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**REPORT ON THE  
1992 EXPLORATION PROGRAM  
ON THE  
DOT COPPER PROJECT**

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N.T.S. 92I/7W

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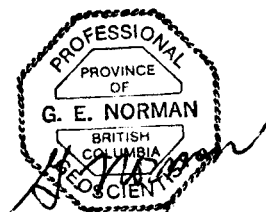
For

**ZAPPA RESOURCES LTD.**  
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By **GEOLOGICAL BRANCH**  
**ASSESSMENT REPORT**  
George Norman, P. Geo.  
Norman Geological

January, 1993

**22,839**



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## 1.0 SUMMARY AND CONCLUSIONS

The Dot Project, consisting of 44 claim units comprising 1,100 ha. is located 15 km. southeast of the Highland Valley porphyry copper district in southern British Columbia. The claims lie 25 km. northwest of Merrit, B.C. at 50° 20' North latitude and 120° 51' West Longitude, NTS 92I/7W (See Figure 1 for location). Good road access and low snowfall allows exploration to proceed on a year-round basis.

The property is underlain by the Guichon Batholith which is host to numerous porphyry copper deposits including Lornex and Valley Copper.

The Dot copper deposit lies within a northwest trending zone of altered intrusive containing disseminated, fracture and vein controlled copper mineralization. The copper mineralization and related I.P. geophysical anomalies occur within an area measuring 150-300 metres wide by 800 metres in length.

Zappa Resources, may earn 100% interest in the Dot claims through an option agreement signed in November, 1992 with the owner of the claims, Larry Ovington. American Bullion Minerals Ltd., have the right to a 50% joint venture interest upon Zappa completing the first \$2.0 million expenditure.

Zappa Resources Ltd. operated and funded the 1992 Exploration Program expending a total of \$51,757 during the period of November 1 to January 31, 1993.

Zappa Resources Ltd. undertook a program of drilling with the completion of 6 reverse circulation drill holes totalling 638.5 metres. The program tested the copper zone over a 255 meter strike length and to a depth of 100 meters. All 6 holes were successful in intersecting copper mineralization. Some of the significant copper grade intercepts of the program include: 73.15 metres of 0.33% Cu in 92RCD-02, 76.2 metres of 0.91% Cu in 92RCD-03 and 64.01 metres of 0.37% Cu in 92RCD-04.

Surface trenches and drill holes have traced copper mineralization for a strike length of 270 metres, a depth of 100 metres to a width of 55 metres. The zone remains open along strike and to depth. Previous geophysical surveys have defined a large northwest trending I.P. anomaly that extends the target zone for an additional 500 metres southeast of the drill tested area. The I.P. anomaly expands from a width of 150 metres within the known copper zone to 300-400 metres in the area to the southeast.

The main copper zone is hosted within structurally controlled intensely hematite altered granodiorite located proximal to a north-northeasterly and northwesterly structural intersection. The zone contains disseminated, vein and fracture controlled native

✓ sulphide and oxide copper mineralization. Primary copper minerals include bornite and minor chalcopyrite with secondary malachite, chalcocite and native copper.

Drill holes have outlined a preliminary geological resource of 2.93 million tonnes grading 0.5% Cu. The primary target for additional reserves occurs within an expanding I.P. chargeability Anomaly located southeast of the drill tested area.

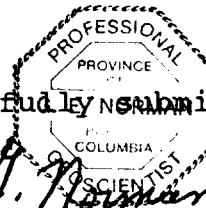
Preliminary metallurgical testing of copper mineralization by Westcoast Biotech of Vancouver, B.C. suggests acceptable levels of acid consumption can be expected in a heap leaching circuit. Additional testing has been recommended by Westcoast to test acid consumption with coarser material (5 cm.) as well as biological leach tests.

Westcoast Biotech anticipates that a 4 month leaching cycle of copper with a 70% extraction rate could produce 8.25 million pound of copper for each million tonnes mined and treated. The cost of solvent extraction and electrowinning is expected to be less than \$0.40 per pound. A mining operation of 1,400 t/d ore containing 0.5% copper would produce 10,000 pounds of cathode copper per day.

In summary, drilling programs have outlined a geological resource of 2.93 million tons grading 0.5% copper on the Dot Property. Potential to increase reserves lies within a large I.P. Anomaly, to the southeast. Preliminary metallurgical testing suggests that the Dot mineralization, which is a combination of oxide/sulphide and native copper, is amenable to heap leaching. If successful, cathode copper will be produced by solvent extraction/electrowinning (SX-EW) technology which eliminates conventional high cost milling and smelting.

Further drilling (500 metres) of three holes totalling 500 metres within the expanding I.P chargeability anomaly is recommended along with continued metallurgical test work to assure the leacheability of the copper mineralization. The recommended program is estimated to cost \$56,000.00.

Respectfully submitted.



G.E. Norman, P. Geo.  
January 21, 1992

## 2.0 INTRODUCTION

The 1992 Dot Exploration Program focused on confirmation drilling of the previously discovered main copper zone. The program was funded and operated by Zappa Resources Ltd. The field program was conducted during November and December, 1992.

## 2.1 LOCATION AND ACCESS

The Dot Property is located in south central British Columbia, approximately 25 kilometres northwest of Merrit, B.C., latitude 50° 20'N, longitude 120° 51'W, NTS 92I/7W. Access is via highway #8, 7 kilometres northwesterly from Merrit to Lower Nicola, then by good pavement 6 kilometres northerly to the Craigmont Mine site at which point the "Aberdeen Mine Road" gives way to an upgraded gravel road. At kilometre "marker 7" northwest from Craigmont, access to claims is gained by travelling northerly an additional 5 kilometres via a unmaintained dirt road.

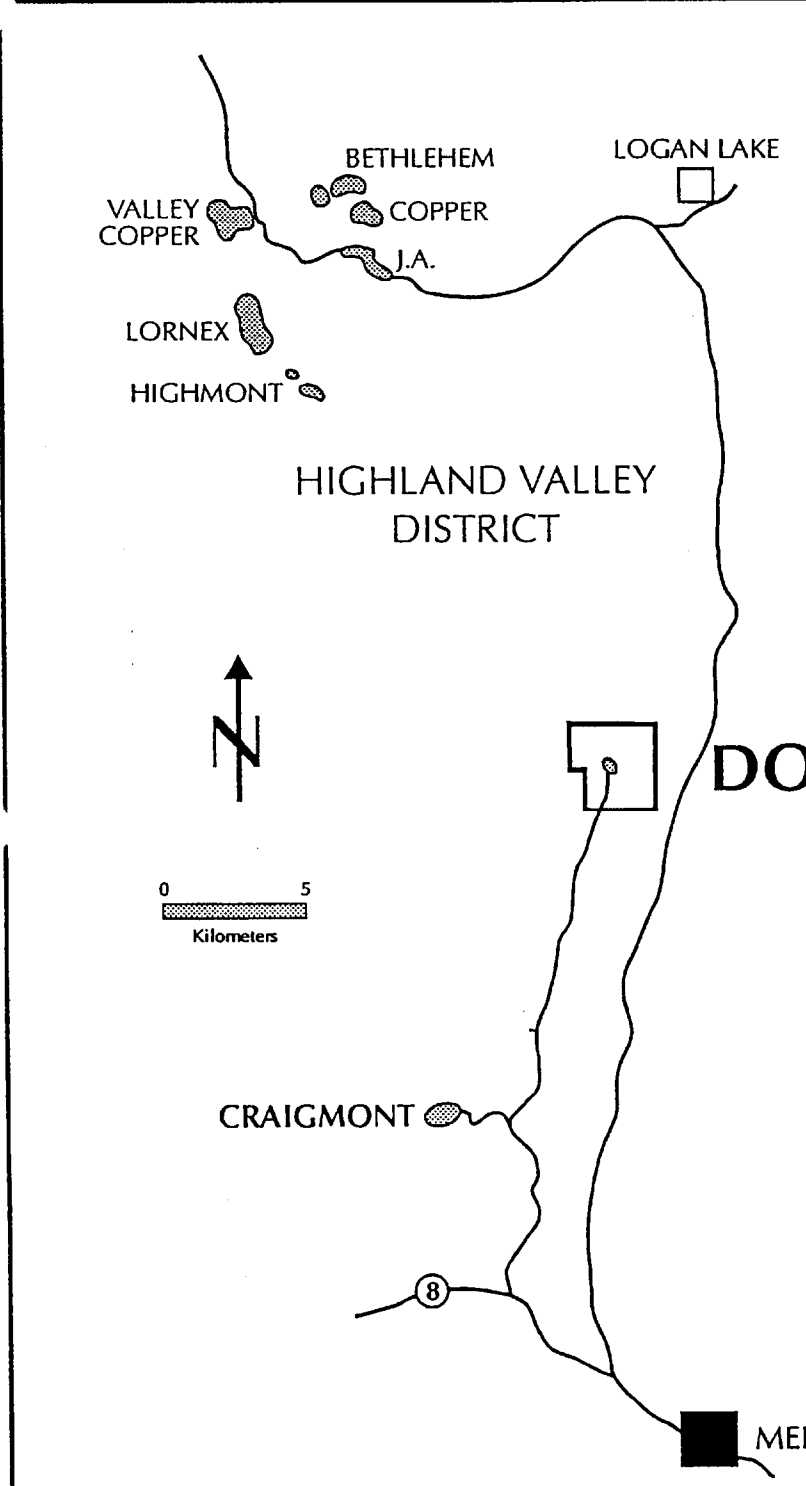
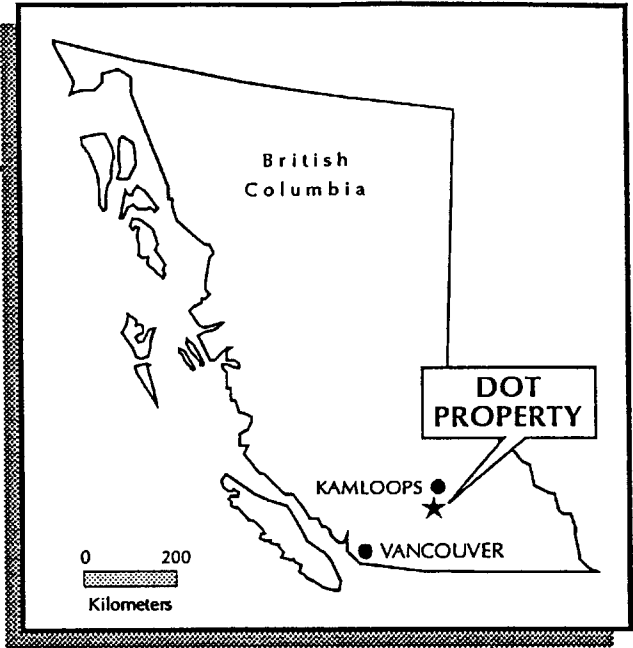
## 2.2 PHYSIOGRAPHY

The area is gently undulating and moderated vegetated with pine conifers. Elevations range from 1000 metres in the southern portion of the property to 1375 metres in the northern extremities. The property is largely overburden covered terrain with variable thickness (up to 34 metres) of glacial drift. An esker ridge, located in the immediate vicinity of the drill area gives local relief of 10-15 metres. A tributary to Broom Creek traverses the southwestern portion of the claims.

## 2.3 CLAIM STATUS

The Dot property consists of 28 mineral claims, the DOT I to DOT X claims and DOT 11 to DOT 28 claims comprising a total of 44 units in 1100 hectares. The claims are currently in good standing until August/November 1993. The registered owner of the claims is Larry Ovington of Kamloops, B.C.. Refer to Table I for record numbers and specific expiry dates.





**DOT PROPERTY**

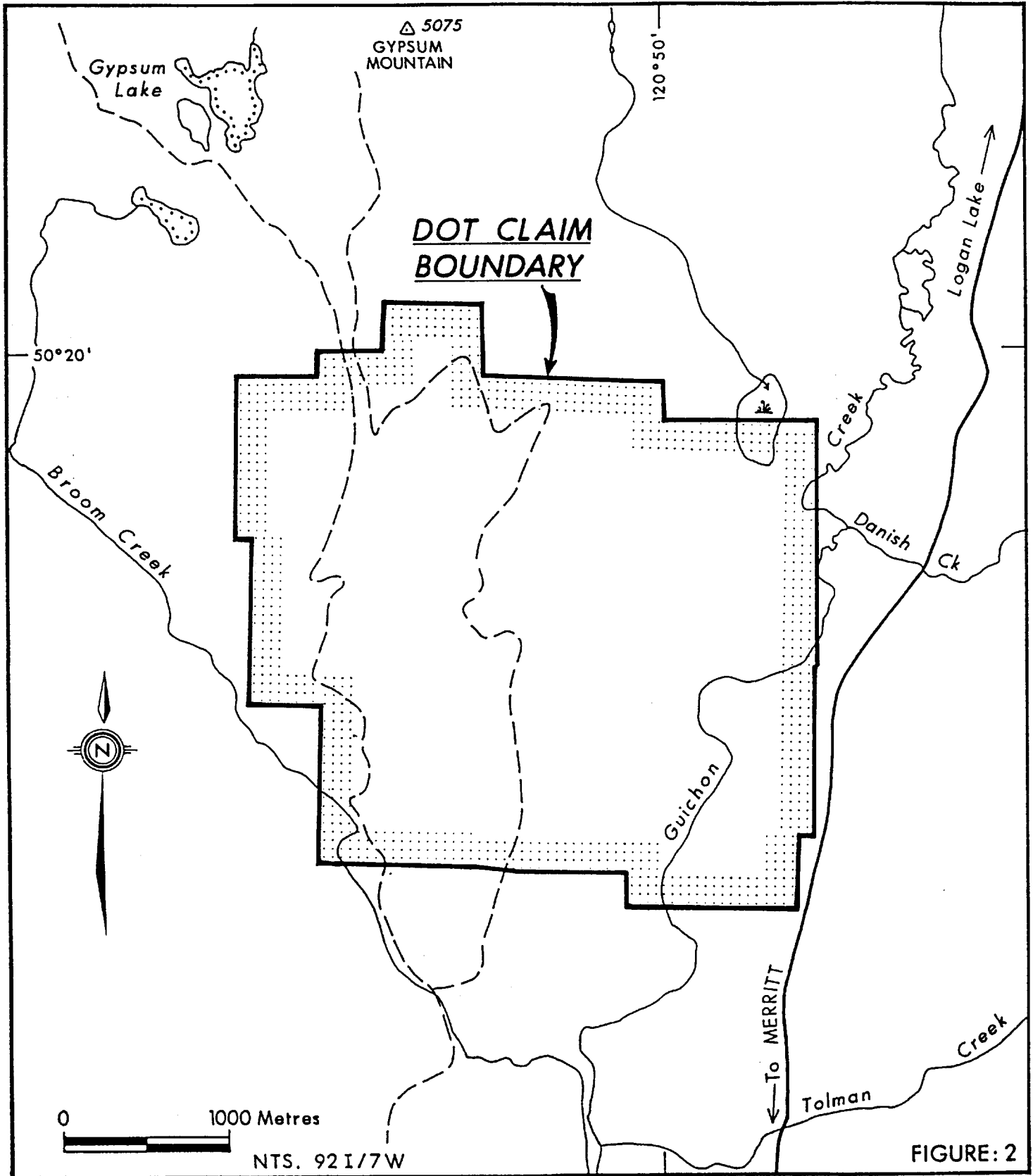


# DOT COPPER PROJECT

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Figure : 1

# DOT PROPERTY CLAIM MAP



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TABLE I  
MINERAL CLAIM STATUS - DOT PROPERTY

CLAIM NAME	RECORD NUMBER	NO. of UNITS	RECORD YEAR	EXPIRY DATE	OWNERSHIP
DOT I	312518	12 (4Nx3W)	08/21/92	08/16/93	L. Ovington
DOT II	312519	6 (35x2W)	08/21/92	08/18/92	"
DOT III	312733	1 (2 POST)	09/1/92	08/24/92	"
DOT IV	312734	1 "	09/1/92	08/24/92	"
DOT V	312735	1 "	09/1/92	08/24/92	"
DOT VI	312736	1 "	09/1/92	08/24/92	"
DOT VII	312737	1 "	09/1/92	08/31/93	"
DOT VIII	312738	1 "	09/1/92	08/31/93	"
DOT IX	312739	1 "	09/1/92	08/31/93	"
DOT X	312740	1 "	09/1/92	08/31/93	"
DOT 11	314782	1 "	11/20/92	11/17/93	"
DOT 12	314783	1 "	11/20/92	11/17/93	"
DOT 13	314784	1 "	11/20/92	11/17/93	"
DOT 14	314785	1 "	11/20/92	11/17/93	"
DOT 15	314786	1 "	11/20/92	11/17/93	"
DOT 16	314787	1 "	11/20/92	11/17/93	"
DOT 17	314788	1 "	11/20/92	11/18/93	"
DOT 18	314789	1 "	11/20/92	11/18/93	"
DOT 19	314790	1 "	11/20/92	11/18/93	"
DOT 20	314791	1 "	11/20/92	11/18/93	"
DOT 21	314792	1 "	11/20/92	11/18/93	"
DOT 22	314793	1 "	11/20/92	11/18/93	"
DOT 23	314794	1 "	11/20/92	11/17/93	"
DOT 24	314795	1 "	11/20/92	11/17/93	"
DOT 25	314796	1 "	11/20/92	11/17/93	"
DOT 26	314797	1 "	11/20/92	11/17/93	"
DOT 27	314798	1 "	11/20/92	11/18/93	"
DOT 28	314799	1 "	11/20/92	11/18/93	"

## 2.4 HISTORY

The Dot property was previously known as the Vimy prospect which was located around the turn of the century. A minor amount of high grade copper ore was mined prior to 1927 from approximately 200 feet of underground workings. During 1956-57 Kennco Explorations completed various surveys including trenching and 3,652 metres of drilling in 30 holes.

From 1965 to 1981 exploration programs were completed on prior claims which are now covered by the present Dot property. This work is summarized below:

1. 1960-67 Chattaway - line cutting, trenching, approximately 50 diamond drill holes (3,658m)
2. 1960-67 Bralorne - Pioneer Mines - line cutting and magnetic surveys, trenching, geochemical surveys, 7 diamond drill holes (341 metres) and 20 percussion drill holes.
- 3.0 1970 Asarco - trenching, 148 percussion holes (5,166 m on a 610 metre grid)
- 4.0 1972 Aselo Industries - Induced Polarization survey
- 5.0 1979-81 Lawrence Mining - induced polarization, 30 BQ diamond drill holes (5,387 metres) and 30 percussion holes (2,288 metres)
- 6.0 1982 Lawrence Mining - 3 diamond drill holes of which the location, results and total metrage is unknown.

