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ELECTROLYTIC ZINC COMPANY OF AUSTRALASIA LIMITED

MINERAL RESOURCES DIVISION

EXPLORATION LICENCE NO. 4/73 - STERLING VALLEY

ANNUAL REPORT ON EXPLORATION ACTIVITY

2ND MAY, 1984 TO 6TH MARCH, 1985.

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E.Z REPORT No. T202.

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1. INTRODUCTION

This is an Annual Report for E.L. 4/73 covering work undertaken up to the Licence renewal date of 6th March, 1985. The report covers work undertaken by Electrolytic Zinc Co. of A'Asia Ltd. and Getty Oil Development Co. Ltd., on behalf of a Joint Venture between E.Z., G.O.D.C., and Aberfoyle Exploration Pty. Ltd., who are the holders of the Licence.

2. PREVIOUS EXPLORATION

Previous exploration on E.L. 4/73 is detailed in E.Z. Report No's 133 (1980), 143 (1981), 146 (1981), 150 (1982), 154 (1982), 161 (1983), 167 (1983) and T 181 (1984).

3. EXPLORATION ACTIVITY COMPLETED 2ND MAY, 1984 TO 6TH MARCH, 1985

3.1. Airborne Geophysics

Almost all of E.L. 4/73, west of the former Murchsion River was covered by a DIGHEM survey. This was flown by Getty Oil Development Co. as part of a larger survey covering a large area of adjacent E.L. 1/62. Lines were flown in an east west direction with a nominal 200m line spacing.

3.2. Geological Mapping

A limited amount of check mapping was undertaken, mainly with the specific aim of more closely locating the trace of the Henty Fault on the ground. Outcrops and areas of float were remapped and sampled on grid lines 3860N, 3740N and 3440N. The track under the E.S.E. trending powerline over Sterling Saddle was remapped and sampled from the 5000E baseline at 2480N to the edge of the E.L. at about 2120N, 4350E. Mapping traverses were undertaken up the Sterling River from about 2400N to about 2070N, and down a tributary creek from its intersection with the Sterling Valley Mine access track at 1960N, 4900E to its conjunction with the Sterling River at 2070N 4590E. These were undertaken to try and locate the Henty Fault trace and to sample rock types in the vicinity.

A creek traverse was mapped from the Sterling Saddle powerline at 2170N 4440E northwards to 2420N 4330E. This was undertaken to check a previous mapping reference to "strongly oxidised ? barite" and to follow up mapping of siltstone float and of bedded tuff in the same general area. The possibility of a stratiform submarine horizon within the volcanics associated with barite could have been very significant.

3.3. Rock Chip Geochemistry

Forty three rock samples were submitted to Analabs Pty. Ltd and analysed for Au by 30 gram fire assay, for As by AAS after nitric-perchloric acid digestion and vapour hydride generation and for Cu, Pb, Zn, Ag by AAS after nitric-perchloric acid digestion. These rocks comprised six grab samples from the costean on line 3260N at 4750E; six grab samples from the costean on line 3260N between 4570E and 4660E (refer to E.Z. Report No. 154 for details), three samples from old Sterling Valley drill holes STP 105 and STP 98, and twenty eight samples collected from mapping traverses carried out mainly in the south-western corner of E.L. 4/73. Location details are given on the data sheets in Appendix 2.

3.4. Thin Section Petrology

Twenty-one samples were submitted to Central Mineralogical Services for thin section identification and description. The samples were collected from the mapping traverses, from the costean on line 3260N and from the relogged old drill core. Location details are contained in the rock ledger sheets in Appendix 2.

3.5. Drill Core Re-Logging

During the years 1959 to 1963 five diamond drill holes were completed by E.Z. Co. as an investigation of the Sterling Valley Mine mineralisation and of geophysical targets along strike to the north. DDH STP 96, STP 98, STP 100 were drilled under the Sterling Valley Mine. STP 98 was relogged and STP 100 briefly re-examined to compare rock types. DDH STP 101 and STP 105 were

drilled under geophysical targets respectively 1,050m and 1,400m north of the old mine adit. Both holes were relogged. Seven very short diamond holes were drilled in 1947 by E.Z. under the Sterling Valley Mine workings. These holes were not re-examined because inspection of the old logs revealed that average core recovery for these holes was about 20%. Core recovery was also a problem for the later deeper holes, especially in the Sterling Vallly Mine lode zone. The combination of strongly cleaved veined mudstone and small core size used in the drilling gave poor recovery through the zones of greatest interest. Original split sampling has resulted in very little core left for some sections. For this reason systematic resampling for assay was not attempted. Some spot samples were taken for analyses or thin section petrology.

3.6. Drill Core Geochemistry

Extensive resampling and reassaying for gold was carried out on pre-existing drill core from holes in the north-western part of E.L. 4/73. Where previous split core sample pulps were recoverable, these were reassayed. Where pulps were not recovered the remaining $\frac{1}{2}$ core was resplit and a $\frac{1}{4}$ core sampled on the old intervals and assayed. Any drill core which had not been previously split, but contained weakly anomalous As or Cu values, or in which alteration or veining or pyrite had been observed, was split, sampled and analysed. All analyses for Au were carried out by Analabs Pty. Ltd. using a 30 gram fire assay. Details of the sampling are as follows.

Drill Hole	No. of Pulps Reanalysed	No. of New Split Core Analysed
STP 217	30	51
STP 218		51
STP 220		53
STP 231	52	
STP 232 A	4	
STP 232 A1	9	
STP 234	12	116

The 116 new split samples from DDH STP 234 were also analysed for Cu, Pb, Zn, Ag, Fe, Mn, Bi by AAS after nitric perchloric leach and vapour hydride generation, and for Sn, W, Sb, Te by X.R.F.

3.7. Soil Sample Geochemistry

An orientation soil sampling exercise was carried out in the area of the costean on line 3,260N between 4,580E and 4,650E. Soil samples were collected by hand auger from the base of the C horizon (taken as the point at which the auger will no longer penetrate) at 10m intervals along the bank above the costean. Where the sample depth was in excess of 0.7m a second sample was collected at half the original depth. All samples were split to produce a total soil fraction sample for analysis. The remaining split was seived to produce -10 +40 mesh; -40 +80 mesh; and -80 mesh fractions. All sub-samples (total 48) were analysed by Analabs Pty. Ltd. for Ag, Bi, Mo, Cu, Pb, Zn, Fe, Mn by A.A.S. after nitric-perchloric acid digestion, for As by A.A.S. after vapour hydride generation, Sb, Ba, W by X.R.F., and for Au by 30 gram fire assay. In twelve of the sub-samples there was insufficient material recovered to carry out the fire assay for Au, and one sample had insufficient material for the X.R.F. analyses. This was despite the collection of much larger than routine samples, specifically to overcome the potential problem of insufficent sample after seiving.

4. RESULTS RECEIVED 2ND MAY, 1984 TO 6TH MARCH, 1985.

4.1. Airborne Geophysics

Appendix 1 contains an extract from Dighem Ltd. Report No. 367. The full report contains the results from the entire Dighem survey which covered principally E.L. 1/62 and part of E.L. 9/66 as well as part of E.L. 4/73. The extract contains all the E.M. anomalies in E.L. 4/73 and the immediately surrounding parts of E.L. 1/62. The E.M. conductor anomalies, together with the Dighem resistivity contours are shown on plan No. AO-526-0037. The aeromagnetic contours from the survey are shown on plan No. AO-526-0036.

No outstanding E.M. anomalies were produced by the Dighem survey. The highest grade conductor within E.L. 4/73 is No. 16, the only grade 3 response (best possibleis grade 6). A grade 4 anomaly, No. 27, occurs outside the E.L. but is attributed to a cultural source. Most significantly, the Dighem failed to respond to the massive pyrrhotite-arsenopyrite mineralisation intersected in

DDH STP 217 and STP 221 (originally reported in E.Z. Reports No."s 133 and 143), even although a flight line ran directly over these boreholes. DDH STP 221 intersected massive pyrrhotite at 30m depth under a cover of 15 to 20m of glacial till. Fresh massive sulphides must sub-crop at the base of the glacial till. The drill intersections lie 170m west of anomaly No. 17. Ground location cannot be a significant factor as the flight line has location fixes 250m west on the highway and 300m east on the powerline track.

The drill hole sulphide zone is picked out by a local low in the resistivity contours but whether this is due to the sulphides or to the scree cover is open to question. The resistivity mapping was undertaken using a 7200Hz channel. The depth penetration of such a high frequency would not be expected to be very great. The resisitivity contour map seems to confirm this. The resisitivity contours outline, fairly clearly, Lake Rosebery and the known areas of glacial till cover extending up the Sterling River Valley. The isolated area of glacial scree cover west of the Sterling Valley Mine is also picked out by a resistivity low.

A comparison of the 7200Hz response with the equivalent co-plannar 900Hz response (see Table Appendix 1.) suggests that the depth penetration of the lower frequency has also been not very great. In all cases the 7200Hz response is greater than the 900Hz response. This is as it should be if both channels are responding to the same source. The 900Hz channel should, in theory, be capable of responding to conductors buried below the penetration limit of the 7200Hz channel. Any such conductor should give a 900Hz response and a negligible 7200Hz response. No conductors with these characteristics were interpreted by Dighem in the area of E.L. 4/73. This suggests that the DDH STP 221 sulphides are below the penetration limit of the 900Hz channel and that the resistivity low is probably a function of weakly conductive glacial cover. A clay rich fluvioglacial horizon could easily provide a conductive source. This also implies that in areas of weakly conductive cover the depth penetration of the Dighem system is only about 20m.

The effect of cover on Dighem response is further evidenced by the anomaly trend No. 1 to No. 21, which lies just east of the Sterling Valley powerline. This corresponds to a horizon of black slatey mudstone with variable disseminated pyrite and pyrrhotite content. The horizon was drilled by DDH STP 220 (see E.Z. Report No. 143) and it outcrops sporadically, particularly towards the southern end of the E.M. trend. Northwards the horizon disappears under glacial cover.

The northern end of the trend is made up of grade 1 anomalies while the southern end of the trend contains the grade 3, and two grade 2 anomalies. It looks very much like an increasing E.M. response to decreasing cover.

The Henty Fault structure appears to have very little E.M. expression. E.M. responses are mostly limited to the Farrell Slates east of the fault. On the northern boundary of the E.L. the anomaly trend No. 2 - No. 13 appears to lie almost along the trace of the Henty Fault as it is known from drilling. Each of the anomalies in this trend is interpreted as a half-space anomaly which is supposedly due to broad shallow conductors or conductive overburden. Possibly the trend is only picking up a thicker glacial or fluvioglacial profile where the old Sterling River Valley has been infilled. Co-incidence of trend No. 2 - No. 13 with the Henty Fault may be fortuitous, or perhaps it is a reflection of pre-glacial geological controls on weathering.

In fairness to the Dighem technique it should be pointed out that the Sterling Valley area represents far from ideal conditions for airborne E.M. surveys. The area contains both steep dipping pyritic black slates and a flat cover of variable thickness glacial and fluiviglacial deposits. Both of these lithologic conductors are traversed by a high tension powerline. As well as being an E.M. conductor in itself, the powerline generates an E.M. signal capable of energising any other conductors in the near vicinity. This has two effects. it creates a lot of background E.M. noise which makes detection of weak (or buried) conductors more difficult. It also means that an anomaly near the powerline may not be due to the line itself, but could be an in-ground conductor energised by the powerline signal. The response from such a conductor may be enhanced or diminished depending on how the powerline and Dighem signals interfere with one another. Despite these "mitigating circumstances", it still seems strange that the Dighem survey failed to respond to the best massive pyrrhotite intersection known from E.L. 4/73.

All the above arguments indicate that it is very difficult to assess the significance of any E.M. conductor in the area. Anomaly No. 16 would suggest that DDH STP 220, which was drilled approximately under anomaly No. 20, did not test the anomaly trend No. 1 - No. 21, in the most conductive part, and that another test 500m to the north is warranted. The grade ratings of anomalies 16 and 20, however, may not be reflecting respective concentrations of conductive material but may be a compounded result of intrinsic conductivity, cover effects, and powerline effects. Grade 1 and even possible grade anomalies may be

very significant. The Sterling Valley Mine lode zone, which sub-crops under a thin soil cover, responds with Dighem anomalies No. 31 and No. 29; a grade 1 and a possible grade anomaly. Another similar grade 1 and possible response is seen in the anomaly pair No. 15 - No. 17. This pair lie under glacial cover, parallel the strike of the Henty Fault, and occur approximately 200m across strike from the pyrrhotite hosted As ± Ag, Au, Cu, Sn mineralisation drilled in holes STP 217, 221, 231, 234 (refer E.Z. Report No. T181). Although low in grade, this might be the most significant anomaly turned up by the Dighem survey.

The aeromagnetic portion of the Dighem survey is far less problematic than the E.M. portion. This is partly because the data is not really new. The area has been previously covered by aeromagnetics and by extensive ground magnetics. The survey clearly shows that the Eastern Volcanics have a higher and more active magnetic signature than the western Mt. Black Volcanics. The dominantly rhyolitic sequence of the Eastern Volcanics is known to contain accessory magnetite and is often haematite stained in outcrop. The prominant magnetic anomalies, over the DDH STP 221 sulphide zone, and over the area of basaltic ?intrusive and granite, have been well defined previously. This survey however does highlight that the peak of the sulphide magnetic anomaly stretches between DDH STP 221 and Dighem E.M. anomaly No. 17. This may enhance the prospectivity of the E.M. anomaly.

4.2. Geology

The mapping traverses and creek mapping carried out was unsuccessful in locating any surface expression of the Henty Fault. In every instance, the area where the fault trace might outcrop is covered by transported scree. Plan no. A3-526-0041 shows the powerline and creek traverses mapped in the south western part of E.L. 4/73. The plan shows, that on the powerline traverse, phyllitic schist and strongly cleaved black mudstone outcrop just east of a small creek situated at 4,985 grid east. From this creek, westwards to the Sterling River at 4,730E, the track runs over glacial scree full of boulders and pebbles of Owen Conglomerate and quartzite. In the Sterling River, very stongly cleaved chloritised andesitic volcanics outcrop. The Henty Fault must lie somewhere between the creek and the river. On the track, at about 4,960E, there exists, what is most probably a large buried boulder, of a cleaved quartzose and volcaniclastic breccia, which was identified as Jukes Breccia (pers. comm. K.D. Corbett). At about

4,905E on the track, there occurs, what is again most probably large buried boulders, of cleaved intermediate volcanics. At both these occurrences, however, the cleavage in the exposed rock is aligned parallel to the regional cleavage. It is probably just fortuitous accident, but if one, or both, of these exposures should be bedrock, then it places the Henty Fault to within 25m either side of 4,930E.

The traverse down the W.N.W. flowing creek commencing at the mine access track at 1,960N 4,900E, and running into the Sterling River at 4,070N 4,590E, should also have crossed the Henty Fault trace. The creek is a small meandering creek, falling gently over a muddy and pebbly bed, through an area devoid of outcrop. The pebbles are all quartzite and look like Owen scree. At 2,000N 4,715E and at 2,060N 4,660E, there occur, in the creek bed, exposures of a very strongly cleaved porphyroblastic rock identified in thin section (sample no. 64338 Appendix 2 and 3) as a mylonitized metaquartzite. Once again these rocks are probably large floaters, but if they should be outcrop they again limit the location of the Henty Fault trace to around 4,630E at 2,060N. This location and the powerline location at 4,930E cannot both be correct as they imply a strike for the fault which has far too much easterly component. In most probability both sets of location indicators are false.

All the rocks mapped on plan A3-526-0041 are cleaved to some degree. The intensity of the cleavage and the degree of shearing in the andesites is greatest in the outcrops in the Sterling River near the powerline track. These outcrops are also very strongly chloritised. This suggests that these outcrops might be quite close to the trace of the fault zone. This is consistent with a change in resistivity located at about 4,750E on line 2,360N (refer to I.P. pseudo sections in E.Z. Report No. 133), which might reflect the fault location.

The creek traverse running north from the powerline track near the western edge of the E.L. was slightly more successful in achieving its aims. Fine grained bedded tuffaceous siltstone was mapped in the creek (samples 64312 and 64317 in Appendix 2 and 3). The sample 64317 outcrop lies adjacent to a dolerite so it may be only a xenolith. It does demonstrate that, in-situ or disrupted, there does exist a stratiform sediment horizon located in this part of the andesitic pile. The attempt to confirm the possible barite occurrence, reported on the old Mt. Black line 372,500N, succeeded only in finding quartz veins in a strongly oxidised andesite (sample 64313 Appendix 2). A primary volcanic foliation, based on flow

alignment of lithic fragments in an andesitic ashflow, was mapped along this creek traverse. The flow foliation has a strike of 31° A.M.G. and dips at 82° N.W. No facing could be determined. In the same creek an andesitic agglomerate containing lithic fragments of fine grained sediment outcrops S.E. of the tuffaceous siltstone. If the lithic fragments relate to the siltstone horizon, this would indicate a S.E. facing and imply that the andesite sequence is slightly overturned. Further evidence of sediment horizons in the volcanic pile comes from sample 64309 on the powerline track. This was identified in thin section (Appendix 3) as a veined and brecciated tuffaceous phyllite.

The check mapping carried out on lines further north along the Sterling Valley grid, together with the thin section report on a sample collected in the 'granite' costean, have resulted in some changes to the previously reported geology of the area. These are shown on plan AO-504-0022. On line 3,860N the eastern limit of andesite was determined by float occurrences at 4,855E (refer to geology plan in E.Z. Report No. 150). Since that mapping a large tree has fallen over and exposed definite andesitic bedrock under its roots at 4,830E (Sample 61295 Appendix 2 and 3). On line 3,740N there was previously reported float of a basaltic volcanic at 4,890E. A re-inspection of this location revealed a typical rainforest floor with a thick cover of decomposed vegetation underfoot. Small pieces of float were found in the roots of a tree at 4,890E. This float was composed of typical Farrell Slates lithologies; dark grey phyllitic mudstone and medium-grained micaceous greywacke (sample 61296 Appendix 2). These two mapping occurrences limit the location of the Henty Fault in the vicinity of 3,800N to about 4,870E to 4,880E.

On line 3,440N at 4850E an outcrop of green strongly cleaved strongly chloritised rock outcrops under the bank of the Sterling River. This was previously mapped as a basaltic rock. Resampling of this outcrop (sample no. 61297 Appendix 2 and 3) and submission for thin section description has shown it to be a trachy-andesitic tuff. The extremely cleaved and chloritised nature of the outcrop suggests it is very close to the Henty Fault contact. This is consistent with the location at about 4,870E implied above and the approximate grid north strike of the fault. The reclassification of this outcrop also reduces the areal extent of the "basaltic intrusive" interpreted from previous mapping.

Several samples were collected from the costeans on line 3,260N (samples no's 64320 to 64331 Appendix 2). These were collected primarily for geochemical analyses in order to establish a larger gold assay data base for the area. One sample (no. 64327) of quartz veined, pyritic, strongly brecciated, fine grained, green rock was submitted for thin section. The report (Appendix 3) describes the green breccia clasts as vaguely tuffaceous altered slate and not as basaltic intrusive as previously ascribed to the country rock in the costean.

A check of E.Z. Reports No's 154 and 161 confirmed that the only other material from the costeans which had been submitted for sectioning were five samples of vein material submitted to determine the sulphide assemblage present in the veins. (E.Z. Report No. 161 Appendix 1 - C.M.S. Report 82/9/12.) This mineragraphic report however describes:

Sample 48701 as containing "ill defined intraclasts of thoroughly silicified sericitic pelite."

Sample 48073A has "variably silicified breccia with clasts of sericitic pelite".

Sample 48073B "gangue is partly recognisable as clays (?sericitie)".

Sample 48075 is "Essentially a mineralised breccia with clasts of vein type quartz and quartz-schorl rock (?altered psammopelite)."

Sample 48076, the final sample describes only vein quartz and sulphide phases.

It would seem that the previous designation of the host rock to the quartz-sulphide veining in the line 3,260N costean as a basaltic intrusive was quite erroneous. The host to the mineralisation appears to be an extremely altered and brecciated sediment. The area of "basaltic intrusive" is therefore further reduced from that indicated by previous mapping.

Apart from samples No's 61300, 64315 and 64318 (Appendix 2 and 3) which are described from thin section as microgabbro in this report, the only other rocks on the Sterling Valley grid positively identified as mafic from thin section are:

Sample No. 48308; 2,480N 4,380E - a uralitised basalt.

Sample No. 48339; 3,320N 4,905E - a picritic microgabbro

Sample No. 48361; 3,740N 4,600E - an altered brecciated basalt lava.

(For these samples refer to E.Z. Report No. 150, Appendix 1 - C.M.S. Report 82/2/16)

Sample No. 48429 was taken from DDH STP 234 at 117.8m and is described in C.M.S. Report 81/12/18 (E.Z. Report No. 146) as a sheared spillite of suspected

original andesitic to basaltic composition. DDH STP 234 contained several narrow bands of this type of rock. It is proposed that the large "basaltic intrusive", previously interpreted for the central Sterling Valley grid area, probably does not exist. The body was interpreted from a few outcrops, some of which now have been positively re-identified, and a lot of scattered float. It is believed that the area more probably contains some thin spillite horizons and thin fine grained tuffaceous sedimentary horizons interbedded with the andesite lavas and pyroclastics, and some intrusive microgabbro dykes and/or sills. If this interpretation is correct it carries with it the implication that the aeromagnetic and ground magnetic anomaly in this part of Sterling Valley cannot be written off as being due to a basaltic intrusive body.

4.3. Rock Chip Geochemistry

The results of geochemical analyses of rock samples are presented in Appendix 2. The most significant samples are those collected from the line 3,260N costeans (sample no's 64320-64331). The only significant gold values reported came from quartz-sulphide veined and brecciated rocks from both eastern and western costeans. The best value was 2.01 g/t Au from Sample No. 64325. This sample is also very high in arsenic at 18.0% As. All the gold bearing samples are high in arsenic. Those from the eastern costean are also elevated in lead and silver. The best example is No. 64322 with 0.99 g/t Au, 8.55% As, 124 g/t Ag, 0.60% Pb. The samples from the western costean are less lead-silver rich, e.g. Sample No. 64328 with 1.63 g/t Au, 3.85% As, 4 g/t Ag, 180 ppm Pb. The precious (and base) metals are very strongly associated with sulphides. Samples of the granite contain no anomalous metal values and neither do non-sulphide country rock such as Sample No. 64329.

The remainder of the rock samples mostly reflect background levels for the elements analysed. Dolerite samples No's 64315 and 64318 contain slightly elevated Cu values reflecting the more basic composition of these rocks. The rhyolite samples from the Eastern Volcanics, No's 64339, 64340, 64343, display lower Zn values reflecting their acid composition.

Two samples collected along the powerline traverse returned anomalous geochemistry. No. 64307, a strongly oxidised andesitic ashflow agglomerate returned 0.16 g/t Au, and No. 64309, a quartz-veined, brecciated, tuffaceous sediment with minor pre-brecciation syngenetic pyrite in the clasts, returned 2,200 ppm As. These

samples lie approximately 100m apart and are almost directly along strike from the costeans on line 3,260N 1km to the north. This raises the possibility that a brecciated stratifrom horizon may exist for over 1km of strike through this part of the grid.

4.4. Drill Core Relogging

The relogging of the old drill core was originally undertaken with the aim of controlling a core resampling programme for gold. This aim was not achieved due to poor core recovery and small core size preventing a systematic resampling of the core. Invariably the best mineralisation, veining or alteration occurred in the zones of poorest receovery. The original logs and the relogs are presented in Appendix 4. All of the old holes lie east of the Henty Fault in Farrell Slates and Eastern Volcanics. The relogging provided a better understanding of the lithologies and associations of these two rock units and of some of the controls on the mineralisation in the Farrell Slates. The main points to arise from the relogging, and a concomitant reappraisal of the available mapping in the area of the drill holes are:-

1. The Eastern Volcanics are an acid felsic sequence, being characteristically cleaved quartz phyric rhyolites. Thin section (Appx. 3) describe the rocks as sericitised divitrified lavas and attribute the vague pyroclastic textures entirely to shearing effects.

2. Quartz + sericitie ± carbonate alteration of varying intensity occurs very frequently throughout the volcanics. The origin of the alteration is obscure. It may be entirely a regional metamorphism effect, or it may be related to the process of devitrification, or it may be a hydrothermal alteration. In STP 105 the more strongly sericitised volcanics contain disseminated pyrite and traces of galena, chalcopyrite and sphalerite, suggesting that the sericitisation may be related to a weak sulphide mineralisation system.

3. The Farrell Slates contain as much acid volcanic and felsic tuff material as 'slate'. The supposedly typical, strongly cleaved phyllitic black mudstones are a minority lithology amongst greywackes, tuffaceous wackes, reworked felsic tuffs and occasional rhyolites.

4. The Farrell Slates pass eastwards gradationally into, and probably interfinger with the Eastern Volcanics. The sequence faces west with the Farrell Slates conformable on the Eastern Volcanics.

5. Significant mineralisation in the Farrell Slates all appears to be epigenetic in type and associated with veins, fracturing and brecciation zones. It can, nevertheless, be regarded as stratabound because almost all the significant mineralisation is very strongly associated with the black mudstone units. Perhaps some component of the mineralisation, such as the source of the sulphur is syngenetic and has only been very locally remobilised.

4.5. Drill Core Geochemistry

The results of reassaying drill core for gold are contained in data sheets in Appendix 5. The results fall into three groups. The first group contains the bulk of the data and consists of six holes drilled into the Henty Fault. These are DDH STP 217, STP 231, STP 234, STP 232A and STP 232 A1 plus STP 221 which was not reassayed but has complete gold assay data from original aqua regia/A.A.S. assay. Comments on the data sheets regarding Arsenic zones 'A' 'B' 'C' 'D' refer to the zones of +1% As mineralisation as defined in E.Z. Report No. T181. Appendix 5 also contains data sheets showing all the gold values from this group of holes greater than 0.1 g/t Au listed in descending order. Twelve samples returned above 1 g/t Au with the highest intersection being:-

STP 234, 313.25-313.85m; 0.60m @ 5.00 g/t Au.

A further 54 samples returned values above 0.1 g/t Au. The following points arise from the assay data on this group of holes.

1. Many gold values are associated with high arsenic and copper but the correlation is not complete; e.g. STP 221 185.5m, 2.08 g/t Au is associated with only 70 ppm As, 250 ppm Cu. This appears to be because the gold distribution is wider than the distribution of the semi-massive sulphides which host the high arsenic zones.

2. The gold is related to sulphide mineralisation but is not proportional to the amount of sulphide present, e.g. in STP 221, at 34.7m, within Arsenic Zone A, high sulphide content containing 10.0% As contains only 0.072 g/t Au while at 52.8m minor sulphide veins containing only 70 ppm As, contains 2.08 g/t Au.

