	825	475
128	3.12	675	475
129	2.39	2325	525
130	2.62	2175	525
131	2.81	2025	525
132	2.96	1875	525
133	3.08	1725	525
134	3.16	1575	525
135	3.20	1425	525
136	3.20	1275	525
137	3.17	1125	525
138	3.09	975	525
139	2.99	825	525
140	2.84	675	525

Worksheet C12B (continued)

141	2.47	2325	575
142	2.67	2175	575
143	2.82	2025	575
144	2.94	1875	575
145	3.03	1725	575
146	3.07	1575	575
147	3.08	1425	575
148	3.05	1275	575
149	2.99	1125	575
150	2.88	975	575
151	2.74	825	575
152	2.56	675	575
153	2.56	2325	625
154	2.72	2175	625
155	2.84	2025	625
156	2.93	1875	625
157	2.98	1725	625
158	3.00	1575	625
159	2.97	1425	625
160	2.91	1275	625
161	2.81	1125	625
162	2.68	975	625
163	2.51	825	625
164	2.30	675	625
165	2.93	1875	675
166	2.95	1725	675
167	2.93	1575	675
168	2.87	1425	675
169	2.78	1275	675

Worksheet C12B (continued)

170	2.65	1125	675
171	2.48	975	675
172	2.28	825	675
173	2.92	1725	725
174	2.87	1575	725
175	2.78	1425	725
176	2.65	1275	725
177	2.49	1125	725
178	2.29	975	725
179	2.54	1275	775
180	2.34	1125	775
181	2.43	1275	825
182	2.20	1125	825
183			
184			
185			

Part 4. Calculation of a confidence interval for the population mean (Notation from Worksheet C2A.WK1)

188	Value	Item
189		
190		
191	112	n
192		
193	3.00	Mean
194		
195	0.74	Standard deviation
196		
197	1.658	t (5%)
198		

Worksheet C12B (continued)

199	0.12	d
200		
201	3.12	UCL
202		
203	2.89	LCL
204		
205		
206	Part 5.	Table of labels
207		
208	A	F56
209	B.1	E62
210	B.2	F62
211	B.3	G62
212	B.4	H62
213	B.5	I62
214	D	C199
215	N	C191
216	S	C195
217	T	C197
218	W.BAR	C193

CHAPTER 13

Ore Concentration and Smelter Settlement

The worksheets in this chapter calculate the concentration of ore after mining and the payments made by smelters for the concentrated products. Two of the worksheets are taken from the comprehensive model in the next chapter; they are separated here so that you can easily work with and modify them.

13.1 ORE CONCENTRATION

After mining, ores are generally separated in a plant termed a mill or a concentrator into two types of products: one or more concentrates containing most of the valuable minerals, and tailings containing most of the other minerals. Because the minerals are processed in an industrial operation, their recovery is imperfect. A materials balance accounts for the ore, named the heads, entering the mill, and the concentrates and tailings leaving the mill.

Worksheet C13A.WK1 is a simplified materials balance for the concentrator at the El Indio mine in Chile (Smith, 1986). It is convenient to consider the concentration of 100 tonnes of 5 percent copper ore; these

numbers are entered in cells D12 and E12. Multiplying tonnes of heads (D12) by grade (E12) gives the contents, in tonnes of copper (F12); we can express this fundamental equation as

$$T * G = C,$$

where T is tonnes, G is grade, and C is the quantity (also named the contents) of contained metal. Multiplying F12 by the percentage recovery of copper (94 percent) gives the contents of copper in the concentrate in F13. Because the concentrate grade is more or less a constant, being largely a function of the efficiency of the concentrator, the tonnes of concentrate in D13 is calculated by dividing F13 by E13 (with a constant to express percentage). Finally, the tonnes and copper contents of the tailings are determined by subtracting the concentrate values from the heads values, and the tailings grade is calculated from the fundamental equation.

Worksheet C13B.WK1 is similar to the previous worksheet with the inclusion of gold, which was omitted for simplicity from the first worksheet. The gold adds no significant additional tonnage to the concentrates.

Worksheet C13C.WK1 is the materials balance that Peters (1987, p. 252) uses for his comprehensive model; the original data, for the Bunker Hill mine in Idaho, are from Sather and Prindle (1970). Two concentrates are produced, a lead concentrate containing most of the lead and silver, and a zinc concentrate containing most of the zinc and some of the silver. In this worksheet, the rows and columns are transposed for convenience in printing. I assumed that the percentages of lead and zinc in the concentrates and the percentage recoveries of all three metals are constants. Therefore, the concentrate weights are linear functions of the grades of lead and zinc in the mined ore. Other assumptions are possible, but making them would require a detailed knowledge of the particular mineralogy and the concentration scheme.