Drilling prior to 1981 indicated that 903,000 tons of 0.35% copper existed within the main copper zone on the Dot prospect. (Information taken from Assessment Report #9699).

## 2.5 1992 EXPLORATION PROGRAM

A total of 6 reverse circulation drill holes in 638.5 metres were completed on the Dot Property during November and December 1992. The drilling was confined to the previously known copper mineralization and was designed to confirm the continuity and grade of previously reported copper intersections.

### 3.0 REGIONAL SETTING

The Dot property occurs within the Highland Valley porphyry copper district within the Intermontane belt of southern British Columbia (Refer to Figure 3, General Geology). Copper deposits occur within the late Upper Triassic Guichon Creek batholith which has a surface area of approximately 1000 km<sup>2</sup> and is a semi-concordant domal body that is elongated slightly west of north. The batholith intrudes sedimentary and volcanic strata of the Permian Cache Creek and late Triassic Nicola groups and is unconformably overlain by sedimentary and volcanic strata ranging in age from Early Jurassic to Middle Tertiary.

The oldest rocks exposed adjacent to the Guichon Creek batholith are Cache Creek Group sedimentary and volcanic rocks of Permo-Carboniferous age that outcrop west of the batholith and west of the Thompson River. The Cache Creek rocks appear to be deposited in two volcanic arcs, separated by a belt of oceanic rocks (Monger, 1972). These rocks include argillite, chert, conglomerate, greywache, tuff, quartzite and greenstone. Metamorphic rocks adjacent to the batholith include Nicola Group and include hornblende - plagioclase gneiss, schist, quartzite and hornfels that occur in a metamorphic halo up to 500 metres in width.

Upper Triassic Nicola Group rocks which envelope the batholith and locally form roof pendants, consist mainly of volcanic flows (basalt and basaltic andesite) and breccias in the north and sedimentary rock (chert, siltstone, sandstone, greywache, limestone and volcanic conglomerates) at the east, south and west edges.

The Guichon batholith is divisible into various concentric phases of related intrusions which have sharp contacts locally, but are generally gradational. The youngest phases form the core of the system although extensive K/Ar dating (Northcote, 1969) has shown that all phases report an approximate age of 198 +/- 8 m.y.

Copper mineralization is structurally controlled during emplacement within the Highland Valley deposits and is associated with copper enriched magma phases of the Guichon batholith. The Bethlehem and Bethsaida phases contain a much higher background copper value than other phases of the batholith. The Highland Valley ore deposits contain approximately 10 million tons of copper metal of which approximately 2 billion pounds of copper has been mined to date. The Dot property lies within the Guichon variety outer Highland Valley phase of the Guichon batholith.

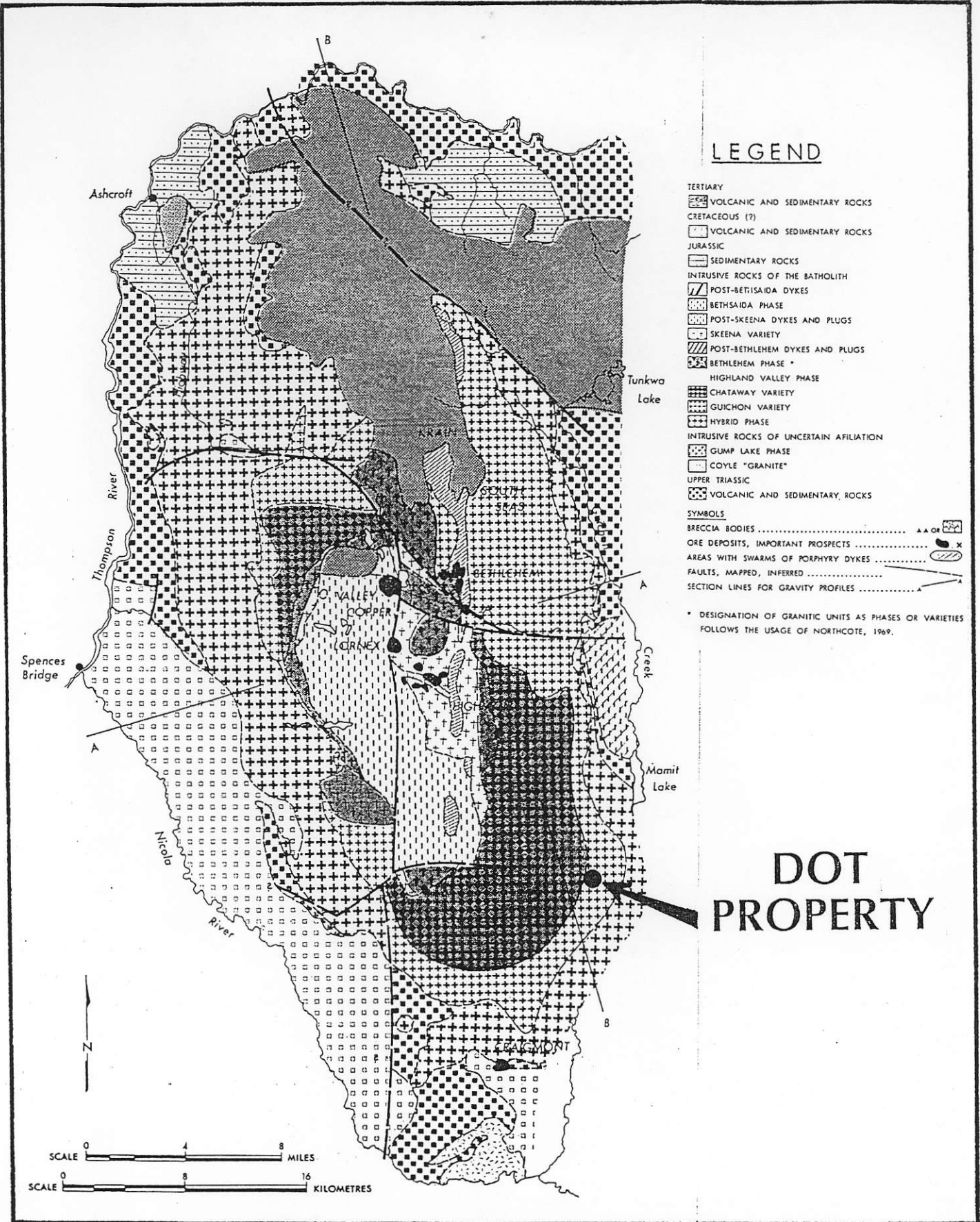


Figure 3 - GENERAL GEOLOGY

#### 4.0 PROPERTY GEOLOGY

The Dot property is located within the eastern portion of the Upper Triassic Guichon Creek batholith. The property is underlain by the Guichon variety Highland Valley phase intrusive rock comprised of fine to medium grained hornblende monzodiorite to granodiorite. Outcrops of a coarser grained granodiorite possibly Chataway variety and younger porphyry intrusives are also noted in the literature. Refer to Appendix I Vancouver Petrographics report by J. Payne for detailed descriptions of several intrusive samples taken from the Dot copper zone.

#### 4.1 MINERALIZATION, ALTERATION AND STRUCTURE

The copper zone within the Dot claims is found within a northwest trending structural zone of altered granodiorite containing disseminated, fracture controlled and vein hosted native copper, sulphide copper (chalcocite, bornite and chalcopyrite and oxide copper (predominately malachite). Better copper grades appear to exist near a north-north easterly trending cross structure which transects the area between 92RCD-03 and 92RCD-04; and 92RCD-05 and 81-11. The surface expression of the fault is exhibiting by a line of poorly drained swamps.

The main copper zone, explored by surface trenches and drill holes has been traced for an approximate length of 270 metres with a width of up to 55 metres and a depth of 100 metres. The deposit remains open along strike and to depth. The zone as depicted in cross sections, Figures 6 to 17, dips steeply at 80° to the northeast and appears in most instances to be fault bound on the hanging wall with unaltered and unmineralized granodiorite.

The main copper zone contains strong hematite alteration which produces a salmon-reddish coloration to the granodiorite. Main copper minerals present are malachite, bornite, native copper, chalcocite and minor chalcopyrite. Bornite and chalcopyrite are hypogene sulphides while malachite, chalcocite and native copper are believed to be supergene. Chalcocite occurs as coatings on bornite and chalcopyrite. The secondary nature of native copper has not been substantiated; although its disseminated nature and spatial proximity to chalcocite suggest a movement from one location to another (personal communication with J. Payne - Vancouver Petrographics). Native copper also occurs as thin platy fracture fillings and in quartz veins. Native copper occurring with quartz veins may either be remobilized or related to a second pulse of mineralization. Specular hematite is present in variable amounts and is usually abundant in high grade copper zones. Quartz and calcite veining is present sporadically throughout the zone. Spectacular high copper grades appear to be associated with silicification and quartz veining as seen in the interval 39.62 to 42.67 metres in hole 92RCD-03 that graded 6 to 10% copper.

TABLE II  
1992 DOT PROJECT  
DRILL HOLE TECHNICAL DATA

Reverse Circ. Drill Hole	Northing (m)	Easting (m)	Azimuth Degrees	Dip Degrees	Total Length(m)	Horz. Proj.(m)	Vertical Proj.(m)
92RCD-01	25+57	20+26	234	-50	120.4	77.39	92.23
92RCD-02	25+22	20+47E	230	-50	100.58	64.65	77.05
92RCD-03	24+81	20+81	230	-50	100.58	64.65	77.05
92RCD-04	24+43	21+20	230	-50	100.58	64.65	77.05
92RCD-05	23+77.5	20+96	50	-50	100.58	64.65	77.05
92RCD-06	23+69.5	21+92	230	-50	115.82	74.44	88.72
					<u>638.5 (m)</u>		



TABLE III  
SUMMARY OF 1992 DOT PROJECT DRILL RESULTS

Reverse Circ. Drill Hole No.	From (m)	To (m)	Width (m)	Native Copper %	Nonsulphide Copper		Total Copper %
					%Cu	% of Total	
92RCD-01	27.43	50.29	22.9		0.06	37.5	0.16
92RDC-02	10.67	83.82	73.15				0.33
	10.67	35.05	24.38				0.24
	35.05	80.77	45.72		0.08	20	0.40
	10.67	21.34	10.67		0.35	82	0.43
	21.34	35.05	13.72	NOT			0.09
	35.05	54.86	19.81	ASSAYED	0.10	15	0.68
	54.86	80.77	25.9				0.18
	24.38	100.58	76.2		0.20	22	0.91
92RDC-03	25.91	83.82	57.91		0.24	12	1.16
	83.82	100.58	16.76		0.06	46	0.13
	35.05	67.06	32.01	0.14	0.30	16	1.85
	35.05	48.77	13.72	0.32	0.53	17	3.05
	35.05	99.06	64.01		0.17	46	0.37
92RCD-04	36.58	45.72	9.14		0.57	54	1.05
	45.72	67.06	21.34		0.10	62	0.16
	67.06	99.06	32.0	NOT	0.11	33	0.33
	22.86	71.63	48.77	ASSAYED	0.16	76	0.21
92RCD-05	22.86	57.91	35.1		0.21	91	0.23
	57.91	71.63	13.72				0.16
	22.86	33.53	10.67		0.31	97	0.32
	33.53	57.91	24.38		0.16	89	0.18
92RCD-06	45.72	48.77	3.05		0.22	100	0.22
	82.30	99.06	16.76		0.20	77	0.26

The main copper zone also contains moderate to weak chloritic alteration of mafics and feldspars which becomes stronger near the narrowing ends of the zone. Sericite alteration is weak and sporadic. Thin section work by John Payne (Appendix I) suggests that K-feldspar is both secondary and primary. The reddish hematite coloration of the granodiorite in hand specimen hinders identification of K-feldspar.

Bulldozer trenching has exposed copper mineralization consisting of malachite and bornite in altered granodiorite at the northwestern end of the deposit. Surface sampling by Zappa Resources returned assays of 1.12% Cu over 32.9 metres. Drilling in this area via 92RCD-01 and 81-15 returned assays of only 0.16 over 22.9 metres and 0.37 over 14.0 metres. As section D-D', Figure 5 depicts, surface enrichment and or structural complexity has increased grade and width of the zone at surface.

Please refer to Section 5.0, Reverse Circulation Drilling Program, for results of 1992 program and Appendix II for a list of previous drill hole intercepts.

## 5.0 REVERSE CIRCULATION DRILLING 1992 PROGRAM

During the period November 27 to December 7, 1992, a total of 638.5 meters in 6 reverse circulation drill holes were completed within the known copper zone on the Dot property. Refer to Figure 4 for drill hole locations, Table II for a summary of drill hole technical data and Table III for a summary of assay results, Appendix III for Geologic Logs and Sample Drill Ledgers, and Appendix IV for Min En analytical results. The drilling was contracted to Dateline Contracting Ltd. of Kelowna, B.C. who supplied a track mounted self contained "Explorer" reverse circulation rig utilizing a 350 psi - 540 CFM compressor and 11.4 cm. diameter down-hole hammer. Conventional reverse circulation equipment was utilized which uses compressed air to force rock chips up the pipe centre via an interchange. All drill hole collars were located on previously built roads therefore minimizing any surface disturbance. No serious drilling problems were encountered. A minimal amount of water was used for wet drilling and was obtained from local small ponds. The average drill penetration rate for the 11 day period was 58 metres/day over an average 9 hour shift.

Drill moves were easily accomplished with the track-mounted rig and generally were completed within 3 hours.

Samples were collected beneath a cyclone and split using a Jones Splitter. Dry samples were collected in plastic bags while wet samples were collected in perforated Hubco bags to allow water drainage. A split sample of approximately 5 kilograms was shipped to Min-En Labs Ltd., North Vancouver, B.C. for assay purposes. Drill intersections, geologically recognized as copper bearing zones, were assayed for total copper and for nonsulphide copper. A limited number of samples were assayed for native copper. Drill intersections with no recognizable visual copper bearing minerals were geochemically analyzed for copper. See Appendix V for a description of analytical techniques.

### RESULTS

A summary of the results of the 6 hole program is given below:

Drill hole 92 RCD-01 was drilled at the north end of the known copper mineralization to test beneath an area where surface trench samples returned an average of 1.12%Cu over 32.9 metres (samples taken by Zappa personnel during initial property exam) [Refer to location map Figure 3, and sections Figures 5 and 6]. 92RCD-01 tested ground previously tested by drill hole 81-5 that assayed 0.37% Cu over 14.0 metres. Hole 92RCD-01 intersected granodiorite throughout that contained a copper zone of native copper with minor bornite. The zone averaged 0.16% total copper over 22.9 metres from 27.43 to 50.29 metres. The

granodiorite within the zone is medium grained and stained by a pink-reddish coloration due to hematite alteration. Weak sericitization is also present within the zone. A strong clay altered zone from 26 to 28 metres is restricted to a fault zone that appears to bound the copper zone to the north.

Drill hole 92RCD-02 was drilled 45 metres to the southeast of hole 92RCD-01 (Refer to Figure 3 for location and Figures 7 and 8 sectional views). The hole was successful in intersecting a thick copper bearing zone approximately 55 metres in true thickness. The zone averages 0.33% Cu over 73.15 metres from 10.67 to 83.82 metres including 45.72 metres averaging 0.40% Cu from 35.05 to 80.77 metres. The zone contains bornite, native copper, malachite and cuprite and is strongly to moderately hematite stained and weak to strongly chloritized. Minor quartz veining was noted from 62.0-67.0 metres. A clay fault zone is again conspicuous near the upper end of the zone.

Drill hole 92RCD-03 was collared 53 metres southeast of 92RCD-02 along the strike of the copper zone. This hole encountered a thick high grade section of copper mineralization. The copper zone averages 0.91% Cu over 76.2 metres (55 metres true thickness) from 24.38 to 100.58 metres with 32.0 metres averaging 1.85% Cu (Refer to Figures 9 and 10). The hole was terminated in copper mineralization (0.106%Cu). The 76.2 metre thick zone averages 0.21% nonsulphide or approximately 22% of the total copper. Main copper minerals are bornite, native copper, chalcocite and malachite. The zone is contained within a very strongly hematite stained salmon-reddish coloured granodiorite. The zone also contains specular hematite and weak epidote alteration. Previous drilling within this area returned assays of 0.62% Cu over 53.9m in drill hole 81-1 and 22.0 metres of 0.36% Cu and 32.3 metres of 0.26% Cu in hole 81-11.

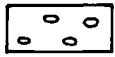
Drill Hole 92RCD-04 is located approximately 55 metres southeasterly from 92RD-03 along the strike of the copper zone. Refer to Figures 3, 11 and 12 for location and sectional views. This hole was successful in intersecting a thick copper zone that assayed 0.37% Cu over 64.01 metres from 35.05 to 99.06 metres including 9.14 metres of 1.05% Cu from 36.58 to 45.72 metres. The overall zone averages 0.17% nonsulphide copper or 45% of the total copper. Copper minerals include malachite, native copper and bornite. The zone is weakly chloritized and strongly hematite stained with some specular hematite noted. A fine grained chloritic rock was intersected from 34.0 to 42 metres and appears to be a fine grained volcanic dyke or xenolith. Previous drilling results in this area include 81-13 with 30.2 metres of 1.32% Cu, 81-18 with 27.4 metres of 2.58% Cu and 81-19 with 20.1 metres of 0.57% Cu.

Drill hole 92RCD-05 was collared 70 metres southwest from 92RCD-04 drilled toward the northeast. (Refer to Figures 13 and 14 for sectional views). 92RCD-05, located adjacent to previous drill hole collars 81-13 and 14, was a 35 metre step out along strike of the copper zone from 92 RCD-04. The hole intersected a section from 22.86 to 71.36 metres averaging 0.21% Cu over 48.77 metres of which 0.16% Cu (76%) is nonsulphide copper. Copper minerals include malachite, native copper, chalcocite and traces of chalcopyrite. Malachite was the dominant copper mineral as shown by the increase in the nonsulphide copper assay. The zone is variably hematite altered and contains an increase in chlorite and epidote alteration. Minor specular hematite was noted.