3. Both STP 221 and 234 contain a low grade gold zone lying between Arsenic Zones 'A' and 'B' which has no base metal correlations.

4. There appears to be no correlation between gold and tin mineralisation.

5. DDH STP 231 split samples have been analysed for gold by both Fire Assay and Aqua Regia dissolution. In every case where gold reports above 0.1 g/t Au, the Fire Assay method gives a higher assay. In some cases the difference is large, e.g. at 56.7m Fire Assay gives 1.29 g.t Au while Aqua Regia gives only 0.276 g/t Au. It is probable that the gold content of STP 221, for which Aqua Regia values only are available, could be much higher.

6. DDH STP 234 contains more +1 g/t Au and more +0.1 g/t Au values than any other hole. This may be just a function of STP 234 being the longest hole drilled, or it may be an indication of a strengthening of gold mineralisation to the north of the zone. STP 234 also contains the highest single assay returned.

7. DDH STP 232A and 232A1 did not properly test the Henty Fault Zone due to unexpectedly deep glacial cover. The holes did not contain any significant Arsenic Zones. Gold, however, reports in both holes with four values above 0.1 g/t Au and a peak of 0.76 g/t Au. This extends the Zone of gold potential 300m north of the arsenic cut-off at STP 234.

8. A gold content for the Sterling River Area Arsenic Zones was calculated using the arsenic "ore reserve" intersections presented in E.Z. Report T181. This gives an ore estimate of:

479,920 tonnes @ 5.02% As, 0.84 g/t Au.

The second group of reanalyses comes from DDH STP 218. This hole was drilled in the N.W. corner of E.L. 4/73 on an I.P. anomaly in the Mt. Black Volcanics. Extensive resampling and reassay failed to return any really significant gold values. Two zones are weakly anomalous in gold:-

outside ETA 4/74

42.0-45.0m; 3.0m @ 0.2 g/t Au.

82.0-84.0m; 2.0m @ 0.12 g/t Au.

Both these zones are associated with thin (about 8mm) fracture-fill veins of carbonate-quartz-epidote-fluorite containing a sulphide assemblage of pyrite, chalcopyrite, arsenopyrite, pyrrhotite (originally reported in E.Z. Report No. 133).

The third group of reanalysis comes from DDH STP 220 which was a hole drilled into a magnetic anomaly in the Farrell Slates. Two weakly anomalous gold values were returned.

89.2-90.2m; 1.0m @ 0.133 g/t Au.

94.7-95.75m; 1.05m @ 0.30 g/t Au.

Both these intervals are associated with pyrite veins with locally minor pyrrhotite, galena and chalcopyrite. Both intervals had been previously split and analysed for gold by aqua regia/A.A.S. and returned values of 0.032 g/t Au and 0.064 g/t Au respectively (originally reported in E.Z. Report No. 143). This hole also bears out the pattern of consistently higher results from fire assay as evidenced by DDH STP 231. The interval of greatest sulphide content and highest base metal assays returned from the original sampling of STP 231 was:

138.0-138.7m; 0.7m @ 3.25% Pb, 3.2% Zn, 0.49% Cu, 0.2% As, 0.15% Sn, 81 g/t Ag.

This interval only returned 0.05 g/t Au on resampling. The original aqua regia gold value was below detection limit (0.008 g/t Au).

4.6. Soil Sample Orientation Geochemistry

An orientation soil sampling exercise was undertaken in the area of the line 3,260N costean. Gold mineralisation had been demonstrated from quartz-sulphide veins in the costean. The results of the sampling are contained in data sheets in Appendix 6. It was hoped that a particular soil horizon or soil size fraction might have an optimum association with gold or a closely associated indicator element. The survey was largely unsuccessful in these aims. The main problem was that too few samples reported any gold content, and the range of gold values which were reported was very low. The inability to produce sufficient sample for a gold assay in some of the fine size fractions further reduced the possibility of finding correlations with gold.

Arsenic shows the best correlation with gold, being very strongly elevated at sample sites which report gold values. The highest value of 0.05 g/t Au also reports the highest arsenic value of 9,300 ppm As. Lead, zinc, copper and iron also report elevated values in sample sites which report gold. In this respect the soils appear to be faithfully reflecting the strong sulphide association of gold as seen in rock sampling from the costean (section 4.3. above). Antimony shows a weak correlation with gold, but along with molybdenum, bismuth, silver and tungsten, it reports values too low to allow any significant correlations to be made. Barium and manganese show no correlations with gold.

The soil profile in the sample area is not well differentiated. A fairly uniform orange-brown, clay rich, soil is developed over the bedrock. Sampling of distinct A, B or C horizons was not possible as these are not developed. Taking a sample at half depth in the thicker soil profiles was the only attempt made to determine vertical dispersion patterns through the soil profile. The results of this trial were inconclusive. For example, at 4,580E higher values of gold and arsenic reported from the sample at 0.60m depth than from the sample at the base of the profile at 1.25m depth, whereas in the adjacent site at 4,590E the opposite situation prevailed with higher gold and arsenic values reporting from the bottom sample at 0.9m.

The different size-fraction sub-samples show a remarkable uniformity in geochemical response at any one sample site. None of the metals analysed show any preferred partitioning into any particular size fraction. This feature, together with the inability of some samples to provide sufficient fine fraction material for a fire assay, suggests that sampling for gold in soils should analyse the -10 mesh fraction instead of the traditional -80 mesh fraction. It is not known how successfully this might be applied across a wider range of soil types.

APPENDIX 1.

REPORT No. 367

AN EXTRACT COVERING THE AREA OF E.L. 4/73

OF

DIGHEM III SURVEY OF THE

ROSEBERY EAST AREA, TASMANIA

FOR

GETTY OIL DEVELOPMENT COMPANY LIMITED

BY

DIGHEM LIMITED

TORONTO, CANADA
1984.

Z. DVORAK,
VICE PRESIDENT.

CONDUCTORS IN THE SURVEY AREA

Anomaly 1 - 21
15 - 17

A pair of well defined bedrock conductors is indicated by these grade 1 to 3 anomalies. The conductors have produced a narrow conductive zone correlating with a narrow magnetic anomaly. The data suggests that the conductive horizon may extend further north.

Anomaly 24

This grade 1 anomaly reflects about 1.7% concentration of magnetite. A weak bedrock conductor may exist at this location which is being masked by magnetite. If real, it may extend further north toward 23.

Anomaly 25, 26

These grade 1 anomalies were classified by the symbol L because they correlate with a road and powerline. However, their EM signatures are quite different from the other responses further north and south. This is well demonstrated by the resistivity patterns at the corresponding locations. The anomalies may reflect the combined effects of culture and bedrock conductor. They should be checked on the ground.

Anomaly 29 - 31

A weak bedrock conductor is indicated by this grade 1 anomaly. The resisitivity patterns suggest that the conductor may extend further south. It is surmized that 29 - 31 reflects a conductive horizon which may also contain anomalies 25 and 26.

Anomaly 32

A weak bedrock conductor is indicated by this grade 1 anomaly.

021

218022

DIGHEM ANOMALIES E.L. 4/73

Refer to plan No. AO-526-0037
(Extract from Report No. 367 by Z. Dvorak, Dighem Limited, 1984)

Anomaly No.	Anomaly Location A.M.G. Co-ords N	E	Interp Symbol	+ Grade	Co-Axial 900hz REAL ppm	Co-Axial 900hz Quad.ppm	Co-Planar 900hz Real ppm	Co-Planar 900hz Quad ppm	Co-Planar 7,200hz Real ppm	Co-Planar 7,200hz Quad ppm	Vertical Dyke Conductivity MHOs	Vertical Dyke * Depth Metres	Horizontal Sheet Conductivity MHOs	Horizontal Sheet Depth Metres	Conductive Earth Resistivity OHM-M	Conductive Earth Depth Metres
1.	5,375,550	384,960	B?	2	2	2	5	3	9	5	8	28	2	141	39	103
2.	5,320	4,560	H	1	1	3	1	3	10	20	1	0	1	45	434	15
3.	5,310	4,640	L?	X												
4.	5,280	4,970	B	1	9	19	9	21	53	67	4	7	1	46	265	7
5.	5,160	4,520	H	1	1	3	1	3	8	15	1	10	1	55	429	25
6.	5,180	4,980	B	1	5	9	4	7	20	17	4	0	1	86	253	30
7.	4,990	4,500	H	1	1	2	1	2	8	18	1	0	1	44	474	14
8.	4,950	4,960	B	1	2	7	1	3	17	15	2	5	1	113	626	15
9.	4,920	3,030	E	X												
10.	4,770	4,410	H?	1	2	3	2	4	16	20	1	0	1	45	274	18
11.	4,840	4,910	B	1	2	5	4	4	12	10	3	13	1	97	1035	0
12.	4,780	2,580	H	X												
13.	4,740	4,390	H?	1	1	3	0	6	26	37	1	1	1	42	230	18
14.	4,550	4,180	E	X												
15.	4,550	4,540	B	1	5	12	1	8	10	4	3	0	2	142	61	99
16.	4,550	4,830	B	3	9	2	4	8	10	6	19	17	1	177	77	73
17.	4,370	4,470	B	X												
18.	4,370	4,620	L	X												
19.	4,370	4,770	B	2	4	7	7	9	18	17	5	17	1	94	127	49
20.	4,220	4,670	B?	2	17	20	2	8	13	16	7	0	1	48	324	0
21.	4,040	4,570	B?	1	0	4	2	6	8	13	1	2	1	68	586	32
22.	3,830	4,440	L	2	4	3	1	2	3	2	8	45	1	189	866	39
23.	3,500	5,140	E?	X												
24.	3,370	5,090	B?	1	1	2	0	2	0	10	1	12	1	74	6434	8
25.	3,170	4,270	L	1	2	8	1	4	7	7	1	19	1	111	393	74
26.	3,000	4,200	L	1	0	4	0	4	5	6	1	12	1	88	850	43
27.	2,460	2,600	L?	4	2	1	4	1	2	2	24	76	3	201	16	170
28.	2,320	2,560	L	2	1	1	3	0	2	2	8	93	2	214	56	167
29.	2,070	3,870	B?	X												
30.	2,030	5,750	L	X												
31.	1,830	3,840	B	1	3	6	1	3	4	5	3	35	1	136	90	91
32.	1,200	3,020	B?	1	1	3	3	3	8	14	1	22	1	81	761	43
33.	0,690	2,890	?	X												

\+ For explanation of Interpretation Symbol and Grade refer to the Legend on Plan No. AO-526-0037.

* Estimated depth may be unreliable because the stronger part of the conductor may be deeper or to one side of the flight line, or because of a shallow dip or overburden effects.

APPENDIX 2.

Rock Sample Data Sheets.

Electrolytic Zinc Co. of A'asia Ltd.
Rosebery, Tas.

ROCK SAMPLE LEDGER

PROJECT: E.L. 4/73 STERLING VALLEY LOCALITY: Sterling River Area Sta.V.grid
COLLECTED BY: I McDONALD DATE: DEC. 1984 → FEB 1985

Sample Number	A.M.G.Co-ordinates N	A.M.G.Co-ordinates E	Sample Type	Geological Description	Metal Content (p.p.m. unless specified) Au	Ag	As	Cu	Pb	Zn	Thin Section
61295	5,373,915	384,325	Grab	Green strongly cleaved and chloritised Andesitic lithic ashflour. from Sta V. grid line 3860N at 4830E	x	x	32	20	20	260	✓
61296	5,373,785	384,345	"	Float in tree roots on rainforest floor Dark grey phyllitic Mudstone and grey mg micaceous wacke' From Sta V. grid line 3740N at 4890E	x	x	26	15	65	25	
61297	5,373,500	384,210	"	Strongly sheared and chloritised andesitic volcanic from outcrop in bank of Sterling River on grid line 3440N at 4850E	x	x	44	110	55	240	✓
61298	5,372,545	384,035		Probably float. Possibly outcrop in an area of scree under the Sterling saddle ~~power~~ powerline. Grey, strongly cleaved volcaniclastic granule breccia. Felsic volcanic and quartz clasts. (Julia Breccia - pers comm K. Corbett)	x	x	2	5	45	40	
61299	5,372,525	383,975		Probably float. Possibly outcrop in an area of scree under powerline. Green cleaved chloritised feldspar phyric Andesitic ashflour. Poss. flow banded lava.	x	x	9	50	35	260	✓
61300	5,372,480	383 760		Outcrop on powerline track. Green fg. non-porphyritic Dolerite ?intrusive	x	x	5	10	30	210	✓
64305	5,372,475	383 740		Mafic to intermediate volcanic. Looks like a feldspar phyric andesitic lava but has large ? xenolith of fg. ? dolerite in it. Float under powerline. Thin Section identifies rock as a dacite.	x	x	3	10	50	170	✓

024

Electrolytic Zinc Co. of A'asia Ltd.
Rosebery, Tas.

ROCK SAMPLE LEDGER

PROJECT: Sterling Valley E.L. 4/73 LOCALITY: Sterling River area
COLLECTED BY: J. McDonald. DATE: FEB 1985

Sample Number	A.M.G. Co-ordinates N	A.M.G. Co-ordinates E	Sample Type	Geological Description	Metal Content (p.p.m. unless specified) Au	Ag	As	Cu	Pb	Zn	Thin Section
64306	5,372,465	383,695	Rock grab	Manganese oxide stained gossanous quartz vein. Float from Powerline Track	x	1·0	18	135	60	185	
64307	5,372,465	383,690	"	Outcrop. Strongly oxidised Andesitic ashflow agglomerate. Lithic up to 50mm long. From Powerline track.	0·16	1·0	4	35	40	170	
64308	5,372,460N	383,675E		White oxidised Feldspar phyric chloritised Andesite lava. from Powerline Track	x	x	x	15	40	210	✓
64309	5,372,450N	383,600		Quartz veined brecciated chloritic tuffaceous phyllite Minor pyrite in quartz veins. From powerline track	x	1·5	2200	140	360	190	✓
64310	5,372,430	383,530		White massive quartz vein from Powerline Track.	x	x	3	10	30	x	
64311	5,372,420	383,470		Pink/green Andesitic lithic ashflow agglomerate from Powerline Track.	x	x	3	10	45	145	✓
64312	5,372,530	383,475		Pale green banded f.g. bedded lithic tuffaceous greywacke Float in creek	x	x	25	110	50	190	✓
64313	5,372,535N	383,400		Strongly oxidised andesitic ashflow with massive white quartz veins from line 372,500N at 383,410E Mt. Sale grid	x	1·5	x	15	40	55	
64314	5,372,550	383,425		c.g. Adesitic ashflow agglomerate with black f.g. sediment and pink quartz phyric rhyolite lithic fragments	x	x	7	55	45	335	✓
64315	5,372,575	383,405		Green f.g. Dolerite in creek	x	x	10	200	60	155	✓

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21,802

025

Electrolytic Zinc Co. of A'asia Ltd.
Rosebery, Tas.

ROCK SAMPLE LEDGER

PROJECT: Sterling Valley E.L. 4/73 LOCALITY: Sterling River area
COLLECTED BY: I. McDonald DATE: Oct. 1984 – Feb 1985

Sample Number	A.M.G. Co-ordinates N	A.M.G. Co-ordinates E	Sample Type	Geological Description	Au	Ag	As	Cu	Pb	Zn	Thin Section
					Metal Content (p.p.m. unless specified)						
64316	5,372,590	383,385	Rockgrab	Pink & green chloritised sheared andesitic lithic lapilli ashflow. - outcrops on creek	x	x	1	35	65	165	✓
64317	5,372,595	383 390	"	green-grey banded fg. lithic & vitric bedded ash tuff. with fg disseminated pyrite and minor arsenopyrite. On edge of dolerite outcrop so possibly a xenolith	x	0·5	43	165	90	205	✓
64318	5,372,600	383,375	"	Grey-green f.g. massive dolerite or microgabbro intrusive.	x	0·5	5	155	30	130	✓
64319	5,372,605	383375		Dolerite with white quartz veins	x	x	5	20	50	205	
				Sample Nos 64320 – 64325 all come from the Costean on Sterling Valley grid line 3260N at 4750E							
64320	5,373,370	384075		Pyritic veined brecciated ?basaltic rock.	x	4·0	3·00%	65	3250	95	
64321	"	"		Pale clay material after ?oxidised granite vein	x	1·0	410	165	905	260	
64322	"	"		Sulphide veined ?basaltic rock.	0·99	124·0	8·55%	420	0·60%	170	
64323	"	"		Pink and green granite possibly with assimilated country rock component.	x	2·0	70	35	145	200	
64324	"	"		Green chlorite altered granite	x	2·0	430	20	195	150	
64325	"	"		Quartz veined sulphide rich ?Basaltic rock	2·01	72·0	13·0%	380	3400	110	

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2136020

Electrolytic Zinc Co. of A'asia Ltd.
Rosebery. Tas.

ROCK SAMPLE LEDGER

PROJECT: Sterling Valley E.L 4/73 LOCALITY: Sterling River area
COLLECTED BY: I McDONALD DATE: DEC 1986 - FEB 1985

Sample Number	A.M.G. Co-ordinates N	A.M.G. Co-ordinates E	Sample Type	Geological Description	Metal Content (p.p.m. unless specified) Au	Ag	As	Cu	Pb	Zn	Thin Section
				Sample Nos 64326 – 64331 all come from the costean on Sterling Valley grid line 3260N between 4570E and 4660E							
64326	5,373,410	383 945	Rockgrab	Quartz-(pyrite) vein material from 4618E	0·18	1·0	1000	65	x	40	
64327	"	"	"	Quartz veined, strongly brecciated pyritic mafic pelitic tuff rock from 4615E	0·09	2·5	1000	170	30	225	✓
64328	"	"	"	Sheared sulphidic ?basaltic rock from about 4620E	1·63	4·0	3·85%	1100	180	200	
64329	5,373,420	383910	"	Strongly sheared basaltic rock from contact with granite vein at 4575E	x	0·5	x	35	20	210	
64330	"	"	"	Quartz-(pyrite) veins from near granite vein at 4575E	x	2·5	550	625	x	30	
64331	"	"		Altered granite from vein at 4575E	x	1·0	95	45	35	70	
64332	5,372,540	383,785	"	Green sheared chloritised weakly feldspar phyric Mafic to Intermediate lava. From Sterling River.	x	2·0	32	50	15	395	✓
64333	5,372,495	383,775	"	Green sheared strongly chloritised mafic to intermediate lava or fg intrusive	x	1·5	23	10	x	255	
64334	5,372,460	383 770	"	Pink/green sheared chloritised weakly feldspar phyric andesite lava.	x	2·0	8	55	x	465	

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027

Electrolytic Zinc Co. of A'asia Ltd.
Rosebery, Tas.

ROCK SAMPLE LEDGER

PROJECT: Starling Valley E.L. 4/73 LOCALITY: Starling River area
COLLECTED BY: I McDONALD DATE: DEC 1984 - FEB 1985

Sample Number	A.M.G.Co-ordinates N	A.M.G.Co-ordinates E	Sample Type	Geological Description	Metal Content (p.p.m. unless specified) Au	Ag	As	Cu	Pb	Zn	Thin Section
64 335	5,372,430	383,770	Rock grab	green aph porphyritic, chloritised, weakly biotitised f.g. Dolerite.	x	1·0	16	150	x	235	
64 336	5 372 410	383 765	"	green sheared chloritic mafic to intermediate lava or f.g intrusive with traces of pyrite and arsenopyrite	x	0·5	26	15	30	170	
64 337	5,372,375	383,735	"	green cleaved chloritic very weakly feldspar phyric mafic to Intermediate lava or f.g. intrusive	x	1·5	5	10	15	165	✓
64 338	5,372,180	383,660	"	Grey strongly sheared porphyroblastic Meta-Quartzite. Mylonitisation of the rock gives a pseudo-quartz porphyritic texture. Probably Float.	0·01	x	52	10	70	40	✓
64339			½ core	DDH STP 105 @ 382 ft depth. grey/white banded pyritic quartz-sericite schist after acid volcanic.	0·08	2·5	86	35	215	40	
64340			½ core	DDH STP 105 @ 420 ft depth. Finely laminated mudstone and volcaniclastics wacke with minor pyrite.	x	1·0	22	40	45	80	
64341			"	DDH STP 101 @ 167 ft depth. Sheared banded quartz rich sericite lava - Thin Section only.							✓
64342			"	DDH STP 101 @ 306 ft depth. Weakly banded sheared quartz phyric sericitic lava - Thin Section.							✓
64343			"	DDH STP 99 @ 243 ft depth. Sheared strongly sericite altered felsic lava.	x	1·0	8	5	40	40	

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218028

028

Electrolytic Zinc Co. of A'asia Ltd.
Rosebery, Tas.

ROCK SAMPLE LEDGER

PROJECT: Sterling Valley E.L. 4/73 LOCALITY:
COLLECTED BY: I McDONALD DATE: FEB 1985

Sample Number	A.M.G.Co-ordinates N	A.M.G.Co-ordinates E	Sample Type	Geological Description	Metal Content (p.p.m. unless specified)										Metal Content (p.p.m. unless specified)					
														Thin Section						
64344			½ core	DDH STP100 from 5.35 ft depth. Fg. Rhyolite lava. - sheared and weakly sericitised - Thin section										✓						

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APPENDIX 3.

Rock Sample Thin Section Descriptions.

C.M.S. Report 85/2/19.

REPORT CMS 85/2/19

Twenty-one rock chip and drill core samples from the Sterling Valley area were received for petrological examination. Representative thin-sections were prepared and examined together with their respective cobaltinitrite-stained offcuts, with carbonate stain tests performed as warranted. Attached tabulated descriptions summarise the microscopic data, and include interpretative comments.

Summary

This suite is a composite of altered and sheared felsic intermediate to acid volcanics, microgabbroic and sedimentary rocks.

Volcanics include andesitic-trachyandesitic and rhyolitic types, characterised by chloritic (± epidote) and sericitic (± quartz, calcite) alteration assemblages respectively. Lava-like facies (including minor intrusives?) predominate, but pyroclastics are also present, and a few rocks exhibit shear-related boudinaging effects ("pseudofragmentals"). Volcanics are typically unmineralised to weakly pyritic.

Sediments include labile (volcanomict) turbiditic types, possibly intercalated with the altered intermediate volcanics on the basis of gross compositional similarities and exhibiting essentially similar alteration patterns.

Vaguely tuffaceous altered pelites appear as clasts in quartz-healed breccias. A single example of mylonitically sheared quartzose psammopelite may be contrasted with the labile turbiditic and tuffaceous pelitic facies.

Microgabbros reflect uralitic alteration and, locally, shear-related "retrograde" chloritisation. These basics include amygdaloidal types and, whilst medium-grained, may represent core zones of thick flows, dependent on field evidence.

D. Cowan, B. Sc.

Sample No.	Classification - Composition	Fabric	Accessories	Comments
S 295 (T.S. 62655)	Andesitic ?Tuff. Frequent albitised/chlorite-stained plagioclase, minor chloritised ferromag phenocrysts in a chloritised/weakly epidote-stained feldspar-microlitic matrix. Sporadic intraclasts of vein-quartz.	Sheared breccia-like with millimetric, angular lava clasts, chlorite matrix. Vague relict lithic fragmental.	Rare quartz-pseudomorphed K-feldspar microphenocrysts. Ultrafine leucoxenic semi-opaques.	Andesitic lithic tuff or lava-breccia (tuff lava), quartz-veined/chloritised subsequently moderately sheared and brecciated.
61297	Trachyandesitic Tuff. Chloritic/variably epidote-stained lava clasts, "megaclasts" in a chlorite-matrixed psammitic fragmental comprising sericitised plagioclase, minor K-feldspar grains, lava clasts.	Phyllitic, with sporadic grit- to pebble-sized angular lava clasts, sheared angular fine to medium sandy matrix.	Minor sheared epidote veinlets. Thinly disseminated leucoxenised opaques (clastic).	Altered/sheared "andesitic" tuff, distinctly K-feldspathic in comparison with 61295, similarly altered. Subaqueous(?), psammitic with sporadic lapilli.
61299	Andesite. Chlorite with pervasive fine albitised plagioclase laths, conspicuous albitised plagioclase phenocrysts/phenocrystal clusters. Disseminated cloudy microcrystalline epidote.	Phyllitic/incipiently crenulated. Relict glomeroporphyritic "andesitic". Weakly (flow-)banded.	Minor microcrystalline quartz, boudinaged quartz veinlets. Fine leucoxenic semi-opaques, trace degraded pyrite.	Chloritised/epidote-stained and relatively sheared andesitic lava, boudinaged on a microscale ("pseudo-fragmental"), incipiently oxidised-pyritic.
61300	Microgabbro. Albitised/chlorite-stained plagioclase laths with a mesostasis of chlorite, including minor chlorite-semi-pseudomorphed ?pyroxene laths, granules, conspicuous fine magnetite.	Weakly plagioclase-porphyritic, medium-grained "basaltic". Moderately sheared.	Traces of epidote, minor sheared discontinuous epidote-, chlorite-quartz veinlets.	Basic minor intrusive or core zone of relatively thick flow, dependent on field evidence. Alteration, shearing pattern analogous to 61295 etc.
64305	"Dacitic" Tuff. Sericite-pseudomorphed feldspar, minor recrystallized quartz, chloritised ferromag crystals/fragments, sporadic chloritised lithic clasts in a matrix of sericitic/chloritic microgranular quartz.	Loose, poorly sorted fragmental with a vaguely relict eutaxitic matrix. Mildly sheared.	Conspicuous leucoxenic semi-opaques, traces variably degraded magnetite.	Silicified-sericitised-chloritised, mildly sheared quartz andesitic-dacitic fragmental. Relict features consistent with an ignimbritic lithic-vitric-crystal tuff.
64308	Trachyandesitic Tuff. Variably sericite-pseudomorphed plagioclase and K-feldspar crystals, fragments, clusters, minor chloritised "trachytic" lava clasts in a matrix of sericitic-chloritic microcrystalline quartzofeldspathic material.	Poorly sorted psammitic fragmental with a vaguely relict shardy matrix. Mildly sheared.	Minor chloritised pumiceous lava clasts. Semi-pervasive fine epidote. Conspicuous leucoxenised opaques.	Crystal-rich trachyandesitic fragmental (lithic-vitric-crystal tuff), poorly diagnostic in terms of subaqueous or subaerial mode of deposition.
64309	Silicified Breccia. Clasts of chloritic muscovite- and muscovite-quartz phyllite in a matrix of fine- to medium-grained vein-type quartz.	Random angular, submillimetric to centimetric clasts. Crudely crustiform, unstressed matrix.	Oxidised fine-grained pyrite in clasts, matrix. Minor phyllite-pseudomorphous sericitic quartz aggregates.	Quartz-matrixed breccia, tectonic characteristics. Phyllite clasts include vaguely tuffaceous sandy pelitic types. Pyrite is pre-brecciation ("syngenetic") in part.
64311	Trachyandesite. Frequent albitised/epidote-stained plagioclase, subordinate sanidine-anorthoclase phenocrysts/phenocrystal clusters in a microcrystalline alkali feldspathic to thoroughly chloritised/epidote-stained groundmass.	Tuff lava-like composite of chloritic and feldspathic porphyritic lavas.	Leucoxenised opaques. Disseminated single to loosely clustered oxidised pyrite anhedra.	Relict features consistent with a trachyandesitic agglomeratic lava (flow-)breccia, probable flow-marginal facies. Relatively altered, weakly stressed.

