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THE SUITABILITY OF THE MERCURY HALO PROSPECTING TECHNIQUE IN SOUTH AUSTRALIA.

bу

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SUMMARY

Background

This project was proposed in 1971 as a consequence of the large number of determinations carried out at Amdel on soils. It seemed that insufficient background data were available on the mercury content of ores and rocks and the proposal was for a systematic study of the mercury content of South Australian ores.

Objective

The purpose of the project was to determine for what ore types and in what conditions the mercury halo technique is suitable for prospecting for base and other metals in South Australia.

Summary of Work Done

After a search of the literature on prospecting using this technique had been carried out, the South Australian Department of Mines submitted samples (mainly core material) from six copper-bearing areas in South Australia.

Determinations of copper, lead, zinc and mercury were made on samples from Kapunda, Parabarana and Kanmantoo. Copper and mercury were determined on samples from Burra, and mercury alone on samples from Mutooroo and Moonta.

Analysis for mercury proved difficult and several analytical methods were used to ensure adequate accuracy and reproducibility.

Conclusions

The results of the experimental work show a low, but positive, correlation of mercury with low levels of copper, and also show some tendency for there to be some concentration of mercury in the near-surface zone above copper ore-bodies. Significant amounts of mercury are associated with the ores of Kapunda and Burra.

The results of this work have demonstrated that, before embarking on a project of this nature, rigid controls must be applied at all stages from sampling through transport, sample preparation and storage to the determination of the mercury. Possibilities for contamination after collection and before analysis are high.

However, the premises on which the project was formulated, namely that mercury is a common associate of base metal ore deposits or of gold or uranium ore deposits and should be a useful guide, are still valid. It may confidently be expected that, in the future, mercury haloes can be used to

determine the position of blind orebodies.

Recommendations

It is recommended that the project be held in abeyance until a completely satisfactory sampling, storage and analytical scheme has been developed.

1. INTRODUCTION

The element mercury is chalcophile and is commonly associated with other (base) metals in hydrothermal and epithermal deposits. Illustrative of these types of occurrence are a number of short articles published in World Mining recently (Argall, 1971), which discussed the recovery of mercury as a by-product during the extraction of:

- (i) copper from copper ores at Rudnany (Czechoslovakia) and Gortdrum (Eire);
- (ii) zinc from zinc ores at Rosebery (Tasmania), at OutokumpuOy (Finland), and at Balmat and Edwards Mines (New York State);
- (iii) gold from gold ores (Nevada).

Interest in mercury lies in its value as an indicator or pathfinder element for buried ore deposits. Elemental mercury has a very high vapour pressure and its compounds decompose at low (geologically speaking) temperatures. This means that mercury has the potential to disperse over a wide area away from ore and elevated values may be found at the surface above a buried sulphide deposit. If the orebody is sufficiently deeply buried, mercury may reach the surface when none of the other metals present in the orebody has been sufficiently mobile to do so. Ideally, the surface expression of anomalous mercury can be mapped and contoured with the highest values overlying the ore.

However, ideal conditions rarely apply, since both primary and secondary dispersion of mercury may have taken place. The surface expression of mercury may be controlled by jointing or faulting as much as by the position of the ore. Alternatively, the orebody, itself deficient in mercury, may mobilise mercury from the country rock and drive it down the temperature and pressure gradient in front of the ore, eventually causing anomalous readings which are difficult to interpret.

At the surface, mercury adsorbed on soil is believed to maintain some sort of equilibrium with mercury in soil air and this in turn is thought to maintain a relationship with mercury in the atmosphere immediately above the earth's surface. Soil mercury, soil air mercury and atmospheric mercury may all be measured, and used to indicate anomalous mercury concentrations.

The present investigation arose out of service work which has been carried out at Amdel over some years. As a consequence of this work, it was suggested

to the Mines Department that a systematic attempt should be made to investigate the mercury content of South Australian ores and to determine for what ore types and in what conditions the mercury halo technique is a suitable prospecting tool.

The Report is divided into two parts. The first part is intended to summarise the main literature which discusses the use of mercury as an exploration tool. The second part reports the results of assays carried out at Amdel and provides some discussion of the data.

PART 1

LITERATURE REVIEW

1. INTRODUCTION

The references collected for, and cited in, this Report have been obtained from Chemical Abstracts, Mineralogical Abstracts and the Index and Bibliography of Geology published by the Amercian Geological Institute.

Where possible, papers cited have been read, but in some circumstances (either because the paper is unobtainable in Australia, or is written in a foreign language), the abstract has been used. In addition, a number of references are cited in the bibliography which are not discussed in the main body of the Report. These papers appear to be relevant to the topic, but neither paper nor abstract have been sighted.

2. THE HISTORICAL DEVELOPMENT OF THE METHOD

The possibility of using a 'mercury halo' technique for detecting hidden sulphide orebodies was postulated by Saukov (1946) in his book 'The Geochemistry of Mercury' (not available in English), but it was not until papers by Fursov (1958) and Ozerova (1959 and 1962) were translated into English that the technique was seriously considered by non-Soviet geologists.

Little interest had been shown in mercury in the past because of its low total concentration in the earth's crust, estimated at only 80 ppb* (Krauskopf, 1967) and the consequent difficulties of determining mercury at this low level. The papers of Fursov (1958) and Ozerova (1959) cited above quoted detection limits of 150 ppb and 300 ppb respectively. Recent developments in analytical chemistry using flameless atomic absorption spectroscopy have brought the detection limits down to 1 to 5 ppb and better views of the patterns of mercury distribution have been obtained.

The method has become of even greater interest because of its alleged universality. Ozerova (1962) discussed primary mercury dispersion patterns around deposits of:

mercury, mercury-antimony,

^{* 1} ppb = 1 part per billion = $10^{-7}\%$ = 0.001 parts per million

a number of authors as XOO (100 - 999) ppb; however, Ozerova and Aidin'yan (1966a) found that over a wide area of the Russian platform the average of 58 samples of clay was 35 ppb. Other sedimentary rocks on the Russian platform appeared to have values lower than the world average.

It is only in recent years that mercury assays have become at all reliable at the low levels of concentration of the element in rocks. 'Early' assays are therfore inherently suspect.

In any investigation in the field it is clear that a local background value for each rock type must be established before consideration can be given to possibly anomalous results.

Table 2 gives selected data on mercury values in rocks and soils of unmineralised regions, while Table 3 gives comparable data for rocks in soils in mineralised regions.

4. MERCURY IN INDIVIDUAL MINERALS

It is pertinent to consider which minerals may be expected to be rich in mercury. This may provide a clue for the choice of suitable ores for testing for their mercury content and indicate potential limits for the use of the mercury halo method.

Mercury tends to be concentrated in minerals where it can diadochically replace divalent cations such as iron, zinc or lead in compounds of these elements. Their respective ionic radii (6-fold co-ordination) are (Mason, 1966, p297-9):

| Ion | Ionic Radius (Å) |
|------------------|------------------|
| Hg ²⁺ | 1.10 |
| Fe ²⁺ | 0.74 |
| Zn ²⁺ | 0.74 |
| Pb ²⁺ | 1.20 |
| Ba ²⁺ | 1.34 |
| Ca ²⁺ | 0.99 |
| Cu ²⁺ | 0.72 |
| Cu ⁺ | 0.96 |
| Sr ²⁺ | 1.12 |

Since the mercuric ion is of different size to those of the most common ore metals there will be strain in the crystal when mercury (II) replaces the

host element, and only limited replacement is expected.

The main minerals which contain mercury in diadochic replacement are discussed by Warren and Thompson (1944), Fleischer (1955), Sergeyev (1961) Ozerova (1962), Hawkes and Williston (1962), James (1964), Chan (1969) and Friedrich and Pluger (1971).

In order of possible preferential replacement the main ore minerals given by James (1964) are as follows:

stibnite, sphalerite, cerussite, galena, chalcopyrite, pyrite.

In addition, mercury can form an amalgam with native gold, silver or copper. The typical range of mercury in the above minerals (and including native gold) is of the order of 0.5 to 25 ppm. Atypically, the concentration of mercury may reach several thousand ppm, particularly in sphalerite and stibnite (Ozerova, 1962).

The mercury content of these minerals may depend on the temperature of formation of the minerals but no comments have been published about temperatures of origin.

The following 'gangue' minerals have been examined and shown to be able to contain varying proportions of mercury:

barite,
calcite,
fluorite,
siderite,
aragonite,
sericite,

These minerals, when occurring as vein 'gangue' in mineralised areas, have a typical mercury content from about 0.01 to 20 ppm, averaging 2 to 5 ppm (Ozerova, 1962; James, 1964). Atypically, the mercury concentration may reach several hundred ppm.

Chan (1969), in an area of lead-silver mineralisation (Shoshone Co., Idaho) measured the mercury contents of various minerals in and near the ore veins of the Galena Mine. His Table 3 (pl56 in the original) is reproduced below:

| Mineral | Mercury Range, ppm | No. of Measurements |
|----------------------|--------------------|---------------------|
| Tetrahedrite | 5.1 - 62.6 | 22 |
| Chalcopyrite | 3.5 - 38.0 | 10 |
| Galena | 3.8 - 10.4 | 10 |
| Pyrite | 1.0 - 2.5 | 10 |
| Arsenopy rite | 0.9 - 2.3 | 8 |
| Siderite | 0.2 - 2.5 | 10 |
| Quartz (vein) | 0.04 - 1.8 | 10 |
| Quartz (wall rock) | 0.08 - 0.5 | 10 |

Friedrich and Pluger (1971) found that samples of barite from various deposits in Germany contained between 1.2 ppm and 250 ppm mercury.

In general, it may be said that mercury is a universal component of rocks and minerals but that the concentrations are normally very low. For the purposes of prospecting it is possible, but not normally economic, to separate-out suitable minerals which may show an elevated content of mercury in the hope of accentuating unusual mercury values.

Table 4 shows the mercury content of common ore and gangue minerals and is taken from Jonasson and Boyle (1972, p34). This represents the latest information available. The high capacity for tetrahedrite to contain mercury as shown in the table is confirmed by other writers.

5. MERCURY DISPERSION HALOES IN ROCKS

The classic papers whose English translations stimulated the work of Western geochemists were those of Fursov (1958) and Ozerova (1959).

Fursov investigated mercury haloes at the Achisai lead-zinc deposit of Kazakhstan. The ores (a mixture of fresh and weathered sulphides of zinc and lead) averaged 79 ppm - approximately 1000 times the estimated crustal average. The oxidised ore averaged 130 ppm against an average of 6.3 ppm in fresh ore. The highest mercury value was found in cerussite sands (1000 ppm). The ores are found in faults and veins in carbonate host rocks. Mercury is concentrated in

cerussite, and to a lesser extent, in galena and sphalerite. The primary dispersion halo is marked, 23 ppm being found in limestone 73 m from the contact. At distances of 140 to 195 m from ore the mercury concentrations were 0.4 to 0.5 ppm. Anomalies occurred over all known mineralisation and over known fracture zones adjacent to ore. In two cases anomalies were present at the surface, 230 m above blind orebodies. As Fursov's analytical technique had a lower limit of detection of 0.3 ppm (300 ppb) the haloes shown were not very wide.

Ozerova (1959) discussed dispersion away from lead-zinc orebodies of the Fergana region of Turkestan. The deposits were found in middle Palaeozoic limestones, dolomites, sandstones and conglomerates. Orebodies crop out, or are buried up to 100 m deep. Traverse lines over the orebodies showed the presence of mercury haloes 1.2 km to 1.8 km wide with values of mercury rising gradually from \$50 ppb (the normal detection limit) to 500 ppb, and with a sharp peak of 1000 ppb directly over the deposit. Only in 15 out of 69 sand and limestone samples assayed was mercury detected, the values being 70 ppb or lower.

Ozerova's (1962) book 'Primary Dispersion Haloes of Mercury' has recently become available (English edition, 1971). She reviews all known Russian orebodies and quotes mercury contents of ores and individual minerals, and background values for different rock types. This is too large a volume to summarise in this Report, but Ozerova's conclusions are that mercury can form extensive, continuous and easily detected haloes around deposits of other chalcophile elements, 'even though there may be no minerals of mercury in the deposits'. In particular, she cites Sb, Cu-Pb-Zn and Ag-Au ores as being rich in mercury. She notes that in the 'Great Basin' mercury is known to penetrate basin-fill more than 300 m above ore deposits; she reports haloes to be commonly more than 1 km wide around hydrothermal deposits. Mercury can penetrate up to 200 m of unfractured rock vertically and at least 600 m along faults.

More general work on haloes, in which mercury is treated as one element among many, has been carried out in the USSR by Dvornikov and co-workers (1961, 1963), Bulkin (1962), Bulkin and Lepilin (1964), Terziev (1966), Lazutin (1966), Kovrigo (1966), Polikarpochkin and Kitaev (1971) and other writers whose work is not available in English.

In general their work supports that of Ozerova; however, there is some disagreement as to whether the haloes obtained are wider (or more meaningful) than those of the base metals or, for example, antimony. Various papers (e.g. 946

Dvornikov, 1962) written about the Nagol'nyi Range 'polymetallic' (lead-zinc) deposit, of the S.E. Donets Basin, show a wide mercury halo (up to 1600 m from ore). In this area the mercury halo is twice as wide as that of lead and zinc, three times that of copper and 2.5 times that of antimony.

Research outside Russia has developed more slowly, but a number of recent papers describe case histories where mercury haloes in rocks have been used in the search for mercury or other metals.

Bercé (1960, 1965), Bradshaw and Koksoy (1968), and Koksoy and Bradshaw (1969) have investigated mercury dispersion away from mercury deposits. The last two papers discuss, respectively, primary and secondary dispersion of mercury away from, not only cinnabar, but also stibnite, deposits in Turkey.

Authors who have investigated mercury in rocks in relation to other types of ore include the following:

| Authors | Date | Ore Type |
|----------------------|--------------|------------------------|
| Friedrich and Hawkes | 1966a, 1966b | Cu, Pb, Zn, Ag |
| Gott and McCarthy | 1966 | Au, Ag, Te |
| Erickson et al. | 1966 | Zn, Pb, Ba, As, etc |
| Gott et al. | 1967, 1969 | Au, Ag, Te, Cu, Pb, Zn |
| Cornwall et al. | 1967 | Ag, Au |
| Jolly and Heyl | 1968 | Pb, Zn |
| Akright et al. | 1969 | Ąu |
| Chan | 1969 | Ag-Pb-Zn |
| Crosby | 1969 | Pb-Zn-Ag |
| Sears | 1971 | Base metals, Au |
| Friedrich and Pluger | 1971 | Ba, CaF ₂ |

These authors show that mercury behaves more erratically than is indicated by the Russians. Friedrich and Hawkes (1966a), for instance, found that although hydrothermal ore veins contained values of mercury up to 26 ppm, the host rock, andesite, a few centimetres away from the ore, rarely contained more than background concentrations (\leq 20 ppb). Despite their observations about the rather erratic behaviour of mercury and the patchy distribution patterns it displays, all the above authors are agreed that mercury haloes exist and they believe that, in the correct circumstances, the halo technique can be useful.

It is clear that some mineralised areas have high backgrounds of mercury; thus Akright et al. (1969) found the background mercury in unaltered sedimentary rocks

(shales, quartzites, limestones) to be 20 ppm while the mineralised rock (Aubearing) itself contained an average of 31 ppm.

Crosby (1969), working in the Coeur d'Alene district, Idaho, sampled underground 'cross-cuts' across various Pb-Zn mineralised veins. He found that mercury behaved in one of two ways; there was either a peak of mercury values directly above the ore or else there was a double peak, one on either side, with a 'low' directly over the ore. The variation in absolute concentrations of mercury was wide, ranging from 10 to 40 ppb in one traverse (which cut ore) to 200-X000 ppb in other traverses. Though the lower values gave the same profile pattern, it is clear that unless the veins had been known previously the low anomalies would have been missed during prospecting.

The double peak with a central low may be interpreted in either of two ways:

- (a) The mercury has dispersed as an envelope in front of the ore, with less mercury inherent in the ore than in the envelope, or
- (b) The fissuring or jointing associated with the mineralised veins has allowed secondary dispersion of mercury with a consequent loss of mercury from the ore zone itself.

Sears (1971) has approached the problem of mercury in ores from a rather different standpoint. He collected approximately 600 samples from 19 producing base-metal mines, three producing gold mines and five previously-producing base-metal mines in Quebec, and assayed the samples in toto for mercury and other ore metals (mainly copper, lead and zinc) as appropriate. He showed (p386) a wide variation of mercury content of the ores even from the same mine. Thus the mercury concentration in ore from the Normetal base metal mine ranged from 10 ppb to 18 000 ppb in 23 samples. Sears plotted mercury contents against copper, lead and zinc concentrations and was unable to find any relationship between mercury and the other elements, except that there was a weak positive correlation between zinc and mercury values. Sears did not assay individual minerals; his conclusions, however, were, firstly, that of copper, lead, zinc and iron sulphides, the zinc sulphides have the highest concentration of mercury and, secondly, that lead sulphides generally have higher mercury contents than iron-copper sulphides.