Drill Hole circulation hole 92 RCD-06 was collared 70 metres along strike to the southeast of the zone from 92 RCD-05 (Refer to Figures 3, 15 and 16 for location and sectional views. Hole 92 RCD-06, collared on an esker penetrated 35 metres of overburden. No problems were encountered drilling the thick gravel/till overburden section. The hole intersected two narrow copper zones; an upper zone from 45.72 to 48.77 metres containing 0.22% Cu over 3.05 metres of which mineralization is all oxide copper and a lower zone from 82.3 to 99.0 metres, an interval of 16.76 metres of 0.26% copper of which 77% is nonsulphide copper. The lower zone contains a high percentage of malachite with some native copper, and bornite. The zone is again strongly hematized and displays weak to strong chlorite and epidote alteration of mafics and feldspars.

# LEGEND

## LITHOLOGIC UNITS



Overburden

### LOWER JURASSIC

GD Granodiorite  
Guichon Batholith

## ALTERATION

CHL - Chloritization  
EP - Epidote  
KSP - Secondary KSP  
SIL - Silification  
HEM - Hematite  
CL - Clay  
SER - Sericite  
CA VN - Calcite Veins  
QTZ VN - Quartz Veins  
VZ - Vein Zone

## DEGREE OF ALTERATION

S - Strong  
M - Moderate  
W - Weak  
V - Very  
T - Trace

## MINERALIZATION

SPEC HEM - Specular Hematite  
N Cu - Native Copper  
BORN - Bornite  
MAL - Malachite  
CHAL - Chalcocite  
CPY - Chalcopyrite  
CUP - Cuprite

0.141 | 1330  
↑  
↓

Total Copper

%Cu | ppmCu

0.91  
76.2 m  
↑  
↓

AVERAGE GRADE

% Total Copper  
Total Length metres

## MISCELLANEOUS

FLT ~ ~ Fault Zone

92RCD-01  
□ — Reverse Circulation Drill Hole ;  
Year, Hole Number

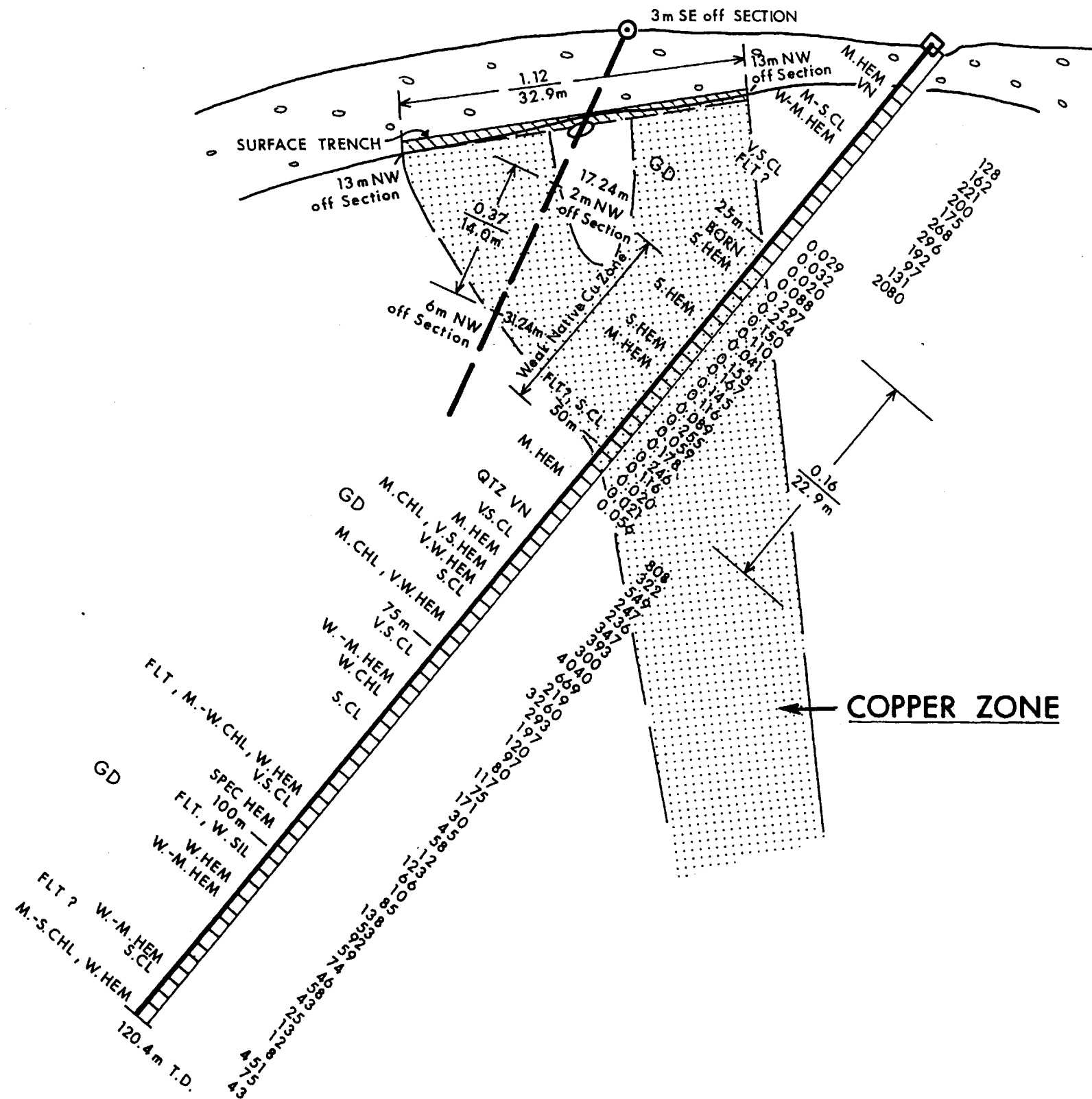
81-5  
○ — Diamond Drill Hole - On Section,  
Off Section  
Pierce Point Of Section



# 92 RCD-01

DIP - 50°  
AZIMUTH: 230°  
LENGTH: 120.4 m

## 81-5



## LEGEND

### LITHOLOGIC UNITS

- Overburden
- LOWER JURASSIC**
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

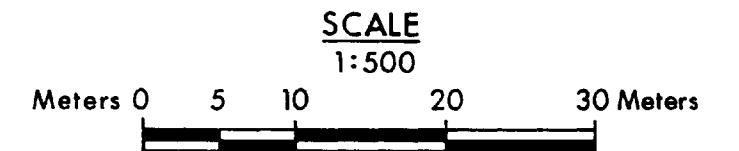
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

Total Copper		%Cu	ppmCu
0.141	1330		
AVERAGE GRADE		% Total Copper	Total Length metres
0.91			76.2 m

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



ZAPPA RESOURCES LTD.  
DOT PROPERTY

## CROSS SECTION D-D' COPPER ZONE

VIEW AZIMUTH 320°

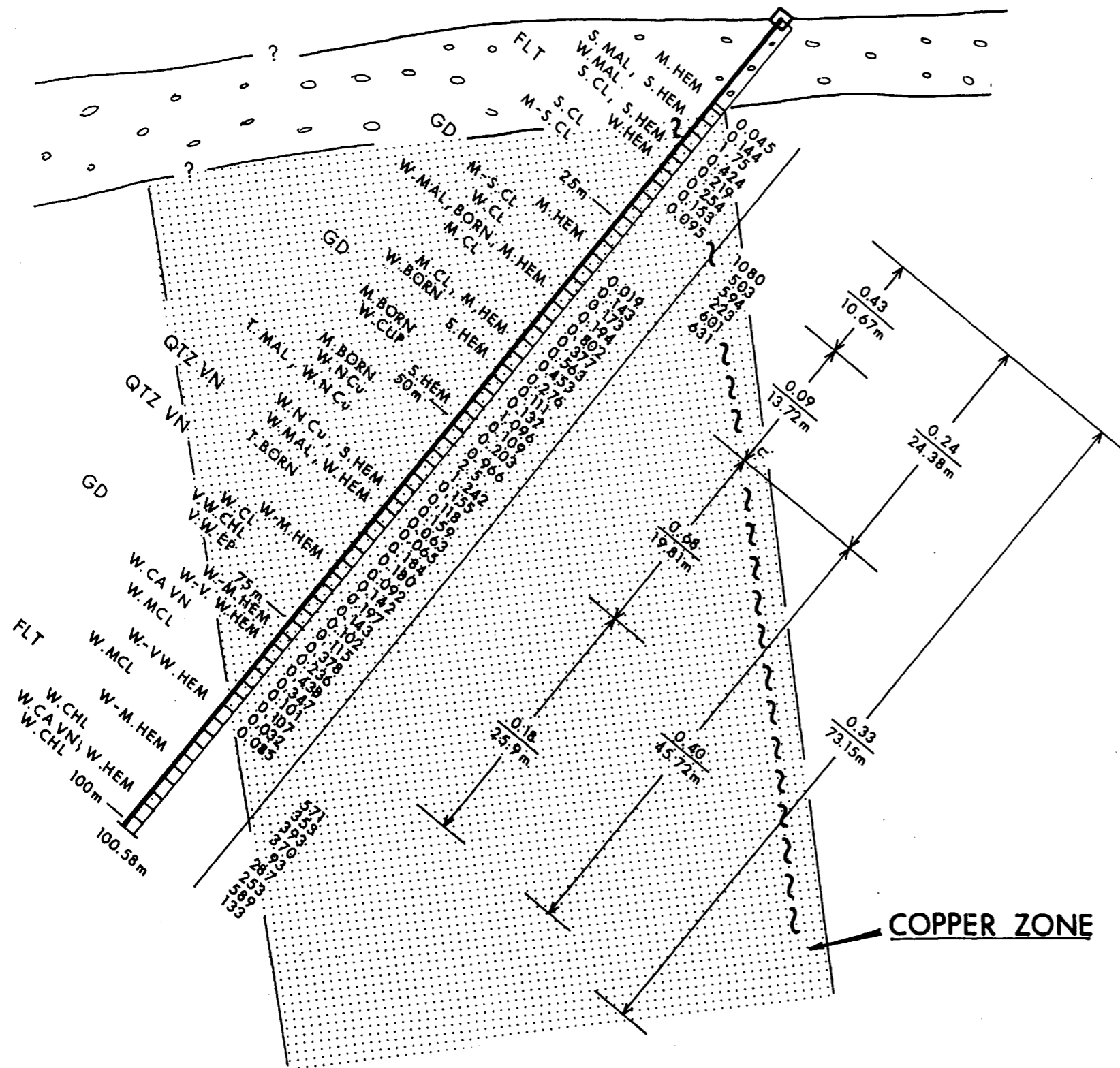
GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	6
DATE: Jan./1993	NICOLA M.D.	





### 92 RCD-02

DIP -50°  
 AZIMUTH 230°  
 LENGTH 100.58m



## LEGEND

### LITHOLOGIC UNITS

- Overburden
- LOWER JURASSIC**
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

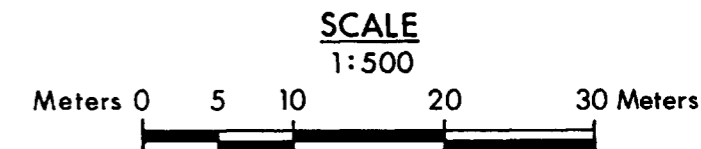
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

0.141	1330	Total Copper
		%Cu   ppmCu
↑		AVERAGE GRADE
0.91		% Total Copper
76.2 m		Total Length metres
↓		

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



**ZAPPA RESOURCES LTD.**  
**DOT PROPERTY**

**CROSS SECTION E-E'**  
**COPPER ZONE**

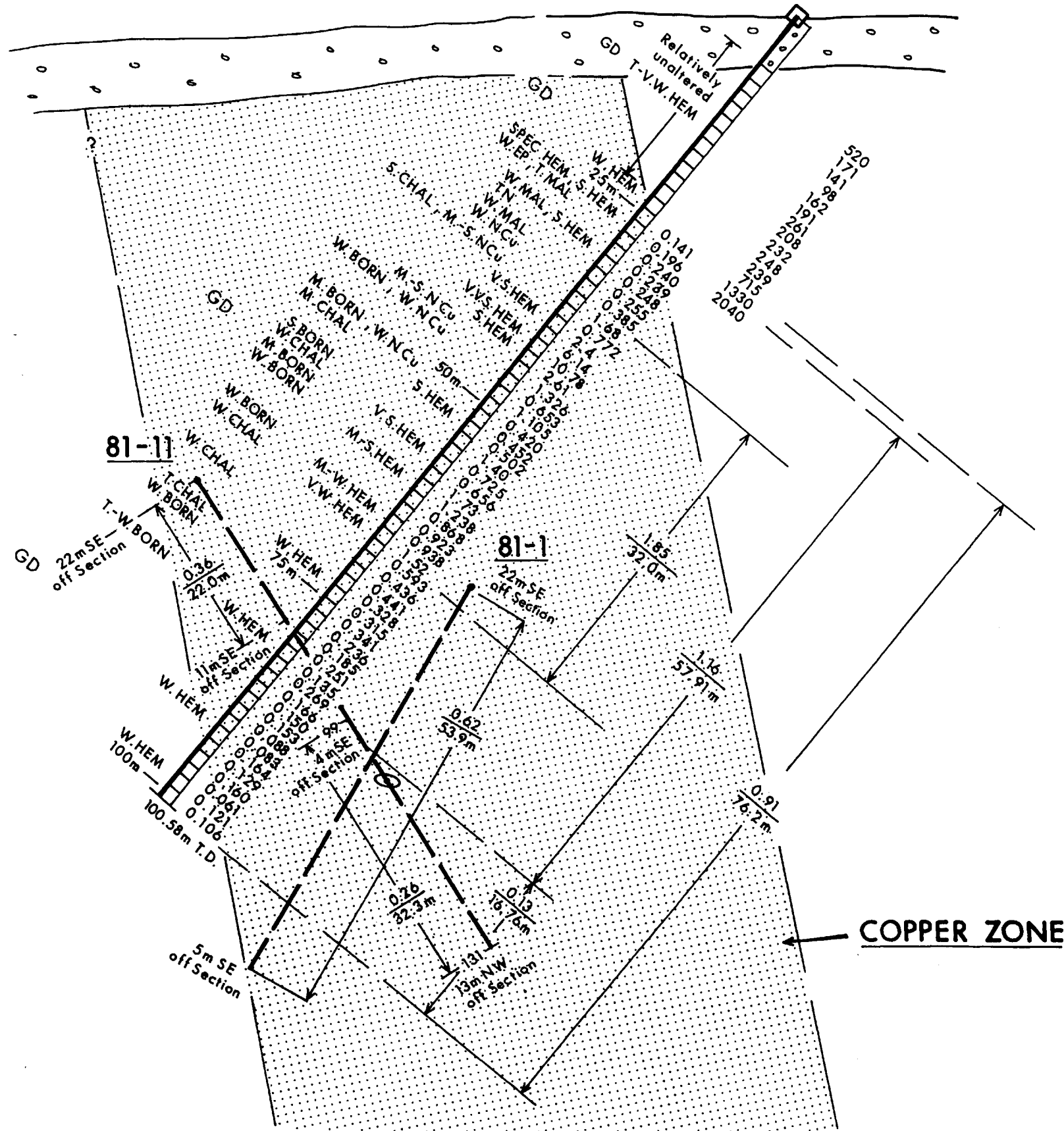
VIEW AZIMUTH 320°

GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	<b>8</b>
DATE: Jan./1993	NICOLA M.D.	



# 92 RCD-03

DIP -50°  
AZIMUTH 230°  
LENGTH 100.58m



## LEGEND

### LITHOLOGIC UNITS

- Overburden
- LOWER JURASSIC**
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

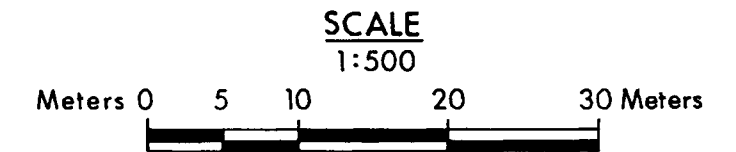
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

0.141	1330	Total Copper
		%Cu   ppmCu
↑		<b>AVERAGE GRADE</b>
0.91		% Total Copper
76.2 m		Total Length metres
↓		

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION F-F'</b>		
<b>COPPER ZONE</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	10
DATE: Jan./1993	NICOLA M.D.	

# 92RCD-04

DIP -50°  
 AZIMUTH 230°  
 LENGTH 100.58m

## LEGEND

### LITHOLOGIC UNITS

- Overburden
- LOWER JURASSIC**
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

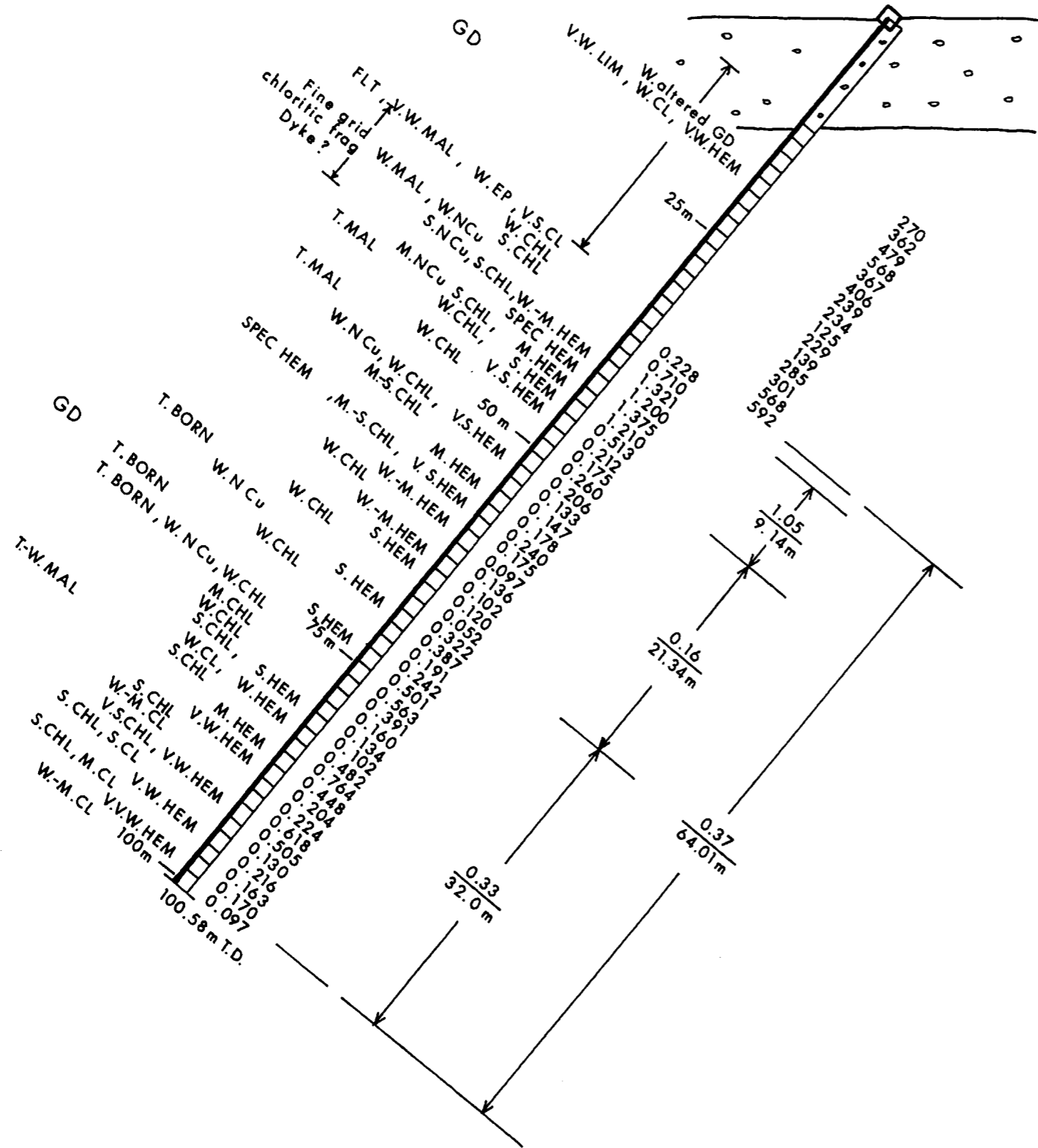
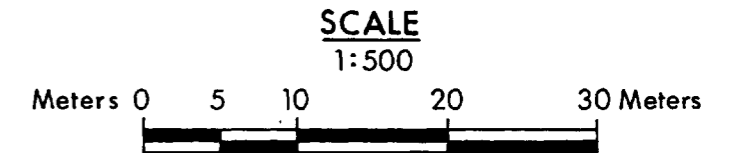
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

0.141	1330	Total Copper
		%Cu   ppmCu
↑		AVERAGE GRADE
0.91		% Total Copper
↓		Total Length metres
76.2 m		

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION G-G'</b>		
<b>92RCD-04-GEOLOGY-ASSAYS</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	11
DATE: Jan./1993	NICOLA M.D.	