Sample No.	Classification - Composition	Fabric	Accessories	Comments
64312	Tuffaceous Siltstone. Framework of silt- to (minor) fine sand-sized, ultrafinely epidote-stained basic-intermediate lava clasts, subordinate feldspar, minor quartz grains. Saussuritic indeterminate matrix.	Variably graded turbiditic silty clastic banded on sub- to millimetric scale. Incipient high-angle cleavage.	Disseminated fine "syngenetic" pyrite. Minor discontinuous veinlets of cloudy epidote, albite, quartz.	Turbiditic labile volcanomict siltstone ("tuffaceous greywacke"). Finer detail obscured by "saussuritic" (cloudy, ultrafine) epidote alteration.
64314	Lithic Tuff. Framework of angular to platy, variably chloritised/epidote-stained trachy-andesitic-trachytic lava clasts. Sparse epidote-stained chloritic matrix. Patchy corroded relics of actinolite in clasts, matrix.	Poorly sorted, variably moulded lithic fragmental with a weakly crenulated phyllitic overprint.	Minor leucoxenised opaques, minor traces of oxidised pyrite.	Felsic-intermediate ("trachytic") lithic tuff ("lapilli tuff") with partly moulded clasts, suggestive of subaerial deposition. Extensively altered, moderately sheared.
64315	Uralitised Microgabbro. Albitised/weakly epidote-stained plagioclase laths and intergrown to interstitial actinolite-pseudomorphed pyroxene with conspicuous fine magnetite. Minor discontinuous epidote veinlets.	Weakly plagioclase-porphyritic, medium-grained, subophitic ("doleritic"). Weakly stressed.	Minor chlorite amygdales, epidote disseminations, actinolite-replacive chlorite aggregates.	Uralitised-saussuritised, weakly porphyritic/amygdaloidal microgabbro with close affinities to 61300; interrelation similarly dependent on field evidence.
64316	Lithic Tuff. Angular to platy clasts of trachy-basaltic to trachytic, variably chloritised lava; subordinte to minor alkali feldspar grains. Chloritic-feldspathic-tuffaceous siltstone/fine sandstone matrix.	Poorly sorted silt- to sand-supported angular clastic. Sheared (phyllitic).	Patchy cloudy microcrystalline epidote stainings, rare clots of prehnite, clastic leucoxenic opaques.	Affinities with conceivably a weakly subaqueously reworked variant of 64314. Includes minor clots, sheared veinlets of ankeritic carbonate.
64317	Tuffaceous Siltstone. Framework of silt- to (minor) fine sand-sized saussuritic/poorly defined lava clasts, subordinate to minor feldspar grains, minor prehnitised mica flakes. Cloudy saussuritic indeterminate matrix.	Closely analogous to 64312.	Thinly disseminated fine to ultrafine "syngenetic" pyrite. Minor pyritic epidote-chlorite-albite veinlets.	Close affinities with 64312, slightly relatively K-feldspathic in comparison; similarly altered, but essentially unsheared.
64318	Uralitised Microgabbro. Albitised plagioclase laths and actinolite-pseudomorphed intergrown to interstitial pyroxene with sporadic veinlets, minor amygdales of epidote, quartz, actinolite.	Analogous to 64315, slightly finer-grained. Mildly stressed.	Fine-grained magnetite (primary).	Close affinities with 61300 and particularly 64315. Weakly glomero-porphyritic with composite clusters of albitised plagioclase, uralitised pyroxene phenocrysts.
64327	Breccia. Vein-type quartz with interspersed clots, vermiform inclusions, millimetric aggregates of chlorite. Interspersed clasts, zones of variably chloritised, locally silicified sericitic slate.	Quartz-veined to brecciated/quartz-healed slate. Restressed.	Thinly disseminated variably oxidised chalcopyrite in vein, replacive chlorite, fine leucoxenic opaques in clasts.	Close affinities with 64309. Clasts are similarly vaguely tuffaceous; relatively altered (chloritised/locally silicified).
64332	Trachyandesite. Phenocrysts of albitised plagioclase, subordinate sanidine-anorthoclase, minor chloritised ?pyroxene. Chlorite groundmass with pervasive albite microlaths.	Phyllitic/relict, weakly glomero-porphyritic, "andesitic".	Traces of cloudy microcrystalline epidote. Minor traces oxidised fine to ultrafine pyrite.	Close affinities with 64311 and related trachyandesitic fragmentals. Typically altered (chloritised, weakly epidotised) and mildly sheared.

Sample No.	Classification - Composition	Fabric	Accessories	Comments
64337	Andesitic ?Tuff Lava. Sericitic, albitised to sericite-pseudomorphed plagioclase phenocrysts, minor chloritised ferromags in an extensively sericitised-chloritised microcrystalline to felsitic-perlitic felsic groundmass.	Phyllitic, with secondary boudinaged-fragmental features. Relict andesitic/vaguely xenolithic, flow-brecciated.	Evenly disseminated leucoxenised opaques. Frequent sheared chlorite films. Disseminations variably oxidised pyrite.	Relatively altered and sheared andesitic ?tuff lava with marked "pseudofragmental" features, reflecting shear-boudinaged perlitic devitrification structures.
64338	"Metaquartzite". Pervasively stressed microcrystalline to medium-grained quartz with interspersed foliae of sericite, semi-pervasive fine to ultrafine variably degraded hematite, ferruginised mica (biotite?) flakes.	Semi-mylonitic. Relict fine to medium sub-angular, shale-parted, sandy clastic.	Traces of relict detrital tourmaline (schorl), rare rounded zircons.	Thoroughly sheared("mylonitised") quartzose psammopelite. Reflects quartz-hematite alteration of shaly bands, partings (pre-shearing).
64341	Biotite Rhyolite. Disseminated quartz, subordinate sericitised biotite, sericite-calcite-stained albite, minor K-feldspar phenocrysts in a pervasively sericitic microcrystalline quartzofeldspathic groundmass.	Phyllitic, relict strongly porphyritic, weakly banded, perlitic-felsitic.	Rare zircons, minor traces of apatite.	Sericite(-calcite-quartz)-altered/sheared porphyritic biotite rhyolite, conceivably a semi-chilled minor intrusive, dependent on field evidence.
64342	Biotite Rhyolite. Frequent relict quartz, subordinate sericitised biotite and sericite (± quartz, carbonate) semi-pseudomorphed feldspar phenocrysts; sericite/microcrystalline quartz groundmass.	Phyllitic, relict coarsely-evenly porphyritic/vaguely flow-banded.	Minor cleavage-concordant weakly stressed veinlets of quartz, calcite with disseminated dark red sphalerite.	Close affinities with 64341, relatively coarsely porphyritic, sericitised, sheared in comparison. Veinlets are broadly syntectonic; cleavage-controlled.
64344 (T.S. 52675)	Biotite Rhyolite. Disseminated quartz, sericitic albite, subordinate silicified ?K-feldspar, sericitised-chloritised biotite phenocrysts. Extensively sericitised felsitic groundmass. Sporadic clots, impregnations of quartz, calcite.	Finely porphyritic, incipiently banded, felsitic; weakly quartz-calcite amygdaloidal, mildly sheared.	Sparse leucoxenic semi-opaques. Rare fine-grained pyrite.	Affinities with 64341, 64342, similarly altered, but relatively mildly sheared. Amygdale quartz, calcite possibly mobilised into veinlets (64342).

APPENDIX 4.

Diamond Drill Hole Logs.

Original Logs	STP 96
	STP 98
	STP 100
	STP 101
	STP 105
Re-logs	STP 98
	STP 101
	STP 105

035

Electrolytic Zinc Company of Australasia Limited

WEST COAST DEPARTMENT

A 90027

Hole No. STP96

Sheet No. 1

RECORD OF DIAMOND DRILL CORE

SPECIFICATIONS

Mine: STERLING MINE
Location: Surface
N. Coord.: 8119N
E. Coord: 5259E
R.L. 725
Direction: 298°StM Grid
Angle: -81°
Length: 785'
Size Hole: AX

SURVEY DATA

Footage	Dir'tion	Angle	Footage	Dir'tion	Angle
0	298°	81	600		59½
100	Local	78	700		56¼
200	Grid	76			
300		73			
400		67½			
500		62½			

OBJECT: To test the Sterling Valley Mine line of lode approximately 450' below surface at the mine workings.

RESULT: Hole intersected 10' width of average ore at 600 ft. below surface

PLOTTED
40 Plan
XS. X01471
L.P. V011
100 Plan
S.
L.P.

1959

PROGRESS				DESCRIPTION			ANALYTICAL DATA									DIPS	
Date	Depth	Advance	Amount of Core	From	To	ROCK TYPE	From	To	Amount of Core	Pb %	Zn %	Cu %	Ag. ozs.	Au dwts.	Fe %		Footage	Angle
May				0	16½	Fine grained sericitic tuff.	693	703	1'6"	2.3	2.2	0.07	1.2	<0.1	8.6		12sh	32
14	10	10	-	16½	24	Tuff with haematite staining.	703	708	2'7"	10.9	8.0	0.20	6.0	0.8	14.0		24sh	27
15	22	12	-	24	27	Tuff.	708	712	3'6"	15.3	8.3	0.35	8.3	0.6	13.6		39sh	20
18	39	17	12	27	35	Tuff with haematite staining	712	721	5'8"	4.7	3.6	0.35	3.3	0.4	16.0		39½b	15
19	90	51	26	35	44	Sericitic tuff											70sh	32
20	105	15	1'6"	44	76	Haematite stained tuff	Average Assays Empirically Reduced for core recovery										88sh	29
21	110	5	5'	76	79	Tuff											143sh	33
26	115	5	1'6"	79	124	Haematite stained tuff											170sh	25
27	125	10	2'6"	124	126	Tuff well sheared	703	712	6'1"	11.2	6.9	0.23	6.1	0.6	11.7		203sh	30
28	155	30	18'	126	170	Altered and haematised tuff	703	721	11'9"	7.5	4.9	0.26	4.4	0.5	12.4		211sh	27½
29	173	18	6'6"			coarser grained with felds-											236c	16½
June						par phenocrysts and resembles											236b	9
1	184	11	3'6"			massive pyroclastics more											245b	26
2	211	27	7'3"			than tuff.											294c	22
3	241	30	17'6"	170	174	Medium grained tuff.											294b	16
4	271	30	22'10"	174	175	Haematised zone											322c	23
5	281	10	-	175	177	Medium grained tuff.											322b	21
9	321	40	28	177	209	Haematised medium grained											362½	18
10	361	40	30			tuff.											b=c	
11	384	23	14'3"	209	231	Fine to medium grained weak-											391	19
12	414	30	9'			ly sheared sericitic tuff											b=c	
17	434	20	10'			with haematite staining in											413sh	17
18	447	13	6			blebs and along veinlets.											444sh	21
19	455	8	2'			Tuff becoming more massive and											452	22
25	466	11	8'			less sericitic.											b=c	
26	471	5	3'	231	246	Grey fine to medium grained											449	19
29	476	5	2'			tuff, weakly sheared and with											b=c	
30	484	8	7'			patches of decomposed tuff.											500	21
July				246	271	Grey medium grained tuff											b=c	
1	495	11	4'			with chloritic streaks and											525	20
2	505	10	4'			blebs.											b=c	
3	510	5	-	271	353	Greenish grey slightly											536	
7	525	15	4'			chloritic tuff.												
8	530	5	2'	353	362	Medium grained tuff with												

Form No. WCD 8

2180036

Hole No. STP96

Sheet No. 2

A 85493

PROGRESS				DESCRIPTION			DIPS		DESCRIPTION			DIPS	
Date	Depth	Advance	Amount of Core	From	To	ROCK TYPE	Footage	Angle	From	To	ROCK TYPE	Footage	Angle
July						carbonate blebs.			767	785	Some slate fragments in beginning but predominantly shattered massive pyroclastics	536 b=c	15
				362	403	Grey fine grained slightly chloritic tuff becoming more sheared and sericitic.					Hole branched 777'	545c	26
9	537	7	3'									551 b=c	22
10	541	4	2'									563 sh	20½
14	546	5	2'	403	449	Grey green somewhat chloritic fine grained tuff well sheared			777	784	Broken massive pyroclastics. This massive pyroclastic slate banding is definitely faulted.	723b	24
15	553	7	2'									727b	45
16	558	5	1'	449	498	Sericitic medium grained tuff fairly well sheared.						737b	43
20	561	3	0'3"										
21	566	5		498	503	Interbedded tuff and slates					MINERALISATION		
23	568	2		503	548	Black slate							
31	570	2				537-548 highly sheared			0	126	Unaltered rock shows traces disseminated fine grained pyrite 89-105 Zone of quartz veins up to 3" wide (poor core recovery)		
Aug 4	572	2	2'	548	566	Black slate							
5	574	2	2			563-566 Badly broken							
6	581	7	4"			553-558 No core recovered							
11	584	3		558	573	Grey slatey tuff well bedded							
12	591	7	6"			and strongly parted on bedding planes.			209	231	Disseminated fine grained pyrite in unaltered tuff. at 232½ leached quartz carbonate vein with traces semi massive fine grained pyrite. at 246, 270, 271½ quartz carbonate veinlets. at 273 quartz carbonate veinlets with traces disseminated fine grained pyrite and sphalerite.		
13	598	7											
14	601	3	6"	573	645	Black slate highly cleaved and shattered.							
18	604	3											
19	611	7	3"	645	668	Contorted black slate not so badly broken.							
20	614	3	3'										
28	620	6		668	703	Fairly broken black slate - more chunky and not so highly cleaved.							
31	622	2	3"										
Sept.													
1	630	8	2"	703	712	Isolated fragments of slate in mineralised zone.							
2	634	4	6"										
3	639	5	3'	712	721	Slates more or less replaced by mineralisation.			273½	277	Quartz carbonate and chlorite gangue with broken tuff throughout - no visible sulphides.		
4	642	3											
14	644	2	2	721	738	Broken and contorted black slates.							
15	649	5	4										
16	652	3	3	738	746	No core recovered			281	283	Quartz carbonate and chlorite gangue with broken tuff throughout - no visible sulphides.		
17	657	5	3	746	750	Broken and contorted black slate becoming decomposed as hole approached 750'							
18	662	5	2'6"										
21	665	3	1						375	376	Quartz carbonate gangue.		
22	673	8	2	750	755	No core - presumably a large fault zone.			443	443½	Quartz vein at 535 traces disseminated medium grained sphalerite, pyrite and galena		
23	683	10	3										
24	693	10	4	755	766½	Broken and decomposed massive Pyroclastics some slatey fragments included.							
25	699	6											
30	701	2	1'6"						537	541	9" core fault zone. at 537 2" quartz and coarse grained pyrite and chalcopyrite in faulted slates, remaining 7" black slate with		
Oct 1	705	4	3"	766½	778	Broken tuff massive pyroclastics. Hole branched 767'							
2	708	3	2½										
13	711	3											

Hole No. STP96

A 85493

Sheet No. 3

PROGRESS				DESCRIPTION			DIPS		DESCRIPTION			DIPS	
Date	Depth	Advance	Amount of Core	From	To	MINERALISATION	Footage	Angle	From	To		Footage	Angle
Oct.						seams and blebs of quartz							
14	713	2				with fine and coarse grained							
15	716	3	3			pyrite.							
16	721	5				at 562 Quartz vein ½"					LOGGED BY ISG:JG		
20	723	2	3	693	703	Disseminated to semi massive					Dist: Mine 1.		
21	724	1	1			fine medium grained pyrite					Geol 1.		
22	727	3	1'6"			throughout - veinlets and					File 1.		
27	731	4	2'6"			blebs of quartz gangue with					Dept. of Mines 1.		
28	733	2	2'			coarse grained galena and							
29	736	3	1			sphalerite traces only.							
30	738	2	1	703	708	Quartz carbonate gangue with							
Nov.						semi massive fine to coarse							
4	743	5	Nil			grained low grade ore.							
				708	712	Quartz carbonate gangue with							
1960						semi massive coarse grained							
March	746	3	Nil			galena - average high grade							
15	750	4	3'0"			ore.							
16	755	5	Nil	712	721	Shattered slates and quartz							
17	760	5	2			carbonate gangue with semi							
18	765	5				massive pyrite and traces of							
						galena and sphalerite							
May						average minor mineralisation							
12	768	3	-	755	778	Quartz veins							
24	773	5	0	767	772	Quartz veins							
25	778	5	1	782	784	Quartz carbonate.							
31	779	9	1										
June													
1	785	6	4										
Hole	branched	at 777'											
2	784	7	2'6"										

038

Electrolytic Zinc Company of Australasia Limited

WEST COAST DEPARTMENT

A 90027

RECORD OF DIAMOND DRILL CORE

Hole No. STP98

Sheet No. 1

SPECIFICATIONS

Mine: Sterling Valley — Length: 878
Location: Sterling Mine — Size Hole: Bx
N. Coord.: 8098N
E. Coord: 4503E
R.L. 611'
Direction: 118° S.T.M.Grid
Angle: -57°

SURVEY DATA

Footage	Dir'tion	Angle	Footage	Dir'tion	Angle
0		-57	600		39*
100		-56½	700		36*
200		-56½	800		33*
300		-51½			
400		-45½			
500		-42			

* Extrapolated

OBJECT: To test the Sterling Mine mineralisation approximately 450' below surface and 300' south of the STP96 intersection.

RESULT: Sterling Mine mineralised zone was intersected from 663-691 as infilling of a shatter zone with quartz carbonate gangue with splashes of pyrite and minor galena and sphalerite.

PLOTTED
40 Plan
X.S. X0154
L.P. V011
100 Plan
S.
L.P.

PROGRESS

Date 1960	Depth	Advance	Amount of Core
July			
18	3	3	-
19	8	5	1
20	13	5	10"
21	23	10	1
27	31	8	1
Aug.			
3	41	10	4"
4	51	10	2"
15	56	5	-
16	66	10	1'6"
17	96	30	0'7"
25	106	10	-
31	116	10	-
Sept.			
7	136	20	4
9	156	20	-
Oct.			
12	171	15	-
13	201	30	-
14	211	10	-
15	231	20	20
1961 Jan 17	285	8	8
18	295	10	3
19	301	6	5
20	321	20	10.6
Feb.			
5	361	20	-
6	391	30	-
7	431	40	-
8	441	10	-
10	461	20	10

DESCRIPTION

To From	From To	ROCK TYPE
0	31	Glacial wash.
31	41	Decomposed black slates.
41	200	Fairly well sheared black slate.
200	210	Well sheared tuff MP with chloritic seams.
210	248	Tuffy MP becoming more massive and less well sheared and tuffy but still with chloritic seams.
248	288	Fairly coarse tuff with some thin slatey bands.
288	341	Dark grey slatey tuff.
341	486	Coarse grey tuff with thin black slate bands.
486	569	Dark grey tuff with thin interbeds of black slate.
569	705	Black slate becoming highly contorted from 660' onward.
705	815	Black slate - well sheared.
815	878	Tuffaceous MP.
		MINERALISATION
191	200	Broken core with limenitic slates.
at	227	Quartz 3" with traces disseminated coarse grained galena.
at	235	Quartz 6"
at	262	Quartz 2"
at	272	Quartz 2"

ANALYTICAL DATA

From	To	Amount of Core	Pb %	Zn %	Cu %	Ag. ozs.	Au. Xg dwts.	Fe %
46	56	Sludge	0.3	0.4	0.10	0.15	Nil	7.6
56	66	"	0.4	0.4	0.07	0.1	Nil	8.0
66	76	"	0.3	0.3	0.05	0.1	Nil	5.9
76	86	"	0.3	0.4	0.07	Tr.	Nil	6.9
86	96	"	0.3	0.5	0.07	0.1	Nil	6.7
645	656	"	0.2	0.1	0.05	Tr.	Nil	6.0
656	666	"	0.15	0.1	0.10	Tr.	Nil	5.6
666	676	"	0.10	0.1	0.10	0.08	Nil	6.5
676	686	"	0.15	0.1	0.10	0.13	Nil	8.9
686	691	"	0.10	0.1	0.10	0.10	Nil	8.6
691	695	"	0.15	0.1	0.1	0.07	Nil	8.8
695	705	"	0.15	0.7	0.12	0.38	Nil	9.3
785	795	"	0.20	0.1	0.07	0.03	Nil	2.8
795	805	"	0.15	0.1	0.07	0.08	Nil	2.3
CORE ASSAYS								
663	666	1'6"	3.0	9.4	1.03	21.0	2.0	37.2
666	676	2'9"	0.1	0.5	0.18	0.2	Nil	20.8
676	682	2'4"	0.15	2.0	0.18	0.6	Nil	18.1
682	686	2'	0.25	1.8	0.13	0.6	Nil	11.5
686	691	2'	0.10	1.6	0.10	0.7	Nil	10.0
AVERAGE								
663	666	4'3"	2.3	7.2	0.79	16.2	1.5	28.6

DIPS

Footage	Angle

Form No. WCD 8

21803

039

Hole No. STP98

Sheet No. 2

A 85493

PROGRESS				DESCRIPTION.			DIPS		DESCRIPTION			DIPS	
Date	Depth	Advance	Amount of Core	From	To	MINERALISATION.	Footage	Angle	From	To	MINERALISATION	Footage	Angle
Mar.				295	302	Zone of quartz carbonate blebs and veinlets.			790	795	Quartz carbonate and pyrite blebs with traces of disseminated coarse grained galena and sphalerite.		
1	466	5	2'6"										
2	486	20	11'6"	at	398	Quartz carbonate and chloritic veins with traces of disseminated fine grained pyrite.							
3	496	10	5										
8	516	20	14										
9	528	12	6	463	466	Zone of quartz carbonate veinlets up to ½" wide.							
10	536	8	8										
13	546	10	-	at	557½	Quartz carbonate vein 1".							
14	566	20	20	569	664	Thin carbonate veinlets in slates with isolated traces of pyrite.							
15	586	20	4										
16	606	20	5										
22	616	10	2'4"	664	664½	Veinlets and splashes of disseminated to semi massive medium grained pyrite.							
24	626	10	2'										
April													
5	679	13	5	666	676	Veinlets and splashes of semi massive medium grained pyrite and 2% iron.							
6	691	12	6										
12	705	14	8								LOGGED BY ISG:JMc		
13	715	10	2	676	682	As above but with traces of galena. Sphalerite 681-682.					Dist: Geol. 1. ✓		
17	719	4	-								Mine 1.		
18	735	16	-	682	686	Pyrite as above but with occasional flecks of galena and sphalerite.					File 1.		
19	755	20	1								Dir. of Mines 1.		
20	775	20	2										
21	785	10	-	686	691	Flecks and veinlets of pyrite and quartz carbonate gangue a semi massive fine grained.							
May													
2	795	10	1'6"										
3	805	10	1'	691	695	Blebs of pyrite in quartz carbonate veinlets - traces of galena and sphalerite.							
4	815	10	1'5"										
8	821	6	-										
9	841	20	2'	695	696	Disseminated to semi massive coarse grained pyrite, sphalerite and galena probably minor mineralisation.							
10	861	20	7'6"										
11	878	17	5'										
				696	705	Veinlets of semi massive pyrite and chalcopyrite with traces of galena and sphalerite semi massive coarse grained.							
				705	790	Quartz carbonate veinlets through-out with disseminated medium grained pyrite - would probably average 1% pyrite.							

218040

040

Electrolytic Zinc Company of Australasia Limited

WEST COAST DEPARTMENT

Hole No. STP100

A 90027

RECORD OF DIAMOND DRILL CORE

Sheet No. 1

SPECIFICATIONS

Mine: Sterling Valley. Length: 716'
Location: Sterling Mine Size Hole: Bx.
N. Coord.: 8877N
E. Coord.: 4783E
R.L. 582'
Direction: 119°G
Angle: -57°

SURVEY DATA

Footage	Dir'tion	Angle	Footage	Dir'tion	Angle
0	119°G	-57°	600		29½°
100		55½°	700		28°
200		52½°			
300		48½°			
400		40½°			
500		34½°			

OBJECT: To test the Sterling Valley mine mineralisation approximately 400' north of STP96 intersected at 450' below surface on the IP and Em anomalies.

RESULT: Hole encountered two pyrite mineralized zone corresponding to Em anomalies. Main zone quartz - carbonate - pyrite infilling of fracture zone in slates. No economic mineralisation was intersected.

PLOTTED
40 Plan
xsV015
LPV011
100 Plan
S.
L.P.

1961 PROGRESS

Date	Depth	Advance	Amount of Core
June			
28	38	34	3
30	40	2	-
	CASING		
July			
31	69	8	-
Aug.			
1	84	15	-
8	114	20	1'6"
9	124	10	6"
10	132	8	4
11	137	5	3
14	142	5	3'6"
30	172	30	5
31	235	63	25
Sept.			
1	249	14	12'8"
4	264	15	8
5	284	20	4
6	304	20	4
7	314	10	2'8"
8	324	10	3'6"
11	334	10	3'
12	364	30	17°
13	384	20	7'6"
14	434	50	26'6"
18	455	21	-
19	485	30	-

DESCRIPTION

From	To	Rock Type.
0	98	Glacial wash - boulders of Owen & Jukes?
98	110	Tuffy MP - some fine grained black patches.
100	212	Dark grey tuffy slate.
212	252	Black slate crumpled and fractured particularly in zone 230 - 252.
252	342	Well cherted black slate cleavage = bedding.
342	348	Coarse grained tuff.
348	354	Black slate.
354	355	Coarse grained tuff.
355	373	Black slate.
373	450	Grey slates, tuff tending towards MP.
450	528	Slightly sheared fractured coarse tuffy MP.
528	545	Broken and contorted black slates.
545	560	Broken and contorted black slates, faulted boundary.
560	650	Medium grained relatively unsheared tuff.
650	716	Massive pyroclastics.

Mineralisation.

From	To	
212	237	Traces of disseminated fine grained pyrite or bedding planes @ 233 quartz 2".