6. MERCURY IN SOILS

Though the practical work of this project has been carried out on ores and adjacent rocks, it is relevant to provide a brief summary of the state of knowledge of the distribution of mercury in soils, since most of the papers which discuss the mercury halo prospecting technique are case histories in which soil or stream-sediment samples have been collected.

6.1 Mercury Content of Soils in Unmineralised Regions

Data on the mercury contents of soils in unmineralised areas are sparse. Figures obtained from the main papers are given in Table 3: these are generalisations and represent background levels. Only two papers have been located which discuss, in detail, mercury in unmineralised soils. These are Stock and Cucuel (1934), summarised by Vinogradov (1959), and Aidin'yan et al. (1964).

Aidin'yan et al. (1964) found, for soils from the USSR (Europe) and from Vietnam, that the upper 10 cm of the soils were lacking in mercury. There was usually a sharp rise to the 'B' horizon or equivalent. This level then held steady or the values gradually decreased through the 'C' horizon. Average soil values were approximately 10 times the values of mercury in the country rocks. Iron-manganese nodules in Vietnam have higher mercury contents than the soils, and mercury is believed to be readily absorbed by sesqui-oxides. The lack of mercury near the surface was attributed to evaporation of mercury from the surface (?followed by redistribution).

Goldschmidt (1954) hypothesises that mercury probably concentrates in forest litter and in the uppermost layers of forest humus. This assertion is repeated by Boyle and Dass (1967).

Some writers, particularly those from the USSR, do not make it clear whether the assays they have reported refer to soil or to the parent rock.

6.2 Mercury Content of Soils in Mineralised Regions

Most of the authors cited as discussing mercury in rocks also discuss mercury in soils. Additional papers concern case history studies related to mercury as an exploration tool. Such papers include those of Fedorchuk (1961), Dvornikov and Petrov (1961), Dvornikov et al. (1963), Bol'shakov (1964), Brown (1966 and 1967), Friedrich and Hawkes (1966a and 1966b), Bradshaw and Koksoy (1968), Koksoy and Bradshaw (1969), and Friedrich and Pluger (1971). Much of the data offered by these and other authors summarised in Table 4.

Dvornikov and co-workers, (1961, 1962, 1964) investigated soils at the Nagol'nyi Range. The soils at 5 to 10 cm and at 10 to 20 cm gave similar values (Dvornikov and Petrov, 1961) with background at 80 to 100 ppb, a plateau of 200 to 300 ppb adjacent to ore, and up to 9500 ppb over ore. No data are given in these papers on the mercury content of the ore, which either crops out or is at shallow depth. The maximum halo width is 1700 m.

Bol'shakov (1964) found that lead-zinc orebodies in the S.E. Donets Basin buried to depths of 200 to 300 m had mercury dispersion aureoles which reached the surface, though there were faults which provided possible access paths for the mercury.

Friedrich and Hawkes (1966a) examined residual soils over volcanic rocks mineralised with Ag, Pb and Cu at Pachuca-Real del Monte, Mexico and found that the soil assay values, though erratic in distribution, were rather more consistent than values obtained for rocks. Where the rock is suitably fractured, mercury anomalies were discovered at the surface above ore buried to a depth of 300 m. In a second paper (1966b), Friedrich and Hawkes found anomalies in residual soils on unmineralised volcanic rocks with ores buried at 16 to 60 m (50 to 200 ft) depth (Shasta District, California). Mercury values peaked at 340 ppb over ore against background values of 20 to 60 ppb.

Brown (1966, 1967) assayed line traverse samples across high temperature molybdenite orebodies in British Columbia and found subdued but definite haloes. Background was in the range 15 to 40 ppb with peaks up to 175 ppb. Over copper ore comparable figures were 25 to 60 ppb and 170 ppb, and over a silver-lead-zinc orebody the comparable figures were 25 to 100 ppb and 500 ppb. Brown believed this area of British Columbia to be similar to much of the Crimean Highlands (quoted by Bulkin, 1962). It has such a high general background level of mercury that shallow peaks associated with molybdenum, or other high temper-ature mineralisation, may be lost in background fluctuations.

Most authors have sampled the upper, humic layers of the soil; however, Friedrich and Hawkes (1966b) found that mercurcy values increased with depth from the surface. Few authors have written about areas with an aridity comparable with South Australia. In the author's experience the 'B' or hardpan horizon has acted as a collector of mercury and the highest values are to be found there. As in so many areas of geochemistry, it appears that thorough orientation is needed before full-scale exploration programmes are carried out.

7. MERCURY IN STREAM SEDIMENTS AND WATERS

Prospecting using mercury in stream sediments has been relatively little used; however, the papers by Missaghi (1966) and Koksoy and Bradshaw (1969) do discuss secondary dispersion of mercury away from mineralisation in the stream sediments. Both the above papers report favourably on the detection of mercury in such sediments. Missaghi's data are summarised in Table 3.

There is little merit in seriously discussing mercury in stream waters with a view to using this method in South Australia. However, papers on prospecting using mercury in streams have been written by Dall'aglio (several, in particular 1971), Dvornikov (1964) and Bayev (1968). Various estimates/measurements of mercury in water are given in Table 6.

It is clear from the literature that mercury can disperse in streams but the mechanism of this dispersion is not clear. Missaghi (1966) and Koksoy and Bradshaw (1969) evidently believe that sulphide minerals with attendant mercury may migrate mechanically downstream. Dall'aglio (1971) believes that mercury moves either in solution, perhaps as HgCl₂ or as a chloride complex, or adsorbed on clay or in some colloidal form.

8. MERCURY IN AIR AND SOIL AIR

A study of mercury in either soil air or the atmosphere may prove to be a worthwhile prospecting technique in South Australia where large areas of ground are covered by moderate thicknesses of relatively loose sand. It has been shown experimentally that sand does not easily retain mercury (by absorption, adsorption or any other process) but allows mercury to pass freely through it. It may therefore be better to explore sandy surface materials by sampling either soil air at a depth of a few metres or by sampling the atmosphere at no great distance from the ground surface.

Discussions of mercury in soil air have been given by Hawkes and Williston (1962), Karasik and Bol'shakov (1965), Fursov et al. (1968), McCarthy et al. (1969a) and Fursov (1970). Discussion of mercury in the atmosphere has been given by Williston (1968), Barringer (1964, 1969), McCarthy et al. (1969a) and Seigel Associates (1971).

All these authors maintain that assay of mercury in air or soil air is possible at the very low levels expected and Karasik and Bol'shakov (1965), Fursov et al. (1968), McCarthy et al. (1969), Seigel Associates (1971) and Fursov (1970) have demonstrated to their satisfaction gaseous dispersion haloes

of mercury associated with ores. Fursov (1970), dealing with gaseous mercury over blind mercury ores, obtained anomalous values up to 1100 times background. (Background values were 10 ng of mercury/litre of soil air.) No depth to ore was given.

9. DISCUSSION

It is clear from the literature that the problems of the potential of the mercury halo prospecting method have never been fully resolved. The early work was hindered by a lack of adequate analytical techniques and only in the last 4 to 5 years has the flameless atomic absorption technique, potentially one of the most sensitive techniques available, become sufficiently reliable to handle all types of sample and to be able to separate true mercury values from false values attributable to contamination or non-atomic absorption.

The use of mercury for prospecting purposes rests on an understanding of the geochemistry of mercury in its relation to ore deposits. Most studies to date have used mercury assays of surface samples of soils, unmineralised rocks or oxidised ores. Comparatively few studies have been made of mercury concentrations in ores and in minerals within or associated with ores, and no paper sighted to date has attempted to relate, other than by inference, the mercury content of ores to their temperatures of formation. Most papers, e.g. Williston (1962), glibly assume that as mercury is commonly associated with mineralised thermal springs as at Ngawha, New Zealand or Sulphur Bank, California, mercury will only be found in, or associated with ores of low temperature. probably true, but the literature reveals that little confirmatory work has been reported on the mercury contents of ores of various types. Cinnabar and mercury ores in general may be truly low temperature in origin, but mercury may be taken into, e.g. a galena lattice, at much higher temperatures.

Establishment of the levels of mercury in various ore types seems, therefore, to be a pre-requisite for intelligent use of the halo method. In this regard Sears' (1971) paper is most instructive for he has attempted this type of work on the ores of Quebec. Points that stand out from his paper are:

- (a) The wide variation of mercury content within the ores, indicating that, in order to characterise a deposit, the assay of a suitably large number of samples is needed.
- (b) The lack of relationship between mercury concentrations and those

of the other ore metals (except for a broad correlation of mercury with zinc).

Sears gave no comparable figures for individual minerals within the ore, nor did he try to relate the mercury content to the temperatures of formation of the various ores. Accumulation of the fundamental data such as Sears has started to collect is vital.

An alternative hypothesis has been raised by Moiseyev (1971). He believes that a non-magmatic source is possible both for the formation of mercury ore deposits and for the formation of mercury haloes. He believes that mercury may be derived from intruded sediments, rather than magmas, and that it can be mobilised by volcanic heat. By his hypothesis, intrusive ore or magma may contain very little mercury but, if the surrounding host rocks contain a sufficient quantity, mercury may be mobilised in front of the ores and be reprecipitated where temperature and pressure conditions are suitable. Moiseyev's paper refers mainly to mercury deposits but he does say (p598) '....if the magma was emplaced in homogeneous rocks the mobilised mercury would deposit in an outer zone where the temperature change was insignificant. This deposition would nearly double or treble the geochemical Clarke (value) and form a 'geochemical anomaly'.'

Thus a complicating factor has been introduced. Moiseyev does not consider in his paper the intrusion of sulphide ores but clearly, if mercury haloes can be related to both orebodies and to magmatic intrusions, then some geochemical haloes (from the prospecting point of view) will be spurious, since a large number of surface halo discoveries will be related to magma intrusion rather than ore.

Additional work is needed to test this hypothesis. Some suggestion of its truth has been obtained in one area by the writer, but much better documented analytical work on samples which represent a passage outwards from a magma to cool country rock is needed.

Various papers have been written on the transport of mercury in hydrothermal solutions/fluids, and the origins of mercury deposits and haloes. The more important papers have been written by Thompson (1954), White (1955, 1967), White and Robertson (1962), Dickson (1968), Barnes et al. (1967), Krauskopf (1951) and Moiseyev (1968). Mechanisms postulated are transported as vapour, in solution as chloride (or chloride complex), carbonate and borate, or as a sulphide complex. Many of the studies have been based on limited observations

of hot springs followed by appropriate thermodynamic calculations. Mercury may then be concentrated in the rocks by condensation as the temperature falls (Moiseyev, 1971, gives a figure of 120°C below which mercury driven off in front of intruding magma is essentially fixed) by adsorption or absorption onto or into clays, limonite, etc., by fixation by diadochic substitution into the lattice of crystallising minerals or by precipitation as mercuric sulphide.

The above discussion centres on what may be termed the primary dispersion characteristics of mercury. Before the halo prospecting method can be used with success it is necessary to consider the behaviour and disperion of mercury in the supergene environment, for this will have a marked influence on the values of mercury obtained by soil or air determinations at or near the ground surface.

If sulphide ore is introduced and the mercury is an integral part of the sulphide component, either as cinnabar or in the lattice of some other sulphide, one or more of the following patterns of mercury dispersion may apply after deposition of the ore.

- (a) The mercury may remain fixed in sulphide either as cinnabar or in other sulphides. Presumably no further dispersion will take place until the sulphides crop out.
- (b) The mercury may remain fixed in the sulphide until the sulphide is oxidised. The mercury may then be released:
 - (i) as vapour, or
 - (ii) in solution

After some dispersion (of unspecified distance) it may then be trapped by one of the processes instanced earlier.

- (c) Mercury may be released at a uniform or slowly decreasing rate (?proportional to its concentration) from unoxidised sulphide and then behave as under (b) above.
- (d) Mercury released under (b) may migrate to the surface soil and then may be:
 - (i) held in the soil
 - (ii) released to the atmosphere as vapour, or
 - (iii) dispersed mechanically (trapped on clay), etc., by normal erosional processes.

(e) An equilibrium of uncertain nature may be set up between mercury in the atmosphere and mercury in the soil which, though possibly owing its origins to mercury released from ore, creates a pattern of mercury apparently unrelated to the position of ore.

If the mercury has been driven out as an envelope in front of the invading ore its behaviour will not only depend on the temperature of the ore but also on the capacity of mercury to volatilise (or remain volatile) or to be taken into solution, or on the presence of clays or other suitable substances to fix it.

It has been shown (by Rinse (1928), Krauskopf (1951) and Dickson (1968)) that there is a significant vapour pressure associated with HgS even when buried under water. If fluids (solutions, etc.) remove mercuric ions/atoms as they form, dissociation may go to completion over a period of time. However, it is not yet clear whether mercury can be released from sulphides which are not being oxidised; nor is it clear whether haloes can be detected when all the sulphide has been completely oxidised.

The interpretation of the behaviour may be rendered more difficult because of contamination. Kurbanyev and Is'kov (1964) discuss mercury values of samples collected in mines and conclude that they have been contaminated with mercury from mercuric fulminate used in detonators. Bradshaw and Koksoy (1968) discuss the same problem though with a different emphasis. Aidin'yan and Ozerova (1969) and Koksoy et al. (1967) point out the fact that mercury from mercury-rich samples may contaminate adjacent mercury-low samples when the two types of samples are stored together.

In recent years considerable prominence has been given to pollution by organic mercury, in particular the methyl-mercury compounds (e.g. Jonasson, 1970). It is more than conceivable that some of the methyl mercury travelling in ground water may be fixed by clays or by sesqui-oxides. In agricultural areas direct contamination by the addition of weedicides and fungicides is possible though the amounts of materials used are quite low. In South Australia, for example, cereals may be treated with 'Ceresan' or 'Leytosan' (1 to 1.5% Hg), at the rate of 2 to 4 ounces of these products per acre. Potatoes may be treated with either methyl-mercuric guanidine (0.61% Hg) or mercuric chloride (approx. 18% Hg). The latter is used at the rate of 3 to 4 ounces per 1000 sq.ft. Pine plantations are commonly treated with 'Lanetan', a spray which contains 3% mercury.

PART 2

EXPERIMENTAL WORK

1. INTRODUCTION

Experimental work to date has involved the assay of a number of cores taken at copper mines or prospects in South Australia. These cores are listed as follows:

| Locality | Drill Hole Number |
|------------|--|
| Parabarana | PDD2 PDD3 NFP22 |
| Kapunda | K6 K10 A4 |
| Kanmantoo | KS31 KS86 |
| Mutooroo | DDMM6 DDMM15D1 DDMM21A DDMM26 |
| Burra | BS4 |
| Moonta | DDM14 |

The Kapunda and Burra ores appear to contain secondary copper minerals (both carbonates and sulphides). Kanmantoo, Mutooroo and Moonta ores appear to contain primary copper (iron) sulphides, whereas the material from Parabarana was received in a crushed state and the copper mineralisation was understood to be primary.

The cores from Parabarana and Kanmantoo were submitted for assay for copper, lead and zinc as well as mercury. Copper values were already known for the Kapunda ores and these samples were submitted for lead and zinc in addition to mercury. The Burra samples were analysed for copper and mercury only.

Samples from the remaining cores were submitted for mercury assay after a binocular microscope examination of the cores to determine whether or not sulphide mineralisation was present.

2. ANALYTICAL METHODS

Copper, lead and zinc were determined by conventional atomic absorption analysis after digestion of finely ground portions of the samples in hot concentrated perchloric acid.

Several techniques were used for the determination of mercury in the samples. The original technique was to use the Lemaire mercury detector. The method here is to heat the powdered sample and drive off mercury. The mercury is diluted with air in constant volume and then driven through a flameless absorption cell. The mercury is determined by reference to known standards, using the absorption line at 2534 Å.

Other techniques now include those of Hatch and Ott (1968) with variations on the technique. The method involves a cold nitric acid digestion of the sample followed by reduction using stannous chloride in combination with another reducing agent such as hydroxylamine hydrochloride. Mercury in the sample is extracted into the acid solution by nitric acid and is reduced to mercury by the reducing agents. An inert gas is then blown through the solution, taking the volatile mercury out of the liquid through an atomic absorption chamber in the position usually occupied by a flame, and the absorption is measured at 2534 Å. Calibration is made by using samples of known mercury concentration and reagent blanks.

The Amdel procedure finally adopted includes digestion with a mixture of cold concentrated nitric and hydrochloric acids (aqua regia), containing free bromine. Aliquots of the solution are treated with stannous chloride as already described and the reduced mercury is determined by the standard flameless atomic absorption method.

Additional confirmatory testing was carried out by emission spectrography.

3. RESULTS

Assay results are presented in Tables 8 to 13.

4. DISCUSSION

4.1 Assay Data

There have been considerable problems in obtaining reliable analytical data. The technique using the Lemaire mercury analyser is satisfactory for the determination of mercury in soils which contain little chloride, sulphate and

carbonaceous material. The actual extent of interference and loss of signal when determining mercury in rocks, and, in particular, sulphide-bearing rocks, had not been fully realised until values were obtained which were clearly well below accepted Clarke values.

