Some practical aspects of the use of lognormal models for confidence limits and block distributions in South African gold mines

D.G. KRIGE
Private Consultant, Florida Hills, Gauteng, South Africa

For the purpose of determining confidence limits for ore grade estimates, an extensive simulated data base was used, which included 1728 ore blocks with 'actual' grades and the corresponding estimates based on ordinary and simple kriging. The practical results show that conditional unbiasedness is essential and that acceptable confidence limits can then be set using Normal theory for the simulated error distribution based on the logarithms of the ratios of 'actuals'/estimates with a variance related to the kriging variance. The mean of the data used to prepare the variogram and the estimated block grade are to be used properly in this approach.

A practical analysis was also done of the actual point and block data from a very large mined out Carbon Leader area. The distributions of the actual grades for two levels of block sizes with different levels of variability were fitted using three models, i.e. 2-, 3- and compound-lognormal. The trends of tonnages, grades and relative profits for a range of pay limits, as well as the usual tonnage-grade curves, show a practical insensitivity for the choice between the three lognormal models considered.

Introduction

Distribution models for observed grade distributions in the South African mining environment have received attention repeatedly, and deservedly so, since the inception of geostatistics more than 50 years ago, and the process is still continuing. Such models are essential for point values and for larger support values such as ore blocks (usually Selective Mining Units—SMUs) as well as for estimator models and the related error distributions of, and confidence limits for, grade estimates for ore blocks and for larger areas such as mine sections still to be exploited.

For point distributions it is essential to compare alternative distributions models for large bases of actual data and in South Africa this led to the use of mainly three lognormal models, i.e. the two, three and compound lognormal distributions — 2PLN, 3PLN and CLN (Sichel 1947, Krige 1960 and Sichel et al. 1992). Experience has shown that before any model is chosen and used, the subdivision of the orebody into geologically homogenous sub-populations is essential. Usually the particular model best suited to the point data has also been accepted for the corresponding block distributions but this practice will be verified for the examples used in this paper.

As far as error distributions models are concerned, Sichel covered the small sample theory for random sampling from the 2PLN (Sichel,1952) with application also to the 3PLN. Some aspects of error distributions for kriged estimates were dealt with by Dowd (1988). However, the cases dealt with by Dowd covered kriging estimates which are seldom of much practical use i.e., estimates of point grades using very limited search routines. These are subject to unavoidable conditional biases and the effects of these were not dealt with. Parkin et al (1990) compared five methods for estimating confidence intervals for random samples drawn from a 2PLN distribution but, as for Sichel's limits referred to above, these will not be applicable to kriging estimates which cater for data with a significant spatial structure and where the error variance is estimated by the kriging variance (KV). An attempt will be made in this paper to address some of the aspects of error distributions and confidence limits based on the KV indicated for kriging estimates.

Error distributions for kriged estimates

In South African gold mines ore reserve blocks are generally valued using a variogram of untransformed grades and ordinary kriging (OK) or simple kriging (SK). This approach is preferred to the use of lognormal kriging which, although theoretically more efficient, can introduce significant biases mainly during the back-transformation process (Sinclair and Blackwell, 2002). The OK or SK block estimates incorporate the corresponding KVs and the use of these to estimate the corresponding theoretical confidence levels will be investigated. If properly interpreted, such confidence levels can assist in the classification of resources and reserves into the appropriate categories. In many circles, however, the use of the KV has been discredited and is still viewed by some as misleading and inappropriate, an issue which requires clarification.

For this purpose, a massive 2PLN data base of 13824 closely spaced point values with a realistic spatial structure, as simulated for a paper by Krige and Assibey-Bonsu (2000), was used. On subdivision into 1728 ore blocks, each block was kriged on the data inside the block as well as sufficient surrounding data to ensure a negligible level of error variances, and thus to provide a set of 'actual' block...
values for comparisons with corresponding block estimates. For these block estimates, a set of 216 point values was selected on a regular grid from the 13824 available values, and these data were used to prepare OK estimates with a limited search as well as SK estimates for each block together with the corresponding KVs.