ANALYTICAL DATA

From	To	Amount of Core	Pb %	Zn %	Cu %	Ag. ozs.	Au dwts.	Fe %	FeCO3
237	244¼	7'3"	1.9	0.2	0.12	1.1	∠0.1	15.0	
528	545	3'6½"	0.1	0.2	0.05	1.0	0.2	9.5	4.6

DIPS

Footage	Angle
124c	27°
132c	31°
140	40°
b =c	
162	60°
b=c	
200b	33°
229b	18°
252b	38°
294b	42°
318b	46°
339b	40°
358b	48°
393b	48°
402b	52°
441b	52°
495b	60°
527b	75°
560sb	62°
650sb	58°
680sb	48°
706sb	70°
715sb	65°

Form No. WCD 8

2180411

Hole No. STP100

A 85493 *Sheet No.* 2

PROGRESS				DESCRIPTION			DIPS		DESCRIPTION			DIPS	
Date	Depth	Advance	Amount of Core	From	To	Mineralisation.	Footage	Angle	From	To		Footage	Angle
Sept				237	244½	Disseminated to semi massive fine grained pyrite in streaks or blebs with quartz carbonate gangue and ¼" seam of massive coarse grained galena at 237½.							
20	505	20	13										
21	515	10	-										
26	528	13	-										
27	540	12	3										
28	545	5	8	251	252	Disseminated to semi massive fine grained chalcopyrite and pyrite in blebs.							
Oct.													
2	552	7	6										
4	571	19	3'6"	264	280	Quartz carbonate mineralisation with blebs of fine grained pyrite.							
5	590	19	8'										
6	600	10	3										
11	610	10	2	292	293	Quartz carbonate mineralisation with blebs of fine grained pyrite.							
12	620	10	1'6"										
16	640	20	6										
17	656	16	15	528	545	Quartz carbonate veinlets with blebs dissem.to semi massive pyrite or traces of galena and sphalerite at 529½ 1½" of quartz carbonate gangue with blebs of semi massive medium grained pyrite and galena.							
24	660	4	2'6"										
25	690	30	19										
26	706	16	9										
27	716	10	7										
Logged by: IG/DY.				At	559	1" quartz carbonate vein with traces semi massive coarse grained pyrite and galena.							
DIST: Mine.			1	695	701	Quartz chlorite rhodochrosite vein with sericite and traces of fine grained pyrite.							
Geology.			1										
File.			1										
Dept. of Mines.			1										

042

Electrolytic Zinc Company of Australasia Limited

WEST COAST DEPARTMENT

A 90027

RECORD OF DIAMOND DRILL CORE

Hole No. StP101

Sheet No. 1

SPECIFICATIONS

Mine: Sterling Valley
Location: Surface
N. Coord.: 7867S
E. Coord: 15630E
R.L. 708
Direction: 301°Rosebery Grid
Angle: -47°
Length: 870'
Size Hole: BX-AX
0-644 BX
644-870 AX

SURVEY DATA

Footage	Dir'tion	Angle	Footage	Dir'tion	Angle
0'	301°G	-47°	500		-13°
100'		-44°	600		- 9°
200		-32¾°	700		- 6°
300		-25°			
400		-15°			

OBJECT: To test the Rio Geohpysical Anomalies at their maximum development on the Sterling Valley grid on line 32S at approx 500' below surface.

RESULT: Hole explained wide anomalies by intersecting three pyritic zones in slates 300' below surface no economic mineralisation was encountered.

PLOTTED
40EPlan ✓
S.
L.P.
100 Plan
X s.yo254
L.P.

1961

PROGRESS

Date	Depth	Advance	Amount of Core
Dec			
19	7	7	2'6"
20	13	6	4
21	33	20	19
22	50½	17½	17
29	70	20	19
1962			
Jan.			
3	90	20	20
4	130	40	39
5	150	20	20
15	156	6	6
17	198	42	42
18	216	18	12
22	221	5	2
23	248	27	22
24	278	30	29
25	288	10	10
29	303	15	15
30	311	8	8
Feb.			
15	341	30	9
16	351	10	2
19	361	10	3
20	381	20	5
21	411	30	19
22	436	25	9
23	446	10	-
27	468	22	8
28	498	30	18
Mar. 1	518	20	9

DESCRIPTION

To	From	ROCK TYPE
0	15	Weathered MP.
15	153	Fairly coarse tuffy MP.
153	163	Sheared coarse grained tuffy MP.
163	170	Finer grained less sheared MP - texture about that of a coarse tuff.
170	186	Grey somewhat sheared tuffaceous MP.
186	216	Coarser grained more sheared chloritic MP.
216	230	Grey siliceous MP.
230	301½	Coarse grained tuffy MP.- slightly sheared - with chloritic seams and carbonate blebs.
301½	321	Slatey tuff - somewhat chloritic.
321	386	Broken and contorted black slates with interbedded tuffs.
386	421	Sheared coarse tuffs with thin black slates interbeds.
421	478	Broken and contorted black slatey tuffs.
478	560	Fine to medium grained tuffs - fairly well sheared.
560	580	Black slates.
580	638	Medium grained tuffs - fairly well sheared and becoming contorted from 625 onward.
638	705	Broken and contorted black slates. 638-644 Broken slates 664-689 Contorted " 689-705 Broken "

ANALYTICAL DATA

From	To	Amount of Core	Pb %	Zn %	Cu %	Ag. ozs.	Equiv Ag dwts.	Fe %	Py%
344'	351	1'3"	1.0	2.8	0.25	0.4	<0.1	17.0	5.1
351	371	5'4"	0.3	0.6	0.10	0.25	<0.1	11.0	9.8
437	444	2'1"	0.2	0.4	0.67	0.15	<0.1	6.0	5.1
444	447	1'3"	0.1	0.2	0.22	2.50	0.4	26.5	35.4
447	456	3'1"	0.1	0.1	0.02	0.25	<0.1	6.9	5.1
456	463	3'6"	0.1	0.1	0.01	0.25	<0.1	7.6	2.9
638	644	2'6"	0.15	0.1	0.10	0.05	Nil	13.3	16.8
644	654	2'9"	0.1	0.1	0.17	0.40	Nil	11.8	20.2
654	664	4'7"	0.1	0.1	0.10	0.20	Nil	10.4	13.7
664	674	4'4"	0.1	0.2	0.20	0.15	Nil	12.1	13.9
674	684	2'6"	0.1	0.1	0.07	0.15	Nil	7.8	10.1
684	694	2'0"	0.1	0.1	0.20	0.20	Nil	9.0	14.6
694	705	0'9"	0.15	0.1	0.55	0.40	Nil	8.0	12.4
AVERAGES.									
344	371	6'7"	0.48	0.17	0.15	0.29	<0.1	12.5	8.5
437	463	9'11"	0.13	0.19	0.23	0.48	0.13	9.8	8.0
638	705	19'5"	0.10	0.10	0.21	0.23	Nil	10.1	14.2

Geol. 1.

DIPS

Footage	Angle
150Sh	35°
161Sh	39°
171Sh	45°
188Sh	52°
201Sh	47°
245Sh	48°
292Sh	43°
310Sh	42°
331Sh	55°
381Sh	67°
401Sh	68°
418Sh	63°
508Sh	63°
545Sh	60°
568b	50°
625	
b=Sh	47°
669	64°
Sh=b	
674	57°
Sh=b	
720Sh	50°
744Sh	80°
754Sh	88°
800Sh	80°
842Sh	76°
849Sh	79°

Form No. WCD

218043

043

Hole No. StP101

A 8089

Sheet No. 2

1962 PROGRESS				DESCRIPTION			DIPS		DESCRIPTION			DIPS	
Date	Depth	Advance	Amount of Core	From	To	ROCK TYPE	Footage	Angle	From	To	MINERALISATION	Footage	Angle
Mar.				705	715	3" core only - Grey medium grained tuff with slatey streaks.			437	440	Veinlets of massive pyrite with traces of pyrrhotite & chalcopyrite.		
2	531	13	5										
7	537	6	3	715	755	Interbedded grey tuff & black slates.			440	442	Quartz carbonate veinlets with traces pyrite chalcopyrite & chlorite.		
8	560	23	5										
12	570	10	3	755	765	Medium grained tuff with slatey beds.							
13	590	20	4										
14	620	30	7	765	819	Coarse grained tuff.			442	444	Traces semi massive fine grained pyrite in gangue veinlets and slates.		
15	638	18	5	819	841	Medium grained tuff.							
19	644	6	3	841	879	Fine grained tuff with interbedded slates.			444	447	Carbonate gangue with semi-massive to massive fine grained pyrite.		
Ax						MINERALISATION							
23	664	20	-	30	156	Occasional quartz carbonate veins up to ½ thick.			447	453	Visible fine grained pyrite in thin seams and films on slates		
28	684	20	11										
29	690	6	2	171	171	Qtz carbonate vein 1"			453	479	Traces pyrite in thin seams & films.		
April				at	174	" " " "							
2	694	4	2'	at	175	" " " 1" with traces galena and pyrite.			479	531	Zone of thin veinlets of carbonate gangue with occasional specks & blebs of fine grained pyrite & traces of chalcopyrite.		
3	715	21	6"										
4	725	10	2'6"	at	176½	1" Quartz Carbonate vein.							
5	745	20	9	at	179	½" veinlet of pyrite with chalcopyrite and traces galena.							
6	755	10	7						531	539	Veinlets of quartz - siderite chlorite gangue with traces of Pyrite.		
10	785	30	14	at	179½	1" Quartz carbonate vein							
11	795	10	12	at	200½	1" " " "			539	587	Isolated veinlets and splashes of fine grained pyrite in semi massive form pyrite less than ½% of total core.		
12	805	10	13	219	227	Isolated quartz veins up to ½" wide.							
13	815	10	-										
17	819	4	6	at	243	Quartz vein ½".							
May				at	243½	Quartz veinlet and traces of chalcopyrite.			638	705	Splashes & blebs of pyrite & phrrhotite with traces of chalco-pyrite in broken slates - veinlets in contorted slates with quartz chlorite and siderite gangue. Sulphides average 1% of core.		
9	839	20	10										
10	860	21	14	at	246	Quartz vein 1".							
11	870	10	10	at	305	¼" seam - semi massive fine grained galena, sphalerite & pyrite in quartz.							
				at	316	¼ veinlet of siderite, pyrite chalcopyrite and pyrrhotite.			at 705		Quartz siderite - chlorite		
				344	351	Zone of veinlets & blebs of fine grained massive to semi massive pyrite with traces of galena and sphalerite.			at 715 at 725½		veinlets with traces of fine grained pyrite.		
									731	735	Zone of quartz chlorite veinlets with blebs & seams of massive fine grained pyrite.		
				351	371	Country rock is broken, contorted and in filled with quartz, chlorite and siderite gangue in ½" to 3" veins, carrying semi massive pyrite & pyrrhotite with traces of chalcopyrite. Sphalerite & galena are notably absent.			735	755	Traces only of fine grained pyrite in thin quartz-siderite-chlorite veinlets		
									755	765	Thin quartz veinlets only.		

Form No. W.C.D. 111

2180441

Hole No. StP101

A 8069

Sheet No. 3

PROGRESS				DESCRIPTION			DIPS		DESCRIPTION			DIPS	
Date	Depth	Advance	Amount of Core	From	To		Footage	Angle	From	To	MINERALISATION	Footage	Angle
									765	860	Carbonate blebs throughout isolated quartz & carbonate veinlets up to ¼" thick at 838 2" zone of thin seams of massive fine grained sphalerite in siderite gangue.		

Form No. W.C.D. 111

Electrolytic Zinc Company of Australasia Limited

WEST COAST DEPARTMENT

RECORD OF DIAMOND DRILL CORE

Hole No. SEP 105

Sheet No. 1

A 90027

SPECIFICATIONS

Mine: Sterling Valley Length: 1062
Location: Line 20S Rio Grid Size Hole: BX
N. Coord.: 6926 S) Rby. Grid.
E. Coord: 17560 E)
R.L. 700'
Direction: 300° Rby Grid
Angle: -53°

SURVEY DATA

Footage	Dir'tion	Angle	Footage	Dir'tion	Angle
0'	300° RG	-53°	500'		-32.5°
100'		-53°	600'		-30°
200'		-49°	700'		-25.0°
300'		-41.5°	800'		-20°
400'		-35.0°	900'		-19.8°
			1000'		-12.5°

OBJECT: Hole to test the IP, EM, SP, and gravity anomalies in this area.

RESULT: The anomalies were adequately explained by the pyrite mineralisation intersected.

PLOTTED
40 Plan
S.
L.P.
100 Plan
S.
L.P.

1962 PROGRESS

Date	Depth	Advance	Amount of Core
Jul.26	38	38	16
27	53	15	13
30	63	10	4
31	87	24	24
Aug.1	108	21	20
2	118	10	8
3	122	4	-
6	132	10	10
7	156	24	15
16	171	15	15
17	181	10	8
21	201	20	-
22	203	2	-
24	221	18	9
27	227	6	6
28	231	4	-
29	241	10	10
30	248	7	5
Sep.5	256	8	10
6	266	10	10
Oct.3	286	20	20
4	296	10	10
8	306	10	9
9	326	20	19
10	358	32	30
15	368	10	10
16	398	30	30
17	428	30	30
25	434	6	1
26	438	4	1

DESCRIPTION

From	To	ROCK TYPE
0	26'4"	White and grey sheared Massive pyroclastics with grit sized Phenocrysts of quartz and plagioclase in finer matrix, chlorite is streaked out along cleavage surfaces.
26'4"	33	Light grey siliceous massive pyroclastics with dark grey band and some agglomeritic material. There is a tendency towards segregation of chloritic zones, feldspar phenocrysts predominate.
33	38	Siliceous medium grey mottled massive pyroclastic with quartz chlorite streaking along cleavage, quartz phenocrysts predominate, agglomeritic around 37'
38'	42'	Siliceous light grey massive pyroclastics with some quartz chlorite streaks.
42'	53'	Banded tuff and light grey massive pyroclastics. High proportion sand size grains in matrix. Banding reflects minor chlorite content.

ANALYTICAL DATA

From	To	Amount of Core	Pb %	Zn %	Cu %	Ag. ozs.	Au dwts.	Fe %	Total Sulphides
327'3"	337'0"	95%	0.2	0.2	0.05	0.02	NIL	5.8	7.2
337	342	95%	0.1	0.2	0.05	0.10	NIL	8.7	12.9
342	350	94%	0.1	0.2	0.03	0.06	NIL	5.2	3.6
350	358	93%	0.1	0.2	0.05	0.26	NIL	5.7	7.5
568	588	18'5"	-0.1	-0.1	0.10	0.4	-0.1	8.4	4.7
763	763'6"	5"	-0.1	-0.1	0.10	0.3	1.4	23.7	11.4
789'	794'6"	4'10"	-0.1	-0.1	0.10	0.3	-0.1	9.5	6.1
855	862	3'	-0.1	0.1	0.02	0.2	NIL	3.2	2.5
862	863'6"	1'5"	0.1	1.1	0.20	1.35	-0.1	25.2	18.2
863'6"	867	3'	0.1	0.2	0.02	0.1	NIL	2.4	3.6
867	870	3'	0.1	0.1	0.07	0.2	NIL	6.2	4.6
870	875'6"	5'6"	0.1	-0.1	0.05	NIL	NIL	6.0	3.1
812'6"	814	1'6"	0.20	0.1	0.07	0.2	NIL	6.0	3.1

DIPS

Footage	Angle
150	7°
30b	20
45ba	17
50ba	17
62ba	10
76b	8
87ba	15
107	25
150Sh	27
163Sh	26
175Sh	20
or ba	
202ba	25
225ba	28
250b	30
or flow bedding	
225ba	28
2 50b	30
or flow bedding	
270ba	27
290flba	30
310b	22
325b	37
341ba	33
349b	45
358ba	33
388Sh	

Form No

Electrolytic Zinc Company of Australasia Limited

Hole No. STP 105

Sheet No. 1

Hole No. STP 105

Sheet No. 2

A 16011

1962 PROGRESS				DESCRIPTION			DIPS		DESCRIPTION			DIPS	
Date	Depth	Advance	Amount of Core	From	To	ROCK TYPE	Footage	Angle	From	To		Footage	Angle
29	446	8	2	53'	63'	White and grey sheared pyroclastics with more carbonate than chlorite in general, and some darker bands, poor core recovery.	?C.	38°	231	256	Very sheared massive pyroclastics comprising interbedded flows and agglomerates at 231' - 232', there is strong banding due to segregation of chlorite and carbonate content parallel to shearing. In general there is a fair amount of replacement by quartz and carbonate. The shearing is less intense between 241' and 256'.		
30	458	12	-				460C	31					
31	478	20	12				469Sh	C36					
Nov.1	498	20	16				493C	58					
6	508	10	7				538C	55					
7	518	10	7	63'	80'	Siliceous light grey to pink massive pyroclastics with quite fine matrix. Contains dark bands up to 1" where matrix replaced by glass (or densely packed shards). In places appears to possess flow banding. Very small fault zones at 66'.	573C	48					
8	528	10	5				610C.	43					
9	538	10	-				640C	61					
12	548	10	20				658C	70					
13	553	5	-				701C	59					
14	568	15	17				745Sh	56					
15	588	20	18										
16	598	10	9						256	266	Green and yellow tuffaceous and agglomeritic massive pyroclastics with high degree of carbonate chlorite replacement. Moderate shearing is somewhat healed.		
19	608	10	2										
20	628	20	16	80'	201'	Massive pyroclastics comprising interbedded flows, agglomerates and crystal vitric tuffs. The tuffs are very dark grey with feldspar phenocrysts, shearing is moderate to strong.							
21	648	20	-										
26	654	6	15										
Dec.7	664	10	4										
11	682	18	13						266	308	Grey, mottled grey and (buff) massive pyroclastics comprising flows, coarse crystal tuffs and some agglomerates. From 290' - 308' there is a considerable quantity of injected quartz carbonate material accompanied by irregular blebs of pyrite and galena and minor sphalerite and chalcopyrite (sulphides -1%).		
12	692	10	9										
13	700	8	5	201	209	Light grey and yellow grey massive pyroclastics, banded in parts. This is a moderately sheared quartz feldspar porphyry volcanic, with blebs and streaks of carbonate (sederite).							
18	710	10	10										
19	722	12	12										
20	7 42	20	20										
21	752	10	-										
24	757	5	5										
26	762	5	5										
27	772	10	10	209	221	Light grey tuffeous massive pyroclastics, parts strongly sheared. Minor streaking of chlorite on shears, together with streaks and small blebs of carbonate. Texture is often quite coarse.							
28	782	10	10										
1963									308	327	Massive pyroclastics comprising green and dark grey feldspathic crystal lithic tuffs. These are predominately coarse grained but there is a finer grained banded phase at 325'. Subordinate agglomerate members occur. There is some		
Jan.3	792	10	10										
9	822	20	10										
10	832	10	10										
11	842	10	10										
14	852	10	10	221	231	Interbedded flow and tuff type. Massive pyroclastics with carbonate occurring as replacement as well as blebs.							
15	862	10	5										
16	872	10	10										
17	882	10	10										

Hole No. STP. 105

Sheet No. 3

A 16011

1963

PROGRESS				DESCRIPTION			DIPS		DESCRIPTION			DIPS	
Date	Depth	Advance	Amount of Core	From	To	ROCK TYPE	Footage	Angle	From	To	ROCK TYPE	Footage	Angle
Jan.													
22	892	10	10			contortion in this section.			388'6"	418	Banded somewhat chloritic medium		
23	902	10	10			This type carries chlorite					grained tuff.		
24	912	10	-			and pyrite, chalcopyrite			418	462	Interbedded and contorted tuffs		
28	-	-	10			smeared on shear surfaces to-					and slates. 428'-441'. Fault		
29	922	10	10			gether with blebs and injec-					Zone poor core recovery and		
30	927	5	-			tions of quartz carbonate and					contorted beds.		
31	-	-	4			chlorite.			462	520	Medium grained tuff with some		
Feb.1	937	10	-	327	358	Fine grained banded quartz					slate interbedded, somewhat		
2	-	-	2			chlorite -(?) sericitic tuff					sheared and occasionally con-		
3	942	5	2'6"			which is strongly sheared.					torted east side up movement.		
5	952	10	10			Grainsize becomes a little			520	605	Black Slate.		
6	962	10	10			finer towards 358'. This			605	676	Coarse tuffs some slate frag-		
7	972	10	10			unit contains throughout					ments, small dragfolds if dips		
8	977	5	-			replacement type sulphide min-					remain steep, west drags now		
11	982	5	10			eralisation of low grade.					indicate west side up but dip		
12	992	10	7			There is an agglomeritic phase					may now be flat east, east side		
13	1002	10	10			at 328'. Bedding and banding					up somewhat remains. This		
14	1012	10	7			show contortions due to small					zone is confused when compared		
15	1022	10	10			scale drag folding.					with drags in black slates		
21	1032	10	10	358	361	Fine grained quartz sericitic					520' - 605'.		
22	1042	10	10			tuff, with minor sulphide			676	690	Coarse tuff siliceous and		
25	1052	10	10			replacement (about 2%).					almost agglomeritic.		
26	1062	10	10	361	366	Medium grained quartz sericitic			690	812	Fine to medium grained tuff,		
						tuff with replacement boundry					grey and uniform but not a		
						of quartz carbonate and some					favourable looking rock type.		
						sulphide (1 - 2%). Banding					Resembles quartz schist from		
						is bedding and is drag folded.					710' - 732'. Becomes more		
				368	371'6"	Interbedded medium grained					siliceous from 790 onwards.		
						tuff and flow rocks with			812	847	Fine to medium grained sandy		
						replacement and injection of					and siliceous tuff.		
						quartz carbonate chlorite and			847	862	Medium grained sandy tuff, dark		
						minor pyrite (-1%).					grey colour.		
				371'6"	372	Sheared coarse grained dacitic			862	872	Fine - medium grained ashy type		
						flow.					tuff, light grey colour.		
				372	388'6"	Grey banded medium grained			872	875	Fine - medium grained siliceous		
						tuffs with a little quartz					tuff.		
						carbonate and minor sulphides.			875	880	Black Slate.		

Form No

Hole No. STP105

Sheet No. 4

A 16011

963

PROGRESS				DESCRIPTION			DIPS		DESCRIPTION			DIPS	
Date	Depth	Advance	Amount of Core	From	To	ROCK TYPE	Footage	Angle	From	To	ROCK TYPE MINERALISATION	Footage	Angle
				880	978	Interbedded black slates and tuffs.			221	231	Minor occurrence of carbonate as blebs and as replacement.		
				893	895	Crumpled zone of tuffy rock with some slate fragments. Rocks gently folded from 912 onwards.			231	256	Replacement by quartz carbonate together with minor chlorite carbonate.		
				978	980	Medium grained sericitic tuff.			256	266	Fair amount of chlorite carbonate replacement.		
				980	985	Interbedded fine grained grey and black slate from 947-978 core in highly sheared and crumpled recovery good.			290	308	Injections of quartz carbonate with minor amount of sulphides Sulphides -1%.		
				985	1040	Medium grained sericitic tuff slightly siliceous black slate band 1024 - 1024'6".			308	327	Chlorite and pyrite chalcopyrite smeared on cleavage and blebs and injections of quartz carbonate and chlorite +1% sulphides.		
				1040	1062	Fine grained banded black slate.			327	358	Replacement by pyrite and chalcopyrite, galena and sphalerite with small 1" richer patches occasionally. Estimated sulphide 5%.		
						MINERALISATION			380	398	Disseminated fine to medium grained pyrite approximately 2% of core.		
				0	26'4"	Streaks of chlorite			466½	467½	Quartz chlorite gangue traces of pyrite fine grained and disseminated.		
				26'4"	33	Some chlorite rich zones.			509	509½	Quartz siderite veins.		
				33	38	Streaks of quartz chlorite smeared on cleavage.			528	598	Specks and blebs of disseminated medium grained pyrite on bedding planes at 529½, 537, 540, 543, 544, 570½'. Fissured zones infilled with quartz chlorite and siderite gangue carrying semi massive fine grained pyrite and pyrrhotite. The whole zone would average 2% sulphide.		
				38	42	Smears of quartz chlorite.							
				42	53	Minor chlorite.							
				53	63	Carbonate and chlorite in minor content.							
				63	66	Chlorite and carbonate in small fault zones.							
				66	127	Thin veins and blebs of galena and sphalerite up to ⅛".							
				127	129	Veins and off-shoots of chalcopyrite and sphalerite up to ½" over 1" zone.							
				201	209	Sporadic blebs and streaks of siderite.							
				209	221	Minor streaks and blebs of carbonate.							

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Hole No. STP 105

Sheet No. 5

ce	Amount of Core	From	To	DESCRIPTION	DIPS Footage	DIPS Angle
		696½	697	½" Quartz veinlets with semi massive fine grained pyrite.		
		697	700	Occasional narrow veinlets of siderite and pyrite at 763½. 6" seam of mixed siderite and pyrite.		
		789	794½	Pyrite becomes now abundant 6% of core.		
		812½	814	Disseminated pyrite and sederite veinlets.		
		814	830	1" seam of carbonate gangue, sphalerite, galena and pyrite.		
		830	841	1" Veinlet of flourite? with quartz carbonate gangue, pyrrhotite, pyrite.		
		855	862	Narrow veinlets of pyrite 5% sulphides.		
		862	863½	Massive - semi massive pyrite and pyrrhotite with some quartz carbonate blebs 70% sulphides.		
		863½	867	Veinlets of pyrite and pyrrhotite 8% sulphides.		
		867	870	Veinlets and splashes of pyrite and pyrrhotite with some siderite veinlets 18% sulphides.		
		870	875	Isolated traces and veinlets of pyrite and pyrrhotite 4% sulphides.		
		875	882	Veinlets of pyrites 8% sulphides.		
		882	886½	Isolated veinlets of pyrite at 884. 6" quartz vein with some sederite.		
		886½	900	Isolated specks of pyrite and some carbonate gangue veinlets.		
		900	903	Narrow veinlets of pyrite and chalcopyrite 15% sulphides.		
		903	913½	Finely disseminated pyrite in blebs and veinlets with some carbonate veinlets from 880 - 906 content of pyrite is greater in the black slate bands.		
		913½	922	Narrow veinlets of pyrite 5% sulphides.		
		922	937½	Quartz veins up to 1' with some veinlets of pyrite.		
		937½	942	Veinlets of pyrite 8% sulphides.		
		947	978	Pyrite and pyrrhotite veinlets 15 - 20% sulphides, numerous carbonate veinlets.		
		978	981	Isolated traces of pyrite.		
		981	985	Veinlets of pyrite 8% sulphides.		
		985	1040	Occasional veinlets of pyrite and odd blebs of fine grained sphalerite.		
		1040	1062	Disseminated veinlets of fine grained pyrite.		