A programme of experimentation followed in which the second technique cited in the section on methods (Section 2), was developed. It was then found that various other elements interfered with analysis for mercury. Of these, gold, selenium and tellurium, when present to the extent of a few parts per million in solution, lowered the apparent mercury content of solutions spiked with mercury very appreciably, because of either the precipitation of mercury selenide or telluride or the formation of a gold amalgam. By reducing aliquots of mercury (and increasing the dilution) satisfactory values were obtained for the spiked samples.

No major problems have been found in analysis of lead and zinc ores using acid digestion followed by reduction of the mercury but, as Band and Wilkinson (1972) discovered, low values were obtained for copper-bearing ores/solutions. Part of the answer to this lies in increasing the digestion time for acid attack on the sample, since copper in chalcopyrite seems more difficult to get into solution than copper in other forms. Copper is relatively easily reduced by stannous chloride solutions and it is believed that the reduced copper may trap some of the mercury. However, the mechanism of entrapment is not fully known.

The samples received in connection with the project have been analysed for mercury in a number of ways. However, the results presented in Tables 8 to 13 are all generated by the same technique of analysis.

It is now believed that the results genuinely represent the mercury content of the ores - with one reservation which will be discussed later. The samples have been assayed in duplicate and the average value is given. The precision of analysis is $\pm 15\%$ relative.

As with all samples in which mercury is sought, there are possibilities of loss of mercury to the air, or of contamination of the samples by mercury accreted from the air, analytical apparatus or chemicals. No adequate assessment can be made of the former, but contamination can be checked by the use of solution blanks.

Recent investigations on a number of containers have shown that glass is the ideal storage medium when mercury analysis is to be undertaken. Most plastic 946

containers (in particular, Duranol containers) also appear to be free of mercury. However, paper sample-bags of the type commonly used for the collection of soil samples contain up to 5 ppm of mercury and will undoubtedly contaminate samples stored in them. It is believed that these bags are deliberately treated with a mercury compound which is a fungicide to enable the bags to store wet samples.

Fortunately the samples of which assays are listed in this Report have been kept mainly in plastic bags. However, in the periods between re-assaying, they have from time to time been in reasonable proximity to samples in paper sample-bags. The effects of possible contamination from this source cannot be assessed at this stage.

4.2 Discussion of Results

The mercury values now finally determined are similar to overall values published for rocks of their respective types around the world. In unmineralised rocks the mercury values range between about 30 and 150 ppb with the exception of DDH14 at Moonta where the overall values of mercury are appreciably higher, commonly well over 150 ppb. However, in all discussion of background values it is necessary to remember the volatility of mercury. The rocks at Moonta are mineralised over considerably greater lengths than most of the rocks from the other cores, and with the extra volatility of mercury the higher values may reflect the greater overall degree of mineralisation — or the possibility of unsuspected mineralisation.

Of interest are the values near the surface. In the upper 15 to 30 m (50 to 100 ft) of each core, a considerable degree of weathering may be expected with some formation of clay. As clay has long been known as a collector of mercury, enhanced values of mercury are to be expected in these parts of the cores if mercury has been able to diffuse upwards. It is therefore particularly interesting to note that at Kanmantoo (Table 10, Figs 10, 11), almost the highest values of each core occur at 1.5 to 7.5 m (5 to 25 ft) depth. Other samples in the possible weathered zones at Kanmantoo do not show particularly high values. Of the other cores, this near-surface enrichment is only shown at Mutooroo (Table 11, Figs 13, 14). In this case the surface mercury may possibly be unrelated to the sulphide mineralisation at depth since, for the most part, mineralisation occurs at depths greater than 300 m (1000 ft) below the surface. A series of surface soil or rock samples over the line of known mineralisation should be made, after assessing the position of possible channelways from the ore to the surface. A high value of mercury at 38 m (126 ft) in the Moonta

core may also represent mercury trapped by clay.

Reference to the down-hole profiles, or the assays themselves, shows that both at Burra (Fig. 15) and at Kapunda (Figs 1, 2), mercury occurs in association with the copper mineralisation. At Burra, mercury has a maximum value of 300 ppb whereas at Kapunda the maximum value recorded in K10-12 is about 1 ppm.

Elsewhere there does not appear to be any significant increase in mercury with the copper mineralisation. However, it is not absolutely clear whether this is due to interference in the determination of the mercury by the high concentration of copper or to inherently low mercury values. Band and Wilkinson (1972) comment that on normal assay there is a negative correlation between mercury and copper in massive sulphides. This they ascribe to purely analytical problems. As far as the present determinations are concerned, the values of mercury found in rocks containing abundant sulphide are believed to be accurate.

The pattern of the relationships of copper and mercury at Kanmantoo KS31 is particularly interesting. Although the correlation coefficient is only +0.18, it will be seen from the down-hole profile (Fig. 8) that there is a decidedly sympathetic behaviour pattern between copper and mercury. In general, when copper increases so does mercury and vice versa. The scatter plot of copper versus mercury (Fig. 11) suggests two trends, both with a simple positive relationship. When the high copper values are removed from consideration and the correlation coefficient is recalculated, it increases to +0.53 and the scatter diagram (Fig. 12) emphasises the relationship.

Overall, at Kanmantoo (when both holes are considered) there is a reasonable correlation of zinc with mercury, and lesser but positive correlations of lead and copper with mercury (Table 15).

The same treatment has been carried out on the results of the determinations made on the Parabarana samples. Down-hole profile plots (Figs. 3, 4, 5) do not indicate any great correlation of copper with mercury or of zinc with mercury. Correlation coefficients derived from the analyses of all the samples indicate absolutely no correlation between mercury and any of the three other metals (Table 14 A). However, recalculation of the correlation coefficients for 'non-mineralised' samples - that is, those samples showing less than 500 ppm copper - does show a positive relationship (and a correlation coefficient of +0.32) between copper and mercury, the coefficients for lead and zinc with mercury remaining much as before (Table 14 B). The scatter diagrams (Figs. 6, 7) show this trend but indicate visually the widely dispersed spread of values.

No copper, lead or zinc determinations were made for the remaining two areas. It was not felt that this was justified until an analytical technique for mercury had been shown to be demonstrably sound. However, down-hole profile plots were drawn for mercury against the depth in the hole with the position of sulphides visible in the hand specimen marked on them.

The sample from Mutooroo (Figs 13, 14) showed some variations and, in many cases, there is an enhancement of mercury at the position of the sulphides at or in the upper portion of the sulphide zone. Moreover, although other mercury peaks occur elsewhere where there are no visible sulphide minerals, the highest values of mercury in DDM6 and DDM7 occur in the vicinity of sulphide mineralisation. The tendency to higher values of mercury at or near the surface at Mutooroo has already been noted, and it does appear that there is a limited but positive relationship between mercury and the presence of sulphides.

At Moonta (Fig. 16), although the mineralised zones are generally richer in mercury, the correspondence is by no means exact and several sulphide-bearing samples show low mercury values. Again, this may be due to interference by copper or other elements.

5. CONCLUSIONS AND RECOMMENDATIONS

With the continuing interest in the possibilities of using volatile elements in prospecting for base metal or gold orebodies, it is too soon to discount the value of the mercury halo prospecting technique without further research; however, it must be said that progress in this project has not been ideal, largely due to analytical difficulties and possible sample contamination. Work is continuing on the development of an adequate, economic method for making geochemical determinations of mercury in rocks, particularly those with sulphides. The method currently in use has been found to be viable for samples of all types but results reflect the sample as presented for analysis and do not take into account possible contamination or loss of mercury between collection and the time of analysis.

The results to date, however, show a low but positive correlation of mercury with low levels of copper, and also show a tendency for there to be some concentration of mercury in the oxidised, near-surface zone above copper orebodies. This is mildly encouraging but not as informative as was hoped when the project commenced.

It is considered that continuation of the project is/will be worthwhile once the problems of sampling, contamination and analysis have been overcome with certainty.

The results have demonstrated that, before embarking on a project of this nature, rigid controls must be applied at all stages from the collection of samples through to the determination of mercury values. However, the premises on which the project was formulated, namely that mercury is a common associate of base-metal ore deposits or of gold ore deposits and should be a useful guide to ore, are still valid; and it may confidently be expected that, in the future, mercury haloes can be used to determine the position of blind orebodies. Fewer problems occurred in the determination of mercury in soils which had been collected and preserved in Duranol containers. The greater part of the results obtained by assay of the soils by Amdel is considered to be valid.

It is recommended that the project be held in abeyance until a completely satisfactory sampling, storage and analytical scheme has been developed. It is anticipated that a further project proposal will be submitted when the analytical method has been upgraded.

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TABLES 1 to 15 FIGURES 1 to 16

TABLE 1: ESTIMATES OR MEASUREMENTS OF MERCURY IN ROCKS (Selected Figures Only)
Values in ppb mercury

| Author | Basic | Inter- mediate | Acid | Lime- stone | Shale/ Clay | Sand- stone | Other | Remarks |
|--|--------|-------------------|-----------------|----------------|----------------|----------------|---|--|
| Preuss (1940) | 100 | | 10 | | 300 | 10 | | Estimated average values |
| Hawkes and Webb (1962) | 90 | | 40 | 30 | 400 | 30-100 | | Estimated average values |
| Turekian and Wedepohl (1961) | 90 | XO | 80 | 40 X0 | 400 X00 | 30 | XO, ultrabasic rocks | Estimated average values Deep sea sediments |
| Krauskopf (1967) | 80 | | 80 | | 400 | | | Estimated average values |
| Vinogradov (1962) | 90 | 50 | 80 | | 400 | | 10, ultrabasic rocks | Estimated average values |
| Afanas'yev and Aidin'yan (1961) | 30-250 | 65-500 | 60-200 | | 40-60 | | 60(1) schist | North Caucasus. These rocks unmineralised but near mineral-isation |
| Ozerova (1962) | | 60-200 | | | | | | Nepheline syenites, USSR |
| Bulkin (1962) | 17600 | | 70-500 | 4000 | 2000 | 5000 | | Average values, USSR Crimea mercury province. |
| Ozerova and Unanova (1965) | • | - 20-200 - | > | | | | | Lavas, unspecified, Kamchatka |
| Aidin'yan et al. (1966) | | 140-580 | | | | | | Alkali syenites, USSR |
| Ozerova and Aidin'yan (1966a) { | | | | 31 | 35 85 | 39 97 | 80-200 limonite | Russian platform, average values Kamchatka, average values |
| Ozerova and Ardin yan (1966) | | | | | 400 | | 70 phosphorites | Combustible shales, Volga River |
| Golobnynya (quoted by Ozerova and Aidin'yan (1966a) | | | | | | | 57(2) quartzite 51(5) paragneiss 47(7) granite gneiss 57(5) orthoamphibolite | Average values, Valdai Series, phyllites to granite |
| | | | | | | | 7–28 | Irtysh, USSR |
| Karasik & Morosov (1966) | | | | | 100-900 400 | | | Kerch Peninsula, average values average values |
| Dvornikov (1967d) | | | | | | | 20-100 coal | Donets Basin, background values |
| Jovanovic & Reid (1968) | | | | | | | 193(1) pelitic schist | Vermont, USA |
| | | | | | | | 18(1) amphibolite | Vermont, USA |
| Aidin'yan et al. (1969) | | 10-75 | <u></u> | • | | | | Estimated average values |

For a much more detailed list of measurements of mercury in rocks see USGS Prof. paper 713, Tables 2 to 12. Figures in brackets in column 'Other' indicate numbers of samples analysed.

TABLE 2: MEASUREMENTS OF MERCURY IN UNMINERALISED SOILS

| Author | Mercury V | alue | Remarks |
|---------------------------------------|--|------------------------------------|---|
| · · · · · · · · · · · · · · · · · · · | Range | mean | |
| Stock & Cucuel (1934) | 30- 34 1- 29 | 70 | Average value, soils Clay soil Sand |
| | 30- 81 100- 290 | • | Humic horizon, forest soil Forest soil |
| Dvornikov & Petrov (1961) | | 30 | Average, background clay soils, Nagol'nyi Range |
| Hawkes & Webb (1962) | 30- 300 | | Average value, soils |
| Dvornikov et al. (1963) | | 20 400 | Average background, Nagol'nyi Range Soil over dispersed mineralisation (Hg province) |
| Aidin'yan et al. (1964) | 15- 75 10- 40 8- 36 10- 30 | 30 | Average values Forest podsol Chernozem Latosol Clays with laterite |
| Warren, Delavault & Barakso (1966) | 10- 50 50-2500 250-2500 | | Unmineralised soils, British Columbia Near mineralisation, British Columbia Above mineralisation, British Columbia |
| Karasik & Morozov (1966) | | 300 | Clay soils of Kerch Peninsula (Hg province) |
| Williston (1968) | 20- 40 100- 200 | | Unmineralised, California Franciscan Formation, California |
| Anderssen (1967) | - 60 - 23 | | Sweden Africa |
| Jonasson & Boyle (1971) | 20- 150 20- 100 60- 200 30- 140 25- 150 50- 200 | 70 50 161 89 96 100 | Unmineralised soils (estimated) Glacial tills Soil, mineralised belt, A horizon Soil, mineralised belt, B horizon Soil, mineralised belt, C horizon Iron enriched, weathered crusts |

TABLE 3: SOME VALUES FOR MERCURY IN ROCKS, SOILS AND STREAM SEDIMENTS IN MINERALISED AREAS (Values in ppb unless otherwise stated)

| Author | Sample Type | No. | Minimum | Maximum | Average | Type Mineralisation | Locality |
|-------------------------------|--|-----|--|--------------------------------------|--|---|---|
| Fursov (1958) | Sedimentary rocks Bedrock above ore | 13 | <300 | >1000 | 1.7 ppm | Pb, Zn, pyrite | Achisai, Kazakhstan |
| Ozerova (1959) | Sedimentary rocks | | < 10 | 500 | | Pb, Zn | Turkestan |
| Sergeyev (1961) | Sediments | | < 20 | 30ppm | | Base metals | Khaidarkanskoe, USSR |
| Fedorchuck (1961) | Limestones Dolomites Marls Shales | | <pre>< lppm < lppm < lppm < lppm < lppm</pre> | 480ppm 180ppm 140ppm 120ppm | 2 ppm 2 ppm 4 ppm 3 ppm | Hg rocks adjacent to orebody background at distance 0.1 ppm | Middle Asia and Pacific provinces USSR |
| Dvornikov & Petrov (1961) | Soil Shale | | 80 | 300 | 830 | Hg Hg | Khpek, Dagestan Manuson, Georgia, USSR |
| | Shales Limestone | | | | 700 300 | (Hg | Ak-Tash, Dagestan |
| | Limestones Sandstones | | | | 200 1.35ppm | (Hg | Chagan-Uzun, Dagestan |
| | Soil, 'A' horizon | | 80 | 9.5ppm | 2100pp | Polymetallic | Nagol'nyi Range, USSR |
| Dvornikov (1962) | Limestone Shale Sandstone | | < 70 Trace 100ppb | 22ppm 4.4ppm 3ppm | 4 ppm 1.9 ppm 700 ppb | <pre>Polymetallic Pb-Zn Hg</pre> | Nagol'nyi Range (one area) Donets Basin, USSR |
| Dvornikov et al. (1963) | Soils | | 20 | 2.7ppm | 400 ppb (Dispersed mineralisation) | Polymetallic and Au | Nagol'nyi Range |
| Dvornikov (1964) | Clay shales } Sandstones } | | | | 20–40 | Background values | S.E. Donets Basin |
| | Limestones } | | 100 | 1000 | 570 | Dispersed mineralisation ?uncertain type | S.E. Donets Basin |
| | Coal | | | | 20–40 600 | Background | S.E. Donets Basin Nagol'nyi Range |
| Karasik & Bolshakov (1964) | Soil | | 5ppm | 60ppm | | Hg | Nikitovka, USSR |
| Bolshakov (1964) | Soil Sandstone Shale | | • | | 1.7 ppm 700 ppb 500 ppb | Dispersed Hg Dispersed Hg Dispersed Hg | Nikitovsk, USSR Nikitovsk, USSR Nikitovsk, USSR |
| Berce (1965) | Shales Dolomite | | < 1ppm | >1000ppm | | Hg | Idrija, Yugoslavia |
| Gott & McCarthy (1966) | Various rocks | 500 | | 70ppm | 180 | Au, Ag, Te | White Pine Co. Nevada |
| Erickson et al. (1966) | Limestones Jasperoids | | 10 . | 3.2ppm | | Au, Sb, W | Cortez, Nevada |