The errors between these estimates and the ‘actuals’ can be defined either as

- **differences**, i.e. ‘actuals’ less estimates, or
- **ratios**, i.e. ‘actuals’/estimates.

The corresponding error distributions were analysed in total, i.e. for 1728 blocks and also for the more homogenous set of 1331 blocks remaining after peeling off the peripheral blocks where the available data were more limited. The results were analysed after subdivision into ten grade estimate categories from the lowest estimates, 0 to 10 per cent cumulative frequencies, to the highest estimates, 90 to 100 per cent cumulative frequencies. Such a split is essential to ensure that the confidence levels indicated do not vary significantly with variations in the block grade levels as estimated.

**Errors defined as differences**

Figure 1 shows the error distributions based on ‘actuals’ less estimates for the OK estimates for several of the ten per cent grade categories and for all 1728 estimates, as well as the theoretical Normal distribution based on the indicated average KV level. It is obvious that the OK distributions for individual grade categories depart radically from each other and from the overall distribution, as well as from the theoretical Normal error distribution. Figure 2 shows the same analyses for the SK estimates; the departures are somewhat less than for the OK estimates, but still unacceptable for the purpose of applying a consistent model for confidence intervals.

**Errors defined as ratios**

The alternative model based on ratios, i.e. ‘actuals’/estimates, should lead to more logical confidence limits expressed as percentage, and not absolute, deviations from the estimates. For the lognormal distributions of the basic block ‘actuals’ and estimates, the log-transformed distributions will be Normal as well as the differences between the log values. The log-differences correspond to the ratios of the untransformed values and thus the distribution of the ratios will also be lognormal with a variance defined as:

\[
\sigma^2_k = m^2 \left[ \exp(\sigma^2_{ln}) - 1 \right],
\]

where \( \sigma^2_k = \text{untransformed error variance} = \text{KV} \)
\( m = \text{mean of data used for the variogram} \)
\( \sigma^2_{ln} = \text{equivalent logvariance of errors} \)
\( \sigma^2_{ln} = \text{variance of deviations of ‘actuals’ from estimates on a log basis.} \)

The ‘logvariance’ can thus be solved for and confidence limits estimated on Normal theory as follows:

Central 90 per cent limits =

\[
\exp \left[ \ln(x) - \frac{\sigma^2_{ln}}{2} + 1.645(\sigma_{ln}) \right]
\]

where \( x = \text{the block estimate, and ln = natural logarithmic variance.} \)

**Note that in formula [1] ‘m’ is not the estimated block grade but the mean of the data on which the variogram is based.** The KV (or error variance) does not depend on the grades accessed for the kriging of a block, but only on the data configuration and the relevant covariances for the data and the ore block. The covariances, in turn, depend on the parameters of the spatial structure as modelled by the variogram and specifically on the total sill (i.e. the population variance) which, due to the proportional effect, will vary with the overall mean grade of the data (see Formula [1]). The KV is thus directly linked to this mean grade and not to the block estimate. The validity of this approach will be tested in some detail on the data set mentioned above. The overall grade distributions for the basic point data, the ‘actual’ block grades and the SK block estimates, can all be accepted as 2PLN. The distribution of the OK ratios of ‘actuals’/estimates are also close to 2PLN. These lognormal patterns were confirmed via log-probability plots (not shown) and supported the so-called ‘permanence’ of the lognormal.

A practical analysis of the block estimates was done on the correlation graphs of the OK and SK estimates versus the corresponding ‘actual’ block values as shown by grey squares on Figures 3 and 4 respectively. The regression trends observed, i.e. the averages of the ‘actual’ grades for the conditional distributions in a range of ten estimate
categories, are shown as filled-in squares, as well as the ideal linear regression with a slope of unity. Within each of the ten grade categories used, the observed 90 per cent central confidence limits for the ‘actual’ distributions are also shown as filled-in circles. Also shown as straight grey lines are the corresponding 90 per cent theoretical limits for the conditional ‘actual’ (Y) distributions based on the lognormal model for the ratios of ‘actuals’/estimates with an untransformed variance accepted at the recorded average KVs for the OK and SK estimates respectively, using formula [2] on the previous page.