LOGGED BY: PSL, ISG, JGD/RJH

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Geology 1.✓
File 1.

050

ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE RECORD

HOLE No. STP 98
SHEET No. 1.

PROJECT: EL. 4/73. STERLING VALLEY J.V.
LOCALITY:
OBJECTIVE: Relog of old hole.
RESULT:

GRID CO-ORDS:
A.M.G. CO-ORDS: 5,371,930mN 383,730mE
COLLAR R.L.:
COLLAR DIP: -57°
AZIMUTH: 116° Mag.
TOTAL DEPTH: 878 ft

HOLE SIZE:
CASING:
COMMENCED:
COMPLETED:
LOGGED BY: I. MACDONALD

Depth (m)	Azimuth (°a.m.g.)	Dip	Depth (m)	Azimuth (°a.m.g.)	Dip

DEPTH ft From	DEPTH ft To	ROCK DESCRIPTION	MINERALISATION	CORE REC'D Run	CORE REC'D Short
65	200	Broken core. Poor recovery. Dark grey to black mudstone and wacke. Typical fine grained to mixed facies Farrell Slates.			
200	248	Rhyolite lava. Weakly banded; accentuated by shearing which gives a pseudo-clastic texture. Variably sericitised. 235-248. Increasingly strong sericitisation; possibly hydrothermal alteration, or possibly a reaction of glassy lava to regional metamorphism.			
248	572.	Dominantly felsic volcanic wacke with some more tuffaceous bands and some more quartz-rich greywacke bands. Essentially all a coarse-grained facies Farrell Slates unit. Variably sheared but not very strong.			
572.	815	Dominantly black mudstones cleaved. Core very broken recovery very poor. A strongly fractured section of typical fine-grained Farrell Slates.			
815	?841	Felsic volcanic wacke. sheared and broken			
?841	878	Strongly altered sericitised Rhyolitic volcanic difficult to classify as an ashflow or a sheared pseudo-clastic lava because core is very broken and recovery is very poor. Probably a unit of the "Eastern Volcanics"			

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE RECORD

HOLE No. STP 101
SHEET No. 1.

PROJECT: E.L. 4/73 STERLING VALLEY J.V.
LOCALITY:
OBJECTIVE: Relog of old hole
RESULT:

GRID CO-ORDS:
A.M.G. CO-ORDS: 5,372,800 mN 384,355 mE
COLLAR R.L.:
COLLAR DIP: -47°
AZIMUTH: 290° AMG.
TOTAL DEPTH: 870 ft.

HOLE SIZE:
CASING:
COMMENCED:
COMPLETED:
LOGGED BY: I McDONALD.

Depth (m)	Azimuth (°a.m.g.)	Dip	Depth (m)	Azimuth (°a.m.g.)	Dip

DEPTH (Ft) From	DEPTH (Ft) To	ROCK DESCRIPTION	MINERALISATION	CORE REC'D Run	CORE REC'D Short
0	153	Porphyritic sheared rhyolite lavas. Possibly some ignimbrite ash-flow units but most of apparent clastic texture is due to shearing effects on the lavas - Typical Eastern Volcanics.			
153	163	Weakly banded rhyolite lava. More sheared and sericitic than above.			
163	170.	Quartz rich rhyolitic lava, strongly sericitised and sheared. Shearing induces a banding and a pseudo clastic texture.			
170	186.	Sheared sericitic rhyolite lava.			
186	301.	Rhyolitic ashflow pyroclastic chloritic patches are crushed pumice fragments. Variably sheared.			
301	321	Weakly banded quartz-phyric sheared sericitic rhyolite lava.			
321	386	Interbedded volcaniclastics sediments and dark grey Mudstones. Typical of the mixed facies Farrell slates but the coarser units have more primary tuffaceous texture than the typical Farrell volcanic wackes.			
386	421	Sheared Felsic Tuffs with thin mudstone partings. Possibly waterlain ashflows.			
421.	478	Felsic Tuffs. Poor core recovery of broken sheared and mineralised			

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE RECORD

HOLE No. STP101
SHEET No. 2

DEPTH From	DEPTH To	ROCK DESCRIPTION	MINERALISATION	CORE REC'D Run	CORE REC'D Short
		rocks Primary textures are obscure but look like primary tuffs rather than epiclastics or typical Forrell lithic wackes.			
478	560	Volcaniclastic Sediment to c.g. Volcanic lithic Wacke More typical of c.g. facies of Forrell Slates			
560	580	Mudstones with wispy interbedded Volcanic Wacke. Typical fine grained facies of Forrell Slates.			
580	638	Laminated crystal rich felsic Tuff. Sheared. Probably waterlain			
638	705	Black Mudstones. Broken and contorted. Very poor core recovery. Typical fine grained facies Forrell Slates.			
705	870	Volcaniclastic Sediments. Coarse grained facies Forrell Slates slightly more tuffaceous than typical			

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE RECORD

HOLE No. STP 105
SHEET No. 1.

PROJECT: E.L. 4/73 STERLING VALLEY J.V.
LOCALITY:
OBJECTIVE: Relog of old hole.
RESULT:

GRID CO-ORDS:
A.M.G. CO-ORDS: S, 373,210 mN 384,515 mE
COLLAR R.L.:
COLLAR DIP: −53°
AZIMUTH: 290° AMG.
TOTAL DEPTH: 1062 Ft.

HOLE SIZE:
CASING:
COMMENCED:
COMPLETED:
LOGGED BY: I McDONALD

Depth (m)	Azimuth (°a.m.g.)	Dip	Depth (m)	Azimuth (°a.m.g.)	Dip

DEPTH ft. From	DEPTH ft. To	ROCK DESCRIPTION	MINERALISATION	CORE REC'D Run	CORE REC'D Short
0	327	Rhyolite lava. generally well sheared which imparts a pseudo-clastic texture. Typical "Eastern Volcanics"			
327	418	Finer grained less "clastic looking" Rhyolite lava Strongly sheared and variably silicified and sericitised - Possibly only regional metamorphism but the unit has the impression of a weakly developed hydrothermal alteration system.	Moderate disseminated pyrite and traces of chalcopyrite, galena and sphalerite.		
418	462	Laminated. Volcaniclastic sediments and Mudstones. Cleavage parallel to ? bedding.			
		428 - 441 Possible Fault. The only evidence is poor core recovery and contortion of bedding	Minor blebby carbonate - pyrite.		
462	520	Weakly bedded Volcaniclastic Sediment. Texture is primary tuffaceous rather than volcanic lithic wacke			
520	607	Black to dark grey cleaved weakly pyritic Mudstones. 595-607. Interbeds of tuffaceous textured Volcaniclastic Sediment. Typical fine-grained facies Farrell Slate" unit.			
607	690	Felsic ashflow pyroclastics containing accidental mudstone clasts. Some banding is probably primary flow-banding but much of the foliation is due to shearing and cleavage development.			
690	790	Felsic ashflow pyroclastics with mixed lithic clasts			

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE RECORD

HOLE No. STP 105
SHEET No. 2

DEPTH (FT) From	DEPTH (FT) To	ROCK DESCRIPTION	MINERALISATION	CORE REC'D Run	CORE REC'D Short
790	812	Fine grained lithic tuffs with accidental mudstone clasts. Possibly waterlain ashflows			
812	862.	Volcaniclastic Sediments. Primary tuffaceous to volcanic lithic wacke texture. Contains minor ashflow bands.			
862	875	Felsic bedded tuff to to felsic volcanic lithic wacke. Typical coarse-grained facies of the "Farrell Slates."			
875	880	Cleaved dark-grey Mudstone - Typical fine-grained facies "Farrell Slates"			
880	985	Volcaniclastic turbidite sequence. Wacke to Mudstone. Typical Farrell Slates.			
985	1040	Tuffaceous Wackes - typicall cg. facies Farrell Slates			
1040	1062	Interbedded black and greg Mudstone and mg. Volcanic lithic Wacke. Typical mixed facies Farrell Slates.			

A 18621

APPENDIX 5.

Drill Core Geochemical Data Sheets.

DDH STP 217
STP 221
STP 231
STP 234
STP 231A
STP 232A1
STP 218
STP 220

ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. STP VARIOUS
SHEET No. 1

GRID CO-ORDS: Sterling Valley
A.M.G. CO-ORDS: Drill holes
COLLAR R.L.: Gold Assays
COLLAR DIP: greater than 0·1g/t Au
AZIMUTH:
TOTAL DEPTH:

LABORATORY						ANALABS														
ANALYTICAL TECHNIQUE						309	APDW REGIA	108	114	118	402	108	108	108	108	108	402	108	402	
						FIRE ASSAY	AAS	AAS	AAS	AAS	XRF	AAS	AAS	AAS	AAS	AAS	XRF	AAS	XRF	
DETECTION LIMIT						0·008	0·008	0·5	1	1	3	5	5	5	50	5	3	10	3	
Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au	Au	Ag	As	Sn	Sn	Cu	Pb	Zn	Fe%	Mn	Sb	Bi	Te	COMMENTS
48427	SPLIT	313·25	313·85	100%	0·60	5·00		29·5	9·00%		x	1·15%	505	50	3·15	2900				STP234. Arsenic Zone 'D'
33074	"	99·65	100·70	"	1·05	4·01		4·5	15·00%	2000	1550	2250	140	125						STP217 Arsenic Zone 'A'
48487	"	185·2	185·5	"	0·30	3·25		11·0	4		150	550	203	32·33%	1150	1550	33	x	14	STP234 Possibly related to Zone 'A'
48417	"	190·0	191·0	"	1·00	2·92		29·5	4·00%		x	4850	310	415	9·10	11·50				STP236 Arsenic Zone 'A'
33079	"	104·55	105·45	"	0·90	2·26		1·0	9400	1120	1500	515	45	130						STP217 Related to Arsenic Zone 'A'
40540	"	51·8	52·8	"	0·10		2·08	x	70		15	250	40	140	6·70	1850				STP221 Related to Arsenic Zone 'A'
48418	"	191·0	191·5	"	0·50	1·96		20·5	5·30%		x	4300	185	285	1·95	625				STP234 Arsenic Zone 'A'
48416	"	189·0	190·0	"	1·00	1·95		16·5	3·20%		x	7100	130	1050	13·50	1900				STP234 Arsenic Zone 'A'
39270	"	56·2	56·7	"	0·50	1·29	0·276	20·0	15·00%	160		2250	200	2000	9·10	780				STP231 Arsenic Zone 'C'
39276	"	78·3	78·8	"	0·50	1·22	0·744	3·0	9·80%	5600		1150	60	70	5·80	660				STP231 Arsenic Zone 'A'
37952	"	232·1	233·1	"	1·00	1·04		2·0	36	4		135	25	140						STP217 Possibly related to Zone 'D'
39273	"	57·7	58·3	"	0·60	1·01	0·436	10·0	8·10%	600		6750	75	190	7·20	465				STP231 Arsenic Zone 'C'
37932	"	159·6	160·6	"	1·00	0·95		2·0	4400	14	20	700	10	95						STP217 Related to Arsenic Zone 'B'
48420	"	261·0	262·0	"	1·00	0·92		2·0	5·20%		1150	2150	30	125	6·00	1350				STP234 Arsenic Zone B
33073	"	98·65	99·65	"	1·00	0·91		4·5	6·00%	3300	3700	2360	40	115						STP217 Arsenic Zone A
39266	"	53·8	53·9	90%	0·10	0·90	0·008	9·5	17·00%	17		850	235	80	11·00	550				STP231 Arsenic Zone C
48421	"	262·0	263·0	100%	1·00	0·85		2·0	3·50%		x	2350	50	200	6·55	1400				STP234 Arsenic Zone B
48406	"	67·7	68·1	"	0·40	0·76		1·5	21	3		490	10	55	5·75	1050				STP232A Unrelated Zone Top of Farrell Slate
40547	"	58·1	59·1	"	1·00		0·704	0·5	47		40	220	25	325	8·25	2350				STP221 Unrelated between Zones A and B
48401	"	66·7	67·7	"	1·00	0·63		0·5	28		3600	355	95	375	3·75	1508				STP232A Unrelated Zone Top of Farrell Slates
39272	"	57·2	57·7	"	0·50	0·62	x	4·0	3·80%	10		2200	60	40	11·00	275				STP231 Arsenic Zone 'C'
39285	"	85·2	85·5	"	0·30	0·58	0·376	5·5	9·00%	6		170	140	155	7·80	520				STP231 Arsenic Zone A
48424	"	253·0	254·1	"	1·10	0·56		0·5	60		4200	250	10	125	13·50	3450				STP234 Possibly related to Zone 'B'
33078	"	103·55	104·55	"	1·00	0·52		x	6000	920	1050	230	15	130						STP217 Related to Zone A
40546	CHIP	57·1	58·1	"	1·00		0·512	0·5	47		110	290	25	140	8·00	2000				STP221 Unrelated between Zones A and B
39271	SPLIT	56·7	57·2	"	0·50	0·47	x	6·0	8·50%	22		2500	65	70	12·50	595				STP231 Arsenic Zone 'C'
40539	"	50·8	51·8	"	1·00		0·456	x	200		25	230	30	175	7·50	2200				STP221 Possibly related to Zone A

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. STP VARIOUS
SHEET No. 2

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified)														COMMENTS
						Au	Au	Ag	As	Sn	Sn	Cu	Pb	Zn	Fe %	Mn	Sb	Bi	Te	
40516	SPLIT	32·6	33·15	100%	0·55		0·412	28·5	3·20%		25	1350	135	165	24·50	1450				STP 221 Arsenic Zone A
48423	"	252·0	253·0	"	1·00	0·40		0·5	3800		3000	865	25	95	7·00	1450				STP 234 Possibly related to Zone B
37951	"	233·1	234·1	"	1·00	0·36		0·3	18	x		80	20	95						STP 217 Possibly related to Zone D
40538	"	49·8	50·8	"	1·00		0·352	1·0	1600		35	155	185	410	17·50	2650				STP 221 Related to Arsenic Zone A
39269	"	55·6	56·2	"	0·60	0·39	0·042	11·0	3·50%	x		1750	95	285	9·70	1200				STP 231 Arsenic Zone C
61245	"	209·0	209·0	"	1·00	0·30		0·5	66		78	185	75	1550	9·85	2200	3	x	x	STP 234 Unrelated between Zones A and B
40537	"	48·8	49·8	"	1·00		0·296	x	1300		10	205	30	245	11·00	2250				STP 221 Related to Zone A
37949	"	225·1	226·1	"	1·00	0·29		2·0	1160	x		280	15	90						STP 217 Possibly related to Zone D
40526	"	38·8	39·5	"	0·70		0·288	4·5	4·30%		3950	1300	15	85	14·00	900				STP 221 Arsenic Zone A
39267	"	53·9	54·9	"	1·00	0·27	0·216	8·5	4·00%	x		735	90	110	6·60	1150				STP 231 Arsenic Zone C
61242	"	205·0	206·0	"	1·00	0·26		x	68		52	195	45	190	7·65	1400	7	x	x	STP 234 Unrelated between Zones A & B
48494	"	194·0	195·0	"	1·00	0·25		x	75		52	150	175	630	5·85	1750	x	x	x	STP 234 Related to Arsenic Zone A
33076	"	101·65	102·55	"	0·90	0·25		3·0	3·50%	1700	1900	1200	20	1050						STP 217 Arsenic Zone A.
40524	"	36·8	37·8	"	1·00		0·240	40·0	3·50%		30	3600	335	235	35·00	465				STP 221 Arsenic Zone A
48497	"	197·0	198·0	"	1·00	0·23		x	25		21	155	25	430	7·35	2150	x	x	x	STP 234 Possibly related to Zone A
61244	"	207·0	208·0	"	1·00	0·23		0·5	100		105	195	80	220	7·65	1750	x	x	x	STP 234 Unrelated between Zones A & B
33067	"	92·35	93·35	80%	1·00	0·21		3·5	1·80%	6400	4950	1550	535	440						STP 217 Arsenic Zone A
48404	"	67·0	68·0	100%	1·00	0·21		1·0	31	3100		325	15	80	5·05	875				STP 232A-1 Unrelated. Top of Farrell Slates
33066	"	90·65	92·35	76%	1·70	0·18		0·5	136	9	x	470	5	130						STP 217 Arsenic Zone A
37933	"	160·6	162·1	100%	1·50	0·18		2·0	7800	27	150	1300	10	110						STP 217 Related to Zone B.
40551	"	65·6	66·6	100%	1·00		0·176	1·5	8000		660	355	40	140	7·55	2100				STP 221 Unrelated between Zones A and B
37950	"	234·1	235·1	"	1·00	0·17		0·5	40	x		90	25	135						STP 217 Possibly related to Zone D.
61263	"	267·0	268·0	"	1·00	0·17		24·0	1·05%		1740	6000	950	385	4·30	1050	20	970	x	STP 234 Possibly related to Zone B
61262	"	266·0	267·0	"	1·00	0·16		7·5	9600		1100	4250	125	190	4·00	885	10	250	x	" " " " " " "
37929	"	156·6	157·6	"	1·00	0·16		1·5	1·40%	x		525	20	145						STP 217 Arsenic Zone B.
40525	"	37·8	38·8	"	1·00		0·152	10·0	8·80%		2650	3050	35	70	15·00	450				STP 221 Arsenic Zone A
39390	Grnd	60·0	65·0	"	5·00	0·15		x	500	20		140	170	835	6·30	1550				STP 231 Related to Zone C.
39277	SPLIT	78·8	79·5	"	0·70	0·14	0·104	4·5	2800	3000		2150	50	100	7·30	1400				STP 231 Related to Zone A
48415	"	188·3	189·0	"	0·70	0·14		1·0	430		52	610	20	360	13·50	2750				STP 234 Arsenic Zone A
37948	"	224·1	225·1	"	1·00	0·14		x	114	x		225	10	85						STP 217 Possibly related to Zone D
40532	"	43·8	44·8	"	1·00		0·120	1·5	55		40	215	80	410	13·50	2850				STP 221 Related to Zone A.
48493	"	193·0	194·0	"	1·00	0·12		x	64		71	70	160	965	5·05	1650	x	x	4	STP 234 Related to Zone A
48458	"	156·0	157·0	"	1·00	0·12		14·0	1300		127	450	635	1·01%	6·85	2750	x	40	x	STP 234 Possibly Arsenic Zone C

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. STP VARIOUS
SHEET No. 3

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified)													COMMENTS
						Au	Au	Ag	As	Sn	Sn	Cu	Pb	Zn	Fe %	Mn	Sb	Bi	Te	
61276	SPLIT	312·3	313·25	100%	0·95	0·12		2·0	1300		59	650	10	250	8·05	1150	x	50	x	STP 234 Arsenic Zone D.
48413	"	112·0	113·2	"	1·20	0·12		1·5	520	1		160	100	190	4·90	1550				STP 232 A1 Unrelated Top of Ferrell Sts
40522	"	35·5	35·8	"	0·30		0·112	14·0	7000		100	1800	60	135	19·50	530				STP 221 Arsenic Zone A
37938	"	155·6	156·6	"	1·00	0·11		2·5	5·60%	x		900	20	120						STP 217 Arsenic Zone B
39393	Grind	75·0	80·0	"	5·00	0·11		x	3800	1100		515	55	240	6·60	1650				STP 231 Possibly related to Zone A.
40548	SPLIT	59·1	59·6	"	0·50		0·104	x	47		60	230	20	150	9·00	1850				STP 221 Unrelated between Zones A & B

ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 217
SHEET No. 1

GRID CO-ORDS: 4280 N ; 4565 E Stg V.
A.M.G. CO-ORDS: 5, 374 393 m N, 384 190 m E
COLLAR R.L.: 176 m
COLLAR DIP: −60°
AZIMUTH: 108°
TOTAL DEPTH: 249.1 m

LABORATORY						ANALABS											
ANALYTICAL TECHNIQUE						309		108	114	118	FUSION	108	108	108			
						FIRE ASSAY		AAS	AAS	AAS	XRF	AAS	AAS	AAS			
DETECTION LIMIT						0.008		0.5	1	1	10	5	5	5			
Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au		Ag	As	Sn	Sn	Cu	Pb	Zn			COMMENTS
33089	CHIP	0.	5.0	2.7	5.0			x	8	9		25	10	30			
82	"	5.0	10.0	0.4	5.0			x	10	1		5	15	30			
83	"	10.0	15.0	0.1	5.0			1.0	16	9		5	10	25			
84	"	15.0	20.0	0.1	5.0			x	46	1		30	50	165			
85	"	20.0	25.0	0.2	5.0			x	10	9		25	15	135			
86	"	25.0	30.0	3.3	5.0			x	7	2		115	10	140			
87	"	30.0	35.0	3.1	5.0			x	10	1		80	10	105			
88	"	35.0	40.0	3.0	5.0			0.5	4	1		80	5	140			
89	"	40.0	45.0	4.5	5.0			1.0	15	2		50	15	125			
33090	"	45.0	50.0	5.0	5.0			x	7	4		200	10	135			
91	"	50.0	55.0	5.0	5.0			x	6	2		145	15	115			
33092	"	55.0	60.0	4.8	5.0			x	6	5		305	10	180			
37954	SPLIT	60.0	61.0	0.8	1.0	0.02		x	10	2		140	5	145			
55	"	61.0	62.0	0.9	1.0	0.01		1.0	10	2		60	x	150			
56	"	62.0	63.0	0.9	1.0	x		1.0	13	210	490	35	x	150			
57	"	63.0	64.0	1.0	1.0	x		x	3	3		55	x	140			
58	"	64.0	65.0	1.0	1.0	x		0.5	4	4		105	20	130			
59	"	65.0	66.0	1.0	1.0	x		x	4	6		220	10	105			
37960	"	66.0	67.0	1.0	1.0	x		x	2	5		110	10	160			
61	"	67.0	68.0	1.0	1.0	x		x	4	8		65	15	215			
62	"	68.0	69.0	1.0	1.0	x		x	8	4		160	20	140			
63	"	69.0	70.0	1.0	1.0	x		x	8	4		10	20	105			
64	"	70.0	71.0	0.7	1.0	0.02		2.5	620	1500	2000	1600	40	175			
65	"	71.0	72.0	0.4	1.0	0.02		x	20	10		30	15	160			
66	"	72.0	73.0	0.5	1.0	x		x	21	11		70	15	165			
67	"	73.0	74.0	1.0	1.0	0.02		0.5	44	4		40	10	215			
68	"	74.0	75.0	1.0	1.0	0.02		1.0	4	6		80	25	170			
37969	"	75.0	76.0	0.9	1.0	x		1.0	4000	520	860	150	50	180			

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 217
SHEET No. 2

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au		Ag	As	Sn	Sn	Cu	Pb	Zn						COMMENTS
37970	SPLIT	76·0	77·0	0·9	1·0	X		X	24	18	45	50	10	155						? Arsenic Zone 'C'
71	"	77·0	78·0	0·9	1·0	X		0·5	24	112	270	90	30	180						
72	"	78·0	79·0	0·9	1·0	X		1·5	4000	760	1000	155	20	100						
73	"	79·0	80·0	0·9	1·0	0·03		2·5	70	40	X	310	30	600						
74	"	80·0	81·0	1·0	1·0	0·01		1·3	12	18	25	100	35	185						
75	"	81·0	82·0	1·0	1·0	X		1·5	14	76	270	40	30	200						
76	"	82·0	83·0	1·0	1·0	X		1·0	4	2	X	15	10	110						
77	"	83·0	86·0	1·0	1·0	X		1·5	10	34	80	25	20	135						
78	"	84·0	85·0	1·0	1·0	X		1·0	8	88	310	45	50	230						
79	"	85·0	86·0	1·0	1·0	X		X	22	200	630	180	40	300						
37980	"	86·0	87·0	1·0	1·0	X		0·5	12	68	330	100	20	180						
81	"	87·0	88·0	1·0	1·0	X		X	5	X		15	20	135						
37982	"	88·0	89·0	1·0	1·0	0·01		X	4	X		20	15	110						
33099	"	89·0	90·0	1·0	1·0	0·02		X	580	5		420	25	185						
33100	"	90·0	90·65	0·65	0·65	X		1·0	8	1		70	20	230						
33066	"	90·65	92·35	1·3	1·70	0·18		0·5	136	9	X	470	5	130						
67	"	92·35	93·35	0·8	1·0	0·21		3·5	1·80%	6400	4950	1550	535	440						Arsenic Zone 'A'
68	"	93·35	94·35	1·0	1·0	X		1·0	305	42	60	900	35	165						
69	"	94·35	95·35	1·0	1·0	X		0·5	160	58	170	325	5	200						
33070	"	95·35	96·35	1·0	1·0	X		X	112	32	60	400	10	210						
71	"	96·35	97·35	1·0	1·0	X		X	76	10	30	350	15	155						
72	"	97·35	98·65	1·3	1·30	0·01		0·5	920	450	360	380	5	155						
73	"	98·65	99·65	1·0	1·0	0·91		4·5	6·00%	3300	3700	2360	40	115						
74	"	99·65	100·70	1·05	1·05	4·01		4·5	15·00%	2000	1550	2250	140	125					→	Gold in Polished Section —
75	"	100·70	101·65	0·95	0·95	0·09		1·5	2700	270	360	465	25	170						Arsenic Zone 'A'
76	"	101·65	102·55	0·90	0·90	0·25		3·0	3·50%	1700	1500	1200	20	1050						
77	"	102·55	103·55	1·0	1·0	0·03		X	560	108	170	285	5	170						
78	"	103·55	104·55	1·0	1·0	0·52		X	6000	920	1050	230	15	130						
79	"	104·55	105·45	0·9	0·9	2·26		1·0	9400	1120	1500	515	45	130						Related to Arsenic Zone 'A'
33080	SPLIT	105·45	106·45	1·0	1·0	0·09		X	250	12	45	40	10	190						
37983	CHIP	106·45	110·0	3·55	3·55			X	6	X		10	20	185						
84	"	110·0	115·0	5·0	5·0			1·0	4	X		100	10	150						
37985	CHIP	115·0	121·0	6·0	6·0			X	14	108	390	60	20	200						

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 217
SHEET No. 3

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au		Ag	As	Sn	Sn	Cu	Pb	Zn						COMMENTS
37910	SPLIT	121·0	122·0	1·0	1·0			x	27	3	x	145	15	175						
911	"	122·0	123·6	1·6	1·6			x	8000	610	1200	900	20	160						
37912	SPLIT	123·6	124·6	1·0	1·0			x	23	6	6	5	25	130						
37986	CHIP	124·6	130·0	5·3	5·4			1·5	8	36	140	95	10	155						
987	"	130·0	135·0	4·9	5·0			2·0	8	100	340	75	220	710						
37988	CHIP	135·0	140·1	5·1	5·1			0·5	4	92		145	70	190						
37913	SPLIT	140·1	141·1	1·0	1·0			x	10	1		30	10	125						
14	"	141·1	142·1	1·0	1·0			x	230	31	200	355	15	105						
15	"	142·1	143·1	1·0	1·0			x	680	21	70	340	5	85						
16	"	143·1	144·1	1·0	1·0			x	14	3	15	45	10	95						
17	"	144·1	145·1	1·0	1·0			1·5	46	26	130	180	10	170						
18	"	145·1	146·6	1·6	1·6			2·5	6	220	440	140	160	370						
19	"	146·6	147·6	1·0	1·0	0·05		2·5	960	330	920	670	80	275						
37920	"	147·6	148·6	1·0	1·0	x		1·5	9	400	740	100	150	400						
21	"	148·6	149·6	1·0	1·0	x		1·0	4	150	470	230	5	170						
22	"	149·6	150·6	1·0	1·0	0·05		1·0	7400	2800	3950	410	10	70						
23	"	150·6	151·6	1·0	1·0	x		1·0	500	920	2250	360	20	180						
24	"	151·6	152·6	1·0	1·0	x		0·5	20	2		5	5	120						
25	"	152·6	153·6	1·0	1·0	x		1·0	18	6		275	15	165						
26	"	153·6	154·6	1·0	1·0	0·01		3·0	840	4		1000	10	360						
27	"	154·6	155·6	1·0	1·0	0·02		3·0	1·40%	x		900	15	150						}
28	"	155·6	156·6	1·0	1·0	0·11		2·5	5·60%	x		900	20	120						} Arsenic Zone 'B'
29	"	156·6	157·6	1·0	1·0	0·16		1·5	1·40%	x		525	20	145						}
37930	"	157·6	158·6	1·0	1·0	0·02		0·5	7400	56	490	210	5	130						
31	"	158·6	159·6	1·0	1·0	0·02		0·5	5200	36	75	320	5	180						
32	"	159·6	160·6	1·0	1·0	0·95		2·0	4400	14	20	700	10	95						} Related To Arsenic Zone 'B'
33	"	160·6	162·1	1·5	1·5	0·18		2·0	7800	27	150	1300	10	110						}
379 34	SPLIT	162·1	163·1	1·0	1·0	x		0·5	180	8	x	50	10	220						
37989	CHIP	163·1	165·0	1·9	1·9			0·5	4	x		15	25	150						
37990	"	165·0	170·0	5·0	5·0			x	2	x		5	10	150						
91	"	170·0	175·0	5·0	5·0			0·5	16	x		10	20	180						
37992	CHIP	175·0	179·3	4·2	4·3			2·5	10	22		15	330	190						
37935	SPLIT	179·3	180·3	0·8	1·0	X		1·0	90	180	650	205	260	560						

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 217
SHEET No. 4.