13.2 CALCULATION OF GRADE FOR A MINED-OUT DEPOSIT

An interesting use of the materials balance in the previous section combines it with Section 11.2 on calculating the specific gravity of ore. Here is a problem for you to consider.

In examining the Veta Grande lead mine in Mexico, you learn from old maps, checked by your observations and measurements, that about 10,000 cubic meters of ore was mined. The record of concentrates shipped is as follows:

<u>Shipments</u>	<u>Tonnes</u>	<u>Assay</u>
Lead concentrate	4,000	65% Pb

The problem is to estimate the grade of ore as mined, because there is a potential for additional ore of a similar grade at depth.

To simplify the problem, we may assume that (1) the lead concentrate consisted entirely of galena plus pyrite, (2) the gangue rejected in milling consisted entirely of limestone, and (3) the recovery of metals was perfect.

You have at hand data on the specific gravities of the minerals and limestone, and the atomic weights of the minerals.

Worksheet C13D.WK1 solves this problem. Part 1 provides constants; the percent of lead in galena (row 13) is obtained from the atomic weights for lead and sulfur.

Part 2 does the calculations. Working from the given data, we first determine the tonnes of ore mined (row 27), equivalent to a specific gravity of 2.89, and then the lead grade (row 29).

This worksheet is an example of problems usually encountered in economic geology. Generally, the problem resolves itself into selecting

the pertinent data and using the basic materials balance relations to determine the required solution. For instance, we might need to work backward from the tailings to determine tonnage and grade of ore as mined.

13.3 SMELTER SETTLEMENT

Concentrates are sold to a smelter, according to negotiated contracts. Peters (1987, p. 255-256) provides “abbreviated examples of typical ‘open’ smelter schedules, the type furnished to independent shippers of ores and concentrates by custom metallurgical plants.” Peters does the calculations (1987, p. 257) for the Bunker Hill ore of worksheet C13C.WK1. His calculations, which are self-explanatory, are given in worksheet C13E.WK1.

Worksheet C13A Materials balance for El Indio mine

1 18 Feb 88, File C13A.WK1 on disks GSK 059-061

2
3 Worksheet C13A.WK1 Materials balance for the El Indio mine

4
5 Part 1. Materials balance for a copper ore

6
7 C D E F

8 -----

9 Material	Tons	Grade	Contents
		Cu, %	Cu, T
10			
11			
12 Heads	100.0	5	5.00
13 Concentrates	14.7	32	4.70
14 Tails	85.3	0.352	0.30
15			

16 -----

17 Cu recovery is 94 percent.

18
19

Worksheet C13A (continued)

20 Part 2. Table of labels

21			
22	CON.TONS		D13
23	CU.%.IN.CON		E13
24	CU.%.IN.HEADS		E12
25	CU.%.IN.TAILS		F14
26	CU.T.IN.CON		F13
27	CU.T.IN.HEADS		F12
28	CU.T.IN.TAILS		F14
29	HEADS.TONS		D12
30	TAILS.TONS		D14

Worksheet C13B Materials balance for El Indio mine with gold included

1 18 Feb 88, File C13B.WK1 on disks GSK 059-061 2
 3 Worksheet C13B.WK1. Materials balance for the El Indio mine
 4 with gold included
 5
 6

7 Part 1. Materials balance for a copper, gold ore
 8

C	D	E	F	G	H
Material	Tons	Grade Au,g/T	Grade Cu,%	Contents Au,g	Contents Cu,T
Heads	100.0	18	5.00	1800	5
Concentrates	14.7	83	32.00	1278	4.7
Tails	85.3	6.119	0.35	522	0.3

19 Au recovery is 71 %. Cu recovery is 94 %.

20
 21

Worksheet C13A (continued)

Part 2. Table of labels

22		
23		
24	AU.G.IN.CON	G15
25	AU.G.IN.HEADS	G14
26	AU.G.IN.TAILS	G16
27	AU.G.P.T.CON	E15
28	AU.G.P.T.HEADS	E14
29	AU.G.P.T.TAILS	E16
30	CON.TONS	D15
31	CU.%.IN.CON	F15
32	CU.%.IN.HEADS	F14
33	CU.%.IN.TAILS	F16
34	CU.T.IN.CON	H15
35	CU.T.IN.HEADS	H14
36	CU.T.IN.TAILS	H16
37	HEADS.TONS	D14
38	TAILS.TONS	D16