**SK estimates**

Figure 4 shows the corresponding results for the SK block estimates. The regression trend is close to unity and the confidence limits agree reasonably well with the theoretical limits as would be calculated using the logarithmic equivalent of the KV and the lognormal distribution model as covered by Formulae [1] and [2] above. Bearing in mind the high level of the data variability and the wide average 90 per cent central confidence levels of about –70 per cent and +125 per cent, this ratio model, although not perfect in this example, is recommended for use in cases where the two- or three-parameter lognormal models apply. In the latter case, the estimate ‘x’ is replaced by ‘x+b’ (with b = 3rd parameter) and ‘b’ is deducted from the limits as calculated.

**Distributions for block grade estimates.**

**Effects of change of support**

In changing from point values to block grades, actual or estimated, the change of support results in a significant reduction in the variance level, as well as a reduction in the practical differences between the shapes of distribution patterns for the three lognormal models under consideration. Apart from the mean and variance, the 2PLN has no other parameter, the 3PLN has a third ‘b’ parameter, and the CLN has a third and a fourth parameter for the logarithmic skewness and kurtosis respectively.

For the purpose of comparing the 2PLN and 3PLN models with the CLN, a detailed analysis was done on actual data from a Carbon Leader mined-out area which is also referred to by Assibey-Bonsu and Krige (2003). The 2PLN proved unsuitable for the point values and the 3PLN required a ‘b’ parameter of 170 cm.g/t. This third parameter did not change significantly for the corresponding block distribution. Figure 5 shows the log-probability plots of the point and 50 x 50 m block values for the ‘actuals’ and the fitted 3PLN and CLN models. The reduction in the differences between these two models for the block values is evident. Changes in the critical parameters for various block sizes are shown in Table I. From Table I it is evident that for these data the critical three parameters, i.e. the logarithmic variance, skewness and kurtosis decrease significantly as the support size increases and effectively the latter two parameters reach Normal levels at the 200 x 200 m block size. Similar trends were observed by Krige and Dohm (1994) for a VCR distribution but not conclusively for a Vaal Reef distribution. The overall effects of these changes for the Carbon Leader in the present case will be studied in some detail.

The observed ‘actual’ distributions in Table I for the 20 x 20 m and 50 x 50 m block sizes show untransformed coefficients of variation (C of V) of 0.65 and 0.47 respectively, and will be used to compare the effects of using the 2PLN, the 3PLN and the CLN models for these distributions in the paragraphs that follow.

**Two-parameter lognormal**

As covered in the example used previously, it is generally accepted that where the point data follow a 2PLN model, the same model will also apply to corresponding distributions for block grade estimates. It should be noted that the latter distributions have much lower variance parameters than those for point data, particularly for deep
level mining areas where the SMUs will be relatively large and block estimates will involve the significant smoothing effect resulting from the essential conditional unbiasedness. In any particular application where individual blocks are valued on an SMU basis (usually for measured and indicated resources) the kriged results provide estimates of the KV and dispersion variance (DV) for each block. If the data used for the block valuations are in their final form, i.e. if no further data will be forthcoming before the final selection of payable blocks is made, the distribution of the individual block estimates will form the basis for the tonnage-grade curve to be used for planning purposes. No mathematical model is then necessary but if a model is required for the observed distribution, the 2PLN model should normally be suitable. For a range of pay limits (PL) the results of the fitted 2PLN and 3PLN models are compared in Figures 6 and 7 with each other and with the ‘actual’ block distributions (grey curves) for the tonnage percentages and the corresponding grades above each PL. The relative profits are also shown, where

\[
\text{relative profit} = \% \text{ tons} \times (\text{grade above PL} - \text{PL}).
\]

Figure 8 shows the usual corresponding tonnage-grade plots.