92RCD-04

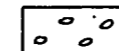
DIP -50°  
AZIMUTH 230°  
LENGTH 100.58m

81-18

81-19  
14m SE  
off Section

# LEGEND

## LITHOLOGIC UNITS



Overburden

0.141 | 1330

Total Copper  
%Cu | ppmCu

## LOWER JURASSIC

GD Granodiorite  
Guichon Batholith

0.91  
76.2 m

AVERAGE GRADE  
% Total Copper  
Total Length metres

## ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

## DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

## MINERALIZATION

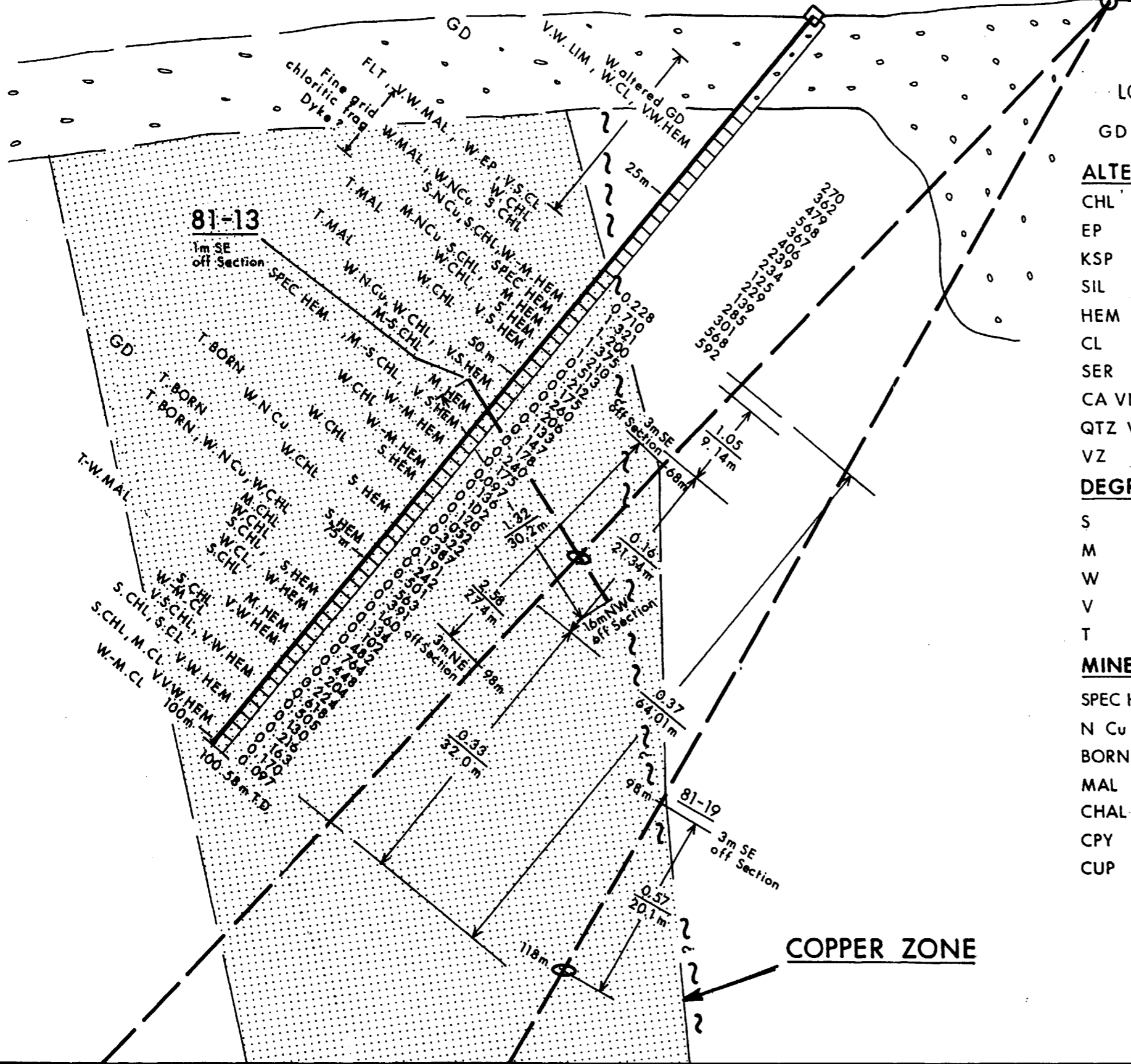
- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

## MISCELLANEOUS

- FLT ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section

## SCALE

1:500

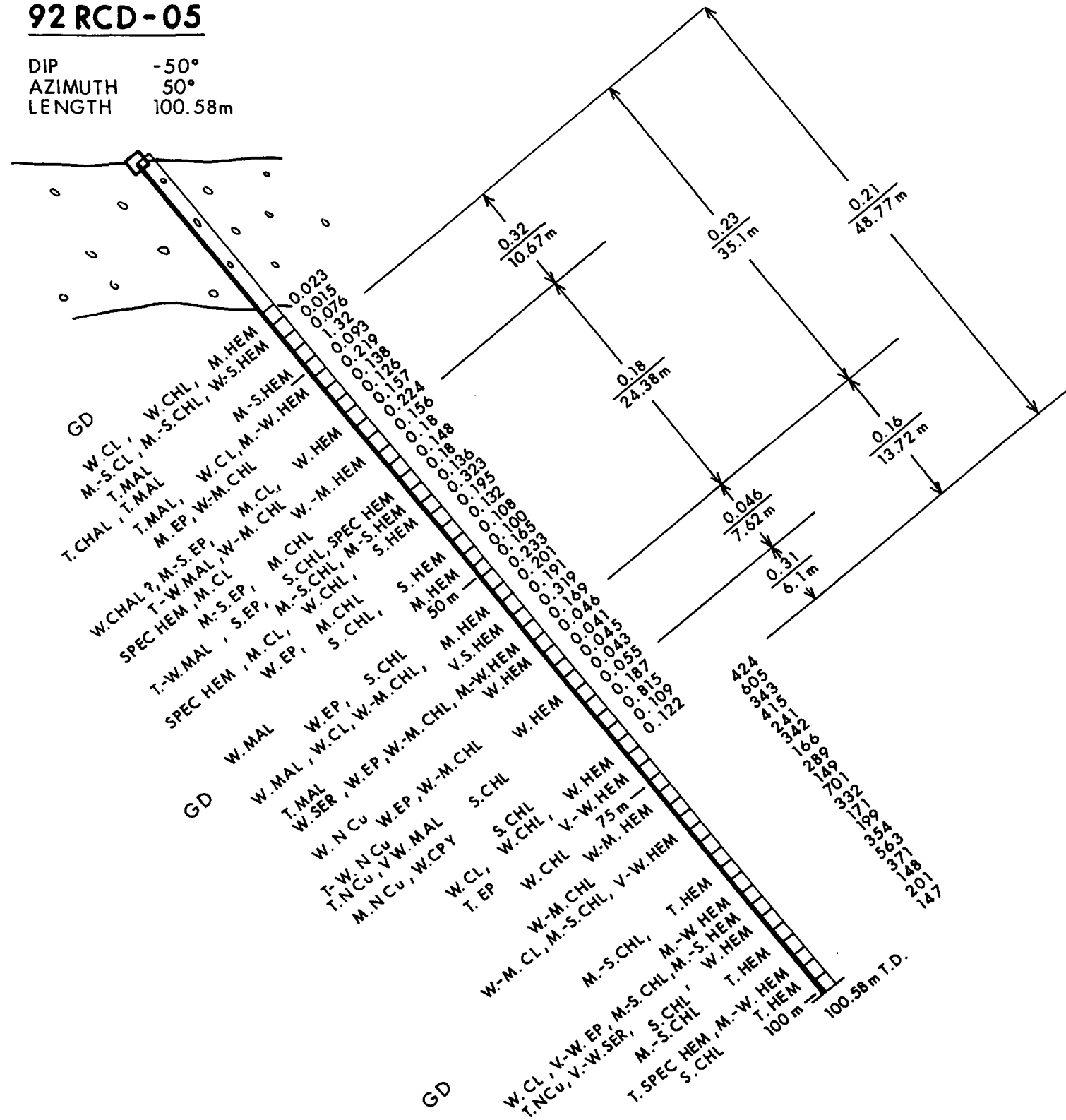


COPPER ZONE

<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION G-G'</b>		
<b>COPPER ZONE</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 921/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	12
DATE: Jan./1993	NICOLA M.D.	

# 92RCD-05

DIP -50°  
 AZIMUTH 50°  
 LENGTH 100.58m



## LEGEND

### LITHOLOGIC UNITS

- Overburden
- LOWER JURASSIC
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

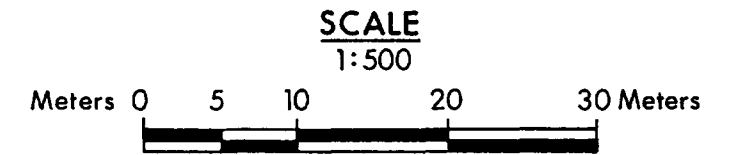
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

Total Copper		%Cu   ppmCu	
0.141	1330		
AVERAGE GRADE		% Total Copper	
0.91		Total Length metres	
76.2 m			

### MISCELLANEOUS

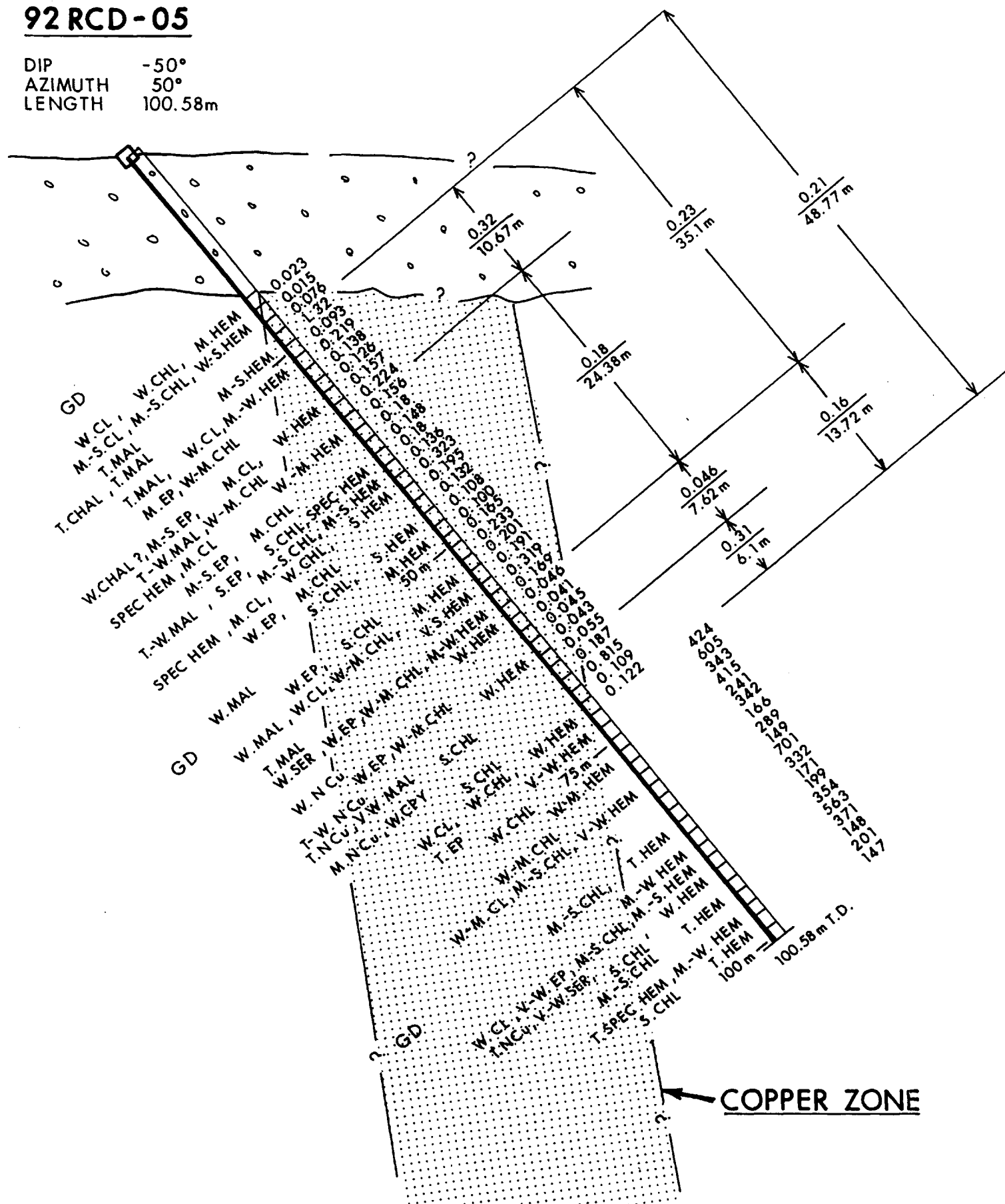
- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION H-H'</b>		
<b>92RCD-05-GEOLOGY-ASSAYS</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	13
DATE: Jan./1993	NICOLA M.D.	

# 92 RCD-05

DIP -50°  
 AZIMUTH 50°  
 LENGTH 100.58m



## LEGEND

### LITHOLOGIC UNITS

- Overburden
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

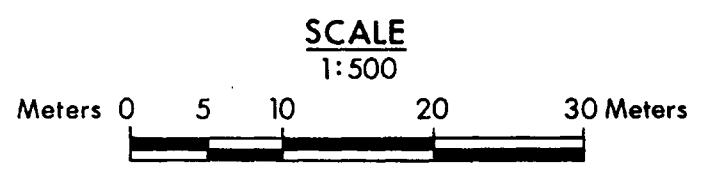
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

Total Copper	
%Cu	ppmCu
0.141	1330
↑	
AVERAGE GRADE	
% Total Copper	
0.91	
↓	
76.2 m	
Total Length metres	

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION H-H'</b>		
<b>COPPER ZONE</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	14
DATE: Jan./1993	NICOLA M.D.	