Worksheet C13C Materials balance for Bunker Hill mine

1	22 Feb 88, file C13C.WK1 on disks GSK 059-061					
2						
3	Worksheet C13C.WK1. Materials balance for the Bunker Hill mine					
4						
5	Part 1. Detail of materials balance					
6	(from Peters, 1987, p. 252)					
7		D	E	F	G	
8						
9		Pb	Zn	Tail-	Heads	
10		Conc.	Conc.	ings		
11						
12	Assays					
13	Percent weight	8.84	7.85	83.31	100.00	
14	Ag, oz/mt	44.57	3.13	0.19	4.34	
15	Ag, g/mt	1385.99	97.25	6.05	135.1	
16	Pb, %	66.59	1.18	0.25	6.18	
17	Zn, %	4.98	54.21	0.25	4.90	
18						
19	Contents					
20	Ag, grams			504.0	13510.0	
21	Ag, grams in Pb concentrate	12245.5				
22	Ag, grams in Zn concentrate		763.5			
23	Pb, kilograms	588.3	9.3	20.7	6659.0	
24	Zn, kilograms	44.0	425.6	20.8	498.0	
25						
26	Percentage distribution					
27	Ag	90.62	5.65	3.73	100.00	
28	Pb	95.16	1.50	3.34	100.00	
29	Zn	8.97	86.78	4.25	100.00	

Worksheet C13C (continued)

30		
31		
32		
33		
34	Part 2. Table of labels	
35		
36	% WT.PB.CON	D13
37	% WT.TAILS	F13
38	% WT.ZN.CON	E13
39	AG.G.IN.HEADS	G20
40	AG.G.IN.PB.CON	D21
41	AG.G.IN.TAILS	F20
42	AG.G.IN.ZN.CON	E22
43	AG.G.P.T.HEADS	G15
44	AG.G.P.T.PB.CON	D15
45	AG.G.P.T.TAILS	F15
46	AG.G.P.T.ZN.CON	E15
47	AG.REC.PB.CON	D27
48	AG.REC.ZN.CON	E27
49	HEADS.TONS	G13
50	PB.% IN.HEADS	G16
51	PB.% IN.PB.CON	D16
52	PB.% IN.TAILS	F23
53	PB.% IN.ZN.CON	E16
54	PB.% REC.PB.CON	D28
55	PB.% REC.ZN.CON	E28
56	PB.KG.IN.HEADS	G23
57	PB.KG.IN.PB.CON	D23
58	PB.KG.IN.ZN.CON	E23

Worksheet C13C (continued)

59	ZN.%.IN.HEADS	G17
60	ZN.%.IN.PB.CON	D17
61	ZN.%.IN.ZN.CON	E17
62	ZN.%.REC.PB.CON	D29
63	ZN.%.REC.ZN.CON	E29
64	ZN.KG.IN.HEADS	G24
65	ZN.KG.IN.PB.CON	D24
66	ZN.KG.IN.TAILS	F24
67	ZN.KG.IN.ZN.CON	E24

Worksheet C13D Estimating lead grade for prospect

1 18 Feb 88, file C13D.WK1 on disks GSK 059-061
 2
 3 Worksheet C13D.WK1. Estimating lead grade for a prospect
 4
 5 Part 1. Constants
 6 -----
 7 Lead, % in concentrate 65
 8 Lead, atomic weight 207.2
 9 Sulfur, atomic weight 32.06
 10 Lead, % in galena 86.6
 11 Galena, specific gravity 7.5
 12 Pyrite, specific gravity 5.0
 13 Limestone, specific gravity 2.65
 14 -----
 15
 16
 17 Part 2. Calculations
 18 -----
 19 Material Tons Cubic Meters
 20 -----
 21 Volume of ore mined 10,000
 22 Lead concentrate produced 4,000
 23 Lead 2,600
 24 Galena 3,002 400
 25 Pyrite 998 200
 26 Limestone 24,910 9,400
 27 Weight of ore mined 28,910
 28 -----
 29 Grade of lead 8.99

Worksheet C13D (continued)

30	-----	
31		
32		
33	Part 3. Table of labels	
34		
35	GALENA.S.G.	D11
36	GALENA.TONS	D24
37	GALENA.VOLUME	E24
38	LIMEST.S.G.	D13
39	LIMEST.TONS	D26
40	LIMEST.VOLUME	E26
41	ORE.VOLUME	E21
42	ORE.WEIGHT	D27
43	PB.%.IN.CON	D7
44	PB.%.IN.GALENA	D10
45	PB.ATOMIC.WT	D8
46	PB.CON.TONS	D22
47	PB.TONS	D23
48	PYRITE.S.G.	D12
49	PYRITE.TONS	D25
50	PYRITE.VOLUME	E25
51	S.ATOMIC.WT	D9

Worksheet C13E Smelter settlement

1 18 Feb 88, file C13.E.WK1 on disks GSK 059-061

2
3 Worksheet C13E.WK1. Smelter settlement

4
5 Part 1. Concentrate grades (from worksheet C13C.WK1)