TABLE 3: CONTINUED

| Author | Sample Type | No. | Minimum | Maximum | Average | Type Mineralisation | Locality |
|--------------------------------|---|-----|-----------------------|---|---|--------------------------|--|
| Missaghi (1966) | Stream sediment Acid volcanics Limestone Quartzite | | <150 < 50 | 1270 196 | | Zn, Pb, Cu, Ag, Au | Magdalena District, New Mexico |
| Friedrich & Hawkes (1966b) | Soil Volcanic rocks (acid) | | < 20 < 30 | 340 1300 | | (Cu, Pb, Zn, As | West Shasta, California |
| Friedrich & Hawkes (1966a) | Andesites Soil | | < 10 < 50 | 2600 1900 | 300 | (Pb, Zn, Cu, Ag, pyrite | Hidalgo, Mexico |
| Brown (1966) | Soil | | < 20 | 200ppm | 1ppm | Hg | British Columbia |
| Brown (1967) | Soil | | 10 | 175 | 50 | Mo and Au, Bi | British Columbia |
| Cornwall <i>et al</i> . (1967) | Altered andesites Soil and altered andesites Soil, sandstones, shale | | 100 20 20 | <pre>> 300 > 200 > 300 (to 2000 when soil also contained silver)</pre> | | Au, Ag Au, Ag Ag | Comstock, Nevada Tonopah, Nevada Silver Reef, Utah |
| Gott <i>et al</i> . (1967) | Altered phonolites | | | | 190 | Au | Cripple Creek, Colorado |
| Bradshaw & Koksoy (1968) | Sericite Schist | | 33 100 | 45 350 > 500 | Background dispersed mineralisation over ore | Hg, Sb | Halikoy, West Turkey |
| | Metasediments Biotite schist Marble Granite Unaltered andesite Altered andesite | | 20 180 23 68 | 35 700 35 > 500 >10000 | Background Dispersed mineralisation √200 | n (Нg | Ivrindi, Turkey |
| Haas (1968) | Soil Soil | 5 | 1000 <110 | > 4000 2230 | | | Secret Canyon, Nevada Roberts Mt. Thrust, Nevada |
| Akright et al. (1969) | Sedimentary rocks | | 20 | 140 | | Au | Carlin Mine, Nevada |
| Crosby (1969) | Greenschist Metasediments Samples in adits Cross cuts | | 20 | > 1000 | | Pb, Zn | Coeur d'Alene, Idaho |

TABLE 3: CONTINUED

| Author | Sample Type | No. | Minimum | Maximum | Average | Type Mineralisation | Locality |
|--------------------|--------------------|------|---------|---------|---------|---------------------|-------------------|
| Koksoy & Bradshaw | Weathered metamor- | • | | | | | |
| (1969) | phics | 19 | 33 | 700 | | (| TI-141 T1 |
| | Soil (-80#) | 22 | 45 | 375 | | ₹ | Halikoy, Turkey |
| | Stream sediments | 15 | 53 | 226 | | Hg | |
| | Soil | 13 | | | | • | |
| | Stream sediments | 5 | | | | | |
| Dall'aglio (1971) | Stream sediments | 1000 | < 50 | 50ppm | | Hg | Tuscany, Italy |
| Friedrich & Pluger | Soils | | < 50 | >2000 | | Ва | Dreislar, Germany |
| (1971) | Soils | | < 20 | > 600 | | CaF ₂ | Osor Mine, Spain |

TABLE 4: MERCURY CONTENTS OF SOME COMMON ORE AND GANGUE MINERALS*

| Mineral | Composition | 'Normal' Range Limits ppm | Highest Reported Content % |
|--------------------|--|---------------------------------|----------------------------------|
| Tetrahedrite | Cu ₁₂ Sb ₄ S ₁₃ | 10 - 1000 | 17.6; 21 |
| Grey copper ores | (Cu,As,Sb) _x S _y | 5 - 500 | 14 |
| Sphalerite | ZnS | 0.1 - 200 | 1 |
| Wurtzite | ZnS | 0.1 - 200 | 0.03 |
| Stibnite | Sb ₂ S ₃ | 0.1 - 150 | 1.3 |
| Realgar | AsS | 0.2 - 150 | 2.2 |
| Pyrite | FeS ₂ | 0.1 - 100 | 2 |
| Galena | PbS | 0.04- 70 | 0.02 |
| Chalcopyrite | CuFeS | 0.1 - 40 | |
| Bornite | Cu ₅ FeS ₄ | 0.1 - 30 | |
| Bournonite | PbCuSbS ₃ | 0.1 - 25 | |
| Chalcocite | Cu_2S | 0.1 - 25 | |
| Marcasite | FeS ₂ | 0.1 - 20 | 0.07 |
| Pyrrhotite | Fe _{1-X} S | 0.1 - 5 | |
| Molybdenite | MoS ₂ | 0.1 - 5 | |
| Arsenopyrite | FeAsS | 0.1 - 3 | |
| Orpiment | As ₂ S ₃ | 0.1 - 3 | |
| Native gold | Au | 1 - 100 | 60 |
| Native silver | Ag | 1 - 100 | 30 |
| Barite | BaSO4 | 0.2 - 200 | 0.5 |
| Cerussite | PbC0 ₃ | 0.1 - 200 | 0.1 |
| Dolomite | CaMg(CO ₃) ₂ | 0.1 - 50 | |
| Fluorite | CaF ₂ | 0.01- 50 | 0.01 |
| Calcite | CaCO ₃ | 0.01- 20 | 0.03 |
| Aragonite | CaCO ₃ | 0.01- 20 | 3.7 |
| Siderite | FeCO ₃ | 0.01- 10 | 0.01 |
| Silica (all forms) | SiO ₂ , SiO ₂ .nH ₂ O | 0.01- 10 | |
| Pyrolusite | $Mn0_2$ | 1 - 1000 | 2 |
| Limonite | Fe ₂ 0 ₃ .nH ₂ 0 | 0.1 - 500 | 0.2 |
| Graphite | . C | 0.5 - 10 | 0.01 |
| Coal | | 0.05- 10 | 2 |
| Gypsum | CaSO4.2H20 | 0.01- 4 | |

^{*}Table 1 of Jonasson and Boyle, 1972, p34

TABLE 5: DEPTH OF BURIAL OF DISCOVERED OREBODIES

| Author | Region | Depth | Mineralisation |
|----------------------------|--------------------|------------|-----------------|
| Friedrich & Hawkes (1966b) | California | 15 - 65 m | Cu, Pb, Zn, Ag |
| Friedrich & Hawkes (1966a) | Mexico | 120 m | Cu, Pb, Zn, Ag |
| Fursov (1958) | Kazakhstan | 25 -300 m | Pb-Zn |
| Bol'shakov (1964) | Nitikovsk, USSR | 200 -300 m | Hg, minor-Pb,Zn |
| Bercé (1965) | Yugoslavia | 100 m | Hg |
| Friedrich (1968) | Idaho | 30 m | Cu, Pb, Zn |
| Ozerova (1959) | Fergana, Turkistan | 100 m | Pb, Zn |

TABLE 6: SOME RECORDED MERCURY CONTENTS OF NATURAL WATERS Values in ppb

| Author | Samples | Minimum | Maximum | Average | Location |
|---|---------------------------------------|--------------|--------------|-------------|---|
| Stock & Cucuel (1934) | | | | 0.1 | Rhine River |
| | | 0.05 0.01 | 0.48 0.05 | 0.03 0.2 | Seawater Rainwater Springs, Germany |
| Hawkes & Webb (1962) | | 10 | 100 | | Average, fresh water |
| Aidin'yan & Belavskaya (1963) | | 1 | 2 | | Danube |
| • | 9 | 0.4 | 1.6 | 1.2 | Atlantic Ocean |
| Karasik <i>et al</i> . (1965) | 26 | <1 | 6.5 | | Permian salt beds, USSR |
| Karasik & Morozov (1966) | | <1 | 2.5 | | Mud volcanoes, Kerch |
| Kraynov et al. (1966) | .37 | <0.005 | 80 | <1 | Springs, Elbrus, Caucusus |
| Krauskopf (1967) | | | | 30 | Average, marine |
| Dvornikov (1967c) | | | | 0.3 | North Sea, average |
| White & Robertson (1962) | | | 20 | | Sulphur Bank Springs (HgS depositing) |
| obta | or 21 valu ained out 70 samples | of | | | |
| | assayed | <1 | 10 | | Donets Basin |

TABLE 7: MAXIMUM MERCURY CONCENTRATIONS IN AIR (Table from Fleischer et al.(1970) with additions)

| Location | Mineralisation | | Maximum Hg Concentration, ng/m³ | | | |
|--|-------------------|----------------|---|------------|----------------|--|
| | | Soil Air | Air Ground Surface | | 0 m (400 ft) | |
| Ord Mine, Arizona | Hg | à | 20,000 (50 readings) | 108 | (4 readings) | |
| Silver Cloud, Nevada | Hg | - | 2,000 (50 readings) | 24 | (8 readings) | |
| Dome Rock, Arizona | Hg | , - | 128 (6 readings) | 57 | (20 readings) | |
| Cerro Colorado Mts. Arizona | Base metals/Au/Ag | - | 1,500 (5 readings) | 24 | (2 readings) | |
| Cortez (Au), Nevada | Au | - | 180 (60 readings) | 55 | (4 readings) | |
| Coeur d'Alene, Idaho | Base metals/Ag | - | 68 (40 readings) | | | |
| San Xavier, Arizona | ? | | - | 25 | (3 readings) | |
| Silver Bell, Arizona | Cu | . - | - | 52 | (3 readings) | |
| Esperanza, Arizona | Cu | | <u>-</u> | 32 | (3 readings) | |
| Vekol, Arizona | Cu | - | - | 32 | (4 readings) | |
| Ajo, Arizona | Cu | | . | 30 | (3 readings) | |
| Mission, Arizona | Cu | - | - | 24 | (3 readings) | |
| Twin Buttes, Arizona | Cu | - | 20 | 22 | (3 readings) | |
| Puma, Arizona | Cu | *** | - | 13 | (3 readings) | |
| Safford, Arizona | Cu | _ | - | 7 | (2 readings) | |
| Blythe, California | None | · | - | . 9 | (20 readings) | |
| Cula Bend, California | None | | - | 4 | (2 readings) | |
| Salton Sea, California | None | | - | 3 | 5 (2 readings) | |
| Los Altos, California (Williston, 1966) | None | - | 1 to 20 | He | ight unknown | |
| Arivaca, Arizona | None · | - | - | 3 | (2 readings) | |
| Nikitovka (Kavasik & Bolshakov, 1964) | Base metal + Hg | Range 2000 to | 16000> 40000 underground air in mine | | | |
| Various, USSR (Fursov, 1970) | Hg | | 10000 - background values soi 30000 in adits of mine | l air abov | e ore | |

TABLE 8: ASSAY OF KAPUNDA DRILL CORES

| Drill Hole | C | D.L. | 7 | |
|-------------|-------|----------|------------------------|------|
| | Cu | Pb | $\mathbf{Z}\mathbf{n}$ | Hg. |
| and Footage | % | ppm | ppm | ppb |
| к6 – 6 | 0.04 | 5 | 45 | 130 |
| 16 | 0.55 | .5 .5 | 120 | 315 |
| 26 | 0.09 | 10 | 5 | 130 |
| 35 | 0.02 | 5 | < 5 | 50 |
| 41 | <0.01 | 5 | 20 | 100 |
| K10 - 4 | 0.60 | 5 | 20 | 100 |
| 12 | 1.20 | 5 | 45 | 1000 |
| 21 | 0.03 | 10 | <5 | 100 |
| 29 | 0.015 | 10 | 10 | 325 |
| 39 | 0.01 | <5 | 20 | 350 |
| 55 | 0.015 | <5 | 20 | 130 |
| | ppm | , J.S. | | |
| A4/6 | ≪80 | 25 | 10 | 100 |
| /12 | <50 | 25 | 45 | 130 |
| /17 | 50 | 25 | 10 | 30 |

TABLE 9: ASSAY OF PARABARANA DRILL MATERIAL

| Drill Hole | Departmental | Cu | Pb | Zn | Hg |
|-------------|--------------|-----------|-----------------------|----------|----------|
| and Footage | Sample No. | ppm | ppm | ppm | ppb |
| PDD2 20 | 5441 | 5 | 15 | 20 | 60 |
| 50 | 5442 | 10 | <5 | 20 | 110 |
| 100 | 5443 | 80 | 20 | 220 | 90 |
| 300 | 5444 | 20 | 15 | 25 | 30 |
| 500 | 5445 | 20 | 5 | 85 | 50 |
| 600 | 5446 | 110 | 15 | 130 | 50 |
| 650 | 5447 | 35 | 5 | 75 | 80 |
| 660 | 5448 | 35 | 5 5 5 5 5 | 90 | 50 |
| 670 | 5449 | 30 | 5 | 120 | 50 |
| 680 | 5450 | 740 | 5 | 70 | 30 |
| 685 | 5451 | 55 | 5 | 75 45 | 80 |
| 700 | 5452 | 2100 | 5 | 40 | 50 |
| 710 | 5453 | 310 | 15 | 70 | 110 |
| 720 | 5454 | 20 | 10 | 50 | 50 |
| 730 | 5455 | 20 | 15 | 50 | 30 |
| 756 | 5456 | 220 | 5 | 40 | 80 |
| 750 | 5450 | 220 | , | 40 | 00 |
| PDD3 500 | 5457 | 25 | 15 | 210 | 25 |
| 700 | 5458 | 110 | 120 | 150 | 40 |
| 800 | 5459 | . 5 | 15 | 55 | 50 |
| 900 | 5460 | 15 | 5 | 20 | 50 |
| 1000 | 5461 | 5 | 5 | 55 | 50 |
| 1020 | 5462 | 50 | 5 | 40 | 110 |
| 1040 | 5463 | 30 | 5 | 70 | 40 |
| 1050 | 5464 | 5 | 20 | 90 | 30 |
| 1060 | 5465 | 50 | 20 | 55 | 170 |
| 1070 | 5466 | 2200 | 5 | 120 | 40 |
| 1110 | 5467 | 2.6% | 40 | -60 | 30 |
| 1150 | 5468 | 310 | 10 | 110 | 70 |
| 1190 | 5469 | 90 | 15 | 30 | 120 |
| 1200 | 5470 | 10 | 10 | 40 | 140 |
| 1210 | 5471 | 140 | 10 | 25 | 120 |
| 1220 | 5472 | 5 | 10 | 20 | 70 |
| 1230 | 5473 | 10 | 35 | 45 | 120 |
| 1250 | 5474 | 100 | 40 | 85 | 170 |
| 1300 | 5475 | 10 | 20 | 80 | 110 |
| 1400 | 5476 | 60 | 45 | 250 | 40 |
| NPP22 5 | 5478 | 25 | 300 | 1100 | 70 |
| 10 | 5476 5479 | 35 120 | 390 2000 | 2600 | 70 80 |
| 20 | 5480 | 95 | | | |
| 50 50 | 5480 5481 | | 460 25 | 1300 | 100 |
| | 5482 | 15 05 | 35 45 | 120 | 100 |
| 100 | | 95 10 | 45 95 | 120 | 140 |
| 150 | 5483 5484 | 10 | 85 120 | 150 | 40 50 |
| 200 | 5484 5485 | ·5 | 120 | 120 | 50 |
| 220 | 5485 | 35 15 | 200 | 310 | 100 |
| 230 | 5486 | 15 | 350 | 310 | 70 |

TABLE 9: CONTINUED

| Drill Hole and Footage | Departmental Sample No. | Cn ppm | Pb ppm | Zn ppm | Hg ppb |
|---------------------------|----------------------------|-----------|-----------|-----------|-----------|
| 235 | 5487 | 95 | 530 | 270 | 70 |
| 240 | 5488 | 4400 | 60 | 140 | 50 |
| 245 | 5489 | 1750 | 20 | 100 | 90 |
| 300 | 5491 | 1600 | 20 | 45 | 90 |
| 325 | 5492 | 4100 | 20 | 55 | 60 |
| 375 | 5493 | 1.1% | 20 | 85 | 70 |
| 425 | 5494 | 1.6% | 35 | 55 | 70 |
| 445 | 5495 | 5300 | 10 | 40 | 80 |
| 465 | 5496 | 1650 | 10 | 30 | 100 |
| 475 | 5497 | 890 | 15 | 25 | 100 |
| 485 | 5498 | 2000 | 35 | 45 | 150 |
| | | • | | | |

| Drill Hole and Footage | Departmental Sample No. | Rock Type | Cu ppm | Pb ppm | Zn ppm | Hg ppb |
|---------------------------|----------------------------|------------------------------------|-------------|-----------|-----------|-----------|
| | | | 170 | 30 | 300 | 170 |
| KS 31 - 5 | G 5499/72 | mica schist | 170 | | | 100 |
| 25 | 5500 | | 90 | 20 | 260 | |
| 50 | 5501 | <u>,</u> | 35 | 20 | 190 | .50 |
| 100 | 5502 | | 35 | 290 | 450 | 50 |
| 300 | 5503 | | 45 | 710 | 2500 | 60 |
| 500 | 5504 | | 70 | 75 | 180 | 80 |
| 700 | 5505 | J | / 15 | 30 | 310 | 50 |
| 900 | 5506 | andalusite } mica schist } | 45 | 10 | 130 | 100 |
| 1000 | 5507 | , | 25 | 10 | 160 | 25 |
| 1050 | 5508 | ferruginised zone | 4.4% | 40 | 90 | 5.5 |
| 1100 | 5509 | andalusite } quartz mica } | 100 | 10 | 65 | 30 |
| 1200 | 5510 | garnet schist | 890 | 15 | 70 | 35 |
| 1300 | 5511 | altered mica schist | 8700 | 15 | 50 | 55 |
| 1375 | 5512 | andalusite | 210 | 10 | 95 | 80 |
| 1400 | 5513 | mica | 10 | 15 | 120 | 30 |
| 1415 | 5514 | schist | 240 | 10 | 65 | .50 |
| 1425 | 5515 | } | 20 | 10 | 90 | 80 |
| 1435 | 5516 | | 20 | 10 | 85 | 80 |
| 1455 | 5517 | J | 30 | 15 | 5.5 | 80 |
| 1475 | 5518 | schist quartzite | 470 | 10 | 55 | 130 |
| 1500 | 5519 | andalusite | 10 | 15 | 110 | 80 |
| 1550 | 5520 | mica | 3700 | 10 | 85 | 60 |
| | 5521 | schist | 100 | 15 | 40 | 80 |
| 1585 | 3321 | chlorite | | | | |
| 1605 | 5522 | schist | 250 | 15 | 75 60 | 80 30 |
| 1615 | 5523 | andalusite | 10 | 5 | | 40 |
| 1625 | 5524 | chlorite | 1200 | 5 | 65 | |
| 1635 | 5525 | schist) | 280 | 5 | 80 | 100 |
| 1655 | 5526 | schist with kyanite | 10 | 5 | 85 | 50 |
| 1675 | 5527 | schist | 150 | 10 | 95 | 130 |
| 1700 | 5528 | } | 110 | 5 | 95 | 160 |
| 1800 | 5529 | | 50 | 5 | 80 | 40 |
| 2000 | 5530 | quartz muscovite schist | 240 | 20 | 140 | 100 |
| 2300 | 5531 | quartz mica amphibole schist | 110 | 15 | 100 | 20 |