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified)													COMMENTS
						Au		Ag	As	Sn	Sn	Cu	Pb	Zn					
37936	SPLIT	180·3	181·3	0·9	1·0	x		1·0	760	72	450	420	30	150					
37	"	181·3	182·3	0·9	1·0	x		1·0	700	76	500	800	30	115					
38	"	182·3	183·3	0·9	1·0	0·08		1·0	1120	460	920	230	35	135					
39	"	183·3	184·3	1·0	1·0	x		1·0	26	128	550	385	30	120					
37940	"	184·3	185·3	1·0	1·0	0·02		1·0	350	6		405	35	110					
41	"	185·3	186·3	0·9	1·0	0·01		2·0	38	x		470	25	100					
37942	SPLIT	186·3	187·3	1·0	1·0	x		3·5	10	x		280	35	85					
37993	CHIP	187·3	190·0	2·5	2·7			1·5	32	x		80	15	100					
94	"	190·0	195·0	5·0	5·0			1·0	40	x		130	20	75					
95	"	195·0	200·0	4·8	5·0			1·0	22	2		140	15	55					
96	"	200·0	205·0	5·0	5·0			x	14	x		30	5	50					
97	"	205·0	210·0	5·0	5·0			1·0	920	x		70	50	200					
98	"	210·0	215·0	5·0	5·0			1·5	20	x		25	10	45					
37999	CHIP	215·0	219·1	5·0	4·1			1·0	30	6		45	10	55					
37943	SPLIT	219·1	220·1	1·0	1·0	x		2·0	3400	8		100	15	75					
44	"	220·1	221·1	1·0	1·0	x		1·0	38	16		10	15	70					
45	"	221·1	222·1	1·0	1·0	x		2·5	24	x		205	25	80					
46	"	222·1	223·1	1·0	1·0	x		2·0	25	x		10	10	60					
47	"	223·1	224·1	1·0	1·0	x		1·0	60	x		15	10	55					
48	"	224·1	225·1	1·0	1·0	0·14		x	114	x		225	10	85					
37949	SPLIT	225·1	226·1	1·0	1·0	0·29		2·0	1160	x		280	15	90					
36701	CHIP	226·1	231·1	5·0	5·0			x	46	3		40	20	85					
37953	SPLIT	231·1	232·1	1·0	1·0	0·03		1·0	24	4		10	20	90					
952	"	232·1	233·1	1·0	1·0	1·04		2·0	36	4		135	25	140					—
951	"	233·1	234·1	1·0	1·0	0·36		0·5	18	x		80	20	95					Strong qtz-carb-~~pot~~-py veining
37950	SPLIT	234·1	235·1	1·0	1·0	0·17		0·5	40	x		90	25	135					Possibly correlate of Arsenic Zone 'D'
36702	CHIP	235·1	240·0	4·9	4·9			x	128	3		50	60	310					
03	"	240·0	245·0	5·0	5·0			x	28	2		25	45	100					
36704	CHIP	245·0	249·1	4·1	4·1			x	34	3		70	70	85					

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 221
SHEET No. 1

GRID CO-ORDS: 4320 N; 4635 E
A.M.G. CO-ORDS: 5,374, 399 m N, 384, 271 m E
COLLAR R.L.: 174m
COLLAR DIP: −60°
AZIMUTH: 76°AMG
TOTAL DEPTH: 203.3

LABORATORY						ANALABS												
ANALYTICAL TECHNIQUE						AQUA REGIA	108	114	118	402	108	108	108	108	108			
						AAS	AAS	AAS	AAS	XRF	AAS	AAS	AAS	AAS	AAS			
DETECTION LIMIT						0.008	0.5	1	1	3	5	5	5	50	5			
Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au	Ag	As	Sn	Sn	Cu	Pb	Zn	Fe%	Mn			COMMENTS
40501	SPLIT	10.5	11.5	0.8	1.0	0.284	1.0	1.00%		310	480	70	295	20.00	3150			
02	"	11.5	12.5	1.0	1.0	x	x	1600		410	100	x	175	19.00	2850			
03	"	12.5	13.5	0.9	1.0	x	x	2100		320	350	10	345	15.00	3100			
04	"	13.5	14.5	1.0	1.0	0.040	x	65		30	130	x	290	7.25	2050			
05	"	14.5	15.0	0.5	0.5	x	1.5	1700		520	205	470	1150	17.20	4150			
06	SPLIT	15.0	16.0	0.9	1.0	x	0.5	40		40	60	75	405	11.00	2450			
07	CHIP	16.0	18.0	1.9	2.0	x	x	60		40	35	x	250	11.00	2400			
08	"	18.0	22.0	3.6	4.0	x	0.5	75		30	25	x	240	7.40	1650			
09	CHIP	22.0	26.0	3.9	4.0	x	0.5	50		3	40	x	180	6.50	1950			
40510	SPLIT	26.0	27.0	0.9	1.0	x	x	14		4	15	x	160	4.60	1100			
11	"	27.0	28.0	1.0	1.0	x	x	20		4	20	x	190	4.50	1500			
12	"	28.0	28.8	0.8	0.8	x	1.5	28		20	205	5	260	12.50	1850			
13	"	28.8	30.8	0.5	2.0	x	8.5	9000		15	1500	110	150	18.00	320			
14	"	30.8	31.6	0.2	0.8	x	43.0	3700		85	1500	1400	240	27.00	645			
15	"	31.6	32.6	0.8	1.0	0.032	4.5	3000		35	730	5	205	20.00	1900			
16	"	32.6	33.15	0.55	0.55	0.412	28.5	3.20%		25	1350	135	165	24.50	1450			
17	"	33.15	33.7	0.55	0.55	0.032	6.5	8000		60	650	5	115	15.00	1300			
18	"	33.7	34.7	1.0	1.0	0.048	42.0	10.00%		40	2250	365	55	35.00	285			
40520	"	34.7	35.2	0.5	0.5	0.072	10.0	4.5%		120	1300	90	180	26.50	1050			
21	"	35.2	35.5	0.3	0.3	0.024	11.0	1.5%		70	1550	75	3350	37.00	675			Arsenic Zone 'A'
22	"	35.5	35.8	0.3	0.3	0.112	14.0	7000		100	1800	60	135	19.50	530			
23	"	35.8	36.8	1.0	1.0	0.072	19.0	13.00%		460	2250	155	2500	36.50	545			
24	"	36.8	37.8	1.0	1.0	0.240	40.0	3.50%		30	3400	335	235	35.00	465			
25	"	37.8	38.8	1.0	1.0	0.152	10.0	8.80%		2650	3050	35	70	15.00	450			
26	"	38.8	39.5	0.7	0.7	0.288	4.5	4.30%		3950	1300	15	85	14.00	900			
27	"	39.5	40.0	0.5	0.5	0.024	0.5	2500		45	90	440	1300	16.50	2000			
28	"	40.0	40.8	0.8	0.8	0.016	2.0	6000		50	205	635	1550	14.00	3050			
40529	"	40.8	41.8	1.0	1.0	x	1.5	45		40	115	300	740	9.75	1950			

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 221
SHEET No. 2

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified)												COMMENTS
							Au	Ag	As	Sn	Sn	Cu	Pb	Zn	Fe %	Mn		
40530	SPLIT	41·8	42·8	1·0	1·0		X	3·0	40		70	300	760	3650	10·50	3800		
31	"	42·8	43·8	1·0	1·0		N.A.	1·0	55	20		100	140	510	12·00	3500		
32	"	43·8	44·8	1·0	1·0		0·130	1·5	55		40	215	80	410	13·50	2850		
33	"	44·8	45·8	1·0	1·0		0·080	1·5	230		50	220	80	455	12·00	2350		
34	"	45·8	46·8	1·0	1·0		0·083	2·0	4500		50	295	145	365	12·50	2700		
35	"	46·8	47·8	1·0	1·0		0·080	2·5	2000		70	250	235	1250	11·50	2600		
36	"	47·8	48·8	1·0	1·0		X	0·5	45		50	140	240	1265	9·60	3100		
37	"	48·8	49·8	1·0	1·0		0·296	X	1300		10	205	30	245	11·00	2250		
38	"	49·8	50·8	1·0	1·0		0·352	1·0	1600		35	155	185	410	12·50	2650		
39	"	50·8	51·8	1·0	1·0		0·496	X	200		25	230	30	175	7·50	2200		
40540	"	51·8	52·8	1·0	1·0		2·08	X	70		15	250	40	140	6·70	1850		Minor sulphide veins ?related to Zone 'A'
41	"	52·8	53·8	1·0	1·0		0·016	X	20		20	80	3	150	6·25	2550		
42	"	53·8	54·4	0·6	0·6		NA	0·5	20	36		65	75	185	5·90	1800		
43	"	54·4	54·8	0·4	0·4		X	0·5	22		X	35	20	150	4·75	1500		
44	SPLIT	54·8	55·8	1·0	1·0		X	X	25		4	70	5	145	3·90	1300		
45	CHIP	55·8	57·1	1·3	1·3		N.A.	X	20	30		145	10	185	6·00	2000		
46	CHIP	57·1	58·1	1·0	1·0		0·512	0·5	47		110	290	25	140	8·00	3000		minor veins Po + Py
47	SPLIT	58·1	59·1	1·0	1·0		0·704	0·5	47		40	220	25	325	8·25	2350		
48	SPLIT	59·1	59·6	0·5	0·5		0·106	X	47		60	230	20	150	9·00	1850		
49	CHIP	59·6	64·6	5·0	5·0		N.A.	1·0	550	34		70	20	175	8·10	1950		
40550	SPLIT	64·6	65·6	1·0	1·0		X	X	17		10	225	10	105	5·70	1450		
51	"	65·6	66·6	1·0	1·0		0·176	1·5	8000		660	355	40	140	7·55	2100		
52	SPLIT	66·6	67·6	1·0	1·0		X	0·5	200		80	65	175	340	6·05	3600		
53	CHIP	67·6	72·6	5·0	5·0			1·5	350	88		215	70	460	8·00	2350		
54	"	72·6	77·0	4·4	4·4			0·5	220	110		125	35	170	9·00	2600		
55	"	77·0	82·0	5·0	5·0			X	10	33		55	10	130	5·35	1700		
56	"	82·0	87·0	5·0	5·0			X	10	72		30	10	120	5·15	1400		
57	"	87·0	92·0	5·0	5·0			X	8	60		45	15	105	4·55	1850		
58	"	92·0	97·0	5·0	5·0			X	12	66		125	80	200	7·70	3250		
59	"	97·0	102·0	5·0	5·0			X	50	62		20	45	155	6·50	1700		
40560	CHIP	102·0	107·0	4·9	5·0			X	3300	200		20	40	125	5·50	1850		
61	SPLIT	107·0	108·0	0·9	1·0		X	X	10	54		10	195	360	4·30	2900		
40562	"	108·0	109·0	0·9	1·0		X	0·5	7	40		10	15	95	3·60	1500		

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 221
SHEET No. 3

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified)											COMMENTS
						Au	Ag	As	Sn	Sn	Cu	Pb	Zn	Fe%	Mn		
40563	SPLIT	109·0	110·0	0·9	1·0	x	x	7	42		5	x	110	4·40	1300		
64	"	110·0	112·8	0·2	2·8	x	x	39	110		15	40	175	4·60	1450		
65	"	112·8	113·65	0·05	0·85	0·032	0·5	12	64		10	30	150	3·20	2600		
66	"	113·65	114·5	0·15	0·85	x	x	56	64		15	115	1000	2·50	2500		
67	"	114·5	115·5	0·4	1·0	0·032	1·0	100	84		20	350	640	4·00	2950		
68	"	115·5	116·5	0·7	1·0	0·024	0·5	70	74		30	45	200	6·00	2850		
69	"	116·5	117·5	0·8	1·0	x	x	210	60		15	40	170	5·25	3450		
40570	"	117·5	118·5	0·7	1·0	0·016	x	55	88		90	30	160	7·40	2700		
71	"	118·5	119·5	0·7	1·0	0·016	3·0	1·30%	38		695	60	115	9·50	1050		Arsenic Zone 'B'
72	"	119·5	120·5	0·8	1·0	x	0·5	140	48		150	20	40	3·70	700		
73	"	120·5	121·5	0·9	1·0	x	0·5	47	58		50	100	230	4·27	1200		
74	"	121·5	122·5	0·4	1·0	0·040	0·5	39	58		185	70	125	6·10	1850		
75	"	122·5	123·5	0·9	1·0	0·008	x	30	52		70	20	50	2·40	1350		
76	SPLIT	123·5	124·5	0·9	1·0	0·016	x	30	56		10	20	40	2·05	950		
77	CHIP	124·5	129·5	4·9	5·0	0·016	x	25	68		15	25	50	3·33	800		
78	CHIP	129·5	132·8	3·1	3·3	0·032	x	25	60		20	15	45	2·80	750		
79	SPLIT	132·8	133·8	1·0	1·0	0·040	1·0	10	44		480	15	50	15·50	790		
40580	SPLIT	133·8	134·8	1·0	1·0	0·024	0·5	35	46		250	25	35	7·60	600		
81	CHIP	134·8	139·8	5·0	5·0	0·008	0·5	180	44		80	30	50	5·56	700		
82	"	139·8	146·8	5·0	5·0	0·008	0·5	27	48		25	55	100	2·80	950		
83	"	144·8	149·8	5·0	5·0	0·008	x	27	62		30	15	65	2·30	630		
84	"	149·8	154·8	5·0	5·0	0·016	0·5	3900	36		140	50	80	5·50	660		
85	"	154·8	159·8	5·0	5·0	0·032	0·5	110	44		130	70	325	4·20	900		
86	CHIP	159·8	164·8	5·0	5·0	0·040	10·0	4000	46		400	145	210	7·80	850		
87	SPLIT	164·8	165·8	1·0	1·0	0·016	1·0	2000	38		245	30	85	5·70	750		
88	"	165·8	166·8	0·9	1·0	0·024	2·5	110	39		260	210	490	13·50	1000		
89	SPLIT	166·8	167·8	1·0	1·0	0·072	23·5	1·00%	34		2200	930	1050	13·50	950		Arsenic Zone 'D'
40590	CHIP	167·8	172·8	4·8	5·0	0·016	0·5	5500	46		260	50	105	6·10	950		
91	"	172·8	177·8	5·0	5·0	x	x	110	38		55	15	90	3·80	800		
92	"	177·8	182·8	5·0	5·0	0·016	x	42	64		30	55	90	3·20	1500		
93	"	182·8	187·8	5·0	5·0	0·032	0·5	38	120		35	305	340	3·46	2150		
94	"	187·8	192·8	5·0	5·0	0·008	0·5	56	38		35	90	315	3·90	8000		
40595	"	192·8	197·8	5·0	5·0	0·016	x	52	52		50	15	125	2·70	1550		

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 221
SHEET No. 4

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified)														COMMENTS
							Au	Ag	As	Sn	Sn	Cu	Pb	Zn	Fe%	Mn				
40596	CHIP	197·8	201·8	4·0	4·0		0·016	0·5	40	52		35	30	100	2·56	1000				
40597	CHIP	201·8	203·0	1·2	1·2		0·016	x	17	64		35	80	230	2·16	625				

ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 231
SHEET No. 1

GRID CO-ORDS: 4160N; 4625E Stg. Vall.
A.M.G. CO-ORDS: 5,374 267 mN 384 216 mE
COLLAR R.L.: 175·4 m
COLLAR DIP: −60°
AZIMUTH: 108° A.M.G.
TOTAL DEPTH: 150·6m

LABORATORY						ANALABS													
ANALYTICAL TECHNIQUE						309	AQUA REGIA	108	114	118		108	108	108	108	108			
						FIRE ASSAY	AAS	AAS	AAS	AAS		AAS	AAS	AAS	AAS	AAS			
DETECTION LIMIT						0·008	0·008	0·5	1	1		5	5	5	50	5			
Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au	Au	Ag	As	Sn		Cu	Pb	Zn	Fe %	Mn				COMMENTS
39379	CHIP	13·30	15·0	0·70	1·7	x		x	46	x		30	70	275	3·40	940				
39380	"	15·0	20·0	2·1	5·0	x		x	920	8		65	70	220	4·60	2250				
81	"	20·0	25·0	4·7	5·0	x		x	1000	x		170	35	210	5·50	1000				
82	"	25·0	30·0	5·0	5·0	x		x	70	x		255	20	170	5·50	1100				
83	"	30·0	35·0	4·9	5·0	x		0·5	220	x		455	1000	270	5·20	1200				
84	"	35·0	40·0	4·0	5·0	x		x	340	x		245	230	360	4·40	1200				
85	"	40·0	45·0	3·3	5·0	0·01		x	35	x		60	35	280	5·90	1700				
86	"	45·0	50·0	5·0	5·0	x		x	32	x		105	10	185	6·00	1400				
39387	CHIP	50·0	53·0	3·0	3·0	x		x	30	x		25	45	250	6·80	1300				
39264	SPLIT	52·1	53·1	1·0	1·0	0·02	0·008	x	170	20		45	190	170	6·70	1850				
65	"	53·1	53·8	0·7	0·7	x	0·016	1	1500	x		140	100	180	6·70	1350				
66	"	53·8	53·9	0·09	0·1	0·90	0·008	9·5	17·0 %	17		850	235	80	11·00	550				
67	"	53·9	54·9	1·0	1·0	0·27	0·216	8·5	4·0 %	x		735	90	110	6·60	1150				
68	"	54·9	55·6	0·7	0·7	x	0·024	2·0	1600	130		450	75	115	7·20	1600				
69	"	55·6	56·2	0·6	0·6	0·35	0·024	11·0	3·8 %	x		1750	95	285	9·70	1200				
39270	"	56·2	56·7	0·5	0·5	1·29	0·276	20·0	15·0 %	160		2250	200	2000	9·10	780				Arsenic Zone 'C'
71	"	56·7	57·2	0·5	0·5	0·47	x	6·0	8·5 %	22		2500	65	70	12·50	595				
72	"	57·2	57·7	0·5	0·5	0·62	x	4·0	3·8 %	10		2200	60	40	11·00	275				
73	"	57·7	58·3	0·6	0·6	1·01	0·436	10·0	8·1 %	600		6750	75	190	7·20	465				—
39274	SPLIT	58·3	59·3	1·0	1·0	x	0·008	0·5	900	520		130	35	105	6·20	1500				
39389	Sand	55·0	60·0	5·0	5·0	0·35		7·5	3·9 %	5		2500	65	185	10·70	850				
390	"	60·0	65·0	5·0	5·0	0·15		x	900	20		140	170	835	6·30	1550				
391	"	65·0	70·0	5·0	5·0	0·04		x	680	x		70	45	260	5·70	1200				
392	"	70·0	75·0	4·7	5·0	0·02		x	1000	20		75	35	750	6·10	1500				
39393	Sand	75·0	80·0	5·0	5·0	0·11		x	3800	1100		515	55	240	6·60	1650				

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 231
SHEET No. 2

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au	Au	Ag	As	Sn		Cu	Pb	Zn	Fe %	Mn				COMMENTS
39275	SPLIT	77·3	78·3	1·0	1·0	x	0·032	0·5	700	1800		75	10	190	8·00	2250	—			
76	"	78·3	78·8	0·5	0·5	1·22	0·764	3·0	9·80%	5600		1150	60	70	5·80	660	—			Arsenic Zone A
77	"	78·8	79·5	0·7	0·7	0·14	0·104	4·5	2800	3000		2150	50	100	7·30	1400	—			
78	"	79·5	80·5	1·0	1·0	0·05	0·080	1·0	2900	180		155	190	1400	7·10	2050				
79	"	80·5	81·5	1·0	1·0	x	0·032	2·0	600	20		60	1000	5550	5·80	4300				
39280	"	81·5	82·2	0·7	0·7	x	x	2·0	1600	4		80	575	2450	6·60	3150				
81	"	82·2	82·6	0·4	0·4	0·03	0·008	2·5	4200	10		225	175	710	5·70	1250				
82	"	82·6	83·6	1·0	1·0	0·01	0·024	2·5	230	2		140	360	815	6·60	1450				
83	"	83·6	84·6	1·0	1·0	0·01	0·016	1·0	500	8		120	75	370	6·20	1400				
84	"	84·6	85·2	0·6	0·6	x	0·016	0·5	4900	6		160	70	100	6·00	1450				
85	"	85·2	85·5	0·3	0·3	0·58	0·376	5·5	9·00%	6		170	140	155	7·80	520	—			Arsenic Zone A
86	"	85·5	86·5	1·0	1·0	x	0·032	0·5	1600	x		20	30	285	4·70	1750				
87	"	86·5	87·5	1·0	1·0	x	0·016	0·5	70	46		15	35	170	4·75	1450				
88	"	87·5	88·0	0·5	0·5	0·02	0·024	1·0	90	28		390	30	175	8·90	2250				
39289	SPLIT	88·0	89·0	1·0	1·0	0·01	x	1·0	70	x		15	35	160	4·40	1500				
39395	GRIND	85·0	90·0	5·0	5·0	0·03		x	3500	12		155	35	210	7·30	1250				
96	GRIND	90·0	95·0	5·0	5·0	x		x	12	x		20	60	140	3·90	1800				
97	CHIP	95·0	100·0	4·9	5·0	x		x	620	42		50	30	270	7·40	3550				
98	"	100·0	105·0	4·1	5·0	x		x	18	x		15	60	560	5·80	1250				
99	"	105·0	110·0	4·5	5·0	x		x	54	x		5	25	405	4·40	1000				
39400	"	110·0	115·0	5·0	5·0	x		x	61	x		x	25	175	5·40	1000				
41851	"	115·0	120·0	5·0	5·0	x		x	37	x		20	15	160	5·20	1400				
52	"	120·0	125·0	5·0	5·0	x		x	31	x		55	35	175	7·30	2000				
53	"	125·0	130·0	4·7	5·0	x		x	20	x		x	5	145	5·40	1050				
54	"	130·0	135·0	4·3	5·0	x		x	16	x		5	35	160	4·90	960				
55	"	135·0	140·0	5·0	5·0	x		x	69	x		30	25	205	5·20	1400				
56	"	140·0	145·0	5·0	5·0															SAMPLE LOST
41857	CHIP	145·0	150·6	5·0	5·0	x		x	27	x		5	100	465	8·70	3900				

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH ST P 234
SHEET No. 1

GRID CO-ORDS: 4340 N ; 4477 E. Sig. Vall.
A.M.G. CO-ORDS: 5,374 400 mN 384 413 mE
COLLAR R.L.: 176 m
COLLAR DIP: -70°
AZIMUTH: 108° AMG
TOTAL DEPTH: 342.5 m.