6	C	D	E	F
7	-----	-----	-----	-----
8	Material	Ag,g/T	Pb,%	Zn,%
9	-----	-----	-----	-----
10	Lead concentrate	1386.0	66.6	5.0
11	Zinc concentrate	97.3	1.2	54.2
12	-----	-----	-----	-----
13	-----	-----	-----	-----
14	-----	-----	-----	-----
15	-----	-----	-----	-----

Worksheet C13E (continued)

16 Part 2. Smelter settlements
 17 (from Peters, 1987, p. 257)

	C	D
	Item	Example
23	Metal prices	
24	Silver	
25	\$ per troy ounce	6.50
26	\$ per gram	0.21
27	Lead	
28	\$ per pound	0.20
29	\$ per kilogram	0.44
30		
31	Zinc	
32	\$ per pound	0.45
33	\$ per kilogram	0.99
34		
35	I. Lead concentrate	
36		
37	Moisture, %	5
38	Freight charge, \$/tonne	10.00
39	Freight charge, \$/dry tonne	10.53
40		
41	Payments	
42	Silver	472.78
43	Lead	261.93
44		210.86

Worksheet C13E (continued)

45	Deductions	
46	Treatment charge	70.00
47	Zn penalty	0.00
48		
49	Value at mine, per dry tonne	392.26
50		
51		
52	II. Zinc concentrate	
53	-----	
54	Moisture, %	6.00
55	Freight charge, \$/tonne	12.00
56	Freight charge, \$/dry tonne	12.77
57		
58	Payments	420.64
59	Silver	10.27
60	Lead	0.00
61	Zinc	410.37
62		
63	Deductions	
64	Treatment charge	156.00
65		
66	Value at mine, per dry tonne	251.88
67		
68		

Worksheet C13E (continued)

Part 3. Table of labels		
69		
70		
71	AG.!.PER.GRAM	D26
72	AG.G.P.T.PB.CON	D11
73	AG.G.P.T.ZN.CON	D12
74	AG.PAY.PB.CON	D42
75	AG.PAY.ZN.CON	D59
76	FREIGHT.PB.CON	D39
77	FREIGHT.ZN.CON	D56
78	PB.!.PER.KILO	D29
79	PB.%.IN.PB.CON	E11
80	PB.%.IN.ZN.CON	E12
81	PB.CON.PRICE	D49
82	PB.CON.TREAT	D46
83	PB.PAY.PB.CON	D43
84	PB.PAY.ZN.CON	D60
85	TOT.PAY.PB.CON	D41
86	TOT.PAY.ZN.CON	D58
87	ZN.!.PER.KILO	D33
88	ZN.%.IN.PB.CON	F11
89	ZN.%.IN.ZN.CON	F12
90	ZN.CON.TREAT	D64
91	ZN.PAY.ZN.CON	D61
92	ZN.PENALTY	D47

CHAPTER 14

Peters's Model for Mineral Property Evaluation

As a financial model for mine evaluation, I use the one presented by Peters (1987, p. 568-570) in a table entitled "Steps in estimating profitability of an undeveloped mineral property," together with accompanying tables from Peters's book. This model presents up-to-date thinking of an expert as well as recent costs.

The Lotus 1-2-3 worksheet for the model is file C14A.WK1. Except for round-off errors, the results match those of Peters.

14.1 THE BASIC MODEL

The basic 1-2-3 model is in four parts, corresponding with Peters's format. These four parts are self-explanatory.

Part 5 is a table of cash flows derived from the cash flows for years 1 to 3, labeled !CASH.FLOW.1, !CASH.FLOW.2, and !CASH.FLOW.3, from row 99. The cash flows for years 4 to 14 are the same as that for year 3, because Peters makes no provision for return of working capital in year 14. Years 15 to 20 are provided for your convenience in changing

the model, if you wish. These cash flows define the label, ANN.CASH.FLOWS, which is the range D136..D155, used to calculate DCFROR, in cell D99.

Part 6 is the materials balance table from Peters (1987, p. 252). In constructing this table, I assumed that the percentages of lead and zinc in the concentrates and the percentage recoveries of all three metals are constants. Therefore, the concentrate weights are taken as linear functions of the grade of lead and zinc in the ore as mined. Other assumptions are possible.

Part 6 is the calculation of percentage depletion from Peters (1987, p. 257), and Part 8 is the net smelter return calculation from Peters (1987, p. 257).

14.2 USING THE MODEL

Using the model will demonstrate to you the outstanding benefit of 1-2-3 for a relatively complex worksheet. By entering new values in selected cells, recalculating, and graphing the results, you can determine quickly how changes affect a particular variable, such as DCFROR.

As an example of the procedure, I changed ore tonnage in place, D13, together with ore grade in place, D15..D17, using increments of plus and minus 10% and 20%. As tonnage increases, grade decreases, with the results graphed in Figures 14.1 and 14.2. Changing tonnage and grade in this way leads to a range in mine life between 10 and 16 years and a cash flow between about \$4 and \$8 million per year (Figure 14.1). DCFROR changes from 31% to 13%, illustrating that maximizing this measure depends on high cash flows early in the life of a project.