TABLE 10: CONTINUED

| Drill Hole | Departmental | Rock | Cu | Pb | Zn | Hg |
|------------|--------------|-------------------------------------|------|-----|-----|-----|
| | Sample No. | Туре | ppm | ppm | ppm | ppb |
| 2600 | 5532 | quartz chlorite schist | 40 | 15 | 150 | 45 |
| 3000 | 5533 | schist | 550 | 10 | 200 | 180 |
| KS 86 - 5 | 5534 | mica schist | 160 | 250 | 510 | 180 |
| 10 | 5535 | } | 80 | 45 | 330 | 100 |
| 25 | 5536 | andalusite mica schist | 180 | 15 | 140 | 55 |
| 80 | 5537 | | 95 | 30 | 250 | 50 |
| 100 | 5538 | } | 55 | 20 | 140 | 50 |
| 200 | 5539 | | 15 | 15 | 120 | 80 |
| 300 | 5540 | | 5 | 15 | 150 | 20 |
| 400 | 5541 | andalusite | 30 | 5 | 75 | 80 |
| | | garnet schist | | | | |
| 500 | 5542 | (sulphides) | 35 | 20 | 80 | 35 |
| 550 | 5543 | } | <5 | 10 | .30 | 40 |
| 565 | 5544 | (sulphides) | 370 | 15 | 40 | 45 |
| 575 | 5545 | | 970 | 10 | 35 | 80 |
| 585 | 5546 | (sulphides) | 430 | 5 | 25 | 30 |
| 600 | 5547 | chalcopyrite rich rock | 3.7% | 20 | 220 | 80 |
| 615 | 5548 | garnet mica schist | 130 | 10 | 25 | 50 |
| 625 | 5549 | quartz garnet mica schist | 500 | 5 | 25 | 35 |
| 635 | 5550 | schist) | 280 | 5 | 25 | 3.0 |
| 650 | 5551 | } | 260 | 5 | 15 | 50 |
| 670 | 5552 | J | 55 | 10 | 15 | 50 |
| 680 | 5553 | schist with visible sulphides | 1800 | 30 | 20 | .80 |
| 690 | 5554 | + | 9800 | 20 | 55 | 65 |
| 700 | 5555 | | 370 | 15 | 45 | 50 |
| 705 | 5556 | schist | 110 | 10 | 40 | 80 |
| 710 | 5557 | | 780 | 10 | 35 | 40 |
| 730 | 5558 | andalusite | | | | |
| 730 | 3330 | mica schist | 130 | 5 | 20 | 50 |
| 760 | 5559 | quartz mica schist | 95 | 10 | 25 | 50 |
| 800 | 5560 | quartz mica | .35 | 15 | 45 | 50 |
| 850 | 5561 | } | 25 | 10 | 35 | 200 |
| 879 | 5562 | | 45 | 15 | 35 | 50 |

TABLE 11: ASSAY OF MUTOOROO CORES

| Orill Hole | Departmental | Rock | Hg |
|---------------|--------------|--------------------------|----------|
| and Footage | Sample No. | Туре | ppb |
| DDMM6 - 12 | G 5567/72 | acid gneiss | 120 |
| 18 | 5568 | 2.5.1.2 S.1.5.1.5.5 | 130 |
| 30 | 5569 | | 100 |
| 50 50 | 5570 | | 80 |
| | | | 100 |
| 150 | 5571 | | 100 |
| 300 | 5572 | | 140 |
| 550 | 5573 | amphibolite | 70 |
| 800 | 5574 | 6 4 4 | 70 50 |
| 950 | 5575 | feldspathic | 50 |
| | • | amphibolite | 20 |
| 1050 | 5576 | feldspar ? | 30 |
| | | 'breccia' | |
| 1065 | 5577 | amphibolite | 70 |
| | | with pyrite | |
| _ 1070 | 5578 | | 80 |
| 1075 | 5579 | amphibolite | 50 |
| 1150 | 5580 | acid gneiss | 50 |
| 1180 | 5581 | schistose acid gneiss | 50 |
| DD 07 - 20 | E E 0.0 | acid gneiss | 50 |
| DMM7 - 30 | 5582 | acid gheiss | 100 |
| 35 | 5583 | | 80 |
| 39 | 5584 | | 100 |
| 80 | 5585 | • | 180 |
| 200 | 5586 | | 80 |
| 400 | 5587 | schistose acid gneiss | |
| 600 | 5588 | amphibolite | 80 |
| 850 | 5589 | acid gneiss | 50 |
| 1100 | 5590 | amphibolite | 100 |
| 1350 | 5591 | | 50 |
| 1450 | 5592 | | 90 |
| 1500 | 5593 | | 80 |
| | 5594 | amphibolite | 290 |
| 1535 | 3394 | with pyrite | |
| 15/0 | EEOE | sulphide rich | 160 |
| 1540 | 5595 | rocks | 100 |
| | | rocks | 130 |
| 1550 | 5596 | | 370 |
| 1555 | 5597 | | |
| 1560 | 5598 | • | 310 |
| 1580 | 5599 | gneiss | 100 |
| 1700 | 5600 | amphibolite | 180 |
| DDMM 15DI - | 0 500 1=0 | | 130 |
| 59 | G 5601/72 | weathered acid | 130 |
| مد ن <i>د</i> | | gneiss | 100 |
| 65 | 5602 | muscovite schist | 100 |

TABLE 11: CONTINUED

| Drill Hole and Footage | Departmental Sample No. | Rock Type | Hg ppb |
|---------------------------|----------------------------|---------------------------------|-----------|
| 70 | 5603 | schistose | |
| | 3003 | micaceous quartzite | 50 |
| 150 | 5604 | acid gneiss | 80 |
| 300 | 5605 | tremolite gneiss | 110 |
| DDMM26 - | | | |
| 345 | G 5643/72 | acid gneiss | 70 |
| 480 | 5644 | ? intermediate gneiss | 50 |
| 680 | 5645 | | 20 |
| 880 | 5646 | | 20 |
| 1200 | 5647 | amphibolite | 20 |
| 1500 | 5648 | • | 120 |
| 1600 | 5649 | | 50 |
| 1700 | 5650 | | 30 |
| 1750 | 5651 | | 120 |
| 1770 | 5652 | amphibolite, pyrite | 50 |
| 1780 | 5653 | "amphibolite" | 50 |
| 1790 | 5654 | amphibolite, mica jarosite | 20 |
| 1795 | 5655 | | 50 |
| 1800 | 5656 | amphibolite | 70 |
| 1805 | 5657 | quartzite, chlorite partings | 50 |
| 1810 | 5658 | quartz, chalcopyrite | .55 |
| 1812 | 5659 | | 80 |
| 1815 | 5660 | quartz, part oxidised sulphide | 30 |
| 1830 | 5661 | quartz, marcasite | 30 |
| 1832 | 5662 | pyrite, chalcopyrite | 100 |
| 1840 | 5663 | amphibolite | 50 |
| 1850 | 5664 | | 20 |
| 1900 | 5665 | | 70 |
| 2000 | 5666 | basic gneiss | 40 |

TABLE 12: ASSAY OF BURRA DRILL CORE

| Drill Hole and Footage | | Departmental Sample No. | Cu ppm | Hg ppb | |
|---------------------------|-----|----------------------------|-----------|-----------|--|
| B54 | 33 | G 5906/72 | 32 | 100 | |
| | 200 | 5907 | 22 | 90 | |
| | 300 | 5908 | 55 | 110 | |
| | 400 | 5909 | 290 | 80 | |
| | 500 | 5910 | 2400 | 310 | |
| | 600 | 5911 | 150 | 80 | |
| | 750 | 5912 | 100 | 140 | |

TABLE 13: ASSAYS OF MOONTA CORE, DDH14

| Footage | Departmental Sample No | Rock Type | Sulphides* | Hg ppb |
|------------------|---------------------------|-----------------------------------|---------------|-----------|
| 21'5" | G6160/72 | Weathered porphyry | | 205 |
| 74.1 | 6161 | Porphyry | - | 245 |
| 126' | 6162 | Porphyry | | 355 |
| 178' | 6163 | Weathered acid porphyry | . | 220 |
| 230 | 6164 | Sheared? acid porphyry | x | 210 |
| 282 ' | 6165 | Sheared amphibole-feldspar rock | x | 245 |
| 334 ' | 6166 | Acid porphyry with sulphides | × | 210 |
| 386' | 6167 | Volcanic rock? | x | 260 |
| 438 ' | 6168 | Actinolite (black) schist | x | 200 |
| 49 0 ' | 6169 | Actinolite schist | | 150 |
| 542' | 6170 | Pyritic actinolite schist | x | 220 |
| 5941 | 6171 | Biotite-amphibole schist | | 230 |
| 646' | 6172 | Actinolite schist | x | 290 |
| 6981 | 6173 | Actinolite schist | x | 215 |
| 750 ' | 6174 | Magnetite rock | x | 310 |
| 802 [†] | 6175 | Amphibolite (diorite) with pyrite | x | 160 |
| 854 ' | 6176 | Amphibolite | x | 225 |
| 906 | 6177 | Amphibolite with calcite | · · · | 115 |
| 958 ' | 6178 | Mica-amphibolite | x | 190 |
| 1010' | 6179 | Schistose amphibolite | x | 150 |
| 1062' | 6180 | Quartz-amphibole-mica rock | × | 200 |
| 1114' | 6181 | Sheared amphibolite | x | 110 |
| 1166' | 6182 | Sheared amphibolite | x | 110 |
| 1218' | 6183 | Amphibolite | · — | 105 |
| 1270' | 6184 | Schistose amphibolite | - | 250 |
| 1322' | 6185 | Actinolite schist | - | 150 |
| 1374' | 6186 | Sulphide-bearing porphyry | x | 200 |
| 1426' | 6187 | Amphibolite | - · . | 150 |
| 1478' | 6188 | Amphibolite | | 110 |
| 1530 ' | 6189 | Amphibolite | x | 100 |
| 1582 ' | 6190 | Magnetite rock | x | 90 |

^{*} x indicates sulphides visible in hand specimen

TABLE 13: CONTINUED

| Footage | Departmental Sample No | Rock Type | Sulphides* | Hg ppb |
|---------------|---------------------------|--------------------------------|---------------------------------------|-----------|
| 1634 | 6191/72 | Schistose amphibolite | | 155 |
| 1686 ' | 6192 | Schistose amphibolite | | 110 |
| 1738' | 6193 | Schistose amphibolite | × | 120 |
| 1790 ' | 6194 | Gneissic amphibolite | · · · · · · · · · · · · · · · · · · · | 190 |
| 1894 | 6195 | Schistose biotite ?diorite | - | 200 |
| 1946' | 6196 | 'Diorite' | | 220 |
| 1998' | 6197 | Amphibolite | - | 240 |
| 2050 | 6198 | Sheared amphibolite | - | 150 |
| 2102 | 6199 | Amphibolite | x | 200 |
| 2154 | 6200 | Sheared acid volcanic rock | - | 300 |
| 2206 | 6201 | Amphibolite | _ | 165 |
| 2258¹ | 6202 | Diorite-amphibolite | - | 215 |
| 2310 | 6203 | Schistose amphibolite | _ | 170 |
| 2362 | 6204 | Mica-diorite | . · · | 380 |
| 2414 | 6205 | Foliated ?diorite/granodiorite | <u> </u> | 330 |

^{*} x indicates sulphides visible in hand specimen

TABLE 14A: PARABARANA CORRELATION COEFFI-CIENTS (All Samples)

| Correlation | Depth | Ċu | Pb | Zn | Hg |
|-------------|-----------|-----------|-----------|-----------|-----------|
| Depth | 1.000/57 | 0.060/57 | -0.574/57 | -0.549/57 | -0.106/57 |
| Cu | 0.060/57 | 1.000/57 | 0.035/57 | -0.008/57 | 0.039/57 |
| Рb | -0.574/57 | 0.035/57 | 1.000/57 | 0.767/57 | 0.089/57 |
| Zn | -0.549/57 | -0.008/57 | 0.767/57 | 1.000/57 | -0.094/57 |
| Нg | -0.106/57 | 0.039/57 | 0.089/57 | -0.094/57 | 1.000/57 |

TABLE 14B: PARABARANA CORRELATION COEFFICIENTS OF SAMPLES WITH LESS THAN 500 PPM COPPER

| Correlation | Depth | Cu | Рb | Zn | Hg |
|-------------|-----------|----------|-----------|-----------|-----------|
| Depth | 1.000/42 | 0.006/42 | -0.582/42 | -0.569/42 | -0.071/42 |
| Cu | 0.006/42 | 1.000/42 | 0.156/42 | 0.309/42 | 0.317/42 |
| Pb | -0.582/42 | 0.156/42 | 1.000/42 | 0.804/42 | 0.082/42 |
| Zn | -0.569/42 | 0.309/42 | 0.804/42 | 1.000/42 | 0.047/42 |
| Hg | -0.071/42 | 0.317/42 | 0.082/42 | -0.047/42 | 1.000/42 |

TABLE 15A: KANMANTOO CORES KS31, KS86 CORRELATION COEFFICIENTS, (All Samples)

| Correlation | Depth | Cu | Pb | Zn | Hg | |
|-------------|-----------|-----------|-----------|-----------|-----------|--|
| Depth | 1.000/64 | 0.088/64 | -0.505/64 | -0.444/64 | -0.211/64 | |
| Cu | 0.088/64 | 1.000/64 | 0.003/64 | -0.162/64 | 0.114/64 | |
| Pb | -0.505/64 | 0.003/64 | 1.000/64 | 0.705/64 | 0.189/64 | |
| Zn | -0.444/64 | -0.162/64 | 0.705/64 | 1.000/64 | 0.259/64 | |
| Hg | -0.211/64 | 0.114/64 | 0.189/64 | 0.259/64 | 1.000/64 | |

TABLE 15B: KANMANTOO CORES KS31, CORRELATION COEFFI-CIENTS (All Samples)

| Correlation | Depth | Cu | Pb | Zn | Hg |
|-------------|-----------|-----------|-----------|-----------|-----------|
| Depth | 1.000/35 | 0.099/35 | -0.469/35 | -0.545/35 | -0.194/35 |
| Cu | 0.099/35 | 1.000/35 | -0.026/35 | -0.262/35 | 0.166/35 |
| Pb | -0.469/35 | -0.026/35 | 1.000/35 | 0.785/35 | 0.013/35 |
| Zn | -0.545/35 | -0.262/35 | 0.785/35 | 1.000/35 | 0.075/35 |
| Hg | -0.194/35 | 0.166/35 | 0.013/35 | 0.075/35 | 1.000/35 |

TABLE 15C: KANMANTOO HOLE KS31 CORRELATION COEFFI-CIENTS FOR NON-MINERALISED SAMPLES (750 ppm Cu or less)

| Correlation | Depth | Cu | Pb | Zn | Hg |
|-------------|-----------|-----------|-----------|-----------|-----------|
| Depth | 1.000/30 | 0.017/30 | -0.472/30 | -0.538/30 | -0.174/30 |
| Cu | 0.017/30 | 1.000/30 | -0.080/30 | -0.121/30 | 0.528/30 |
| Pb | -0.472/30 | -0.080/30 | 1.000/30 | 0.823/30 | -0.028/30 |
| Zn | -0.538/30 | -0.121/30 | 0.823/30 | 1.000/30 | 0.025/30 |
| Hg | -0.174/30 | 0.528/30 | -0.028/30 | 0.025/30 | 1.000/30 |