LABORATORY							ANALABS												
ANALYTICAL TECHNIQUE							309		108	114		402	108	108	108	108	108	402	108	402	W 402
							Fire Assay		AAS	AAS		XRF	AAS	AAS	AAS	AAS	AAS	XRF	AAS	XRF	XRF
DETECTION LIMIT							0.008		0.5	1		3	5	5	5	50	5	3	10	3	10

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au		Ag	As		Sn	Cu	Pb	Zn	Fe%	Mn	Sb	Bi	Te	W	COMMENTS
41687	Grnd	20.0	25.0	100%	5.0			X	13		16	85	30	150	6.00	1650					
88	"	25.0	30.0	"	5.0			X	24		6	65	20	150	5.85	1750					
89	"	30.0	35.0	"	5.0			X	7		14	105	20	185	7.15	2100					
41690	"	35.0	40.0	"	5.0			X	7		X	70	10	145	6.00	1500					
91	"	40.0	45.0	"	5.0			X	9		X	55	15	175	6.50	1600					
92	"	45.0	50.0	"	5.0			X	10		10	80	20	155	6.15	1450					
93	"	50.0	55.0	"	5.0			X	15		8	105	10	125	5.50	1200					
94	"	55.0	60.0	"	5.0			X	15		22	60	25	190	7.60	1950					
95	"	60.0	65.0	"	5.0			X	17		10	130	15	150	6.30	1950					
96	"	65.0	70.0	"	5.0			X	11		X	160	30	160	5.65	945					
97	"	70.0	75.0	"	5.0			X	7		6	100	5	125	5.95	1300					
98	"	75.0	80.0	"	5.0			X	6		X	85	10	160	7.55	1700					
99	"	80.0	85.0	"	5.0			X	6		4	85	15	135	5.50	1400					
41700	"	85.0	90.0	"	5.0			X	2		X	60	15	135	5.90	1600					
45801	"	90.0	95.0	"	5.0			X	1		X	25	10	155	5.65	1750					
02	"	95.0	100.0	"	5.0			X	4		26	30	15	160	8.15	2400					
03	"	100.0	105.0	"	5.0			X	2		X	25	20	145	5.70	1750					
04	"	105.0	110.0	"	5.0			X	2		X	40	10	160	5.30	1550					
05	"	110.0	115.0	"	5.0			X	9		X	65	15	195	5.55	1450					
06	"	115.0	120.0	"	5.0			X	3		X	60	20	305	6.95	2100					
07	"	120.0	125.0	"	5.0			X	3		X	50	30	275	6.65	2300					
45808	Grnd	125.0	130.0	"	5.0			X	4		X	75	5	165	6.65	2050					
48432	SPLIT	130.0	131.0	"	1.0	X		X	10		X	90	10	150	4.90	1500	X	X	X	X	
33	"	131.0	132.0	"	1.0	X		X	4		X	85	5	145	4.90	1950	X	X	X	X	
34	"	132.0	133.0	"	1.0	X		X	5		X	115	X	165	5.60	1850	X	X	X	X	
35	"	133.0	134.0	"	1.0	X		X	6		X	95	10	155	5.15	1850	X	X	X	X	
36	"	134.0	135.0	"	1.0	X		X	11		265	520	X	205	9.25	2750	X	X	X	X	
48437	SPLIT	135.0	136.0	"	1.0	X		X	900		433	195	X	215	16.00	2950	X	X	X	26	

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 234
SHEET No. 2

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au		Ag	As		Sn	Cu	Pb	Zn	Fe %	Mn	Sb	Bi	Te	W	COMMENTS
48438	SPLIT	136·0	137·0	100%	1·0	x		x	4		x	20	5	165	5·30	1900	x	x	23	x	
39	"	137·0	138·0	"	1·0	x		x	4		x	5	5	200	5·75	2200	x	x	x	15	
48440	"	138·0	139·0	"	1·0	x		x	4		87	40	x	205	8·35	3150	x	x	13	x	
41	"	139·0	140·0	"	1·0	x		x	10		19	165	10	180	6·55	2550	x	x	x	x	
42	"	140·0	141·0	"	1·0	x		x	72		474	245	35	265	12·50	3300	x	x	x	20	
43	"	141·0	142·0	"	1·0	x		x	12		13	145	10	155	5·25	2150	3	x	x	x	
44	"	142·0	143·0	"	1·0	x		x	400		730	70	15	120	5·30	2100	x	x	x	10	
45	"	143·0	144·0	"	1·0	x		x	8		30	15	30	155	4·80	1950	x	x	x	23	
46	"	144·0	145·0	"	1·0	x		x	4		18	10	x	175	4·70	1700	x	x	x	x	
47	"	145·0	146·0	"	1·0	x		x	86		147	75	x	305	7·65	2400	x	x	x	15	
48	"	146·0	147·0	"	1·0	x		x	1000		286	5	x	135	6·70	1950	x	x	x	20	
49	"	147·0	148·0	"	1·0	x		x	75		87	10	30	225	4·25	3250	x	x	x	11	
48450	"	148·0	149·0	"	1·0	x		x	8		10	5	5	145	5·50	1700	x	x	x	11	
51	"	149·0	150·0	"	1·0	x		x	10		66	25	10	175	7·70	2500	x	x	18	15	
52	"	150·0	151·0	"	1·0	x		x	8		126	10	20	190	5·90	3000	x	x	x	20	
53	"	151·0	152·0	"	1·0	x		x	1900		612	170	15	805	8·25	3100	x	x	x	16	
54	"	152·0	153·0	"	1·0	x		x	98		91	40	10	290	7·20	5750	x	x	x	x	
55	"	153·0	154·0	"	1·0	x		x	17		63	40	50	285	8·90	3500	6	x	x	15	
56	"	154·0	155·0	"	1·0	x		x	5		23	45	10	175	5·25	2600	x	x	x	x	
57	"	155·0	156·0	"	1·0	x		x	8		3	9	x	160	5·60	2000	x	x	x	x	
58	"	156·0	157·0	"	1·0	0·12		1·40	1300		127	450	635	1·01%	6·85	2750	x	40	x	x	Possibly Arsenic Zone 'C'
59	"	157·0	158·0	"	1·0	x		x	9		48	10	5	150	4·10	2600	x	x	x	x	
48460	"	158·0	159·0	"	1·0	x		x	7		x	10	x	175	4·85	2100	x	x	x	x	
61	"	159·0	160·0	"	1·0	x		x	4		x	5	x	155	4·90	1800	x	x	x	x	
62	"	160·0	161·0	"	1·0	x		x	4		x	5	x	150	4·70	1900	x	x	x	x	
63	"	161·0	162·0	"	1·0	x		x	7		16	10	5	175	7·50	2350	x	x	x	15	
64	"	162·0	163·0	"	1·0	x		x	7		49	35	20	240	8·05	2500	x	x	x	x	
65	"	163·0	164·0	"	1·0	x		x	4		x	20	x	175	5·75	1700	x	x	x	x	
66	"	164·0	165·0	"	1·0	x		x	3		x	35	5	160	4·75	1800	x	x	x	x	
67	"	165·0	166·0	"	1·0	x		x	4		x	10	x	190	5·90	2200	x	x	33	x	
68	"	166·0	167·0	"	1·0	x		x	3		x	5	15	205	5·95	2800	x	x	23	x	
69	"	167·0	168·0	"	1·0	x		x	3		x	10	x	190	5·60	1850	x	x	x	x	
48470	SPLIT	168·0	169·0	"	1·0	x		x	3		x	10	x	155	4·80	1650	x	x	5	x	

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 234
SHEET No. 3

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au	Ag	As	Sn	Cu	Pb	Zn	Fe %	Mn	Sb	Bi	Te	W	COMMENTS
48471	SPLIT	169·0	170·0	100%	1·0	x	x	1400	313	30	10	180	9·00	3700	x	x	x	10	
72	"	170·0	171·0	"	1·0	x	x	7	11	5	x	180	6·15	3150	x	x	x	x	
73	"	171·0	172·0	"	1·0	x	x	520	5	10	5	200	6·80	3250	x	x	x	x	
74	"	172·0	173·0	"	1·0	x	x	12	x	10	5	190	6·15	2300	x	x	23	x	
75	"	173·0	174·0	"	1·0	x	x	10	103	5	x	210	6·80	2100	x	x	x	36	
76	"	174·0	175·0	"	1·0	x	x	100	207	25	10	175	9·45	3400	4	x	8	42	
77	"	175·0	176·0	"	1·0	x	x	13	39	5	x	205	7·45	2300	x	x	x	16	
78	"	176·0	177·0	"	1·0	x	x	26	21	5	x	180	7·15	2100	x	x	7	34	
79	"	177·0	178·0	"	1·0	x	0·5	570	169	10	45	205	7·75	2150	x	x	x	26	
48480	"	178·0	179·0	"	1·0	x	x	12	9	10	x	145	5·10	1750	x	x	x	x	
81	"	179·0	180·0	"	1·0	x	x	16	16	10	20	190	6·15	1950	x	x	x	x	
82	"	180·0	181·0	"	1·0	x	x	12	375	5	5	205	7·85	2350	x	x	x	x	
83	"	181·0	182·0	"	1·0	x	x	77	417	75	x	125	6·80	2700	x	x	x	17	
84	"	182·0	183·0	"	1·0	x	x	17	739	75	110	385	7·20	2450	x	x	x	x	
85	"	183·0	184·0	"	1·0	x	x	18	82	10	x	210	8·45	2750	x	x	x	x	
86	"	184·0	185·2	"	1·2	x	x	30	164	35	x	350	8·95	2750	x	x	8	x	
87	"	185·2	185·5	"	0·3	3·25	11·0	4	150	550	205	32·33%	11·50	1550	33	x	14	x	? related to Arsenic Zone 'A'
88	"	185·5	186·5	"	1·0	0·03	0·5	18	148	110	40	6400	9·55	3050	x	x	x	12	
89	"	186·5	187·5	"	1·0	0·06	2·5	1200	250	608	60	3000	13·50	2850	x	x	8	33	
48490	"	187·5	188·3	"	0·8	x	x	17	123	85	65	425	13·00	3650	x	x	x	12	
48415	"	188·3	189·0	"	0·7	0·14	1·0	430	52	610	20	360	13·50	2750					
16	"	189·0	190·0	"	1·0	1·95	16·5	3·20%	x	7100	130	1050	13·50	1900					Arsenic Zone 'A'
17	"	190·0	191·0	"	1·0	2·91	29·5	4·00%	x	4850	310	415	9·10	1150					Gold in Polished Section.
48418	"	191·0	191·5	"	0·5	1·96	20·5	5·30%	x	4300	185	285	1·95	625					2·5m @ 2·34g/t Au
48491	"	191·5	192·0	"	0·5	0·02	x	100	45	245	5	400	4·85	1550	x	x	x	x	
92	"	192·0	193·0	"	1·0	0·04	x	13	14	45	50	360	4·40	1300	x	x	x	x	
93	"	193·0	194·0	"	1·0	0·12	x	64	71	70	160	965	5·05	1650	x	x	4	21	
94	"	194·0	195·0	"	1·0	0·25	x	75	52	150	175	630	5·85	1750	x	x	x	12	? related to Arsenic Zone 'A'
95	"	195·0	196·0	"	1·0	x	x	14	5	15	15	115	4·25	1550	x	x	8	13	
96	"	196·0	197·0	"	1·0	0·03	0·5	13	61	105	90	4200	7·00	2200	x	x	x	x	
97	"	197·0	198·0	"	1·0	0·23	x	25	21	155	25	430	7·35	2150	x	x	x	13	
98	"	198·0	199·0	"	1·0	0·07	x	16	35	85	55	465	5·10	1850	x	x	10	10	
48499	SPLIT	199·0	200·0	"	1·0	x	x	11	9	95	x	215	3·75	1450	x	x	3	13	

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 234
SHEET No. 4

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au		Ag	As		Sn	Cu	Pb	Zn	Fe%	Mn	Sb	Bi	Te	W	COMMENTS
48500	SPLIT	200·0	201·0	100%	1·0	x		x	7		x	25	x	105	3·25	1250	x	x	17	x	
61238	"	201·0	202·0	"	1·0	x		x	8		x	85	x	105	3·65	1800	x	x	3	x	
39	"	202·0	203·0	"	1·0	x		x	7		x	65	x	90	3·10	1200	x	x	x	x	
61240	"	203·0	204·0	"	1·0	x		x	7		x	25	x	110	3·80	1300	x	x	x	x	
41	"	204·0	205·0	"	1·0	x		x	13		x	15	10	110	4·25	1500	x	x	x	11	
42	"	205·0	206·0	"	1·0	0·26		x	68		52	195	45	190	7·65	1400	7	x	x	12	
43	"	206·0	207·0	"	1·0	0·04		0·5	54		89	115	70	800	6·75	1500	x	x	x	25	
44	"	207·0	208·0	"	1·0	0·23		0·5	100		105	195	80	220	7·65	1750	x	x	x	31	
45	"	208·0	209·0	"	1·0	0·30		0·5	66		78	185	75	1550	9·85	2200	3	x	x	44	
61246	SPLIT	209·0	210·0	"	1·0	x		x	16		13	60	25	275	6·65	2150	x	x	x	x	
45825	Grind	210·0	215·0	"	5·0			x	23		6	75	35	175	5·60	1800					
26	"	215·0	220·0	"	5·0			x	14		6	55	90	240	6·35	2250					
27	"	220·0	225·0	"	5·0			x	20		8	25	5	165	7·20	1800					
28	"	225·0	230·0	"	5·0			x	6		8	35	5	125	5·85	1800					
29	"	230·0	235·0	"	5·0			x	4		x	25	10	130	6·30	1650					
45830	"	235·0	240·0	"	5·0			x	3		x	100	10	175	7·20	2000					
45831	"	240·0	245·0	"	5·0			x	3		32	85	20	205	8·20	2200					
61247	SPLIT	245·0	246·0	"	1·0	x		x	6		x	5	x	135	3·90	1650	x	x	x	x	
48	"	246·0	247·0	"	1·0	0·06		x	3		x	20	x	140	3·70	1400	x	x	x	40	
49	"	247·0	248·0	"	1·0	x		x	6		x	5	x	140	4·25	805	x	x	12	63	
61250	"	248·0	249·0	"	1·0	0·02		1·0	19		678	60	10	155	7·50	1700	x	X	x	42	
51	"	249·0	250·0	"	1·0	0·11		2·0	900		2840	80	110	245	9·15	2150	x	100	x	33	
52	"	250·0	251·0	"	1·0	x		1·0	11		355	80	40	280	9·85	2350	x	x	x	29	
61253	"	251·0	252·0	"	1·0	0·02		1·5	1100		2470	515	15	140	9·35	2050	x	x	x	56	
48423	"	252·0	253·0	"	1·0	0·40		0·5	3800		3000	865	25	95	7·00	1450					} ?related to Arsenic Zone 'B'
48424	"	2530	254·1	"	1·1	0·56		0·5	80		4200	250	10	125	13·50	3458					
61254	"	254·1	255·0	"	0·9	x		x	12		23	10	25	200	4·65	2000	5	x	x	13	
55	"	255·0	256·0	"	1·0	x		1·0	17		413	100	40	115	9·50	2300	x	x	9	37	
56	"	256·0	257·0	"	1·0	0·01		1·5	12		102	125	85	210	6·95	1750	3	x	11	13	
57	"	257·0	258·0	"	1·0	0·05		x	8		5	5	x	110	3·00	1150	3	x	x	11	
58	"	258·0	259·0	"	1·0	x		x	8		x	5	5	110	2·85	780	4	x	x	x	
61259	"	259·0	260·0	"	1·0	x		0·5	15		7	135	5	195	6·45	1650	x	x	18	x	
48419	SPLIT	260·0	261·0	"	1·0	0·04		0·5	1100		10	2150	10	140	6·55	1358					

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DD H STP 234
SHEET No. 5

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au	Ag	As	Sn	Cu	Pb	Zn	Fe%	Mn	Sb	Bi	Te	W	COMMENTS
48420	SPLIT	261·0	262·0	100%	1·0	0·92	2·0	5·20%	1150	2150	30	125	6·00	1350					} Arsenic Zone 'B'
21	"	262·0	263·0	"	1·0	0·85	2·0	3·50%	x	2350	50	200	6·55	1400					
48422	"	263·0	264·0	"	1·0	0·01	1·0	380	6	655	15	100	4·90	950					
61260	"	264·0	265·0	"	1·0	x	2·0	15	27	610	35	159	9·00	1300	7	40	7	24	
61	"	265·0	266·0	"	1·0	0·02	3·5	63	37	1700	45	190	5·35	900	10	60	x	13	
62	"	266·0	267·0	"	1·0	0·16	7·5	9600	1100	4250	125	190	4·00	885	10	250	x	x	} ? related to Arsenic Zone 'B'
63	"	267·0	268·0	"	1·0	0·17	24·0	1·05%	1740	6000	950	385	4·30	1050	20	970	x	x	}
64	"	268·0	269·0	"	1·0	0·01	1·0	1300	179	235	5	95	6·25	1050	9	x	x	x	
65	"	269·0	270·0	"	1·0	x	0·5	190	12	55	10	70	3·15	1600	5	x	x	x	
66	"	270·0	271·0	"	1·0	0·03	1·0	1·15%	166	225	10	65	5·25	1150	10	150	x	x	
67	"	271·0	272·0	"	1·0	x	0·5	140	x	25	5	60	2·80	1050	4	x	x	x	
68	"	272·0	273·0	"	1·0	x	1·0	100	x	10	5	60	2·20	1050	x	x	x	x	
69	"	273·0	274·0	"	1·0	x	x	47	x	10	x	55	2·65	1200	x	x	x	x	
61270	SPLIT	274·0	275·0	"	1·0	0·03	x	61	5	10	x	60	2·75	1150	4	x	x	x	
45838	GRIND	275·0	280·0	"	5·0		x	39	8	20	10	70	4·50	1050					
39	"	280·0	285·0	"	5·0		x	280	x	130	10	80	5·20	1050					
45840	"	285·0	290·0	"	5·0		x	40	x	15	10	85	4·00	860					
41	"	290·0	295·0	"	5·0		0·5	20	4	15	10	65	3·00	635					
42	"	295·0	300·0	"	5·0		0·5	20	4	20	10	80	3·30	620					
45843	Grind	300·0	305·0	"	5·0		0·5	23	x	15	25	105	3·25	770					
61271	SPLIT	305·0	306·0	"	1·0	0·02	0·5	90	x	25	10	95	3·55	1550	x	x	11	x	
72	"	306·0	307·0	"	1·0	x	x	100	x	10	5	110	3·60	1250	4	x	x	x	
73	"	307·0	308·0	"	1·0	x	1·0	100	x	115	x	90	4·45	870	8	x	8	x	
61274	"	308·0	309·3	"	1·3	0·01	0·5	100	x	50	x	85	3·15	620	4	x	6	x	
48425	"	309·3	310·3	"	1·0	0·02	2·0	3100	17	455	15	85	5·05	578					
48426	"	310·3	311·3	"	1·0	0·02	0·5	9900	x	225	15	75	5·10	578					
61275	"	311·3	312·3	"	1·0	x	0·5	510	7	25	5	85	3·95	1350	4	x	7	x	
61276	"	312·3	313·25	"	0·95	0·12	2·0	1300	59	650	10	250	8·05	1150	x	50	x	x	
48427	"	313·25	313·85	"	0·60	5·00	29·5	900%	x	1·15%	505	50	3·15	2900					Arsenic Zone 'D' Gold in P/Section
61277	"	313·85	314·5	"	0·65	x	0·5	100	6	40	5	125	4·35	1200	7	x	7	x	
78	"	314·5	315·5	"	1·0	x	0·5	55	x	10	5	85	3·45	1700	4	x	12	x	
79	"	315·5	316·5	"	1·0	x	0·5	59	x	10	x	95	4·05	1250	x	x	14	x	
61280	SPLIT	316·5	317·5	"	1·0	x	0·5	51	x	10	5	95	3·80	1300	x	x	x	12	

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 234
SHEET No. 6

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au		Ag	As		Sn	Cu	Pb	Zn	Fe %	Mn	Sb	Bi	Te	W	COMMENTS
61281	SPLIT	317·5	318·5	100%	1·0	x		0·5	48		x	20	10	80	3·20	1000	4	x	17	x	
82	"	318·5	319·5	"	1·0	x		10·0	7100		3	205	75	85	5·45	650	10	130	x	x	
83	"	319·5	320·5	"	1·0	0·06		10·5	8200		4	70	70	75	4·00	1000	5	140	x	x	
61284	"	320·5	321·5	"	1·0	x		3·0	110		x	70	20	85	5·35	970	11	x	x	x	
45847	Grind	320·0	325·0	"	5·0			0·5	55		x	40	10	80	4·40	1050					
48	"	325·0	330·0	"	5·0			1·0	42		6	30	45	120	4·40	1050					
49	"	330·0	335·0	"	5·0			x	24		x	30	10	70	3·50	840					
45850	"	335·0	340·0	"	5·0			0·5	31		x	10	10	50	3·45	845					
45851	"	340·0	341·5	"	1·5			0·5	61		x	15	50	105	3·45	1350					

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074

21803

ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. D.D.H. STP 232 A
SHEET No. 1

GRID CO-ORDS: 4640N, 4645E Stg. Vall.
A.M.G. CO-ORDS: 5,374,728·5 m N, 384,395·8 m E
COLLAR R.L.: 174 m.
COLLAR DIP: −60°
AZIMUTH: 108° AMG
TOTAL DEPTH: 92·8m.

LABORATORY						ANALABS											
ANALYTICAL TECHNIQUE						309		108	114	118		109	108	108	108	108			
						FIRE ASSAY		AAS	AAS	AAS		AAS	AAS	AAS	AAS	AAS			
DETECTION LIMIT						0·008		0·5	1	1		5	5	5	50	5			
Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au		Ag	As	Sn		Cu	Pb	Zn	Fe %	Mn				COMMENTS
48401	SPLIT	66·7	67·7	1·0	1·0	0·63		0·5	28	3600		355	95	375	3·75	1508				Wispy po-pg-stringers in
48406	SPLIT	67·7	68·1	0·4	0·4	0·76		1·5	21	3		490	10	55	6·75	1050				Fe-mall slates unit
48402	SPLIT	77·0	76·4	0·4	0·4	0·02		6·5	38	x		875	38	85	5·35	565				Stringers and veinlets of po-pg
48403	SPLIT	77·9	78·25	0·35	0·35	0·02		3·8	26	18		155	80	7800	5·50	725				associated with quartz-carb veining

ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 232A-1
SHEET No. 1

GRID CO-ORDS:
A.M.G. CO-ORDS: S, 376,728·5mN; 384,395·8mE
COLLAR R.L.: 174m
COLLAR DIP: −60°
AZIMUTH: 108° A.M.G.
TOTAL DEPTH: 198·2m

LABORATORY						ANALABS													
ANALYTICAL TECHNIQUE						309	AQUA REGIA	108	114	118		108	108	108	108	108			
						FIRE ASSAY	AAS	AAS	AAS	AAS		AAS	AAS	AAS	AAS	AAS			
DETECTION LIMIT						0·008	0·008	0·5	1	1		5	5	5	50	5			
Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au	Au	Ag	As	Sn		Cu	Pb	Zn	Fe%	Mn				COMMENTS
41624	Grind	63·0	65·0	2·0	2·0		x	0·5	58	x		25	435	100	2·00	560				
41625	Grind	65·0	70·0	5·0	5·0		x	3·5	92	x		130	605	510	6·20	2250				
48404	SPLIT	67·0	68·0	1·0	1·0	0·21		1·0	31	3100		325	15	80	5·05	875				Stringers of Po+Py
41626	Grind	70·0	75·0	5·0	5·0		x	3·0	240	68		180	280	1850	6·50	2750				
41627	Grind	75·0	80·0	5·0	5·0		x	1·0	47	x		200	5	145	7·10	1850				
48405	SPLIT	74·7	75·7	1·0	1·0	0·01		1·5	16	5		370	30	220	5·50	855				
48407	SPLIT	75·7	76·7	1·0	1·0	0·02		2·0	200	45		485	15	105	6·45	825				
48408	SPLIT	76·7	77·7	1·0	1·0	0·03		1·5	14	10		365	5	85	6·70	730				
48409	SPLIT	77·7	78·7	1·0	1·0	0·01		0·5	6	18		325	10	80	5·85	710				
41628	Grind	80·0	85·0	5·0	5·0		x	0·5	210	x		20	30	120	3·50	2550				
29	Grind	85·0	90·0	5·0	5·0		x	1·0	86	10		40	90	215	5·30	1400				
30	Grind	90·0	95·0	5·0	5·0		x	0·5	38	x		20	30	70	3·05	1300				
41631	Grind	95·0	100·0	5·0	5·0		x	1·0	68	x		45	50	120	4·55	2650				
48410	SPLIT	98·9	99·9	1·0	1·0	0·05		0·5	44	30		125	25	135	8·90	1200				
48411	SPLIT	99·9	100·9	1·0	1·0	0·02		0·5	35	x		75	35	135	9·80	1400				
48632	Grind	100·0	105·0	5·0	5·0		x	2·0	1500	x		85	50	110	5·20	1050				
48412	SPLIT	104·6	104·9	0·3	0·3	0·05		14·5	4·5%	x		845	345	115	5·55	685				50% Stringers Po+Py+Asp
4862	Grind	105·0	110·0	5·0	5·0		x	0·5	80	x		20	35	150	2·95	1750				
4863	Grind	110·0	115·0	5·0	5·0		x	1·0	280	x		95	200	630	4·75	2350				

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. DDH STP 232 A1
SHEET No. 2

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au	Au	Ag	As	Sn		Cu	Pb	Zn	Fe %	Mn				COMMENTS
48413	SPLIT	112·0	113·2	1·2	1·2.	0·12		1·5	520	1		160	100	190	4·90	1550				
48635	Grind	115·0	120·0	5·0	5·0		0·008	0·5	160	10		55	145	2250	6·0	2650				Blebs and wispy stringers Sph. rpg
36	"	120·0	125·0	5·0	5·0		X	X	39	X		10	15	160	5·0	1500				
37	"	125·0	130·0	5·0	5·0		0·008	X	58	X		20	220	2000	3·0	1650				
38	"	130·0	135·0	5·0	5·0		X	X	74	X		20	65	185	2·85	1900				
39	"	135·0	140·0	5·0	5·0		X	X	24	X		10	20	35	2·10	705				
48640	"	140·0	145·0	5·0	5·0		X	X	37	X		40	75	105	3·50	2200				
41	"	145·0	150·0	5·0	5·0		X	X	47	X		45	20	30	3·25	1900				
42	"	150·0	155·0	5·0	5·0		0·008	X	49	X		35	45	90	3·75	3100				
43	"	155·0	160·0	5·0	5·0		X	X	41	X		35	105	195	3·10	2200				
44	"	160·0	165·0	5·0	5·0		X	X	26	X		35	25	55	2·15	1100				
45	"	165·0	170·0	5·0	5·0		X	X	7	X		35	45	180	2·30	755				
46	"	170·0	175·0	5·0	5·0		X	X	21	X		25	55	100	1·95	590				
47	"	175·0	180·0	5·0	5·0		X	0·5	21	X		25	20	25	1·85	565				
48	"	180·0	185·0	5·0	5·0		X	X	41	X		65	25	365	3·05	1000				
49	"	185·0	190·0	5·0	5·0		X	X	22	X		230	20	35	2·50	1000				
48650	"	190·0	195·0	5·0	5·0		X	X	73	X		70	195	415	2·90	1400				
48651	Grind	195·0	198·2	3·2	3·2		X	X	30	X		70	30	25	2·20	500				

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. STP 218
SHEET No. 1

GRID CO-ORDS:
A.M.G. CO-ORDS:
COLLAR R.L.:
COLLAR DIP:
AZIMUTH:
TOTAL DEPTH:

LABORATORY						309													
ANALYTICAL TECHNIQUE						Fire Assay													
DETECTION LIMIT						0·008													
Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au													COMMENTS
64201	½ core	24·0	25·0		1·0	x													
02	"	25·0	26·0		1·0	x													
03	"	26·0	27·0		1·0	x													
04	"	27·0	28·0		1·0	x													
05	"	28·0	29·0		1·0	x													
06	"	29·0	30·0		1·0	x													
07	"	30·0	31·0		1·0	x													
08	"	31·0	32·0		1·0	0·017													
09	"	32·0	33·0		1·0	0·017													
64210	"	33·0	34·0		1·0	x													
11	"	34·0	35·0		1·0	0·017													
12	"	35·0	36·0		1·0	0·008													
13	"	36·0	37·0		1·0	x													
14	"	37·0	38·0		1·0	0·032													
15	"	38·0	39·0		1·0	0·017													
16	"	39·0	40·0		1·0	0·032													
17	"	40·0	41·0		1·0	0·025													
18	"	41·0	42·0		1·0	0·008													
19	"	42·0	43·0		1·0	0·125													Calcite-quartz-epidote-fluorite-pyrite-chalcopyrite ± pyrrhotite ± arsenopyrite veins
64220	"	43·0	44·0		1·0	0·258													
21	"	44·0	45·0		1·0	0·217													
64222	"	45·0	46·0		1·0	0·082													
64223	½ core	78·0	79·0		1·0	0·017													
24	"	79·0	80·0		1·0	0·017													
25	"	80·0	81·0		1·0	0·017													
64226	"	81·0	82·0		1·0	0·067													

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. STP 218
SHEET No. 2.