# 92RCD-06

DIP -50°  
AZIMUTH 230°  
LENGTH 115.82

## LEGEND

### LITHOLOGIC UNITS

- Overburden
- LOWER JURASSIC**
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

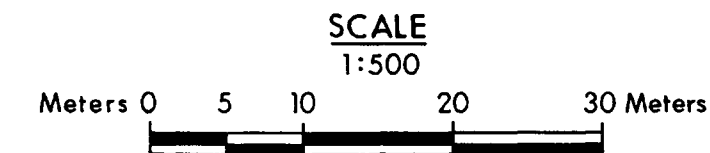
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

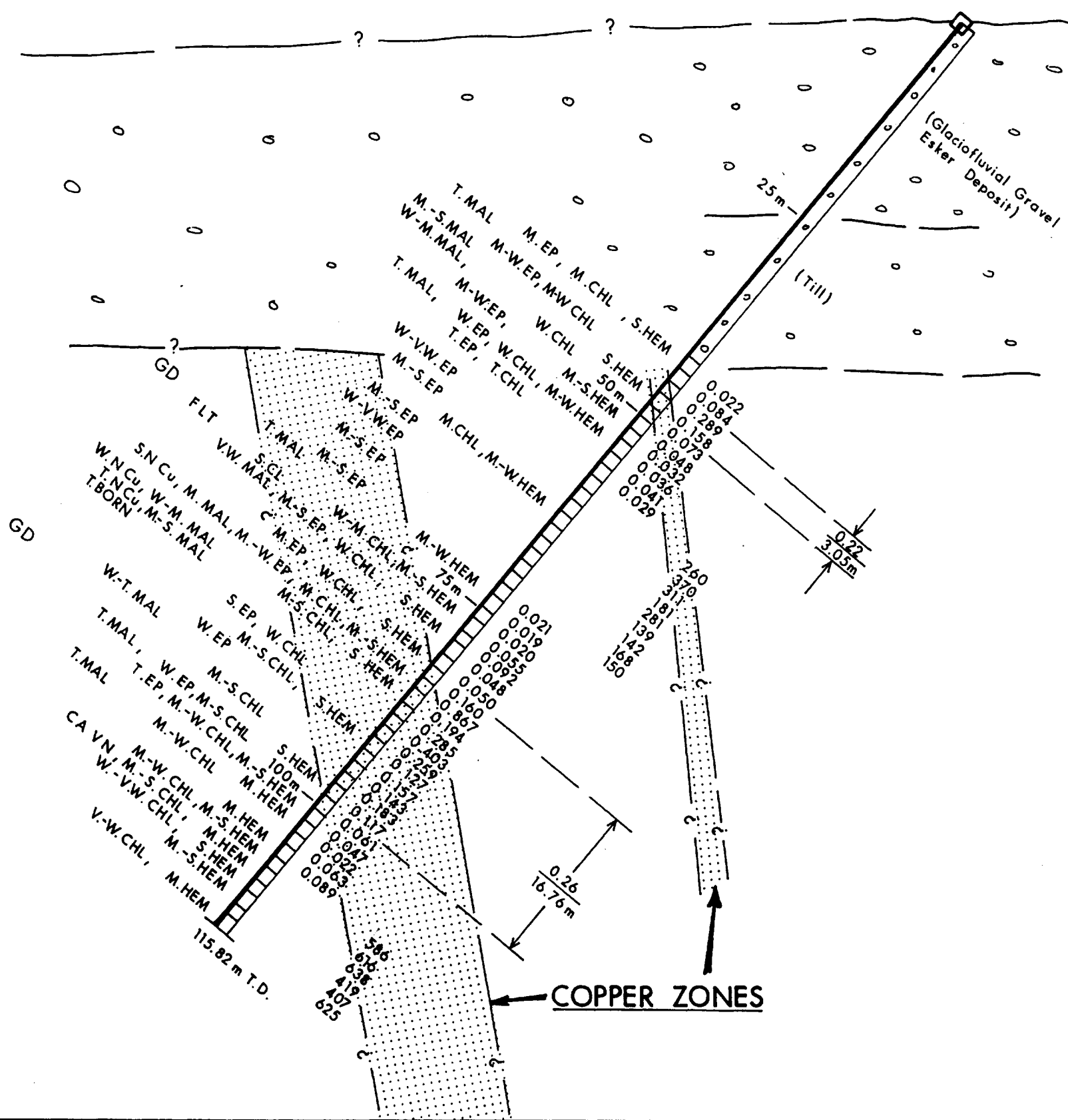
Total Copper	
0.141	1330
% Cu	ppm Cu
AVERAGE GRADE	
0.91	% Total Copper
76.2 m	Total Length metres

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



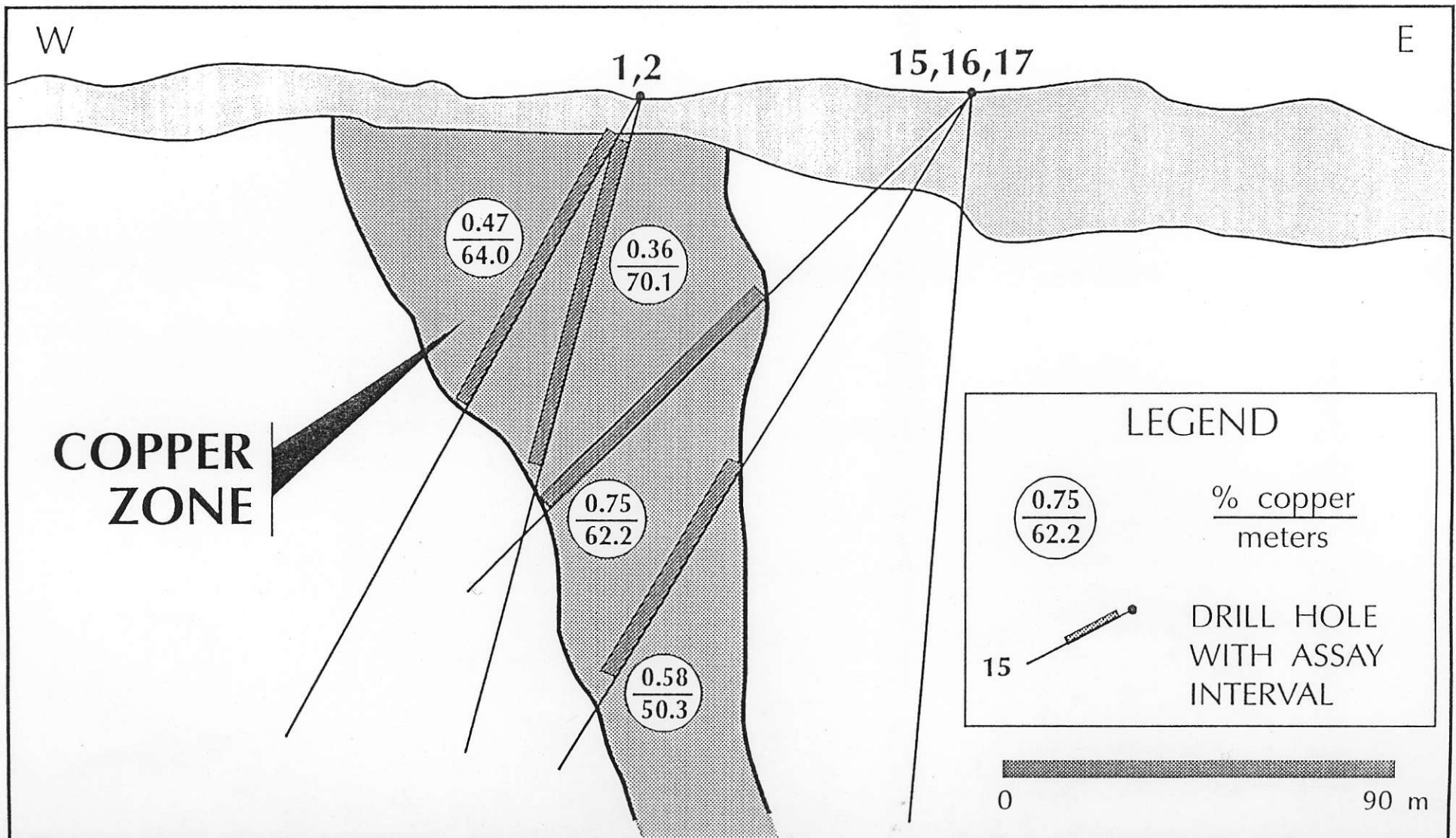
<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION I-I'</b>		
<b>COPPER ZONE</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 921/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	16
DATE: Jan./1993	NICOLA M.D.	



# ZAPPA RESOURCES LTD.







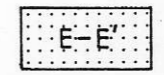
## DOT PROPERTY CROSS SECTION COPPER ZONE

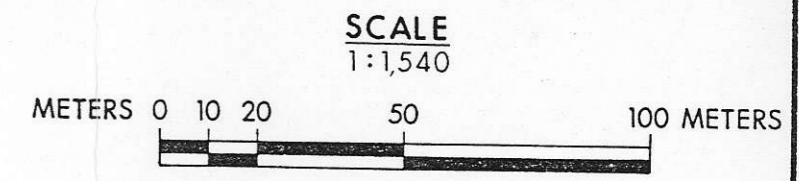
Figure 17



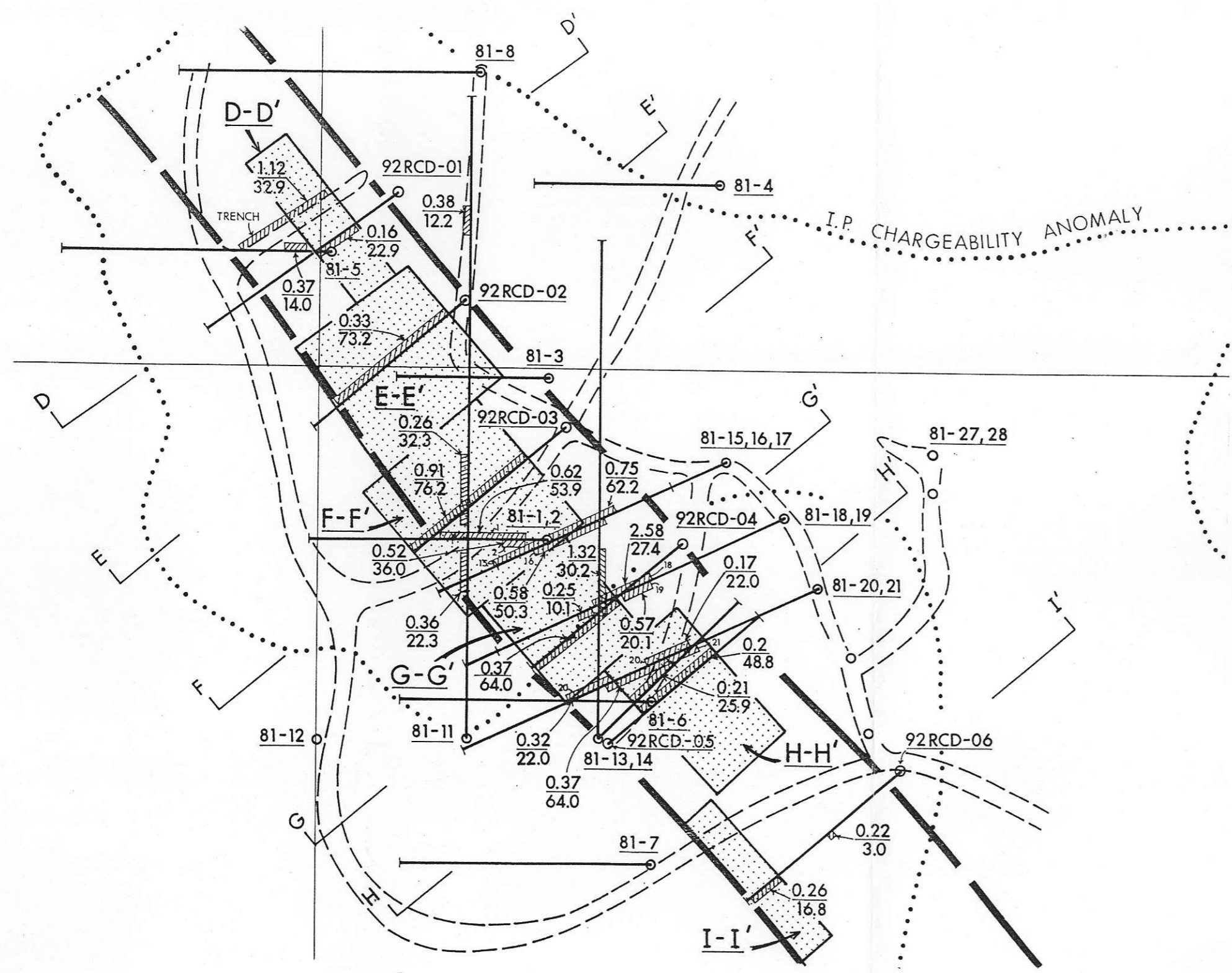


### LEGEND

- $\frac{2.58}{27.4}$  -  $\frac{\% \text{ COPPER}}{\text{INTERVAL (Meters)}}$   
 81-18,19  
 DRILL HOLE  
 81-18 DIAMOND DRILL HOLE  
 P 81-9 PERCUSSION DRILL HOLE  
 92RCD-01 REVERSE CIRCULATION DRILL HOLE  
 HOLE COLLAR - UNIDENTIFIED  
 SECTION LINES  
 TONNAGE BLOCK



ZAPPA RESOURCES LTD.		
DOT PROPERTY		
<b>TONNAGE BLOCKS</b>		
GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:1540	<b>18</b>
DATE: Jan./1993	NICOLA M.D.	



## 6.0 METALLURGICAL TESTING

The Dot Copper deposit contains both primary and secondary copper mineralization. Metallurgical testing was initiated to test for the viability of copper extraction by leaching techniques. Initial acid bottle roll test completed by MinEn Labs on samples from drill hole 92RCD-03 indicated that approximately 12% of the copper is nonsulphide copper. (Refer to Table V).

Further metallurgical testing of Dot mineralization was initiated to determine if biological leaching would be a viable method of low cost copper extraction. Gibraltar Mines located 60 km. north of Williams Lake is currently biochemical leaching a stock pile of low-grade sulphide/oxide copper mineralization. At Gibraltar, 5,000 tons per year of high-purity copper cathode is produced from rock containing grading 0.15% copper. The process consists of heap leaching with sulphuric acid coupled with a solvent extraction plant and use of electro-winning to yield saleable copper cathodes grading 99.99% copper. The leaching process relies on bacteria (*thiobacillus ferro-oxidans*) that break down the sulphides within the rock. These bacteria occur naturally in sulphide mine waters and flourish under conditions of low acidity and moderate temperatures. The organisms eat away at the iron sulphides in the rock to liberate the copper by oxidation.

A. Bruynesteyn of Westcoast Biotech was commissioned to assess the chemical and biological leaching characteristics of the Dot oxide-sulphide-native copper bearing rock from the Dot property. Refer to Appendix VI Report by A. Bruynesteyn, Westcoast Biotech - "Preliminary Evolution of the Leacheability of Ores from the Dot Property". Testing by Westcoast Biotech was completed on a composite sample obtained from mixing -200 mesh pulp material (25 mg.) from 25 samples taken throughout the Dot Copper Zone. Refer to Appendix VII for the listing of samples used. The average head grade of the sample was 0.48% total copper. The amount of copper leached with initial acid leaching tests was 0.187% or 39% which compares well with 38% average as shown in Table V for total nonsulphide copper within the deposit.

Initial testing of the Dot copper material indicates that acid consumption of the ore in a dump/heap leach will be in the order of 17 kg/t. The 18.7 mg. of copper leached from the 10.0 gram sample is probably readily acid soluble copper such as copper oxide but also may be partly chalcocite and native copper. The sulphide copper minerals include chalcocite-bornite and minor chalcopyrite. Thin section work by Vancouver Petrographics (Refer to Appendix I) on samples taken from 92RCD-03 (ore block F-F') shows that sample 48410 (41.14 - 42.67 metres) assaying 10.78% total copper contains at least 3-5% chalcocite (4.5% Cu) and 0.3% bornite/chalcopyrite where as sample 48421 (57.91-59.44 metres) assaying 1.73% total copper and 0.145% nonsulphide copper contains 5-7% bornite, 0.3% chalcocite and 0.1% chalcopyrite. The wide range of percentages

TABLE IV  
 ACID BOTTLE ROLL TESTS - Dot Project  
 By Min En Labs

Drill Hole	Sample No.	*Cu %	**Cu %	Nonsulph Cu %	Total Cu %	Native Cu %	% Oxide Leached	% Leached of Total
92RCD-03	48410	0.905	.908	1.123	10.78	0.628	80	8.4
92RCD-03	48412	0.266	0.263	0.395	1.326	0.366	67	20
92RCD-03	48421	0.112	0.115	0.145	1.730	.008	77	6.5

\* 10 gram sample, 100 ml. of 3% H<sub>2</sub>SO<sub>4</sub>-Leached for 3 hours  
 \*\* 10 gram sample, 100 ml. of 10% H<sub>2</sub>SO<sub>4</sub>-Leached for 3 hours

TABLE V  
 1992 DOT PROJECT  
 Approximate Percent Oxide Within Tonnage Blocks

Sectional Block	Approx. % Oxide in Block	x1000 (metric) Tonnage of Block	x1000 (metric) Tonnes Oxide/Block
D-D'	37%	286.2	105.89
E-E'	20%	701.9	140.38
F-F'	20%	758.43	151.686
G-G'	45%	601.02	270.459
H-H'	76%	378.95	288.002
I-I'	77%	206.7	
	Totals	2,933.2	1,115.57
	Average % Oxide of Deposit = 38%		

indicate the variable content of the different sulphides, chalcocite and bornite, in the more sulphide rich portion of the deposit. Bruynesteyn states that chalcocite leaches rapidly (70-80% extraction in a 120 day period) but chalcopyrite leaches slowly. Bruynesteyn suggests bornite would leach relatively easily.

Westcoast Biotech concludes the Dot ore will be highly amenable to dump/heap leach process with reservations about acid consumption. To that end Bruynesteyn has recommended an inexpensive column test to be carried out to determine acid consumption of 5cm. material. If acid consumption is favourable after 1-2 months, further testing of the column with bacteria leaching will commence. Column testing will require procurement of a representative sample of the deposit through further drilling.



## 7.0 TONNAGE RESOURCE CALCULATION

A preliminary tonnage calculation of the geological resource outlined on the Dot property has been calculated to be 2.933 million tons grading 0.5% Cu. Refer to Table VI for details of the calculation and Figure 18 for location of tonnage blocks. The parameters of reserve calculations include:

1. Section lines are determined by 1992 drill hole section lines
2. Tonnage Blocks are defined by half distances between drill holes and 25 meter sphere of influence from drill holes at ends of deposit
3. Average copper grade of block is determined by weighted average of drill intersections within block.

TABLE VI  
PRELIMINARY GEOLOGICAL RESOURCE  
DOT PROPERTY

Section	Drill Hole	Grade Intersection % Cu/meters	Average Grade of Section % Cu	Dimensions Block Meters			X1000 M3 Volume	2.65+M3 X1000 Tonnes/Section	Ave. Grade/ Block x Tonn. (x1000)
				Length	Width	Depth			
D-D'	92RCD-01	.16/22.9	0.24	60	18	100	108	286.2	64.688
	81-5	.37/14							
E-E'	92RCD-02	.33/73.2	0.33	48	54	100	264.6	701.9	231.627
F-F'	81-11	.36/22.3							
	81-11	.26/32.3							
	81-1	.62/53.9							
	81-2	.52/36.0	0.64	53	54	100	282.2	758.43	485.4
	81-15	.75/62.2							
	81-16	.58/50.3							
G-G'	92RCD-03	0.91/76.2							
	81-15	.75/62.2							
	81-16	.58/50.3							
	81-13	1.32/30.2							
	81-18	2.58/27.4	0.87	42	54	100	226.8	601.02	522.89
H-H'	81-19	.57/20.1							
	92RCD-04	.37/64.0							
	81-20	.32/22							
	81-20	.17/22							
	81-21	.37/64.0							
	81-14	.21/25.9	0.27	55	26	100	143.0	378.95	102.317
	92RCD-05	.2/48.8		60	13	100	78.10	206.7	53.74
TOTALS								2,933.2	1460.662

**8.0 SUMMARY OF EXPENDITURES, DOT PROPERTY**

FOR THE PERIOD

NOVEMBER 1, 1992 TO JANUARY 31, 1993

<u>EXPLORATION FUNCTION</u>	<u>EXPENDITURE</u>
ANALYSIS - ASSAYS	\$5,660.00
ACCOMODATION	970.72
CONSULTING - GEOLOGICAL	10,837.50
DRAFTING, MAPS & PRINTS	1,438.22
EXPEDITING	128.35
DRILLING	21,635.00
EQUIPMENT - CONSUMABLES	1,436.04
FUEL	241.57
SALARY & WAGES	3,250.00
TRANSPORTATION - AIRLINES	614.18
TRANSPORTATION - VEHICLE	170.03
TRANSPORTATION - FREIGHT	656.78
MISCELLANEOUS - INDIRECT	14.25
PROJECT MANAGEMENT FEES	<u>4,704.44</u>
TOTAL EXPLORATION EXPENDITURES	<u><u>\$51,757.08</u></u>

## 9.0 PROPOSED 1993 EXPLORATION PROGRAM 1993 DRILL PROGRAM

The 1992 exploration program focused on confirmation drilling within the main copper zone. It is recommended that the 1993 program be directed to reserve expansion by drill testing the large I.P anomaly which extends the target zone 500 metres to the southeast. Three angle holes (approximately 150 metres in length) would be required to drill a fence across the central portion of the anomaly. Refer to Compilation Map, Figure 4 for location of proposed drill holes. An airphoto lineament interpretation of the area is recommended to locate any significant cross-structures to the general northwesterly trend. Placement of the fence of holes proximal to a cross structure would enhance the probability of discovering additional copper reserves.