FIG.1: DOWN-HOLE PROFILE K6 KAPUNDA

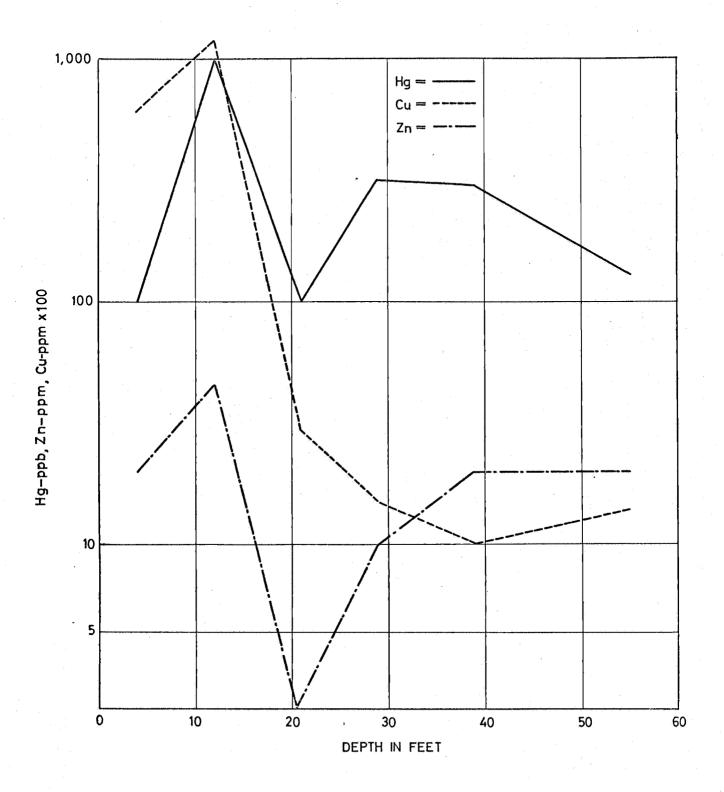


FIG. 2: DOWN-HOLE PROFILE K10 KAPUNDA

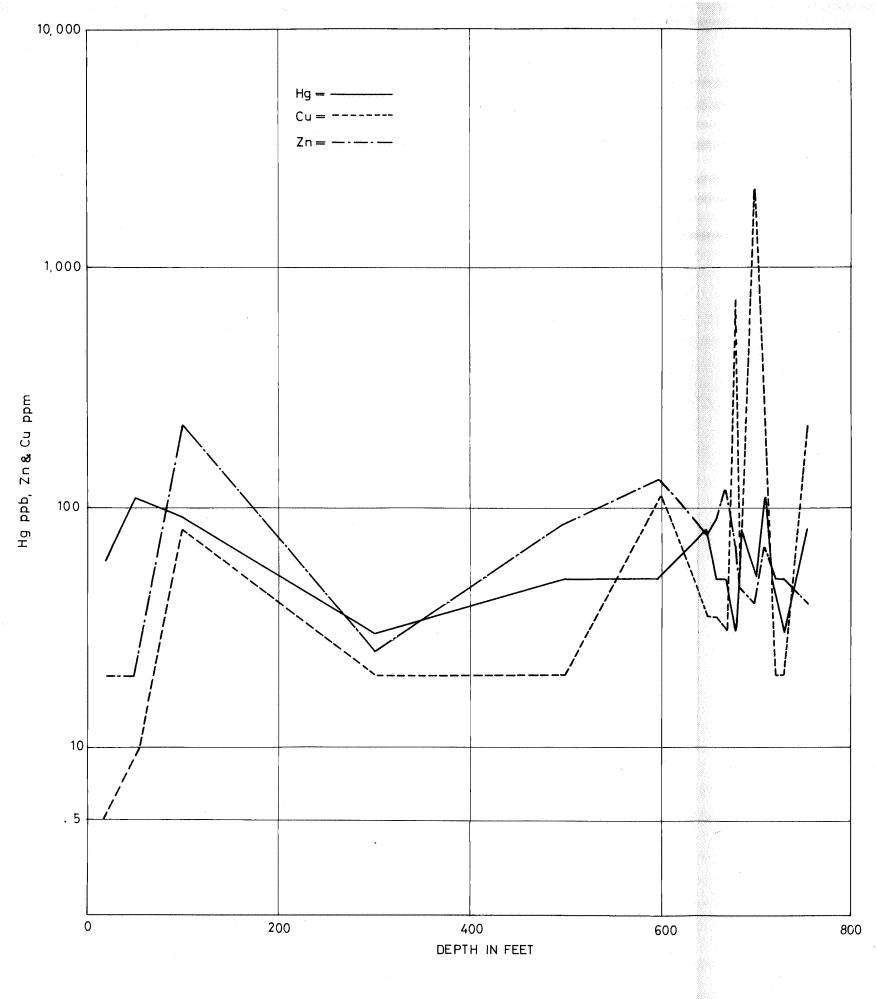


FIG. 3: DOWN-HOLE PROFILE OF PDD 2 PARABARANA

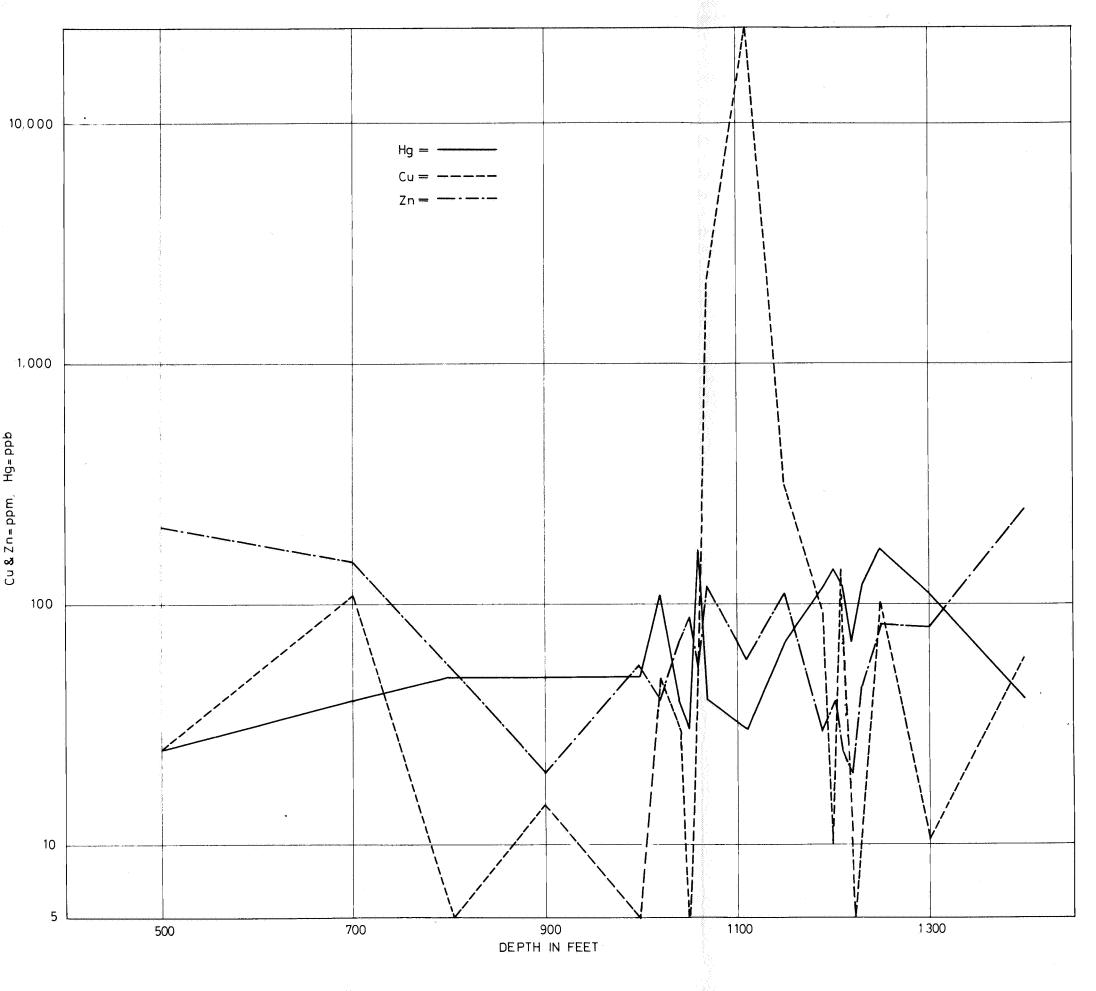


FIG. 4 : DOWN-HOLE PROFILE OF PDD3 PARABARANA

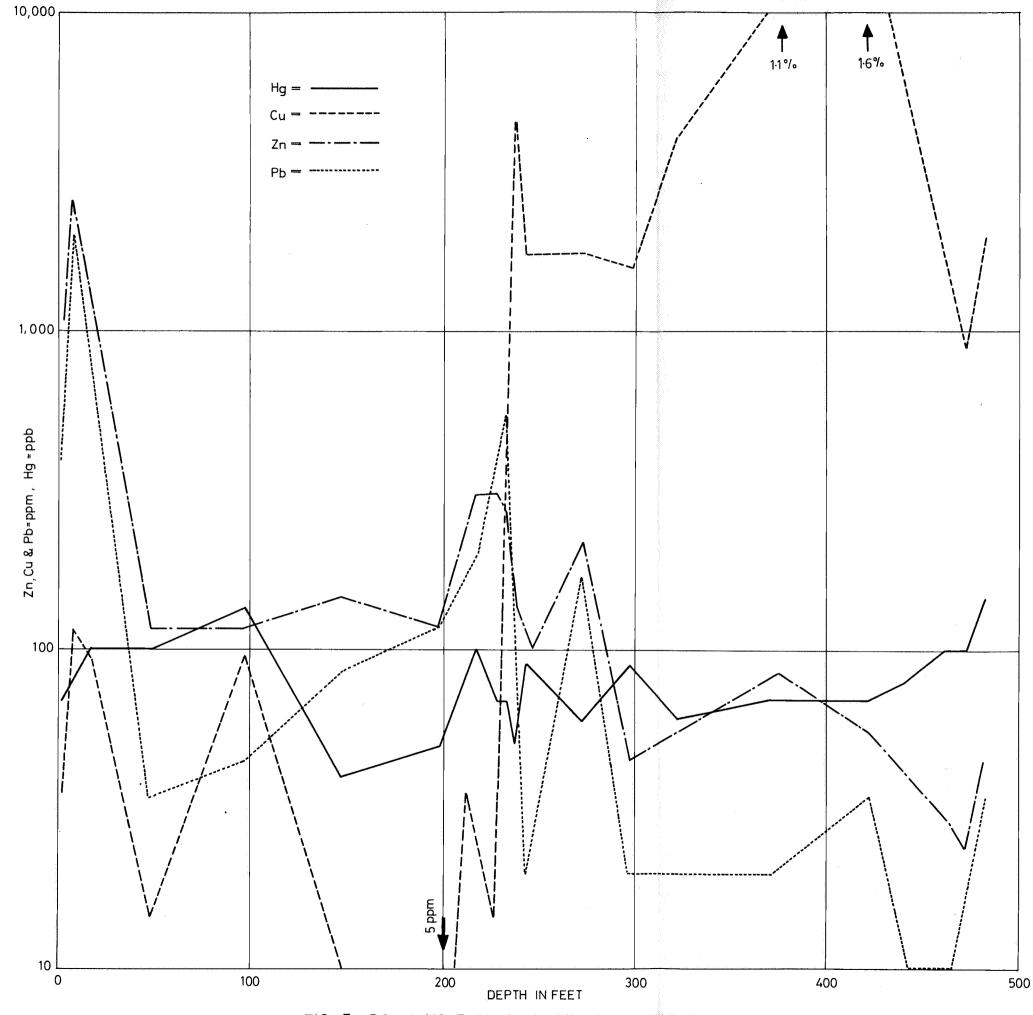
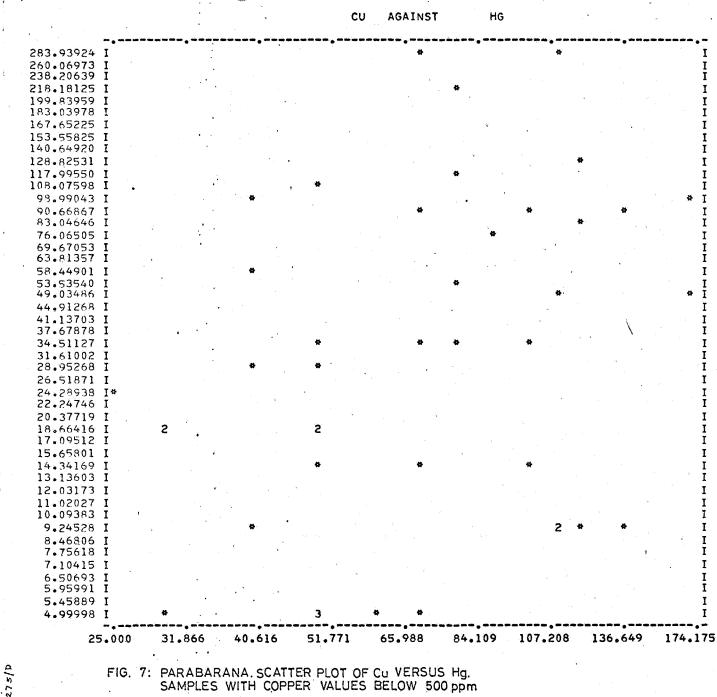


FIG. 5: DOWN HOLE-PROFILE OF NDP 22 PARABARANA

| 1672.52302 8065.32145 5058.51746 | I . | | | | | | | | | | | | | | | |
|--|------------|----|----|-----|-------|--------|-----|---|-----|-----|------|----|-----|---|----------|---|
| 2552.15406 | | | | | | | | | | | v. | | | | | |
| 2352•13406 0462•97550 | | | , | | | | | | 4 | | | | | | . * | |
| 8721.51136 | | | | | | | | | | | | | | | | |
| 7269.89508 | | | | | | | • | | | | | | • • | | | |
| 6059.88741 | | | • | | | | | | | | | | | | | |
| 5051.27192 | | | | | | | | | | * | | | | | | |
| 4210.53290 | | | · | | | # | | | | | | | | | | |
| 3509.72772 | | | | | | | | # | | | | | | | | |
| 2925.56167 | | | , | | | | | | | | | 4 | | | | |
| 2438,62748 | I | | • | | | | | | | | | | | | | |
| | I | | | # | • | # | | | | | | | | | | |
| 1694.40389 | I | | • | | · . | 4 | | | | | . 18 | | | | * | |
| 1412.38570 | I | | ٠. | | | | | | | | * | * | | | | |
| 1177.30498 | | | • | | • . | | | | | | | | | | | |
| 981.35352 | | | | • | | | | | | | | | | | | |
| 818.01498 | | | 14 | | | | | | | | | 12 | | | | |
| 681.86283 | | R | | | | • | | | • | • . | | | | | | |
| 568.37296 | | | | | | | | | | | | | | | | |
| 473.77169 | | | | | | | | | | | | | | | <i>(</i> | |
| 394.91618 329.18584 | | | ٠. | | | | | | | | | | | | \ | |
| 274.39547 | | | • | | . * * | | | | | | | | | | | |
| 228.72460 | | | | | | | | | | | | | | | | |
| 190.65535 | | | | | ٠ | | | | | # | | | :" | • | | |
| 158.92231 | I | | | | | | | | | | | | | | | |
| 132.47097 | I | | | | | | | | | | | | -#1 | • | | |
| 110.42237 | | | | | | | | | | * | | | | | | |
| 92.04347 | | | | . * | | * | | | * | | | * | | | # | |
| 76.72355 | | | | | | | | | | | • | | , 4 | | | • |
| 63.95357 | | | | | | | | | | | | | | | | |
| 53.30907 | | | 4 | . # | | | | | | ₩. | | | | | | |
| 44.43622 | | | | | • | | • | | | | | ¥ | ~ | | | |
| 37.04019 | | | | | | | | | | | | # | | | | |
| 30.87518 | | 3 | • | | | W M | | | ~ | ~ | | - | | | | |
| 25.73626 21.45270 | | | | ** | | * | | | | | | | | | | |
| 17.88207 | | 2 | ¥ | | | 2 | | | | | | | | | | |
| 14.90576 | _ | ۲. | | | × . | 4 | | | * | | | # | | | | |
| 12.42483 | | | | | | | | | | | | | 44 | | 4 | |
| 10.35682 | | | | | | | | | | | | | | | | |
| 8.63302 | | | | . * | | | - ' | | | | | | 2 * | | * | |
| 7.19613 | | | • | | | | | | | | | | | | | |
| 5.99839 | | | | | | | | | | | | | | | | |
| 5.00001 | | | | | | | | | | | | | | | | |
| 4.16780 | | # | ٠, | | | 3 | | 4 | . # | | | | + | | | |