You can alter the model to suit yourself. Here are some examples of problems that may interest you:

1. Suppose that the ore tonnage in place, ORE.TONS.PLACE, changes inversely with the percentage extraction, %EXTRACTION, so that as the tonnage decreases the extraction rate increases.

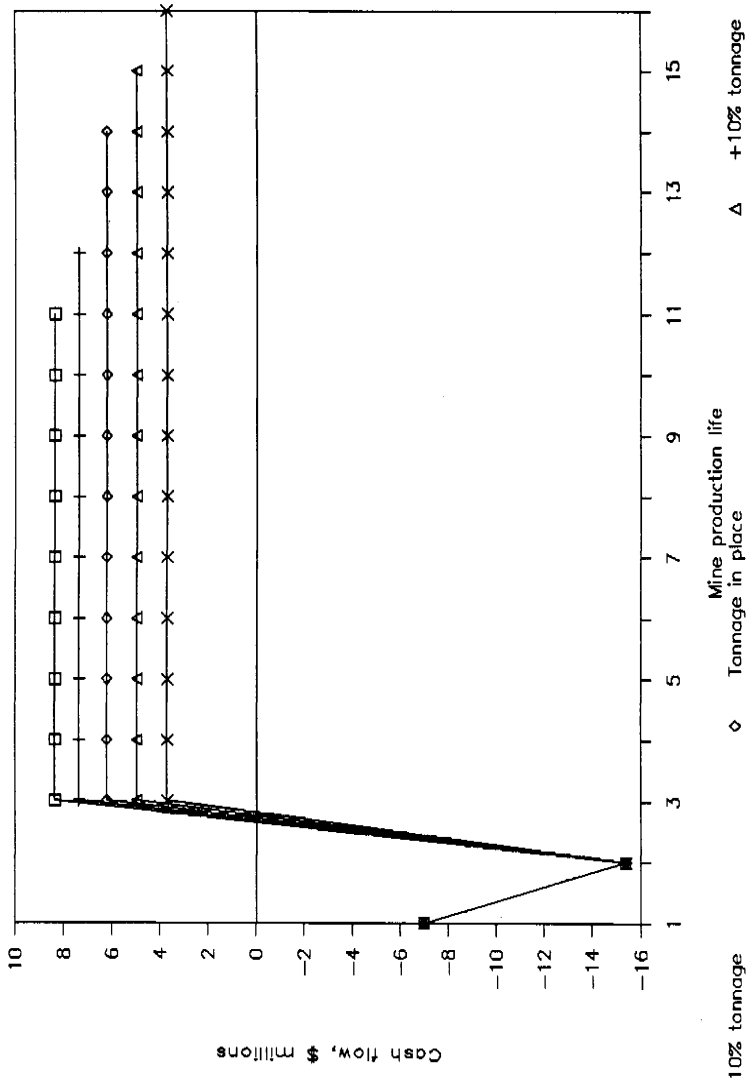


Figure 14.1 Effect of changing tonnage and grade

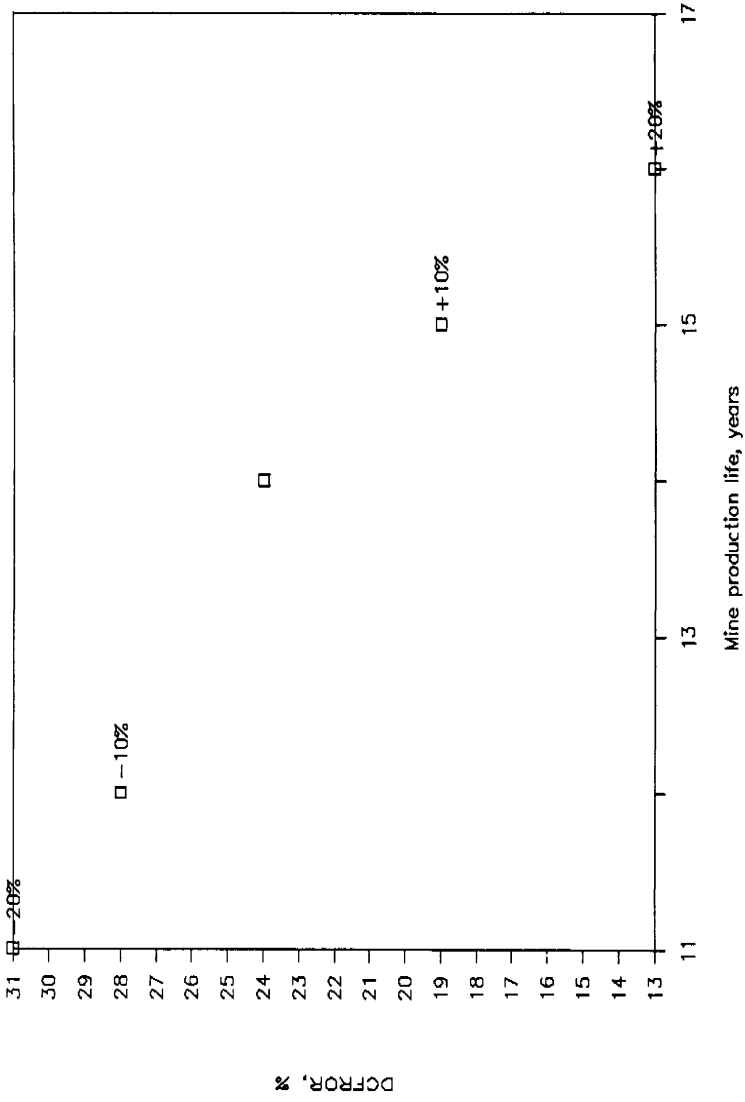


Figure 14.2 Peters's model, effect on DCFROR of changing tonnage

2. What happens if the dilution factor, %DILUTION, changes?
3. What happens if mill recovery, as indicated by the concentration ratios, PB.CONC.RATIO and ZN.CONC.RATIO, increases together with an increase in operating cost, OP.COST.PER.TON, required to achieve this improved mill recovery?
4. Suppose that tonnage, ORE.TONS.PLACE, changes along with mining rate, TONS.PER.DAY, so that the ratio of these two variables remains the same. Also, you may want to change the capital investment values—TOT.CAP.INVEST., DEPREC.YEAR.1, and DEPREC.YEAR.2—assuming that a larger capital investment is needed for a higher mining rate.
5. Suppose that you change income tax, %INCOME.TAX, to correspond to situations where industry lobbyists are doing better or worse with the legislature.
6. Change revenue per ton broken ore, REV.PER.TON.ORE, accompanied by a change in operating costs, OP.COST.PER.TON, because the union has agreed to make wages change according to price.
7. Assume the same situation as in (6), except that the union president is difficult to work with so that wages and therefore operating costs, OP.COST.PER.TON, increase with lower revenue per ton, REV.PER.TON.ORE. At what point do you shut down the whole operation?