### 1993 Metallurgical Testing

Continued metallurgical testing to ensure leacheability of the Dot copper mineralization is recommended in accordance with work proposed by Westcoast Biotech. In order to procure a proper sample for column testing several large diameter holes will be necessary if insufficient quantity of core can be obtained from previous drilling. Because of the variability of oxide/sulphide, more than one column test is warranted. Additional polished thin section work is required within the sulphide rich zones to quantify percentages of chalcocite and bornite, chalcocite being leacheable.

### MISCELLANEOUS

It is recommended that the drill locations and assay information for diamond drill holes 81-21 to 81-30 and 82-31 to 82-33 completed by Lawrence Mining Corporation be acquired before additional drilling is initiated on the Dot property. This information will be important for locating future drill holes.

10.0 PROPOSED 1993 EXPLORATION BUDGETA. DRILLING PROGRAM

1.	Reverse Circulation Drilling	500m @ \$40.00/meter	\$20,000.00
2.	Analytical Costs Geochemical/Assays	300 samples @ \$15/ea.	\$4,500.00
3.	Geological Consulting	20 days @ \$250.00/day	\$5,000.00
4.	Accommodation Hotels and Meals	10 days @ \$50.00/day	\$500.00
5.	Drafting Maps Prints		\$1,500.00
6.	Expediting-Telephone		\$500.00
7.	Equipment-Consumables Field Equipment, Sample Bags		\$1,000.00
8.	Fuel	10 days @ \$20.00/day	\$200.00
9.	Salaries and Wages	20 days @ \$200.00/day	\$4,000.00
10.	Truck Rental	2 wks. @ \$250.00/wk.	\$500.00
11.	Transportation Freight	2000 lbs. @ \$0.50/lb.	\$500.00
12.	Miscellaneous		\$500.00
		Subtotal	\$39,200.00
13.	Management Fee @ 10%		\$3,920.00
		TOTAL ESTIMATED BUDGET	<u>\$43,120.00</u>

B. METALLURGICAL TESTING

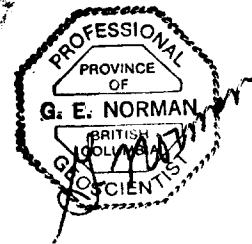
1.	Phase I Acid Leach		\$2,250.00
2.	Phase II Biological Leach		\$6,250.00
3.	Metallurgical Consulting	5 days @ \$500.00/day	\$2,500.00
4.	Thin Section Work	5 thin sections @ \$125.00 ea.	\$625.00
5.	Salaries and Wages	10 days @ \$200.00/day	\$2000.00
		Subtotal	\$11,825.00
6.	Management Fee @ 10%		\$1,180.00
		<b>TOTAL ESTIMATED COSTS</b>	<b><u>\$13,000.00</u></b>
		<b>TOTAL OF PROPOSED EXPLORATION AND METALLURGICAL PROGRAMS</b>	<b><u>\$56,120.00</u></b>

11.0 STATEMENT OF QUALIFICATIONS

I, George E. Norman of 12252 North Park Crescent, Surrey, B.C. certify that:

1. I am a graduate of the University of Alberta with a B.Sc. in Honours Geology, 1973.
2. I am a self-employed geologist d.b.a. Norman Geological
3. I am a registered Professional Geoscientist with the Association of Professional Engineers and Geoscientists of B.C. and a Professional Geologist with the Association of Professional Engineers, Geologists and Geophysicists of Alberta (APEGGA) as well as a Fellow with the Geological Association of Canada (FGAC).
4. I have actively practised my profession as a geologist for the past 19 years.
5. I have personally supervised the fieldwork on the Dot Property for Zappa Resources Ltd. between November 27 and December 7, 1992 and under the supervision of Wayne J. Roberts, Vice-President Exploration.
6. I have no interest or future interest either directly or indirectly in the Dot Property or in Zappa Resources Ltd..
7. I consent to the use of this report in or in connection with a prospectus related to raising of funds.

  
George Norman, P. Geol., P. Geo.



## REFERENCES

12.0 REFERENCES

Anderson, J.A. (1982): Characteristics of Leached Capping and Techniques of Appraisal, Advances in geology of the Porphyry Copper Deposits, Southwestern North America

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McMillan, W.J. (1976): Geology and Genesis of the Highland Valley Ore Deposits and Guichon Creek Batholith, Porphyry Deposits of the Canadian Cordillera, CIM Special Volume 15.

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APPENDIX I

Report on Dot Project Rock Samples

by J. Payne

Vancouver Petrographics Ltd.



# Vancouver Petrographics Ltd.

8080 GLOVER ROAD, LANGLEY, B.C. V3A 4P9  
PHONE (604) 888-1323 • FAX (604) 888-3642

Report for: George Norman,  
Zappa Resources Ltd.,  
1500 - 675 West Hastings Street,  
VANCOUVER, B.C., V6C 2X8

Job 920096  
December 1992

Samples: 48410, 48421  
48576 (reference)

## Summary:

**Sample 48576** is a hornblende monzodiorite composed of medium to coarse grained, subhedral plagioclase and ragged fine to medium grained hornblende with interstitial grains and patches of medium to coarse grained K-feldspar and patches of acicular tremolite. Quartz forms minor interstitial patches, commonly intergrown intimately with K-feldspar. In a few patches, extremely fine grained K-feldspar replaces plagioclase. Accessory minerals include minor magnetite and sphene and trace zircon.

The percussion drill samples contain relatively fresh fragments of quartz diorite and granodiorite moderately similar to Sample 48576. K-feldspar is in part of primary and in part of secondary origin, and in many fragments its origin is uncertain.

In **Sample 48140** fragments are of two main types: slightly altered quartz diorite to granodiorite and moderately to strongly altered rocks. The fresher rocks are dominated by plagioclase with interstitial quartz and K-feldspar and much less hornblende and magnetite and locally minor biotite. The altered rocks are dominated by quartz with patches of K-feldspar, chlorite and sericite. Several fragments are dominated by aggregates of platy grains of specular hematite. Some fragments contain patches of Cu-sulfides: original bornite-(chalcopyrite) is replaced slightly in a few fragments and completely in many to intergrowths of chalcocite and another secondary Cu-sulfide of similar composition.

In **Sample 48421** fragments are of slightly and strongly altered quartz diorite to granodiorite. The fresher fragments are similar to those in Sample 48410. The altered fragments are dominated by quartz with less abundant K-feldspar, chlorite and sericite/muscovite. Bornite, in part with minor chalcopyrite, forms disseminated patches which are altered slightly to chalcocite. Specular hematite forms scattered clusters of platy grains.

*John G. Payne*  
John G. Payne  
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**Sample 48421****Slightly and Strongly Altered Quartz Diorite to  
Granodiorite; Alteration: Quartz-Chlorite-Sericite-  
Bornite-(Chalcopyrite-Specular Hematite-Rutile)  
Minor Secondary Chalcocite**

host rock (40% of sample)

plagioclase	60-65%	magnetite	1- 2%
quartz	20-25	sphene	1- 2
K-feldspar	5- 7(?)	apatite	trace
hornblende	2- 3	zircon	trace
biotite	1- 2		

The sample contains fragments of slightly altered host rock dominated by fine to medium grained, subhedral plagioclase with interstitial fine to medium grained quartz and minor hornblende, magnetite, and sphene. Plagioclase is altered slightly to sericite. In a few fragments, plagioclase is replaced by patches of very fine grained, interlocking K-feldspar grains. Epidote forms a few irregular patches averaging 0.1-0.2 mm in size as a replacement of plagioclase.

K-feldspar occurs in a few fragments as anhedral, interstitial grains up to 1.5 mm in size. Some of it may be of replacement origin.

Hornblende forms a few subhedral grains up to 0.7 mm in size.

Biotite forms a few flakes up to 0.4 mm in size; alteration is complete to pseudomorphic chlorite and lenses of Ti-oxide parallel to cleavage. Magnetite forms equant grains up to 0.4 mm in size; alteration is slight to moderate to hematite, mainly along grain borders. Sphene forms patches up to 0.8 mm long; it is altered completely to extremely fine grained aggregates of Ti-oxide. Apatite forms a few subhedral prismatic grains up to 0.2 mm long. Zircon forms a prismatic grain 0.07 mm long. A few brecciated zones contain aggregates of strongly interlocking, extremely fine grained quartz.

secondary assemblage (60% of sample)

quartz	60-65%		
chlorite	10-12	specular hematite	0.5
sericite/muscovite	8-10	chalcocite	0.3
bornite	5- 7	chalcopyrite	0.1
K-feldspar	5- 7 (?)	rutile	0.1

In altered fragments, plagioclase is replaced strongly to completely by aggregates of very fine grained sericite/muscovite and minor to moderately abundant patches of calcite.

Chlorite forms anhedral patches of extremely fine to very fine grains. A few fragments are dominated by intergrowths of very fine grained quartz and extremely fine grained chlorite. In some of these, quartz forms euhedral grains enclosed in chlorite.

Bornite forms irregular patches averaging 0.05-0.3 mm in size. In many fragments it is free of chalcopyrite, and is altered slightly to moderately in irregular patches to chalcocite. A few grains of bornite are rimmed by thin zones of hematite. In a few fragments it contains irregular, exsolution(?) patches and lenses of chalcopyrite, which are concentrated mainly along grain borders of bornite.

Specular hematite forms patches up to 0.6 mm in size of platy grains averaging 0.05-0.15 mm long.

Rutile forms a few clusters of subhedral grains up to 0.15 mm in size enclosed in a patch of muscovite.

Late veinlets are of calcite. In several fragments, they are more abundant in plagioclase than in quartz. The largest fragment of bornite, 2 mm across, is cut by a few intersecting veinlets up to 0.1 mm wide of carbonate.

Sample 48410

**Fresh and Altered Quartz Diorite/Granodiorite:  
Alteration: Quartz-Chlorite-K-feldspar-  
Sericite-Specular Hematite-Bornite/Chalcopyrite  
Secondary Chalcocite-Mineral X-(Native Copper)**

Fragments are of two main types: slightly altered quartz diorite to granodiorite and moderately to strongly altered rocks. The former are dominated by plagioclase with interstitial quartz and much less hornblende and magnetite and locally minor biotite. The latter are dominated by quartz with patches of K-feldspar, chlorite and sericite. A few fragments are dominated by specular hematite. Some contain patches of Cu-sulfides: original bornite-(chalcopyrite) is replaced moderately to completely by intergrowths of chalcocite and another secondary Cu-sulfide of similar composition.

host rock	
plagioclase	50-55%
quartz	12-15
K-feldspar	5- 7 (?)
hornblende	1- 2
magnetite	1
biotite	0.2
Ti-oxide	0.2

Several fragments of less altered, fine to medium grained quartz diorite are dominated by subhedral plagioclase grains and interstitial patches of quartz and much less hornblende. Plagioclase is altered slightly to disseminated, dusty to extremely fine grained sericite and moderately abundant dusty hematite. One fragment was brecciated moderately.

Magnetite forms disseminated equant grains averaging 0.1-0.2 mm in size; larger grains are altered completely to hematite; a few small grains are preserved within quartz grains.

One fragment contains a flake of biotite 0.4 mm long replaced completely by pseudomorphic chlorite.

Ti-oxide forms patches averaging 0.05-0.15 mm in size and a few up to 0.3 mm across.

secondary mineralogy	
quartz	60-65%
K-feldspar	10-12
chlorite	5- 7
sericite	4- 5
chalcocite, Mineral X	3- 5
bornite/chalcopyrite	0.3
calcite	minor

Altered fragments are dominated by fine grained aggregates of quartz and less commonly K-feldspar. It is uncertain how much of the K-feldspar is primary and how much is secondary. Sericite forms patches of subradiating flakes, in part secondary after plagioclase grains. Other, more strongly altered fragments contain intimate intergrowths of sericite and chlorite more coarsely intergrown with very fine grained quartz grains; in these original textures were destroyed.

Chlorite (2-3%) forms patches averaging 0.05-0.2 mm in size of extremely fine to very fine grained flakes. In a few fragments chlorite forms extremely fine grained aggregates. Pleochroism is from light to medium green.

(continued)

Deep red to opaque, specular hematite forms aggregates of slender, platy to tabular grains averaging 0.1-0.3 mm long, and locally up to 0.5 mm long. These are concentrated in several of the fragments, in which they comprise from 20-90% of the fragment. Interstitial to hematite are patches of very fine grained quartz and muscovite. In one fragment, the interstitial material (20%) is mainly chalcocite with minor quartz.

One fragment dominated by extremely fine grained chlorite contains several replacement? patches up to 1 mm across of bornite-quartz in which bornite patches average 0.05-0.15 mm in size. Some bornite patches contain minor to moderately abundant, slender exsolution lenses of chalcopyrite oriented in two nearly-orthogonal crystallographic directions. Bornite is altered slightly to moderately along borders of patches to chalcocite and covellite.

One fragment 2 mm long is dominated by extremely fine grained chalcocite containing 10-20% grains of slightly lower reflectivity and slightly deeper purplish grey color. It probably is a secondary Cu-sulfide with a composition similar to that of chalcocite. The fragment also contains a patch 0.5 mm across of radiating clusters of sericite flakes. A few fragments contain patches up to 0.5 mm in size of the two secondary Cu-minerals. Textures suggest that both Cu-minerals were formed by replacement of bornite or bornite-chalcopyrite; the variation in color may reflect incomplete replacement.

Native copper occurs in a few fragments as grains and clusters of a few grains averaging 0.01-0.03 mm in grain size. Only in one fragment is it associated with a patch of chalcocite.

Calcite forms scattered ragged grains averaging 0.05-0.1 mm in size.

APPENDIX II

List of Pre-1992 Drill Hole

Copper Intercepts

LIST OF PRE-1992 DRILL HOLE COPPER INTERCEPTS

DOT PROPERTY

<u>Drill Hole</u>	<u>Thickness</u>	<u>Grade % Cu</u>
81-1	53.9	.62
81-2	36.0	.52
81-5	14.0	.37
81-11	22.3	.36
81-11	32.3	.26
81-13	30.2	1.32
81-14	25.9	.21
81-15	62.2	.75
81-16	50.3	.58
81-17	10.0	.37
81-17	10.0	.35
81-18	27.4	2.58
81-19	20.1	.57
81-20	22.2	.32
81-21	64.0	.37



APPENDIX III  
1992 Drill Hole Geologic Logs  
and  
Sample Ledgers





















## DIAMOND DRILL LEDGER

DDH No. 92RCD-01

SSAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Copper ppm.	Non sulph Copper %	Total Copper Assay%	DESCRIPTION
	Metres	Feet	Metres	Feet					
48251	4.57	6.10	1.52	5.0'		128			0-4.57 <u>OVER BURDEN.</u> 4.57-120.4 <u>GRANODIORITE</u>
48252	6.10	7.62	"	"		162			4.57-13.72 med gr'd. reddish-pinkish
48253	7.62	9.14	"	"		221			med hematized granodiorite, minor calc.
48254	9.14	10.67	"	"		200			med-wk calc.
48255	10.67	12.19	"	"		175			
48256	12.19	13.72	"	"		268			
48257	13.72	15.24	"	"		296			- 13.72 - 27.43 - Strong clay altered gnd.
48258	15.24	16.76	"	"		192			
48259	16.76	18.29	"	"		97			
48260	18.29	19.81	"	"		131			
48261	19.81	21.34	"	"		2080			- 19.81-21.34 Very strong clay mineral.
48262	21.34	22.86	"	"			6.014	0.029	
48263	22.86	24.38	"	"			.016	.032	
48264	24.38	25.91	"	"			.008	.020	
48265	25.91	27.43	"	"			.029	.088	
48266	27.43	28.96	"	"		∧	.079	.297	∧ 27.43 - 50.29 <u>Copper Zone</u> -
48267	28.96	30.48	"	"			.102	.254	- str hematized w/upt.
48268	30.48	32.0	"	"			.082	.150	3-5% frog w/ native copper, & minor bornite,
48269	32.0	33.53	"	"			.052	.110	becoming weaker in hem. alter. form.
48270	33.53	35.05	"	"			.030	.047	38.10 - 50.29.