FIG. 6: PARABARANA. SCATTER PLOT OF Cu VERSUS Hg (ALL SAMPLES)

FSTATS



RE.7275/D

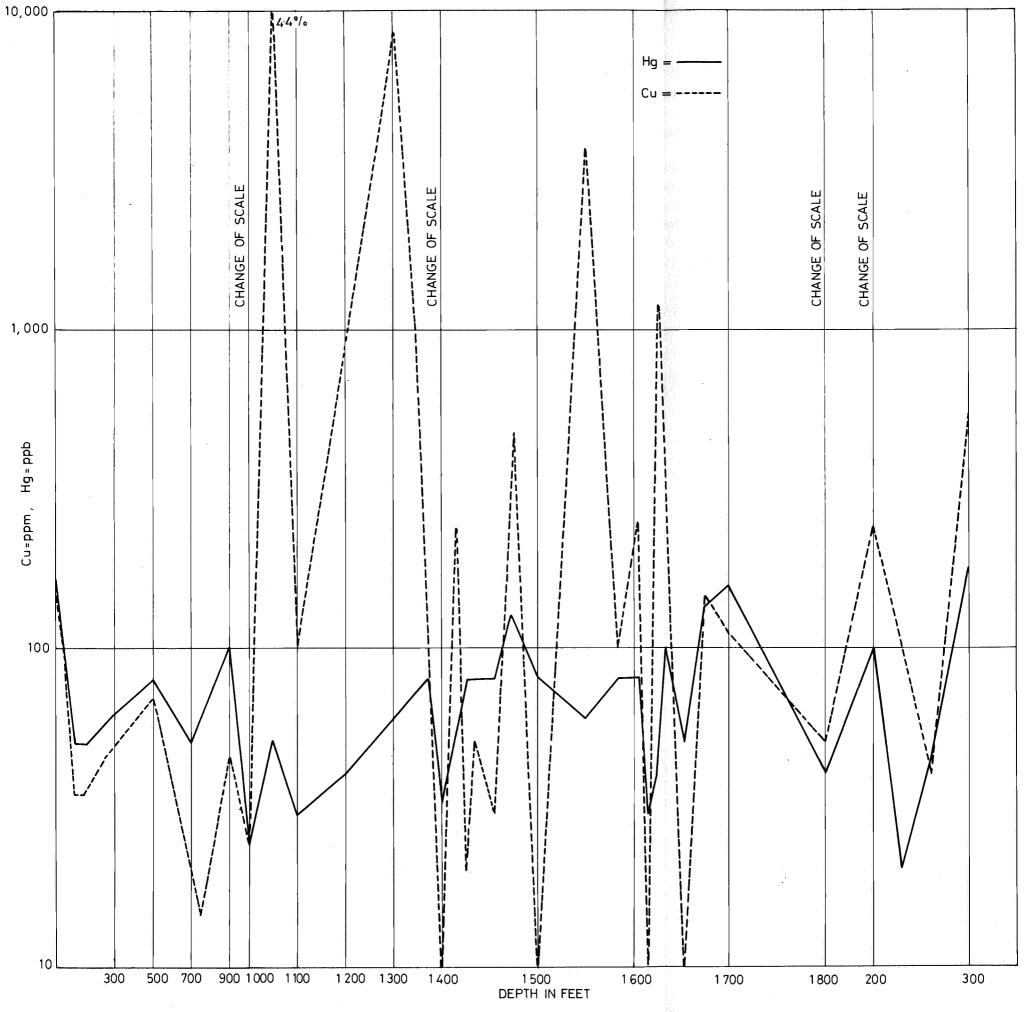
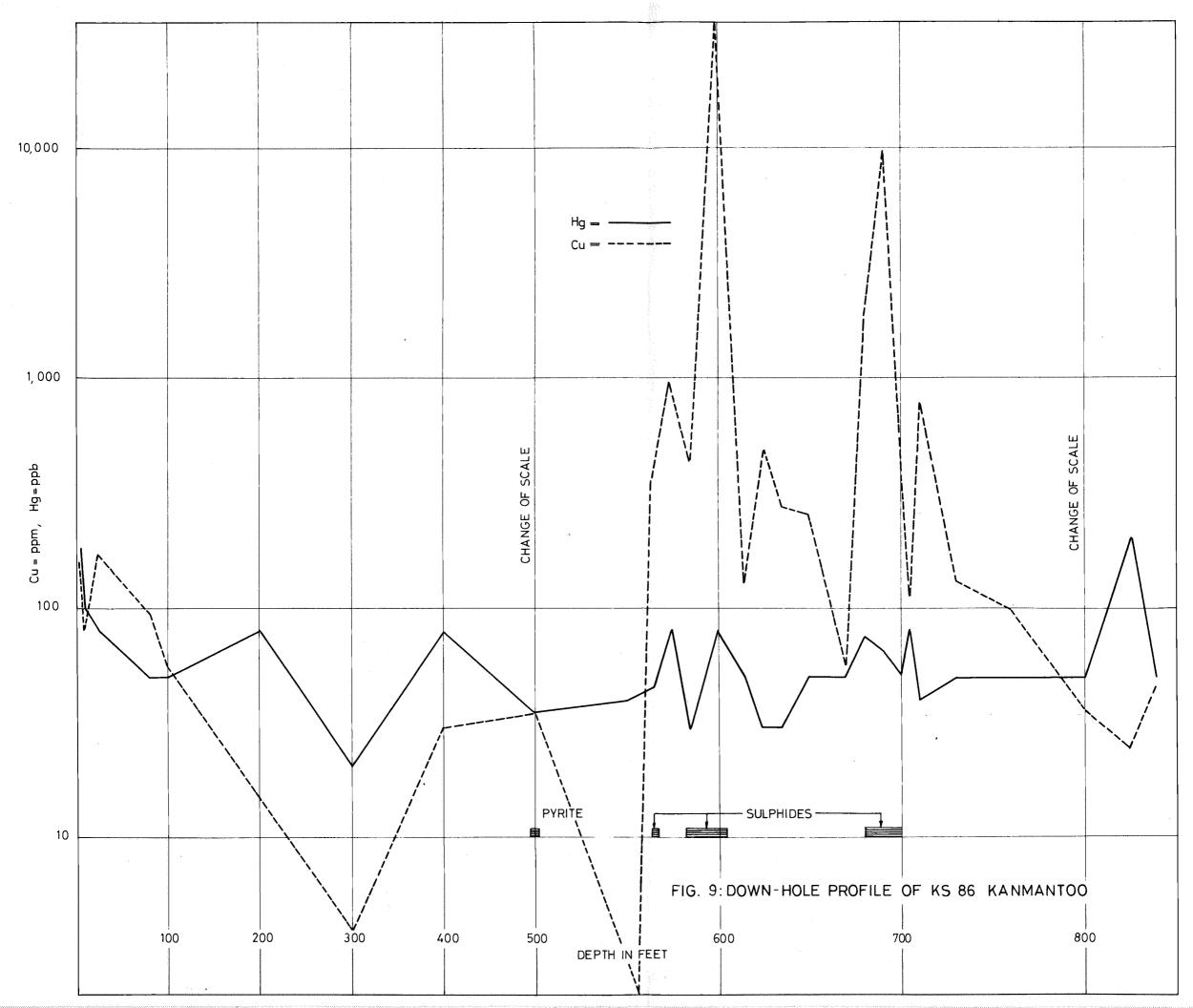
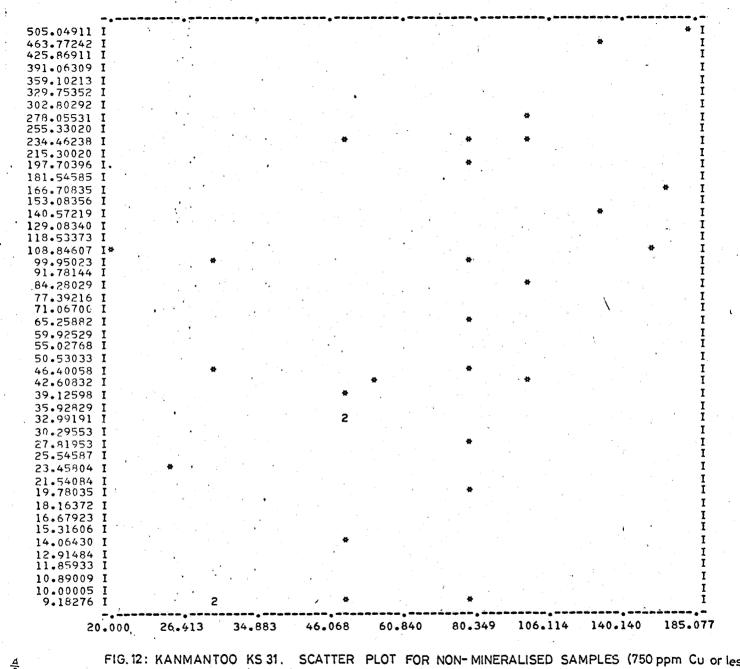


FIG. 8: DOWN-HOLE PROFILE OF KS 31 KANMANTOO



| 36807,19137 I 25756,80017 I 21546,24462 I 18024,00351 I 15077,54326 I 12612,76007 I 10550,00427 I 8826,11752 I 7383,28099 I 6176,30959 I 5166,63909 I 4322,03054 I 3024,45531 I 2316,44053 I 1770,4582 I 1481,03595 I 1286,39364 I 1886,39364 I 1866,39364 I 1866,39102 I 725,24345 I 1686,59102 I 725,24345 I 1686,59103 I 725,24345 I 1686,68516 I 1686,68516 I 1686,68516 I 1686,68516 I 1687,50928 I 1248,52014 I 207,89373 I 173,90847 I 145,47896 I 121,6895 I 101,80271 I* 85,16071 I 71,23914 I 59,59343 I 49,85148 I 41,70212 I 34,88492 I 29,88218 I 24,41167 I 20,4200 I 17,08273 I 14,29015 I 11,95409 I 9,99992 I 2 2 * 20,000 26,413 34,883 46,068 60,840 80,349 106,114 140,140 | 30790.18593 I 225756.80017 I 21546.24462 I 18024.00351 I 15077.54326 I 12612.76007 I 10550.90427 I 8826.11752 I 7383.28099 I 6176.30959 I 5166.63999 I 4322.03054 I 4322.03054 I 4322.03054 I 3024.45531 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.39364 I 866.97102 I 725.24345 I 606.66516 I 507.50828 I 424.54410 I 335.14235 I 297.08600 I 248.52014 I 207.89373 I 173.99847 I 145.47896 I 121.69695 I 101.80271 I* 85.16071 I 71.23914 I 59.95343 I 49.85188 I 44.70212 I 34.88482 I 34.88482 I 34.88482 I 34.88482 I 34.88482 I 34.88482 I 36.9015 I 11.96023 I 14.29015 I 11.960273 I | 24007 10127 7 | | | | <u>-</u> | | | | |
|--|--|---------------|------------|----------|--------|----------|--------|---------|------------------------------------|-----|
| 28756,80017 I 18024.00351 I 18024.00351 I 12612.75007 I 18550.90427 I 18826.11752 I 7383.28099 I 6176.30959 I 5166.63909 I 4322.03054 I 3615.49258 I 3024.45531 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.93936 I 866.97102 I 725.22435 I 606.68516 I 507.50928 I 424.54410 I 355.14235 I 277.08600 I 248.52014 I 207.89373 I 145.47896 I 121.89695 I 101.80271 I* 85.16071 I 71.23914 I 59.99343 I 49.85148 I 49.85148 I 49.85148 I 40.8218 I 40.82 | 28756,80017 I 18024,00351 I 18024,00351 I 18074,5436 I 12612,76007 I 18266,11752 I 7383,28099 I 6176,30959 I 5166,63999 I 6176,30959 I 3024,45531 I 2510,30716 I 2510,30716 I 2510,30716 I 2618,4055 I 1770,4582 I 1481,03595 I 1238,02546 I 1038,39344 I 866,97102 I 728,20346 I 600,68516 I 500,68516 I 500,68516 I 500,68516 I 500,68516 I 1018,0271 I 1277,08600 I 248,52014 I 267,79373 I 173,90847 I 145,47896 I 121,66967 I 185,16071 I 77,23914 I 79,93933 I 49,85148 I 41,70212 I 34,88492 I 29,8818 I 20,4107 I 20,2000 26,413 34,883 46,068 60,840 80,349 106,114 140,140 18 | | | | | | | | $(x_i, x_i) \in \mathcal{X}_{i+1}$ | |
| 18024.00351 I 12612.76907 I 10550.90427 I 8826.11752 I 7383.26099 I 6176.30959 I 5166.63909 I 4322.03054 I 3615.49258 I 3024.49531 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.0359 I 1036.39364 I 866.97102 I 725.24345 I 606.68516 I 507.50828 I 424.54410 I 355.14235 I 297.0800 I 248.52014 I 127.68695 I 101.80271 I* 85.16071 I 77.123914 I 59.95934 I 49.85148 I 41.70212 I 34.88492 I 34.88492 I 34.88492 I 32.4810 I 32.48280 I 34.88492 I 34.88492 I 34.88492 I 37.999992 I 32.899999 I 32.89999 I 32.899999 I 32.89999 I 32.899999 I 32.89999 I 32.8999 I 32.8999 I 32.89999 | 18026.00351 I 12612.76007 I 18262.1752 I 7383.28099 I 6176.30959 I 5166.63999 I 6176.30959 I 5166.63990 I 4322.03054 I 4322.03054 I 4322.03054 I 1036.549258 I 3024.45531 I 2239.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.02546 I 1038.39364 I 866.97102 I 786.24346 I 606.68516 I 507.50828 I 424.54410 I 327.08600 I 248.22014 I 207.789373 I 1207.89373 I 121.69691 I 85.16071 I 71.23914 I 71.23915 I 71.398273 I 71.398291 I | | | • | | | | | | |
| 15077,54326 I 12612,76007 I 10550.90427 I 8826.11752 I 7383,28099 I 6176.30959 I 5166.63999 I 4322.03054 I 3024.45531 I 2330.30716 I 2116.44053 I 1770.45822 I 1481.33595 I 1238.92546 I 13036.39354 I 866.97102 I 725.24345 I 606.68516 I 507.50828 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 207.89373 I 173.90847 I 145.47896 I 121.69695 I 101.80271 I 85.16071 I 71.23914 I 99.99992 I 2 | 15077,54326 I 10550.00427 I 8826.11752 I 7383.28099 I 6176.3999 I 5166.43909 I 4322.03054 I 3024.45531 I 2330.3716 I 2116.44053 I 1770.45822 I 1481.03595 I 1036.39364 I 866.97102 I 725.24345 I 606.65516 I 507.50828 I 424.5410 I 355.14235 I 1273.90847 I 148.47896 I 121.69695 I 101.8027 I 85.16071 I 71.23914 I 59.59343 I 49.85148 I 41.70212 I 34.88492 I 29.8218 I 24.41167 I 20.4010 I 17.08273 I 14.29095 I 11.95409 I 9.99992 I 2 20.000 26.413 34.883 46.068 60.840 80.349 106.114 140.140 18 | | • | * . | | | | | | |
| 12612-76007 I 1050-09427 I 8826-11752 I 7383-28099 I 6176-30959 I 5166-63909 I 4322-03054 I 3615-49258 I 3024-45531 I 2530-03716 I 2116-44053 I 1770-45822 I 1481-03595 I 1238-92546 I 1036-39364 I 866-97102 I 725-24345 I 606-66516 I 507-50828 I 424-54410 I 355-14235 I 297-08600 I 248-52014 I 207-88937 I 145-47896 I 121-68695 I 101-80271 I 85-51691 I 71-23914 I 71-23915 I 71-68695 I 101-80271 I 71-23914 I 71-23914 I 71-23915 I 72-441167 I 72-4210 I 71-68273 I | 12612.76007 I 1050.90427 I 1826.11752 I 7383.28099 I 6176.30959 I 5166.63909 I 4322.30354 I 3615.49258 I 3024.49531 I 12530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.39346 I 866.97102 I 725.24345 I 606.68516 I 507.50928 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 207.99373 I 173.90847 I 145.47896 I 121.69695 I 101.80271 I* 85.16071 I 71.23914 I 59.59343 I 49.85148 I 41.70212 I 34.88492 I 32.48218 I 29.18218 I 20.42100 I 17.08273 I 14.2015 I 11.95409 I 9.99999 I 2 ** 20.000 26.413 34.883 46.068 60.840 80.349 106.114 140.140 18 | | .* | | | | | 4. | | |
| 10550.90427 I 8826.11752 I 7383.28099 I 6176.30959 I 5166.63909 I 4322.03054 I 3024.45531 I 231.03053 I 1770.45822 I 1481.03595 I 1238.92546 I 1306.39394 I 866.97102 I 725.24345 I 606.68516 I 507.50828 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 207.89373 I 173.90847 I 145.47896 I 121.69695 I 101.80271 I* 85.16071 I 71.23914 I 75.59343 I 49.85148 I 41.70212 I 34.88492 I 29.18218 I 20.42100 I 17.08273 I 11.98097 I 11.98097 I 11.08027 I 12.08028 I 12.080 | 10550.90427 I 8826.11752 I 7383.28099 I 6176.30959 I 5166.63909 I 4322.03054 I 3024.45531 I 2230.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.39344 I 866.97102 I 725.24345 I 666.68516 I 507.50828 I 424.54410 I 335.14235 I 297.08600 I 248.52014 I 207.69373 I 113.90847 I 145.47896 I 121.6965 I 101.80271 I* 85.16071 I 71.23914 I 59.59343 I 49.65148 I 41.70212 I 34.88492 I 29.18218 I 24.41167 I 20.42100 I 17.08273 I 11.95409 I 29.99992 I 20.000 26.413 34.883 46.068 60.840 80.349 106.114 140.140 18 | | | | | | | * * | | |
| 7383.28099 I 6176.30959 I 5165.63909 I 4322.03054 I 3024.45531 I 2230.03716 I 2116.44053 I 1770.45822 I 1481.0359 I 1238.92546 I 1036.39344 I 866.97102 I 725.24345 I 606.68516 I 507.50828 I 424.54410 I 355.14235 I 297.0800 I 248.52014 I 207.89373 I 145.47896 I 101.80271 I* 85.16071 I 71.23914 I 59.59343 I 49.85148 I 41.70212 I 34.88492 I 29.48273 I 24.41167 I 20.4810 I 27.08273 I 24.29015 I 11.95409 I 9.99992 I 2 2 ** | 7383,28099 I 6176,10959 I 5166,63909 I 4322,03054 I 33615,49258 I 3024,45531 I 22530,03716 I 2116,44053 I 1770,45822 I 1481,03595 I 1238,92546 I 1036,339364 I 866,97102 I 725,24345 I 606,668516 I 507,50828 I 424,54410 I 335,14235 I 297,08600 I 248,52014 I 207,89373 I 145,47896 I 121,6695 I 101,80271 I* 85,16071 I 71,23914 I 59,59343 I 49,85148 I 41,70212 I 34,88492 I 35,99992 I 20,000 26,413 34,883 46,068 60,840 80,349 106,114 140,140 18 | | • | • | | | | • | | |
| 6176.30959 I 5166.63909 I 4322.03054 I 3615.49258 I 3024.45531 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.39364 I 866.97102 I 725.24345 I 606.68516 I 507.50828 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 207.89373 I 145.47896 I 121.68695 I 101.80271 I* 85.16071 I 77.123914 I 59.593343 I 49.85148 I 41.70212 I 34.88492 I 29.18218 I 29.18218 I 20.42100 I 17.08273 I 1.08273 I 1.08273 I 1.08273 I 1.08273 I 20.42100 I 17.08273 I 1.08273 I 1.08274 I 2.08274 I 2.08 | 6176.30959 I 5166.63909 I 4322.03054 I 3615.49258 I 3024.45531 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.39364 I 866.97102 I 725.24345 I 606.68516 I 507.50828 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 107.089373 I 173.90847 I 145.47896 I 121.69695 I 101.80271 I* 85.16071 I 71.23914 I 59.59343 I 49.85148 I 41.70212 I 34.88492 I 34.88492 I 34.88492 I 20.100 I 17.08273 I 14.29015 I 17.08273 I 14.29015 I 17.08273 I 14.29015 I 11.95409 I 9.99992 I 2 4 9 20.000 26.413 34.883 46.068 60.840 80.349 106.114 140.140 18 | | | | | | | • | | |
| 5166.63909 I 4322.03054 I 3615.49258 I 3024.45551 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.39364 I 866.97102 I 725.24345 I 606.68516 I 507.50828 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 207.89373 I 145.47896 I 121.69695 I 101.80271 I* 85.16071 I 71.23914 I 59,59343 I 49.85148 I 49.85148 I 41.70212 I 34.88492 I 29.18218 I 24.41167 I 20.42900 I 17.08273 I 14.59099 I 9,99992 I 2 * * * | 5166,63909 I 4322,03054 I 3615,49258 I 3024,45531 I 22530,03716 I 2116,44053 I 1770,45822 I 1481,03595 I 1238,92546 I 1036,39364 I 866,97102 I 725,24345 I 606,68516 I 507,50928 I 424,54410 I 355,14235 I 297,08600 I 248,52014 I 207,89373 I 173,90847 I 145,47896 I 101,80271 I* 85,16071 I 71,23914 I 59,59343 I 49,85148 I 41,70212 I 34,88492 I | | | | | | • | | | |
| 3315.49258 I 3615.49258 I 3024.45531 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.39364 I 866.97102 I 725.24345 I 606.68516 I 507.50828 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 207.89373 I 173.90847 I 145.47896 I 121.69695 I 101.80271 I* 85.16071 I 71.23914 I 59.59343 I 49.85148 I 41.70212 I 34.88492 I 32.48210 I 17.08273 I 14.2010 I 17.08273 I 14.2010 I 17.08273 I 14.2015 I 11.95409 I 9.99992 I 2 ** | 4322.03054 I 3015.49258 I 3024.45531 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.39364 I 866.97102 I 725.24345 I 606.68516 I 507.50928 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 207.89373 I 173.90847 I 145.47896 I 121.69695 I 101.80271 I* 85.16071 I 71.23914 I 59.59343 I 49.85148 I 41.70212 I 34.858492 I 34.858492 I 34.858492 I 34.858492 I 34.858492 I 34.858492 I 34.858491 I 35.46107 I 71.23914 I 59.59343 I 49.85148 I 41.70212 I 34.858492 I 34.858492 I 34.858492 I 34.858493 I 173.08273 I 14.29015 I 11.95409 I 9.99992 I 20.000 26.413 34.883 46.068 60.840 80.349 106.114 140.140 18 | | • • | | | | * * | _ | | |
| 3024.45531 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.39364 I 866.97102 I 725.24345 I 606.68516 I 507.50928 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 207.89373 I 173.90847 I 145.47896 I 121.69695 I 101.80271 I* 85.16071 I 71.23914 I 59.59343 I 49.85168 I 41.70212 I 34.88492 I 34.89492 I 35.89492 I 36.89492 I 36.89492 I 37.89492 I 38.89492 I 38.99492 I 38.99 | 3024.45531 I 2530.03716 I 2116.44053 I 1770.45822 I 1481.03595 I 1238.92546 I 1036.33934 I 866.97102 I 725.24345 I 606.68516 I 507.50828 I 424.54410 I 355.14235 I 297.08600 I 248.52014 I 207.89373 I 145.47896 I 101.80271 I* 85.16071 I 71.23914 I 59.59343 I 49.85148 I 41.70212 I 34.88492 I 29.18218 I 24.41167 I 20.42100 I 17.08273 I 11.95409 I 9.99992 I 2 | | | | • | | | | | |
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| 11.95409 I 9.99992 I 2 * * | 11.95409 I 9.99992 I 2 * * 20.000 26.413 34.883 46.068 60.840 80.349 106.114 140.140 18 | 17.08273 I | • | • | • | | | | | |
| 9.99992 I 2 * * * | 9.99992 I 2 * * * 20.000 26.413 34.883 46.068 60.840 80.349 106.114 140.140 18 | | | | | | | | | |
| 20,000 26,413 34,883 46,068 60,840 80,349 106,114 140,140 | | | 7 | 2 | * | | | | | |
| ZO.000 ZN.415 54.885 46.008 NU.840 NU.847 IUD.114 14U.14U . | | | | 97.000 | | | | 106 174 | 140 140 | 101 |
| 200000 200000 | FIG.11: KANMANTOO CORE, KS 31. SCATTER PLOT (ALL SAMPLES) | 20.00 | JU -25.413 | 34.883 | 40.008 | 00.040 | 0V+347 | 100-114 | 140-140 | 10: |