8. Assume that the plant is built at a lower price (changing TOT.CAP.INVEST., DEPREC.YEAR.1, and DEPREC.YEAR.2) but that the less expensive plant gets poorer mill recovery, as measured by PB.CONC.RATIO and ZN.CONC.RATIO.

Worksheet C14A Mineral property evaluation

1 24 May 89, file C14A.WK1 on disks GSK 059-061
 2
 3 Worksheet C14A.WK1. Mineral property evaluation
 4 (from Peters, 1987, p. 568-570)
 5
 6
 7 Part 1. Summary data from working estimates
 8
 9 C D
 10 -----
 11 Item Example
 12 -----
 13 Ore tonnage in place, millions 4.2
 14 Ore grade in place, percent
 15 Pb, percent 7.11
 16 Zn, percent 5.64
 17 Ag, g/T 155.4
 18 Percentage extraction 80.0
 19 Dilution factor, percent 15.0
 20 Broken ore tonnage, millions 3.9
 21 Mining grade
 22 Pb, percent 6.18
 23 Zn, percent 4.90
 24 Ag, g/T 135.1
 25 Concentration ratio
 26 Pb concentrate (100/PB%.CONC) 11.32
 27 Zn concentrate (100/ZN%.CONC) 12.74

Worksheet C14A (continued)

28	Operating cost, \$ per ton	\$28.50
29	Mining rate	
30	Tons per day	1000
31	Tons per year (thousands)	320
32	Preproduction devel. period, years	2
33	Total capital investment, \$M	22.4
34	Depreciable portion, \$M	20.0
35	Year 1, \$M	7.0
36	Year 2, \$M	13.0
37	Working capital (12%)	
38	Year 2, \$M	2.4
39	Mine production life	12
40	Revenue per ton broken ore	\$54.43
41	Pb concentrate	34.66
42	Zn concentrate	19.77
43	-----	
44		
45		

Worksheet C14A (continued)

46	Part 2. Fixed (or assigned) parameters		
47	-----		
48	C	D	
49	-----		
50	Item	Example	
51	-----		
52	Equity 100% (unleveraged)		
53	Constant dollars at project start		
54	Royalty, percent	3.0	
55	Percentage depletion (in USA)	19.6	
56	(See Appendix 3)		
57	Federal and state income tax, %	51.0	
58	Depreciation and amortization	straightline	
59	Assigned hurdle rate for DCFROR, %	15.0	
60	-----		
61			
62			