DIAMOND DRILL LEDGER

DDH No. 92 RCD-C

VSAV TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu ppm	Non Sulph Cu %	Total Cu %	DESCRIPTION	
	Metres	Feet	Metres	Feet						
48271	35.05	36.58	1.52	5.0'			.072	.155	<u>GRANODIORITE CONT'D</u> 35.05 - 36.58 w/upto 5-10% frag w/. native copper or bornite.	
48272	36.58	38.10	"	"			.067	.167		
48273	38.10	39.62	"	"			.064	.145		
48274	39.62	41.15	"	"			.046	.116		
48275	41.15	42.67	"	"		$\frac{.06}{22.9}$	.045	.089		0.166 (27.43 - 50.29m) 22.9m
48276	42.67	44.20	"	"			.087	.255		
48277	44.20	45.72	"	"			.031	.059		
48278	45.72	47.24	"	"			.060	.178		
48279	47.24	48.77	"	"			0.10	.246		
48280	48.77	50.29	"	"			.055	.116		50.29 - 67.06 - mod-str lam, altered
48281	50.29	51.82	"	"			.015	.020	granodior. w/ strong clay alterin	
48282	51.82	53.34	"	"			.011	.021		
48283	53.34	54.86	"	"			.036	.056		
48284	54.86	56.39	"	"		808				
48285	56.39	57.91	"	"		322				
48286	57.91	59.44	"	"		549				
48287	59.44	60.96	"	"		247				
48288	60.96	62.48	"	"		236				
48289	62.48	64.0	"	"		347				
48290	64.0	65.53	"	"		393				

## DIAMOND DRILL LEDGER

DDH No. 92RCD-01

SAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu ppm	Non- Sulph. % Cu	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet					
48291	65.53	67.06	1.52	5.0'		300			GRANODIORITE CONT'D
48292	67.06	68.58	"	"		4040			67.06 - 73.15 - v. wk hem. altered, relative
48293	68.58	70.10	"	"		669			unox. gndior. w/ mod-str clay. alterin
48294	70.10	71.63	"	"		219			and mod chloritic alterin. Possible.
48295	71.63	73.15	"	"		3260			fault at. 70.1 - 71.63.
48296	73.15	74.68	"	"		293			73.15 - 86.87 - v. wk to mod. hem. alterin,
48297	74.68	76.20	"	"		197			str - v. str clay alterin ' wk to v. str.
48298	76.20	77.72	"	"		120			chl. altered gndiorite.
48299	77.72	79.25	"	"		97			
48300	79.25	80.77	"	"		80			
48301	80.77	82.30	"	"		117			
48302	82.30	83.82	"	"		75			
48303	83.82	85.34	"	"		171			
48304	85.34	86.87	"	"		30			
48305	86.87	88.39	"	"		45			86.87 - 120.4 - mod - wk chlorite alterin
48306	88.39	89.92	"	"		58			mod-wk hem. alterin, v. str clay alterin
48307	89.92	91.44	"	"		12			from 94.49 - 96.01 & 99.06 - 105.16 these
48308	91.44	92.96	"	"		123			could be faulted zones.
48309	92.96	94.49	"	"		66			
43310	94.49	96.01	"	"		10			





















## DIAMOND DRILL LEDGER

DDH No. 92RCD-02

SAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu Ppm	Non- Sulph Cu %	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet					
									0 - 10.06 <u>OVER BURDEN</u>
48327	10.06	10.67	0.61	2.0'		.039	.045		10.06 - 100.58 <u>GRANODIORITE</u>
48328	10.67	12.19	1.52	5.0'		.131	.144		10.06 - 12.19 reddish - pinkish med gr'd.
48329	12.19	13.72	"	"		1.66	1.75		gr'dior. w/ mod hem. stain, 15% bio.
48330	13.72	15.24	"	"		.386	.424		12.19 - 16.74 - reddish brn, str hem altered w/
48331	15.24	16.76	"	"		.133	.219		mal stain on frag (12.19 - 13.72. 60% of frags.
48332	16.76	18.29	"	"		.068	.254		w/ mal.
48333	18.29	19.81	"	"		.051	.153		16.76 - 21.34 - unoxidized grey clay zone - Fault Zone.
48334	19.81	21.34	"	"		1080	.095		
48335	21.34	22.86	"	"		503			21.34 - 39.62 - mod hem stained, gndior., traces
48336	22.86	24.38	"	"		594			of biotite at 35.05 - 36.58, mod-wk
48337	24.38	25.91	"	"		223			chlorite alterin.
48338	25.91	27.43	"	"		601			
48339	27.43	28.96	"	"		631			
48340	28.96	30.48	"	"			.008	.019	
48341	30.48	32.0	"	"			.057	.143	
48342	32.0	33.53	"	"			.083	.173	
48343	33.53	35.05	"	"			.096	.194	
48344	35.05	36.58	"	"			.105	.802	
48345	36.58	38.10	"	"			.106	.377	

## DIAMOND DRILL LEDGER

DDH No. 92 RCD-04

DAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu ppm	Non Sulph Cu	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet					
48346	38.10	39.62	1.52	5.0'			.120	.563	Granodiorite cont'd.
48347	39.62	41.15	"	"			.133	.453	39.62-51.82 - Str leached w/ bornite poss.
48348	41.15	42.67	"	"			.116	.276	some bornite & cuprite
48349	42.67	44.20	"	"			.061	.111	
48350	44.20	45.72	"	"			.068	.137	
48351	45.72	47.24	"	"			.097	1.096	
48352	47.24	48.77	"	"			.046	.109	
48353	48.77	50.29	"	"			.068	.203	
48354	50.29	51.82	"	"			.099	.966	
48355	51.82	53.34	"	"			.101	2.5	51.82-53.34 relatively unoxidized w/ str clay.
48356	53.34	54.86	"	"			.177	1.242	-mod chlorite - numerous drogs of bornite.
48357	54.86	56.39	"	"			.069	.155	53.34-60.96 - Str-u-str hem. alteration w/
48358	56.39	57.91	"	"			.054	.118	native copper. - 3mm wire at 53.34-54.86.
48359	57.91	59.44	"	"			.055	.159	wk mal 56.39-57.91 & qtz veining 56.39-59.44
48360	59.44	60.96	"	"			.032	.063	
48361	60.96	62.48	"	"			.031	.065	60.96-62.48 mod hem granitic w/wk mal.
48362	62.48	64.0	"	"			.060	.184	62.48-68.58 Quartz Veining - upto 80% of sample
48363	64.0	65.53	"	"			.049	.180	Trace bornite 64-65.53.
48364	65.53	67.06	"	"			.031	.092	
48365	67.06	68.58	"	"		1430	.066	.142	



## DIAMOND DRILL LEDGER

DDH No. 92RCD-02

SAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu ppm	Non- Sulph Cu %	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet					
48366	68.58	70.10	1.52	5.0		1960	.095	.197	Granodiorite Cont'd.
48367	70.10	71.63	"	"		1690	.066	.143	68.58. - 100.59. granodiorite w/ wk - mod.
48368	71.63	73.15	"	"		1000	.052	.102	hem staining { wk - mod. chlorite
48369	73.15	74.68	"	"		1120	.055	.115	alteration.
48370	74.68	76.20	"	"		3950	.125	.378	
48371	76.20	77.72	"	"		2290	.102	.236	
48372	77.72	79.25	"	"		4530	.074	.438	
48373	79.25	80.77	"	"		3430	.088	.347	
48374	80.77	82.30	"	"		948	.038	.101	
48375	82.30	83.82	"	"		1050	.043	.107	
48376	83.82	85.34	"	"		308	.012	.032	2.164 / 15.9802
48377	85.34	86.87	"	"		1000	.031	.085	
48378	86.87	88.39	"	"		571			
48379	88.39	89.92	"	"		353			
48380	89.92	91.44	"	"		393			
48381	91.44	92.96	"	"		370			
48382	92.96	94.49	"	"		93			
48383	94.49	96.01	"	"		287			
48384	96.01	97.54	"	"		253			
48385	97.54	99.06	"	"		589			
48386	99.06	100.58	"	"		133			



















## DIAMOND DRILL LEDGER

DDH No. 92 RCD-03

No's checked Rec 22

SAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu ppm	Non- Sulph c% %	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet					
									0-6.10 OVER BURDEN.
48387	6.10	7.62	1.52	5.0'		520			6.10 - 106.58 GRANODIORITE
48388	7.62	9.14	"	"		171			6.10 - 27.43 relatively fresh biotite
48389	9.14	10.67	"	"		141			granodiorite w/v. wk hematite stains
48390	10.67	12.19	"	"		98			
48391	12.19	13.72	"	"		162			
48392	13.72	15.24	"	"		191			
48393	15.24	16.76	"	"		261			
48394	16.76	18.29	"	"		208			
48395	18.29	19.81	"	"		232			
48396	19.81	21.34	"	"		248			
48397	21.34	22.86	"	"		239			
48398	22.86	24.38	"	"		715			
48399	24.38	25.91	"	"		1330	.130	.141	
48400	25.91	27.43	"	"		2040	.175	.196	
48401	27.43	28.96	"	"			0.226	0.24	27.43 - 30.48. - mod strong. hematite.
48402	28.96	30.48	"	"			.210	.239	alterin, traces of mal
48403	30.48	32.0	"	"			.202	.248	30.48 - 51.82. Zone of very strong hematite.
48404	32.0	33.53	"	"			.208	.255	alterin; native copper, malachite
48405	33.53	35.05	"	"			.295	.385	chalcocite ! poss. cuprite

## DIAMOND DRILL LEDGER

DDH No. 92RCD-0

SAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Total Cu ppm.	Non- Sulph Cu %	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet				
48406	35.05	36.58	1.52	5.0'	.392	.734	1.68	
48407	36.58	38.10	"	"	.127	.397	.772	
48408	38.10	39.62	"	"	.209	.557	2.4	
48409	39.62	41.15	"	"	.345	.779	6.14	- 39.62- 41.15- 30-40% Frag of specular hematite
48410	41.15	42.67	"	"	.628	1.123	10.78	- 50% Frag of Quartz - Vein material
48411	42.67	44.20	"	"	.585	.399	2.61	- native copper
48412	44.20	45.72	"	"	.366	.375	1.326	
48413	45.72	47.24	"	"	.160	.266	.653	
48414	47.24	48.77	"	"	.028	.153	1.105	
48415	48.77	50.29	"	"	.009	.113	.420	
48416	50.29	51.82	"	"	.008	.098	.452	
48417	51.82	53.34	"	"	.012	.135	.502	51.82-67.06 u-str hematite stained gndwite
48418	53.34	54.86	"	"	.007	.097	1.400	with 1-5% specular hematite with
48419	54.86	56.39	"	"	.012	.114	.725	bornite (upt. 2-3% of frag). and chalcocite
48420	56.39	57.91	"	"	.007	.111	.656	
48421	57.91	59.44	"	"	.008	.145	1.73	
48422	59.44	60.96	"	"	.012	.143	1.238	
48423	60.96	62.48	"	"	.024	.154	.868	
48424	62.48	64.0	"	"	.014	.115	.923	
48425	64.0	65.53	"	"	.012	.107	.938	

## DIAMOND DRILL LEDGER

DDH No. 92RCD-05

SAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu %	Non Sulph Cu %	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet					
48426	65.53	67.06	1.52	5.0'		.078	.131	1.52	Granodiorite Cont'd
48427	67.06	68.58	"	"			.163	.593	67.06-76. wk hematite alterin with
48428	68.58	70.10	"	"			.085	.436	mod - str chlorite alterin, trace -
48429	70.10	71.63	"	"			.142	.441	wk blk oxide? on Srogs. (chlorite)
48430	71.63	73.15	"	"			.094	.328	
48431	73.15	74.68	"	"			.156	.315	
48432	74.68	76.20	"	"			.078	.341	
48433	76.20	77.72	"	"			.100	.236	76.20-100.58. - wk - mod chlorite alterin
48434	77.72	79.25	"	"			.057	.185	and wk hem. alterin.
48435	79.25	80.77	"	"		2410	.121	.251	
48436	80.77	82.30	"	"		1190	.058	.135	
48437	82.30	83.82	"	"		2560	.116	.269	
48438	83.82	85.34	"	"		1570	.068	.166	
48439	85.34	86.87	"	"		1320	.082	.150	
48440	86.87	88.39	"	"		1340	.083	.153	
48441	88.39	89.92	"	"		769	.044	.088	
48442	89.92	91.44	"	"		783	.034	.083	
48443	91.44	92.96	"	"		1480	.081	.164	
48444	92.96	94.49	"	"		1260	.048	.129	
48445	94.49	96.01	"	"		1530	.091	.160	





















## DIAMOND DRILL LEDGER

DDH No. 92RCD-04

ASSAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Total Cu ppm	Non Sulph gCu %	Total Cu%	DESCRIPTION
	Metres	Feet	Metres	Feet				
								0 - 12.19 <u>OVERBURDEN</u>
48449	12.19	13.72	1.52	5.0'	270			12.19 - <u>GRANODIORITE</u>
48450	13.72	15.24	"	"	362			12.19 - 33.53 <i>Relatively unaltered beige-</i>
48451	15.24	16.76	"	"	479			<i>whitish med grained biotite granodiorite</i>
48452	16.76	18.29	"	"	568			<i>15% blk bio, fsp altered w/ky to clay</i>
48453	18.29	19.81	"	"	367			<i>v. wk hem stain ~1% of fsp.</i>
48454	19.81	21.34	"	"	406			
48455	21.34	22.86	"	"	239			
48456	22.86	24.38	"	"	234			
48457	24.38	25.91	"	"	125			
48458	25.91	27.43	"	"	229			
48459	27.43	28.96	"	"	139			
48460	28.96	30.48	"	"	285			
48461	30.48	32.0	"	"	301			
48462	32.0	33.53	"	"	568			
48463	33.53	35.05	"	"	592			33.53 - 35.05. - str clay altered - Fault?
48464	35.05	36.58	"	"		.223	.228	<i>w/ky alter of bio.</i>
48465	36.58	38.10	"	"		.582	.710	35.05 - 38.1 - str clay - mod hem, mod-str
48466	38.10	39.62	"	"		.582	1.321	<i>limonite - weak malachite</i>
48467	39.62	41.15	"	"		.363	1.20	

## DIAMOND DRILL LEDGER

DDH No. 92 RCD-04

ASSAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu ppm	Non Sulph %Cu	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet					
48468	41.15	42.67	1.52	5.0'			.749	1.375	GRANODIORITE CONT'D
48469	42.67	44.20	"	"			.740	1.210	38.1-51.82. Zone of strong native copper.
48470	44.20	45.72	"	"			.407	.513	mod-wk malachite - High grade
48471	45.72	47.24	"	"			.167	.212	Sample 39.62-44.5 up to 5% native Cu.
48472	47.24	48.77	"	"			.146	.175	38.1-44.20 .5% - 5% Cu.
48473	48.77	50.29	"	"			.206	.260	- str hem and str chl
48474	50.29	51.82	"	"			.170	.206	
48475	51.82	53.34	"	"			.096	.133	51.82-67.0 Zone of weaker native
48476	53.34	54.86	"	"			.085	.147	Copper, mod-wk chlorite alterin
48477	54.86	56.39	"	"			.101	.178	v. str specularite hematite
48478	56.39	57.91	"	"			.104	.240	
48479	57.91	59.44	"	"			.086	.175	
48480	59.44	60.96	"	"			.050	.097	
48481	60.96	62.48	"	"			.080	.136	
48482	62.48	64.0	"	"			.063	.102	
48483	64.0	65.53	"	"			.069	.120	
48484	65.53	67.06	"	"			.034	.052	67.06-77.72 Traces of native copper,
48485	67.06	68.58	"	"			.163	.322	bornite; chalcopyrite - str hem
48486	68.58	70.10	"	"			.137	.387	stain; wk-mod chlorite alterin
48487	70.10	71.63	"	"			.096	.191	

## DIAMOND DRILL LEDGER

DDH No. 92RLD-04

ASSAY TAG No.	SAMPLE Metres	INTERVAL Feet	SAMPLE Metres	LENGTH Feet	Au	Total Cu ppm.	Non Sulph % Cu	Total Cu %	DESCRIPTION
48488	71.63	73.15	1.52	5.0			.122	.242	GRANODIORITE
48489	73.15	74.68	"	"			.118	.501	
48490	74.68	76.20	"	"			.162	.563	
48491	76.20	77.72	"	"			.137	.391	
48492	77.72	79.25	"	"			.077	.160	77.72 - 85.34 - Str hem. alterin, mod-str
48493	79.25	80.77	"	"			.076	.134	chlorite v. wk ep., No visible
48494	80.77	82.30	"	"			.052	.102	Copper minerals
48495	82.30	83.82	"	"			.119	.482	
48496	83.82	85.34	"	"			.105	.764	
48497	85.34	86.87	"	"			.134	.448	85.34 - 89.92 - Str chlorite, wk to
48498	86.87	88.39	"	"			.077	.204	str clay alterin and weak-trace
48499	88.39	89.92	"	"			.112	.224	of malachite.
48500	89.92	91.44	"	"			.104	.618	89.92 - 100.58. Str chlorite, mod-str.
48501	91.44	92.96	"	"		4850	.146	.505	clay alterin, & very weak hematite
48502	92.96	94.49	"	"		1220	.052	.130	alterin - no visible copper minerals.
48503	94.49	96.01	"	"		2200	.117	.216	ization.
48504	96.01	97.54	"	"		1560	.065	.163	
48505	97.54	99.06	"	"		1490	.078	.170	
48506	99.06	100.58	"	"		884	.046	.097	





















## DIAMOND DRILL LEDGER

DDH No. 92RCD-05

ASSAY TAG No.	SAMPLE INTERVAL Metres	INTERVAL Feet	SAMPLE LENGTH Metres	LENGTH Feet	Au	Total Cu ppm	Non Sulph % Cu	Total Cu %	DESCRIPTION
48507	18.29	19.81	1.52	5.0'			.018	.023	0-18.29 <u>OVERBURDEN</u> 18.29- <u>GRANODIORITE</u>
48508	19.81	21.34	"	"			.013	.015	18.29-39.62 mod-wk hem staining
48509	21.34	22.86	"	"			.070	.076	w/ some spec. hem. (33.5 - 36.6).
48510	22.86	24.38	"	"			1.256	1.32	- mod clay alteration, and w/ traces
48511	24.38	25.91	"	"			.086	.093	to wk. malachite staining
48512	25.91	27.43	"	"			.204	.219	22.86-24.38 3/4 frag of mal., odd frag
48513	27.43	28.96	"	"			.126	.138	w/ blk oxide w/ mal.
48514	28.96	30.48	"	"			.117	.126	
48515	30.48	32.0	"	"			.146	.157	
48516	32.0	33.53	"	"			.211	.224	
48517	33.53	35.05	"	"			.139	.156	
48518	35.05	36.58	"	"			.161	.18	
48519	36.58	38.10	"	"			.132	.148	
48520	38.10	39.62	"	"			.157	.18	39.62-56.39 mod-str hem. stained.
48521	39.62	41.15	"	"			.123	.136	mod-str chloritically altered, wk-mod.
48522	41.15	42.67	"	"			.284	.323	clay altered w/ wk - trace malachite
48523	42.67	44.20	"	"			.173	.195	staining, wk spec. hematite, & some frag
48524	44.20	45.72	"	"			.119	.132	w/ blk oxide? or chalcocite?
48525	45.72	47.24	"	"			.096	.108	
48526	47.24	48.77	"	"			.092	.100	