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au													COMMENTS
64227	½ core	82·0	83·0		1·0	0·108													Carbonate-fluorite-epidote-pyrite-chalcopyrite-arsenopyrite veins.
28	"	83·0	84·0		1·0	0·133													
29	"	84·0	85·0		1·0	0·017													
64230	"	85·0	86·0		1·0	0·017													
31	"	86·0	87·0		1·0	x													
32	"	87·0	88·0		1·0	x													
33	"	88·0	89·0		1·0	x													
34	"	89·0	90·0		1·0	x													
35	"	90·0	91·0		1·0	0·008													
36	"	91·0	92·0		1·0	x													
37	"	92·0	93·0		1·0	x													
38	"	93·0	94·0		1·0	x													
39	"	94·0	95·0		1·0	0·008													
64240	"	95·0	96·0		1·0	x													
41	"	96·0	97·0		1·0	x													
42	"	97·0	98·0		1·0	x													
43	"	98·0	99·0		1·0	0·017													
44	"	99·0	100·0		1·0	x													
45	"	100·0	101·0		1·0	x													
46	"	101·0	102·0		1·0	0·008													
47	"	102·0	103·0		1·0	x													
64248	"	103·0	104·0		1·0	0·017													
64249	"	104·0	105·0		1·0	0·017													
64250	"	105·0	106·0		1·0	x													
64251	"	106·0	107·0		1·0	x													

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. STP 22D
SHEET No. 1

LABORATORY: Analabs
ANALYTICAL TECHNIQUE: 309 Fire Assay
DETECTION LIMIT: 0.008

GRID CO-ORDS:
A.M.G. CO-ORDS:
COLLAR R.L.:
COLLAR DIP:
AZIMUTH:
TOTAL DEPTH:

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au													COMMENTS
64252	1/4 core	88.1	89.2		1.1	x													
53	1/4 core	89.2	90.2		1.0	0.133													Pyrite veins with locally Po+Gn.
54	1/2 core	90.2	91.4		1.2	0.032													
55	"	91.4	92.5		1.1	0.050													
56	"	92.5	93.6		1.1	0.017													
57	"	93.6	94.7		1.1	0.108													
58	1/4 core	94.7	95.75		1.05	0.300													Veins and dissem. of Py ± cpy
59	1/4 core	95.75	96.0		0.25	0.017													
64260	1/4 core	96.0	96.25		0.25	0.017													
61	1/2 core	96.25	97.1		0.85	x													
62	"	97.1	98.0		0.9	0.017													
63	"	98.0	99.0		1.0	x													
64	"	99.0	100.0		1.0	x													
65	"	100.0	101.1		1.1	0.008													
66	"	101.1	102.2		1.1	0.008													
67	"	102.2	103.3		1.1	x													
68	"	103.3	104.4		1.1	x													
69	"	104.4	105.5		1.1	x													
64270	1/2 core	105.5	106.5		1.0	x													
71	1/4 core	106.5	107.5		1.0	x													
64272	1/4 core	107.5	108.5		1.0	0.032													
64273	1/4 core	137.0	138.0		1.0	x													
74	"	138.0	138.7		0.7	0.050													
75	"	138.7	139.7		1.0	0.017													
76	"	139.7	140.7		1.0	0.017													
64277	1/4 core	140.7	141.7		1.0	0.017													

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ELECTROLYTIC ZINC COMPANY OF A'ASIA LTD.
MINERAL RESOURCES DIVISION — TASMANIA

DIAMOND DRILL CORE GEOCHEMICAL ANALYSES RECORD

HOLE No. STP 220
SHEET No. 2

Sample No.	Sample Type	From	To	Core Rec'd	Sample Length	METAL CONTENT (ppm unless specified) Au												COMMENTS
64278	¼ core	141·7	142·7		1·0	x												
79	¼ core	142·7	143·1		0·4	x												
64280	½ core	143·1	144·1		1·0	x												
81	"	144·1	145·1		1·0	x												
82	"	145·1	146·1		1·0	x												
83	"	146·1	147·1		1·0	x												
84	"	147·1	148·1		1·0	x												
85	½ core	148·1	148·8		0·7	x												
86	½ core	148·8	149·8		1·0	x												
87	½ core	149·8	150·8		1·0	0·032												
88	½ core	150·8	151·2		0·4	x												
89	"	151·2	152·2		1·0	x												
64290	"	152·2	153·2		1·0	x												
91	"	153·2	154·2		1·0	x												
92	"	154·2	155·2		1·0	x												
93	"	155·2	156·2		1·0	x												
64294	½ core	156·2	157·5		1·3	x												
64295	½ core	229·1	230·0		0·9	x												
96	"	230·0	231·0		1·0	x												
64297	½ core	231·0	232·0		1·0	x												
64298	½ core	250·5	251·65		1·05	x												
99	"	251·65	252·7		1·05	x												
64300	"	252·7	253·8		1·1	x												
01	"	253·8	254·8		1·0	x												
02	"	254·8	255·8		1·0	0·017												
03	"	255·8	256·8		1·0	x												
64304	½ core	256·8	257·8		1·0	x												

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APPENDIX 6.

Soil Sample Data Sheets.

Orientation sampling on line 3,260N.

E.Z. Co. of A'Asia Ltd.,
ROSEBERY, Tasmania

GEOCHEMICAL SAMPLE DATA SHEET

PROJECT: E.L. 4/73
LOCALITY: Sterling Valley Costean
GRID NAME: Sterling Valley
NOMINAL GRID AZIMUTH: Grid North = 28°AMG.

MATERIAL: Soil
SAMPLE METHOD: Hand Auger
SAMPLED BY: W. Mogle, D. Hopton, I. Mc.
DATE: 4 Dec. 1984

SIZE FRACTION ANALYSED: Total; -10+40 mesh; -40+80 mesh; -80 mesh.
ANALYSED BY: Analabs
METHOD: Au by Fire Assay; Sb, Ba, W by ARF; rest by A.A.S.

218084

SAMPLE NUMBER	SAMPLE LOCATION DATA: GRID LINE NO.	A.M.G. CO-ORDINATES: GRID EASTING	NORTHING	EASTING	DEPTH (cm)	SAMPLE COMPOSITION DATA: COLOUR	Clay	Sand	Rock Frags.	Organic	SIZE FRACTION	METAL CONTENT (ppm unless specified): Au	Ag	As	Sb	Ba	Bi	Mo	W	Cu	Pb	Zn	Fe %	Mn
43748	3260N	4650			0-70	OR	95		5		TOTAL	x	x	560	7	1500	10	x	10	50	50	70	10.00	190
											-10+40	0.008	x	580	4	1190	x	x	15	50	50	65	9.95	180
											-40+80	x	x	490	6	240	x	x	15	50	60	70	9.30	170
											-80	I/S	x	520	9	200	10	10	x	50	40	70	8.60	155
43749		4650			0-35	OR	95		5		TOTAL	x	x	670	7	200	10	x	10	45	40	65	9.70	130
											-10+40	x	x	570	9	165	x	x	x	40	35	60	8.25	150
											-40+80	x	x	490	4	160	x	x	15	35	50	55	6.60	120
											-80	x	x	440	7	160	x	x	30	35	20	50	6.35	100
43750		4640			0-30	BROR	75	10	15		TOTAL	x	x	950	4	160	x	x	10	20	35	50	11.50	90
											-10+40	x	x	730	7	120	x	x	x	25	30	45	10.00	90
											-40+80	I/S	x	330	4	140	x	x	35	20	45	45	6.65	80
											-80	I/S	x	390	7	130	x	x	20	20	15	40	6.25	70
43751		4630			0-50	OR	90	5	5		TOTAL	x	x	1000	8	180	10	x	10	35	80	65	11.50	130
											-10+40	x	x	1100	9	190	x	x	x	35	70	60	9.05	125
											-40+80	x	x	800	3	150	x	x	10	30	80	55	8.65	110
											-80	x	x	1000	9	165	x	20	15	30	45	50	27.50	95
43752		4620			0-65	OR	90		10		TOTAL	0.017	x	1700	15	75	30	x	10	95	115	130	18.00	600
											-10+40	0.025	x	2200	15	165	x	x	10	100	125	145	9.55	445
											-40+80	x	x	1700	15	180	10	x	x	95	160	135	18.00	475
											-80	0.017	x	1600	15	165	20	x	x	25	115	130	18.50	620
43753		4610			11-25	BROR	70	15	15		TOTAL	0.017	x	1500	20	190	x	x	x	85	255	460	12.00	1800
											-10+40	0.020	x	2200	20	120	x	x	x	90	270	495	12.50	1800
											-40+80	0.031	x	1600	20	120	x	x	x	85	250	440	11.00	1450
											-80	0.012	x	1500	20	130	x	x	x	80	215	410	9.50	1500
43754		4610			0-60	OR	99	1			TOTAL	0.017	x	6400	55	120	x	x	x	100	430	220	17.50	190
											-10+40	x	x	6500	70	100	x	x	x	100	430	230	17.50	190
											-40+80	I/S	x	5000	40	140	x	x	x	95	365	185	15.00	210
											-80	I/S	x	4600	40	110	x	x	x	85	305	160	14.00	170
43755		4600			0-30	BRDKBR	70	10	15		TOTAL	x	x	1500	30	65	10	x	20	45	130	100	6.85	200
											-10+40	x	x	1700	25	85	x	x	25	50	135	105	6.90	155
											-40+80	I/S	x	1400	I/S	I/S	x	x	I/S	40	130	90	5.60	175
											-80	I/S	x	800	20	125	x	x	30	30	95	65	5.15	145

084

E.Z. Co. of A'Asia Ltd.,
ROSEBERY, Tasmania

GEOCHEMICAL SAMPLE DATA SHEET

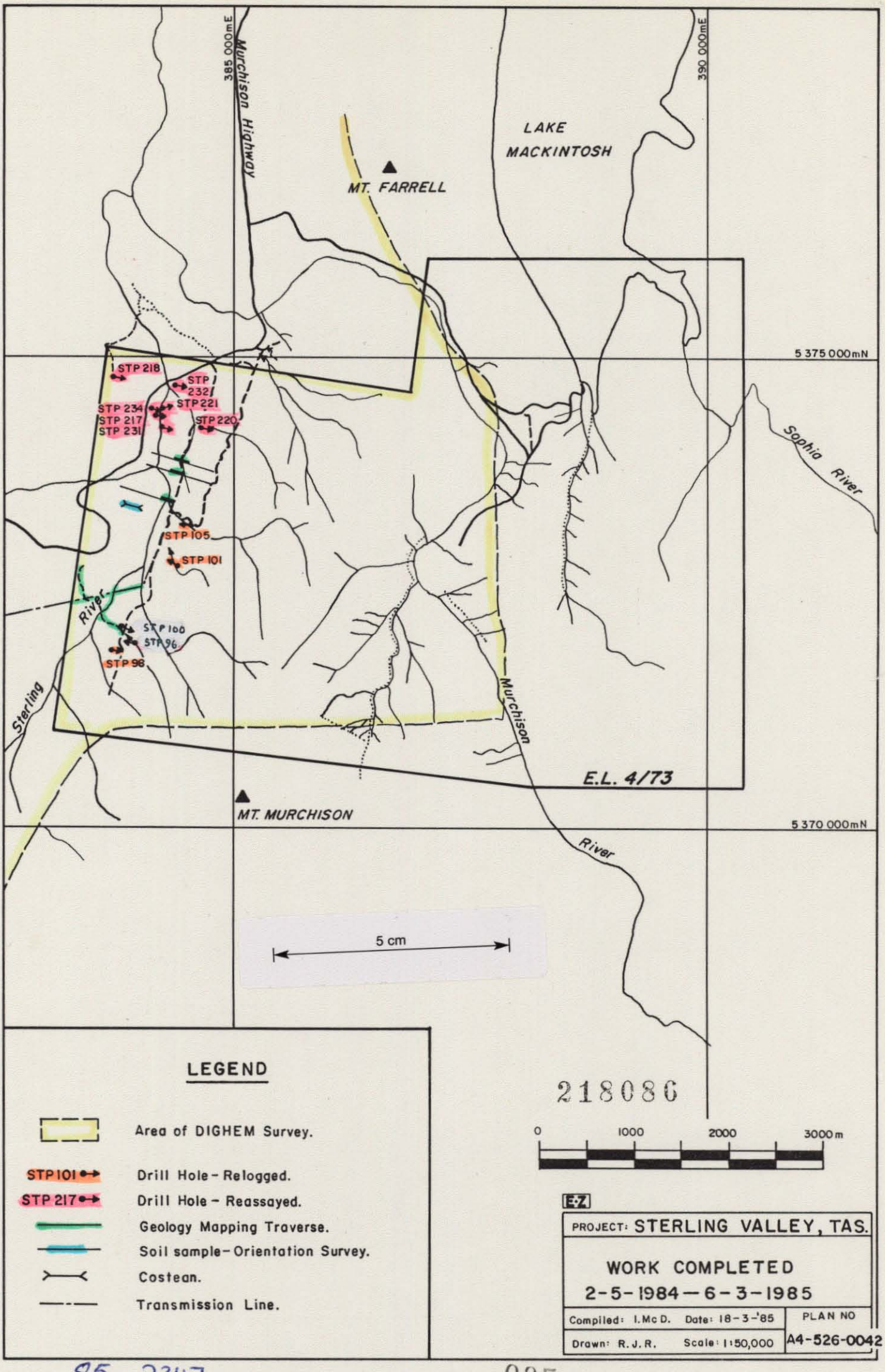
PROJECT: E.L. 4/73
LOCALITY: Sterling Valley Costean
GRID NAME: Sterling Valley
NOMINAL GRID AZIMUTH: Grid North : 28° AMG.

MATERIAL: Soil
SAMPLE METHOD: Hand Auger
SAMPLED BY: W. Mogle, D. Hopton, I. McD.
DATE: 4 Dec. 1984

SIZE FRACTION ANALYSED: Total ; -10+40 mesh ; -40+80 mesh ; -80 mesh.
ANALYSED BY: Analabs
METHOD: Au by Fire Assay ; Sb, Ba, W by ARF ; rest by A.A.S.

218085

SAMPLE NUMBER	SAMPLE LOCATION DATA: GRID LINE NO.	GRID EASTING	A.M.G. CO-ORDINATES: NORTHING	EASTING	DEPTH (cm)	SAMPLE COMPOSITION DATA: COLOUR	Clay	Sand	Rock Frags.	Organic	SIZE FRACTION	METAL CONTENT (ppm unless specified): Au	Ag	As	Sb	Ba	Bi	Mo	W	Cu	Pb	Zn	Fe %	Mn
43756	3260N	4590			0-90	ORBR	85	10	5		TOTAL	0.050	X	8700	15	120	X	X	X	315	275	130	23.00	115
											-10+40	0.050	X	9300	15	120	20	X	X	320	275	120	23.50	110
											-40+80	I/S	X	8900	20	170	10	10	X	305	275	115	22.00	110
											-80	I/S	X	7900	15	210	X	X	X	290	245	115	23.50	120
43757		4590			0.45	BROR	90	5	5		TOTAL	0.025	X	5300	10	45	10	X	X	115	180	100	16.00	170
											-10+40	0.033	X	5400	15	45	10	X	X	150	155	100	16.50	100
											-40+80	I/S	X	4600	10	75	10	X	10	145	175	100	16.00	175
											-80	I/S	X	4700	9	125	X	10	X	145	125	100	15.50	220
43758		4580			1.25	GNOR	70	15	15		TOTAL		X	5400	5	3300	X	X	X	100	110	195	17.50	320
											-10+40		X	6000	8	3200	X	X	X	100	105	190	17.85	315
											-40+80	I/S	0.5	6700	X	3200	X	X	X	95	120	190	17.25	325
											-80		0.5	5200	X	3700	X	X	X	90	90	200	16.75	335
43759		4580			0.60	OR	99	1			TOTAL	0.017	X	21600	8	1000	X	10	15	80	105	75	13.00	100
											-10+40	0.008	X	2200	7	1000	X	X	X	60	90	75	13.00	95
											-40+80	X	X	2300	10	1120	X	X	X	95	110	75	13.00	95
											-80	X	X	2300	6	900	X	20	X	85	90	70	13.00	95


385 000mE
390 000mE
Murchison Highway
LAKE MACKINTOSH
MT. FARRELL
5 375 000mN
STP 218
STP 232
STP 221
STP 234
STP 217
STP 231
STP 220
Sophia River
STP 105
STP 101
River
STP 100
STP 96
STP 98
Sterling
Murchison
E.L. 4/73
MT. MURCHISON
5 370 000mN
River
5 cm
LEGEND
Area of DIGHEM Survey.
STP 101 Drill Hole - Relogged.
STP 217 Drill Hole - Reassayed.
Geology Mapping Traverse.
Soil sample - Orientation Survey.
Costean.
Transmission Line.
218086
0 1000 2000 3000 m
E·Z
PROJECT: STERLING VALLEY, TAS.
WORK COMPLETED
2-5-1984 — 6-3-1985
Compiled: I. Mc D. Date: 18-3-'85
Drawn: R. J. R. Scale: 1:50,000
PLAN NO
A4-526-0042

85-2347

085

086

A0-526-0037

- LEGEND -

Anomaly Grade — 2/B — Interpretive Symbol

Identification No.

Anomaly Grade	Conductance Range (mhos)
4	20-49
3	10-19
2	5-9
1	< 5.
X	Indeterminate

Interpretive Symbol	Conductor ("Model")
B	Bedrock conductor.
S	Conductive cover. ("horizontal thin sheet").
H	Broad conductive rock unit, deep conductive weathering, thick conductive cover. ("half space").
E	Edge of broad conductor. ("edge of half space").
L	Culture, e.g. power line, building, fence.

N.B. - All contours are measured in Ω m

E.L. 4/73
E.L. 1/82

SCALE
0 100 200 300 400 500m
5 cm

EZ

ELECTROLYTIC ZINC COMPANY OF AUSTRALASIA LIMITED
MINERAL RESOURCES DIVISION

PROJECT STERLING VALLEY

218087

RESISTIVITY CONTOURS
and
E.M. ANOMALIES 086

Compiled I.McD.	Date 2-1-'85	Scale 1:5000	
A.M.G.	Lat/side	Long/side	
Drawn R.J.R.	File No.	PLAN NO. A0-526-0037	

85-2347

087

A0-526-0036

ELECTROLYTIC ZINC COMPANY OF AUSTRALASIA LIMITED
MINERAL RESOURCES DIVISION

PROJECT STERLING VALLEY

218088

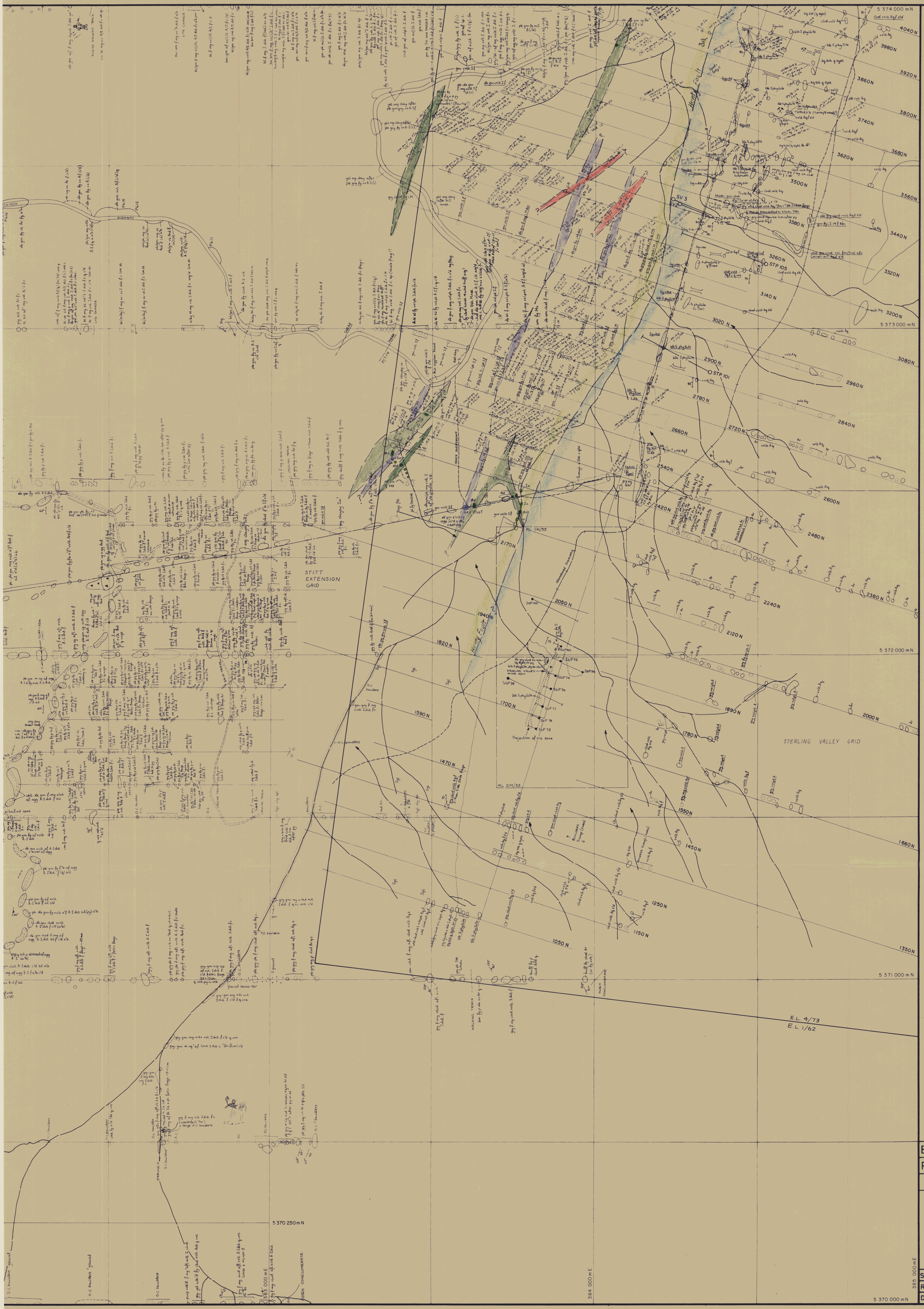
AERO MAGNETICS

087

Compiled I. Mc D.	Date 7-11-'84	Scale 1:5000
AMG.	Latitude.	Longitude.
Drawn R.J.R.	File No.	PLAN NO:

A0-526-0036

85-2347



NOTE 1 (Sterling Valley J.V. E.L. 4/73) Underlined rock descriptions = W.S.F. opinion from tramping around out bush (also A.J.M. later)

NOTE 2 Source of further information, Asarco (Aust.) Pty Ltd., E.L. 4/73 Sterling Valley Progress Report to June 1974 by R.G. Barker

N.B.- CORRESPONDING SHEET, READ-ROSEBERY- AO-520-0108

ELECTROLYTIC ZINC CO. OF A'ASIA. LTD.

PROJECT: MT. BLACK E.L.1/62 | TAS.

STERLING VALLEY J.V. E.L.4/73

GEOLOGY

218089

5 cm

088

SCALE: 1:5000	Survey: P.K., W.F., A.M., G.I., R.S.	Revised: 19-3-1985
Reference:	Date: February 1979	REF NO.
Drawn: R.P.T.	Checked:	AO-526-0043

85-2347

EZ PLAN CABINET TRIM LINE

A3-526-0041

383 500mE

384 000mE

372 500N

5 372 500mN

E.L. 1/62

E.L. 4/73

Sterling River

H.E.C. POWERLINE

5000E

STERLING VALLEY grid baseline

2660N
2600N
2540N
2480N
2420N
2360N
2240N
2170N
2050N
1940N
1820N

af Ia
iMd
bd'd fg VSsilt
af lap Ia
cg lap,af,tIa shale liths
pk/gm lap, tt Ia
ox af Ia
ox af Ia qtz vns
flt bdd fg VSsilt
af agg Iaf
p fbx'd Il c'd
flt Sqtz (erratic)
pIaf
clv'd p Iaf
bx'd Ia sil'd
clv'd pIaf
pIaf qtz vns
wht qtz vns
Bx'd VSsst sil'd
py q vns
ox af agg Ia
p xlv,af,tIl
ox p Iaf(q)
flt gos q vns
fg ves Ia
p Adcf f
shr'd Ia c'd
scree
grn stg clv'd Il c'd
clv'd p Ia f c'd
shr'd Ia c'd (tr py + asp)
clv'd p fbx'd Ia f c'd
clv'd Ia(f) c'd
float? gry stg shr'd Sqtz
float? gry stg shr'd mylonitised Sqtz
o/c clv'd p Ia c'd
VS bx qtz, felsic volc liths
float?
stg clv'd Sbs
SCH q vns

353° 48°
78° 43°
338° 50°
312° 56°
358° 76°
82° 38°
335° 64°
12° 87°
14° 87°
22° 78°
003° 65°
360° 82°
105° 71°
008° 84°
002° 85°
10° 78°
356° 77°

5 cm

KEY ?

SCALE
0 50 100 150 200m

218090

EZ

ELECTROLYTIC ZINC COMPANY OF AUSTRALASIA LIMITED
MINERAL RESOURCES DIVISION

PROJECT STERLING VALLEY E.L. 4/73

MAPPING TRAVERSES
S.W. E.L. 4/73

085A

Compiled I. McD.	Date 14-3-'85	Scale 1:2500	
Latitude	Longitude	PLAN NO. A3-526-0041	
Drawn R.J.R.	File No.		