Worksheet C14A (continued)

63 Part 3. Annual estimate of cash flow (\$, Millions)				
		D	E	F
		Year		
	Item	1	2	3-14
64	Revenue			\$17.42
65	Royalty			0.52
66	Gross income			16.90
67	Operating costs			9.12
68	Operating income			7.78
69	Depreciation and amortization			1.67
70	Net income before depletion			6.11
71	Percentage depletion			3.31
72	50% limit			3.05
73	Selected depletion			3.05
74	Taxable income			3.05
75	Income tax			1.56

Worksheet C14A (continued)

90	Net income		1.50
91	Add depreciation and amortization		1.67
92	Add depletion		3.05
93			
94	Operating cash flow		\$6.22
95			
96	Capital expenditure	-7.0	-13.0
97	Working capital		-2.4
98			
99	Cash flow	(\$7.0)	(\$15.4)
100			\$6.22
101			
102			
103	Part 4. Profitability expressions		
104			
105		C	
106			
107		Item	
108		Example	
109	Payback period, years	3.6	
110			
111	DCFROR, percent	24	
112	Present value of income, \$M	\$15.7	
113	P/A(I,N)	3.871	
114	P/F(I,2)	0.652	
115			
116	Present value of capital invest., \$M	\$15.7	
117	P/F(I,1)	0.807	
118	P/F(I,2)	0.652	

Worksheet C14A (continued)

119		
120		
121	Net present value (@ hurdle rate), \$M	\$7.75
122	P/A(H,N)	5.421
123	P/F(H,2)	0.756
124	P/F(H,1)	0.870
125	Present value of capital invest. (\$17.73)	
126	Present value of income	\$25.49
127	-----	
128		
129		

Part 5. Table of cash flows

130			
131			
132			
133			
134			
135			
136			
137			
138			
139			
140			
141			
142			
143			
144			
145			
146			
147			

C

D

Year \$, Millions

1	-7.00
2	-15.40
3	6.22
4	6.22
5	6.22
6	6.22
7	6.22
8	6.22
9	6.22
10	6.22
11	6.22
12	6.22

Worksheet C14A (continued)

176	Ag, grams in Pb concentrate	12245.5			
177	Ag, grams in Zn concentrate		763.5		
178	Pb, kilograms	588.3	9.3	20.7	618.3
179	Zn, kilograms	44.0	425.6	20.8	490.4
180					
181	Percentage distribution				
182	Ag	90.62	5.65	3.73	100.00
183	Pb	95.16	1.50	3.34	100.00
184	Zn	8.97	86.78	4.25	100.00
185					
186					
187					

Part 7. Calculation of percentage depletion
(from Peters, 1987, p. 257)

	C	D	E	F
191				
192				
193				
194				
195				
	Item	Pb Conc.	Zn Conc.	Totals
196	Concentrate selling price, \$/ton	392.26	251.88	
197	Ag, \$	217.31	6.15	223.46
198	Pb, \$	174.94		
199	Zn, \$		245.72	
200	Pb + Zn, \$			420.67
201	Ag, % of net smelter return (NSR)			35
202	Pb + Zn, % of NSR			65
203	Percentage depletion from Ag	5.2		

Worksheet C14A (continued)

204 Percentage depletion from Pb 14.4
 205 Percentage depletion 19.6
 206 -----
 207 -----
 208 -----

209 Part 8. Net smelter return calculation
 210 (from Peters, 1987, p. 257)

211 -----
 212 C D
 213 -----

214 Item Example
 215 -----

216 Metal prices

217 Silver

218 \$ per troy ounce 6.50

219 \$ per gram 0.21

220 Lead

221 \$ per pound 0.20

222 \$ per kilogram 0.44

223 -----

224 Zinc

225 \$ per pound 0.45

226 \$ per kilogram 0.99

227 -----

228 I. Lead concentrate

229 -----

230 Moisture, % 5

231 Freight charge, \$/tonne 10.00

232 Freight charge, \$/dry tonne 10.53

Worksheet C14A (continued)

233			
234	Payments		472.78
235	Silver		261.93
236	Lead		210.86
237			
238	Deductions		
239	Treatment charge	70.00	
240	Zn penalty	0.00	
241			
242	Value at mine, per dry tonne		392.26
243			
244			
245	II. Zinc concentrate		
246	-----		
247	Moisture, %	6.00	
248	Freight charge, \$/tonne	12.00	
249	Freight charge, \$/dry tonne	12.77	
250			
251	Payments		420.64
252	Silver		10.27
253	Lead		0.00
254	Zinc		410.37
255			
256	Deductions		
257	Treatment charge	156.00	
258			
259	Value at mine, per dry tonne		251.88
260			
261			

Worksheet C14A (continued)