It is clear that within the range of ‘actual’ grade variability levels covered, the results are not sensitive to variations in the ‘b’ value between zero and 30%, i.e. between the 2PLN and 3PLN models respectively. This conclusion will apply even more forcibly to the usual case where block ‘actuals’ are not known and modelling is done on SMU estimates. Note also that the results for the 50 m ‘actuals’ are expected to be closer to the position expected for block estimates in deep level gold mines.

### Compound lognormal model

Apart from the fitted 2PLN and 3PLN models, Figures 6 to 8 also show the results of fitting the CLN model to the ‘actuals’. It is evident that the three fitted models agree well with the ‘actuals’, particularly for the 50 m block case. It is thus concluded that the modelling of these block

---

**Table I**

<table>
<thead>
<tr>
<th>Support</th>
<th>In mean</th>
<th>In variance</th>
<th>In skewness</th>
<th>In kurtosis</th>
<th>Data number</th>
</tr>
</thead>
<tbody>
<tr>
<td>Points</td>
<td>7.715</td>
<td>1.035</td>
<td>-0.670</td>
<td>5.430</td>
<td>59951</td>
</tr>
<tr>
<td>20 x 20 m blocks</td>
<td>8.008</td>
<td>0.323</td>
<td>-0.221</td>
<td>5.640</td>
<td>4965</td>
</tr>
<tr>
<td>50 x 50 m blocks</td>
<td>8.058</td>
<td>0.202</td>
<td>-0.162</td>
<td>4.189</td>
<td>1315</td>
</tr>
<tr>
<td>100 x 100 m blocks</td>
<td>8.069</td>
<td>0.139</td>
<td>-0.044</td>
<td>3.140</td>
<td>403</td>
</tr>
<tr>
<td>200 x 200 m blocks</td>
<td>8.106</td>
<td>0.098</td>
<td>+0.021</td>
<td>3.021</td>
<td>117</td>
</tr>
</tbody>
</table>

---

Figure 5. Showing the CLN model for ‘actual’ point values (grey curve), the corresponding 3PLN model with \(b = 170\) (black squares) and the ‘actual’ grade distribution for 50 x 50 m blocks (black triangles).

---

**Figure 6**

Showing the ‘actual’ pay tonnage percentages, pay grades and profit for 50 x 50 m blocks with a grade coefficient of variation of 0.47 and the fitted three lognormal models with the same C of V

---

**Figure 8**

Showing the usual corresponding tonnage-grade plots.
distributions is not sensitive to the choice between the three lognormal models, particularly within the practical limits of selective mining percentages. For deep level areas selective mining decisions will be made on the basis of larger SMUs, the KVs will be higher, and lower variability levels can be expected for block estimates. This will result in the choice of the relevant model being even less significant than in the practical examples dealt with.

Conclusions

The following conclusions are drawn from these analyses:
• In estimating confidence limits for grade estimates on a lognormal model the mean of the data used for the variogram modelling must feature correctly as covered in a previous paragraph, the limits should be on a per cent basis and will be skew.
• Any distribution model has its own limitations and can never be accepted as a perfect and final representation of the behaviour of the variable concerned.
• Where alternative models present themselves some practical tests of sensitivities and applicability are required. For SMU block distributions in deep level gold mines the choice of a specific model does not appear to be critical.
• The best validation of a model, apart from any theoretical considerations, is to be obtained from practical follow-up tests wherever possible.

Acknowledgements

The author expresses his appreciation for the assistance provided by Dr W. Assibey-Bonsu in handling of the Compound Lognormal Model and for the permission of Gold Fields Limited for the use of the relevant data base.

References


KRIGE, D.G. and ASSIBEY-BONSU, W. Limitations in accepting repeated simulations to measure the uncertainty in recoverable resource estimates particularly for sections of an ore deposit. 6th International Geostatistics Congress, South Africa, April 2000.