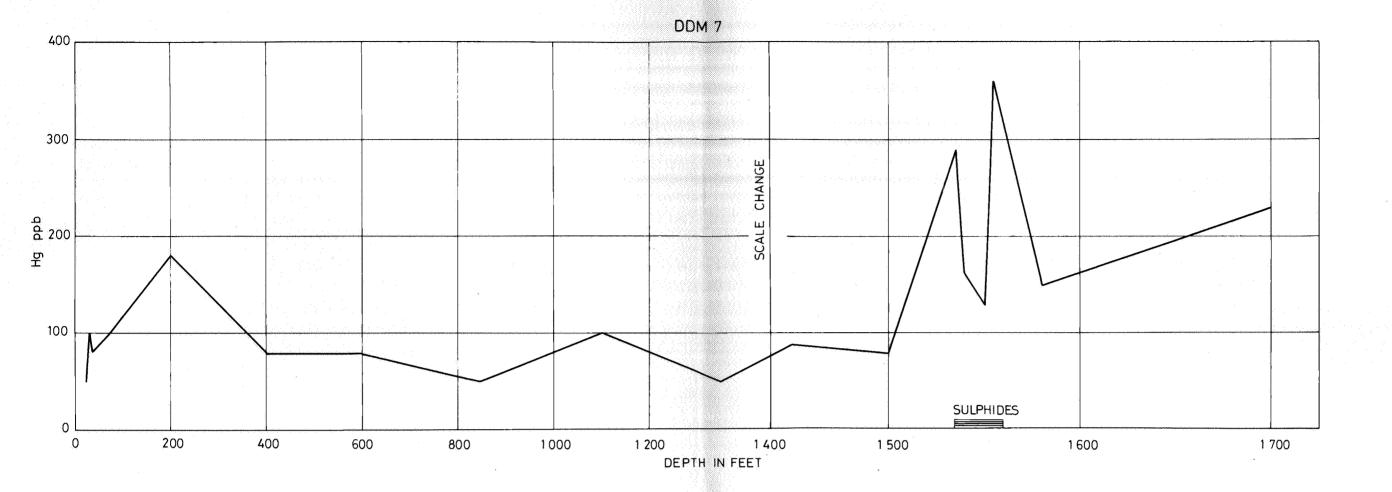


AGAINST

CU

HG

SCATTER PLOT FOR NON-MINERALISED SAMPLES (750 ppm Cu or less)



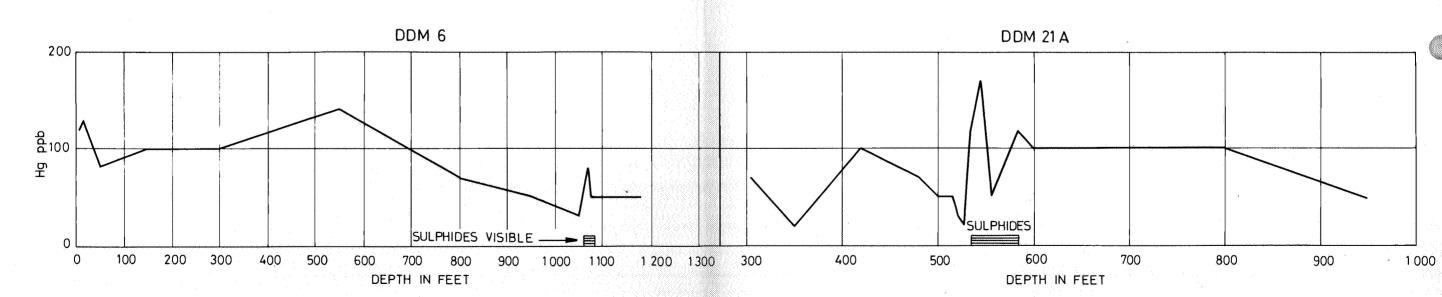


FIG. 13: DOWN-HOLE PROFILES MUTOOROO

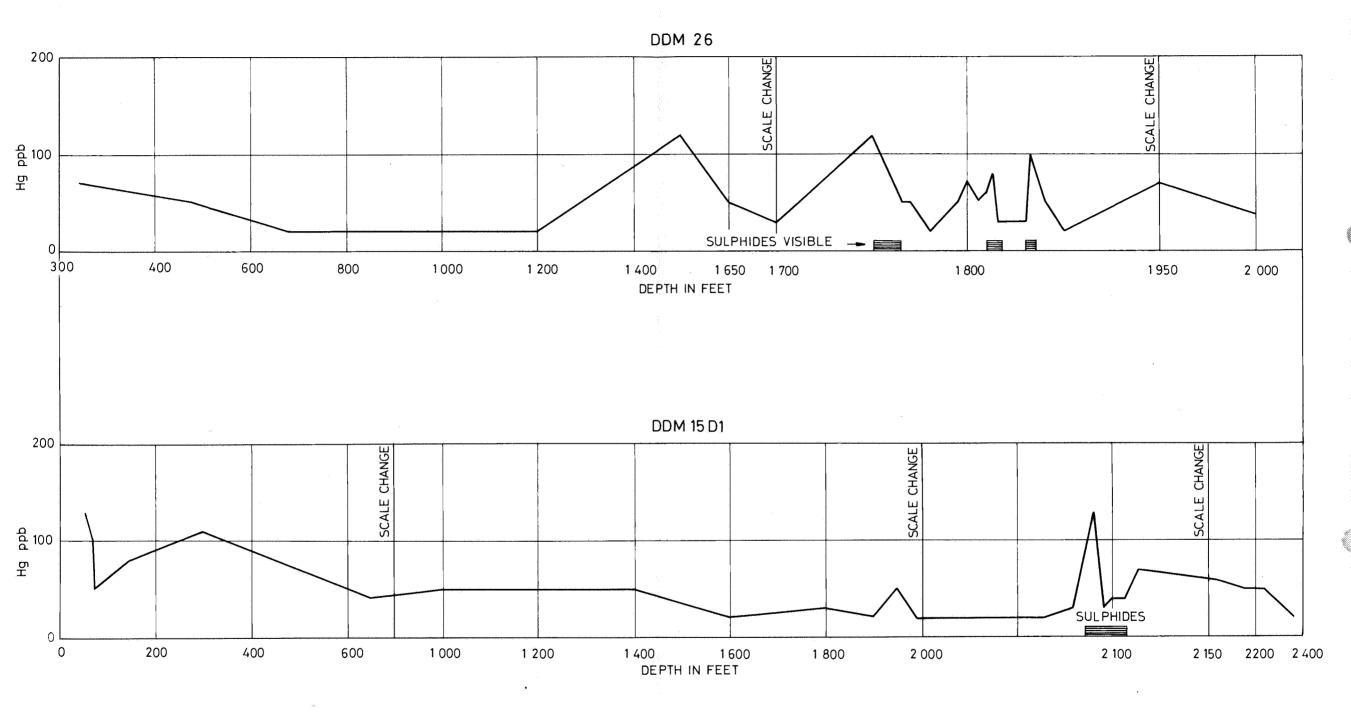


FIG.14: DOWN-HOLE PROFILES MUTOOROO



FIG. 15: DOWN-HOLE PROFILE BURRA BS 4

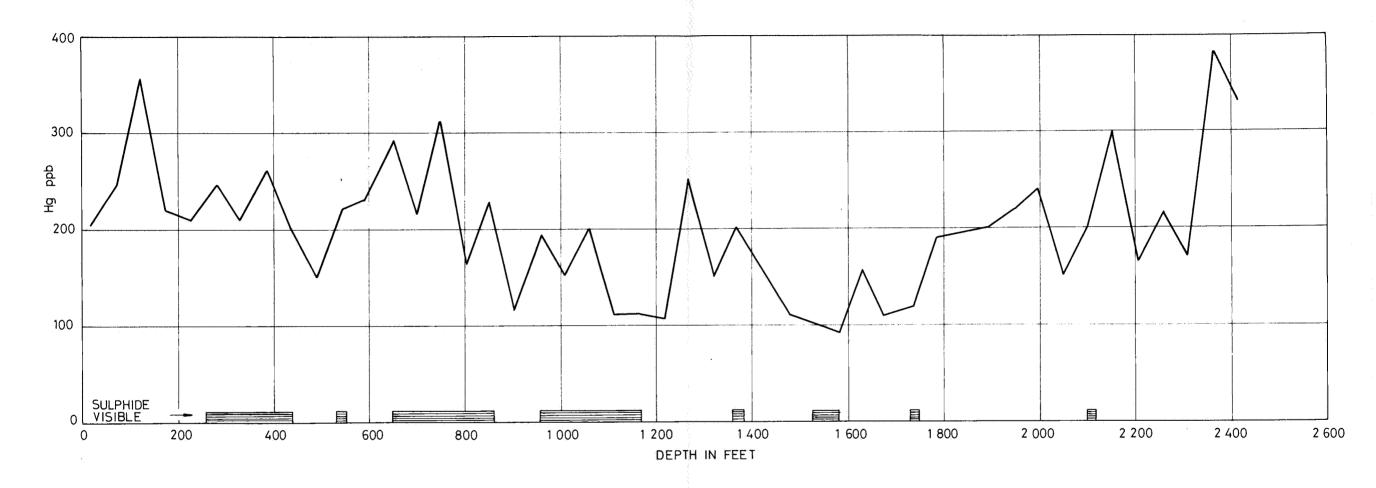


FIG. 16: DOWN - HOLE PROFILE DDH 14 MOONTA