262	III, Net smelter return (NSR) on ore		
263	mined		
264	-----		
265	NSR per tonne of ore	54.43	E99
266	Lead concentrates	34.66	F99
267	Zinc concentrates	19.77	F84
268	-----		F85
269			F75
270			F88
271	Part 9. Table of labels		F81
272			F90
273	!CASH.FLOW.2		F94
274	!CASH.FLOW.3		F78
275	!DEPRE.&.AMORT.		F76
276	!DPL.50%.LIMIT		F83
277	!DPL.SELECTED		F72
278	!GROSS.INCOME		F73
279	!INCOME.TAX		F87
280	!NET.BEFORE.DPL		D122
281	!NET.INCOME		D113
282	!OP.CASH.FLOW		
283	!OP.INCOME		
284	!OPERATING.COST		
285	!PCT.DPL		
286	!REVENUE		
287	!ROYALTY		
288	!TAX.INCOME		
289	#P.A.H.N		
290	#P.A.I.N		

Worksheet C14A (continued)

291	#P.F.H.1	D124
292	#P.F.H.2	D123
293	#P.F.I.1	D117
294	#P.F.I.2	D114
295	%DCFROR	D111
296	%DEPLETION	D205
297	%DILUTION	D19
298	%EXTRACTION	D18
299	%HURDLE.RATE	D59
300	%INCOME.TAX	D57
301	%ROYALTY	D54
302	AG.!.PER.GRAM	D219
303	AG.G.IN.HEADS	G175
304	AG.G.IN.PB.CON	D176
305	AG.G.IN.TAILS	F175
306	AG.G.IN.ZN.CON	E177
307	AG.G.P.T.PB.CON	D170
308	AG.G.P.T.TAILS	F170
309	AG.G.P.T.ZN.CON	E170
310	AG.G.PER.T.ORE	D17
311	AG.ORE.MINED	D24
312	AG.PAY.PB.CON	D235
313	AG.PAY.ZN.CON	D252
314	AG.REC.PB.CON	D182
315	AG.REC.ZN.CON	E182
316	ANN.CASH.FLOWS	D136..D155
317	DEPR.CAP.INVEST	D34
318	DEPREC.YEAR.1	D35
319	DEPREC.YEAR.2	D36

Worksheet C14A (continued)

320	FREIGHT.PB.CON	D232
321	FREIGHT.ZN.CON	D249
322	HEADS.TONS	G168
323	LIFE	D39
324	NSR.PB.CONC	D266
325	NSR.TOTAL	D265
326	NSR.ZN.CONC	D267
327	OP.COST.PER.TON	D28
328	ORE.TONS.BROKEN	D20
329	ORE.TONS.PLACE	D13
330	PB%.ORE	D15
331	PB%.ORE.MINED	D22
332	PB.!.PER.KILO	D222
333	PB.%.IN.PB.CON	D171
334	PB.%.IN.TAILS	F178
335	PB.%.IN.ZN.CON	E171
336	PB.%.REC.PB.CON	D183
337	PB.%.REC.ZN.CON	E183
338	PB.CON.TREAT.	D239
339	PB.CONC.PRICE	D242
340	PB.CONC.RATIO	D26
341	PB.CONC.REVENUE	D41
342	PB.KG.IN.HEADS	G178
343	PB.KG.IN.PB.CON	D178
344	PB.KG.IN.ZN.CON	E178
345	PB.PAY.PB.CON	D236
346	PB.PAY.ZN.CON	D253
347	PB.WT.%.OF.CON	D168
348	PV.CAP.INVEST.	D125

Worksheet C14A (continued)

349	PV. INCOME	D126
350	REV. PER. TON. ORE	D40
351	TAILS. WT. %. CON	F168
352	TONS. PER. DAY	D30
353	TONS. PER. YEAR	D31
354	TOT. CAP. INVEST.	D33
355	TOT. PAY. PB. CON	D234
356	TOT. PAY. ZN. CON	D251
357	WORKING. CAPITAL	D38
358	ZN%. ORE	D16
359	ZN%. ORE. MINED	D23
360	ZN. !. PER. KILO	D226
361	ZN. % IN. PB. CON	D172
362	ZN. % IN. ZN. CON	E172
363	ZN. % REC. PB. CON	D184
364	ZN. % REC. ZN. CON	E184
365	ZN. CON. TREAT.	D257
366	ZN. CON. PRICE	D259
367	ZN. CONC. RATIO	D27
368	ZN. CONC. REVENUE	D42
369	ZN. KG. IN. HEADS	G179
370	ZN. KG. IN. PB. CON	D179
371	ZN. KG. IN. TAILS	F179
372	ZN. KG. IN. ZN. CON	E179
373	ZN. PAY. ZN. CON	D254
374	ZN. PENALTY	D240
375	ZN. WT. %. OF. CON	E168

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