## DIAMOND DRILL LEDGER

DDH No. 92 RCD-05

ASSAY TAG No.	SAMPLE INTERVAL Metres Feet	SAMPLE LENGTH Metres Feet	Au	Total Cu ppm	Non Sulph % Cu	Total Cu %	DESCRIPTION
48527	48.77 50.29	1.52 5.0			.151	.165	<u>GRANODIORITE CONT'D.</u>
48528	50.29 51.82	" "			.207	.233	
48529	51.82 53.34	" "			.182	.201	
48530	53.34 54.86	" "			.173	.191	
48531	54.86 56.39	" "			.283	.319	
48532	56.39 57.91	" "			.142	.169	56.39-68.58 Zone w/ traces - wk natural
48533	57.91 59.44	" "			.034	.046	copper, mod-wk hematite alteration
48534	59.44 60.96	" "			.031	.041	wk-strong chlorite alter and wk.
48535	60.96 62.48	" "			.027	.045	epidote, wk chalcoprite from
48536	62.48 64.0	" "			.027	.043	65.5-68.6.
48537	64.0 65.53	" "			.038	.055	
48538	65.53 67.06	" "			.066	.187	
48539	67.06 68.58	" "			.095	.815	68.58-86.9 - weakly hematitically
48540	68.58 70.10	" "			.026	.109	altered and weakly chloritized
48541	70.10 71.63	" "			.041	.122	no visible copper minerals.
48542	71.63 73.15	" "		424			
48543	73.15 74.68	" "		605			
48544	74.68 76.20	" "		343			
48545	76.20 77.72	" "		415			
48546	77.72 79.25	" "		241			









# ZAPPA RESOURCES LTD.

## DOT PROPERTY

DRILL HOLE No. 92RC0-06

LITHOLOGIC LOG

PAGE 3 of 8

FROM	TO	ASSAY NO.	LITHOLOGY	ALTERATION								MINERALIZATION						ZONE	
				SI	QV	SE	CY	CH	EP	CB	KF	hem	PY	CP	AS	PO	VG		MG
30.48	32.0		COBBLE OVERBURDEN TILL.																
			- high % of rock frag - clast frag up to 3cm. - ~20% silt rest cobbles; pebbles variety of types																
32.0	33.53		- as per last sample.																
33.53	35.05		- as per prev. sample																
35.05	36.58		- as per prev. sample																
36.58	38.10		- as above																
38.10	39.62		- as above																
39.62	41.15		- as above - similar to above w/ hem frag grs possibly close to bedrock.																
41.15	42.67																		
42.67	44.20	40561	GRANODIORITE - 30% pebbles contain? , 70% reddish hematitic gndior. - 2% ep alt. frag.																str
44.20	45.72	40562	- reddish - salmon colored gndior. w/ must frags w/ salmon colored hematite, of those ~80% w/ chl alt. ~1-2% ep alt. frag. - traces of mel - 2 frag., trace of spl. hem.																str IV spec.











# ZAPPA RESOURCES LTD.

## DOT PROPERTY

DRILL HOLE No. 92 RCD-06

LITHOLOGIC LOG

PAGE 8 of 8

FROM	TO	ASSAY NO	LITHOLOGY	ALTERATION							MINERALIZATION					ZONE			
				SI	QV	SE	CY	CH	EP	CB	KF	hem	PY	CP	AS		PO	VG	blks.
106.68	108.20	48603	GRANODIORITE CONT'D 38 ft grey clay altered frag, 90% salmon colored hem altered frag, 5% frag, esp: blk bro, bro overall is w/ly chl w/ some blk.				uk	uk											
108.20	109.73	48604	100% of frag of salmon colored hem, bio. mod-str altered text not clear - alter chl., some calc frag - UNTS.					n to str		uk		str							
109.73	111.25	48605	95% uk-mod hem stained frag., bro - starting to become blk w/ only uk chl., acid calc. frag.					uk		uk		M.							
111.25	112.78	48606	v. similar to above sample 100% of frag w/ hem stain. bro - relat blk w/ v. uk chl.					uk				n to str							
112.78	114.3	48607	98% frag v. whight salmon colored hem stained frag (esp w/ fm blk - lt green w/ly chl. frag)					uk				M							
114.3	115.82	48608	v. similar to above w/ 100% of frag uk-mod lt pink hem stain w/ black bro ~ 15%, some blk oxide? on frag. fis					uk				M							uk
			End of hole 115.82.																

## DIAMOND DRILL LEDGER

DDH No. 92RCD-06.

ASSAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu ppm	Non Sulph Cu %	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet					
									0 - 42.67 <u>OVERBURDEN</u>
48561	42.67	44.20	1.52	5.0		.019	.022	42.67	<u>GRANODIORITE</u>
48562	44.20	45.72	"	"		.079	.084		42.67 - 51.82. reddish-salmon colored.
48563	45.72	47.24	"	"		.28	.289	↑	str. hem. stained granodiorite w/ traces
48564	47.24	48.77	"	"		.151	.158	$\frac{.22(.22)}{3.05}$ ↓	to mod malachite staining.
48565	48.77	50.29	"	"		.069	.073		
48566	50.29	51.82	"	"		.045	.048		
48567	51.82	53.34	"	"		.030	.032		51.82 - 77.72 - mod. hem. stained gdiomite
48568	53.34	54.86	"	"		.034	.036		rubly chloritized w/ wk to strong
48569	54.86	56.39	"	"		.037	.041		epidote alterin - trace of malachite
48570	56.39	57.91	"	"		.028	.029		at. 53.34 - 54.86.
48571	57.91	59.44	"	"		260			
48572	59.44	60.96	"	"		370			
48573	60.96	62.48	"	"		311			
48574	62.48	64.0	"	"		181			
48575	64.0	65.53	"	"		281			
48576	65.53	67.06	"	"		139			
48577	67.06	68.58	"	"		142			
48578	68.58	70.10	"	"		168			
48579	70.10	71.63	"	"		150			

## DIAMOND DRILL LEDGER

DDH No. 92RCD-06

ASSAY TAG No.	SAMPLE INTERVAL		SAMPLE LENGTH		Au	Total Cu ppm	Non. Sulph Cu %	Total Cu %	DESCRIPTION
	Metres	Feet	Metres	Feet					
48580	71.63	73.15	1.52	5.0'		.021	.021	GRANODIORITE Cor'd.	
48581	73.15	74.68	"	"		.019	.019		
48582	74.68	76.20	"	"		.020	.020		
48583	76.20	77.72	"	"		.050	.055		
48584	77.72	79.25	"	"		.086	.092	77.72-79.25 Fault - 45% clay clasts w/str.	
48585	79.25	80.77	"	"		.043	.048	hem. - w/ 2% of frag w/ malachite anh.	
48586	80.77	82.30	"	"		.044	.050	1% of frag w/ blk oxide or chalcocite.	
48587	82.30	83.82	"	"		.14	.160	79.25-82.30 strongly hematitic (wkly)	
48588	83.82	85.34	"	"		.663	.867	chloritized ~1% blk frag (oxide-chalcocite)	
48589	85.34	86.87	"	"		.177	.194	82.30-89.92 Zone of native copper,	
48590	86.87	88.39	"	"		.264	.285	malachite, blk fragments (oxide-chalcocite)	
48591	88.39	89.92	"	"		.358	.403	with mod-strong chlorite, wk-strong.	
48592	89.92	91.44	"	"		.187	.259	hematite	
48593	91.44	92.96	"	"		.088	.127	89.92-96.01 - Strong hem. stain,	
48594	92.96	94.49	"	"		.097	.157	mod-strongly chloritized w/ wk-traces	
48595	94.49	96.01	"	"		.106	.143	.26 (.20) of malachite stain. traces of bornite	
48596	96.01	97.54	"	"		.095	.183	16.76 at 91.44-92.96	
48597	97.54	99.06	"	"		.056	.117	96.01-112.78. mod-str hem. stain, wk-mod.	
48598	99.06	100.58	"	"		.027	.061	chlorite alterin, traces of mal to	
48599	100.58	102.11	"	"		.032	.047	106.68.	



APPENDIX IV

1992 Min En Analytical Results





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FAX (604) 980-9621

**SMITHERS LAB.:**  
3176 TATLOW ROAD  
SMITHERS, B.C. CANADA V0J 2N0  
TELEPHONE (604) 847-3004  
FAX (604) 847-3005

**Assay Certificate**

**2V-1304-RA1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-08-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 22 RC CORE samples submitted DEC-03-92 by G.NORMAN.

Sample Number	NON-SUL CU%	TOTAL CU%
48262	.014	.029
48263	.016	.032
48264	.008	.020
48265	.029	.088
48266	.079	.297
48267	.102	.254
48268	.082	.150
48269	.052	.110
48270	.030	.047
48271	.072	.155
48272	.067	.167
48273	.064	.145
48274	.046	.116
48275	.045	.089
48276	.087	.255
48277	.031	.059
48278	.060	.178
48279	.100	.246
48280	.055	.116
48281	.015	.020
48282	.011	.021
48283	.036	.056

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FAX (604) 847-3005

**Geochemical Analysis Certificate**

**2V-1304-RG1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-08-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

*We hereby certify* the following Geochemical Analysis of 24 RC CORE samples submitted DEC-03-92 by G.NORMAN.

Sample Number	TOTAL-CU PPM
48251	128
48252	162
48253	221
48254	200
48255	175
48256	268
48257	296
48258	192
48259	97
48260	131
48261	2080
48284	808
48285	322
48286	549
48287	247
48288	236
48289	347
48290	393
48291	300
48292	4040
48293	669
48294	219
48295	3260
48296	293

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FAX (604) 847-3005

**Geochemical Analysis Certificate**

**2V-1304-RG2**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-08-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

*We hereby certify* the following Geochemical Analysis of 24 RC CORE samples submitted DEC-03-92 by G.NORMAN.

Sample Number	TOTAL-CU PPM
48297	197
48298	120
48299	97
48300	80
48301	117
48302	75
48303	171
48304	30
48305	45
48306	58
48307	12
48308	123
48309	66
48310	10
48311	85
48312	138
48313	53
48314	92
48315	59
48316	74
48317	46
48318	58
48319	43
48320	25

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FAX (604) 847-3005

**Geochemical Analysis Certificate**

**2V-1304-RG3**


Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-08-92**

Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Geochemical Analysis of 6 RC CORE samples submitted DEC-03-92 by G.NORMAN.

Sample Number	TOTAL-CU PPM
48321	13
48322	12
48323	8
48324	451
48325	75
48326	43

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**SMITHERS LAB.:**

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FAX (604) 847-3005

Assay Certificate

2V-1306-RA1

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-10-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 24 RC CORE samples  
submitted DEC-04-92 by G.NORMAN.

Sample Number	NON-SUL CU %	NON-SUL CU %	TOTAL CU %	TOTAL CU %
48327	.039		.045	
48328	.131		.144	
48329	1.660		1.750	
48330	.386		.424	
48331	.133		.219	
48332	.068		.254	
48333	.051	.048	.153	.157
48340	.008		.019	
48341	.057		.143	
48342	.083		.173	
48343	.096		.194	
48344	.105		.802	
48345	.106		.377	
48346	.120		.563	
48347	.133		.453	
48348	.116		.276	
48349	.061	.058	.111	.108
48350	.068		.137	
48351	.097		1.096	
48352	.046		.109	
48353	.068		.203	
48354	.099		.966	
48355	.101		2.500	
48356	.177	.172	1.242	1.251

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FAX (604) 847-3005

**Assay Certificate**

**2V-1306-RA2**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-10-92**

Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 8 RC CORE samples  
submitted DEC-04-92 by G.NORMAN.

Sample Number	NON-SUL CU %	NON-SUL CU %	TOTAL CU %	TOTAL CU %
48357	.069		.155	
48358	.054		.118	
48359	.055		.159	
48360	.032	.030	.063	.060
48361	.031		.065	
48362	.060		.184	
48363	.049		.180	
48364	.031		.092	

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**SMITHERS LAB.:**

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FAX (604) 847-3005

**Geochemical Analysis Certificate**

**2V-1306-RG1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-10-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Geochemical Analysis of 24 RC CORE samples submitted DEC-04-92 by G.NORMAN.

Sample Number	TOTAL-CU PPM
48334	1080
48335	503
48336	594
48337	223
48338	601
48339	631
48365	1430
48366	1960
48367	1690
48368	1000
48369	1120
48370	3950
48371	2290
48372	4530
48373	3430
48374	948
48375	1050
48376	308
48377	1000
48378	571
48379	353
48380	393
48381	370
48382	93

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**SMITHERS LAB.:**  
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SMITHERS, B.C. CANADA V0J 2N0  
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**Geochemical Analysis Certificate**

**2V-1306-RG2**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-10-92**

Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

*We hereby certify* the following Geochemical Analysis of 4 RC CORE samples submitted DEC-04-92 by G.NORMAN.

Sample Number	TOTAL-CU PPM
48383	287
48384	253
48385	589
48386	133

---

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**SMITHERS LAB.:**  
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FAX (604) 847-3005

Assay Certificate

2V-1306-PA1

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-15-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 15 RC CORE samples submitted DEC-04-92 by G.NORMAN.

Sample Number	CU TOTAL %	NON-SUL CU %
48334	.095	.042
48365	.142	.060
48366	.197	.095
48367	.143	.066
48368	.102	.052
48369	.115	.055
48370	.378	.125
48371	.236	.102
48372	.438	.074
48373	.347	.088
48374	.101	.038
48375	.107	.043
48376	.032	.012
48377	.085	.031

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FAX (604) 980-9621

**SMITHERS LAB.:**  
3176 TATLOW ROAD  
SMITHERS, B.C. CANADA V0J 2N0  
TELEPHONE (604) 847-3004  
FAX (604) 847-3005

Assay Certificate

**2V-1307-PA3**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-16-92**  
Copy 1. **ZAPPA RESOURCES LTD., VANCOUVER, B.C.**

We hereby certify the following Assay of 3 PULP samples submitted DEC-04-92 by GEORGE NORMAN.

Sample Number	* CU %	** CU %
48410	.905	.908
48412	.266	.263
48421	.112	.115

\* 10 GRAM SAMPLE, 100 ML OF 3% H2SO4 LEACHING FOR 3 HOURS.  
\*\*10 GRAM SAMPLE, 100 ML OF 10% H2SO4 LEACHING FOR 3 HOURS.

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**SMITHERS LAB.:**  
3176 TATLOW ROAD  
SMITHERS, B.C. CANADA V0J 2N0  
TELEPHONE (604) 847-3004  
FAX (604) 847-3005

**Assay Certificate**

**2V-1307-RA1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-04-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 24 RC CORE samples submitted DEC-04-92 by GEORGE NORMAN.

Sample Number	NON-SUL CU %	NON-SUL CU %	TOTAL CU %	TOTAL CU %
48401	.226		.240	
48402	.210		.239	
48403	.202		.248	
48404	.208		.255	
48405	.295		.385	
48406	.734	.718	1.680	1.690
48407	.397		.772	
48408	.557		2.400	
48409	.779		6.140	
48410	1.123		10.780	
48411	.399		2.610	
48412	.395		1.326	
48413	.266		.653	
48414	.153		1.105	
48415	.113	.107	.420	.417
48416	.098		.452	
48417	.135		.502	
48418	.097		1.400	
48419	.114		.725	
48420	.111		.656	
48421	.145		1.730	
48422	.143	.142	1.238	1.227
48423	.154		.868	
48424	.115		.923	

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**SMITHERS LAB.:**  
3176 TATLOW ROAD  
SMITHERS, B.C. CANADA V0J 2N0  
TELEPHONE (604) 847-3004  
FAX (604) 847-3005

***Geochemical Analysis Certificate***

**2V-1307-RG1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-10-92**

Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

*We hereby certify* the following Geochemical Analysis of 24 RC CORE samples submitted DEC-04-92 by GEORGE NORMAN.

Sample Number	TOTAL-CU PPM
48387	520
48388	171
48389	141
48390	98
48391	162
48392	191
48393	261
48394	208
48395	232
48396	248
48397	239
48398	715
48399	1330
48400	2040
48435	2410
48436	1190
48437	2560
48438	1570
48439	1320
48440	1340
48441	769
48442	783
48443	1480
48444	1260

Certified by \_\_\_\_\_

**MIN-EN LABORATORIES**



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**SPECIALISTS IN MINERAL ENVIRONMENTS**  
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**VANCOUVER OFFICE:**  
705 WEST 15TH STREET  
NORTH VANCOUVER, B.C. CANADA V7M 1T2  
TELEPHONE (604) 980-5814 OR (604) 988-4524  
FAX (604) 980-9621

**SMITHERS LAB.:**  
3176 TATLOW ROAD  
SMITHERS, B.C. CANADA V0J 2N0  
TELEPHONE (604) 847-3004  
FAX (604) 847-3005

---

---

**Geochemical Analysis Certificate**

**2V-1307-RG2**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-10-92**

Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

*We hereby certify* the following Geochemical Analysis of 4 RC CORE samples submitted DEC-04-92 by GEORGE NORMAN.

Sample Number	TOTAL-CU PPM
48445	1530
48446	564
48447	1010
48448	982

---

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TELEPHONE (604) 847-3004  
FAX (604) 847-3005

**Assay Certificate**

**2V-1307-RA2**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

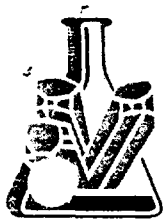
Date: **DEC-15-92**  
Copy 1. **ZAPPA RESOURCES LTD., VANCOUVER, B.C.**

We hereby certify the following Assay of 10 RC CORE samples  
submitted DEC-04-92, by **GEORGE NORMAN.**

Sample Number	NON-SUL CU %	NON-SUL CU %	TOTAL CU %	TOTAL CU %	NATIVE CU %
48425	.107		.938		.012
48426	.131	.140	1.520	1.510	.078
48427	.163		.593		
48428	.085		.436		
48429	.142		.441		
48430	.094		.328		
48431	.156		.315		
48432	.078		.341		
48433	.100		.236		
48434	.057		.185		

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TELEPHONE (604) 847-3004  
FAX (604) 847-3005

Assay Certificate

2V-1307-PA1

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-15-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 12 RC CORE samples submitted DEC-04-92 by GEORGE NORMAN.

Sample Number	CU TOTAL %	NON-SUL CU %
48399	.141	.130
48400	.196	.175
48435	.251	.121
48436	.135	.058
48437	.269	.116
48438	.166	.068
48439	.150	.082
48440	.153	.083
48441	.088	.044
48442	.083	.034
48443	.164	.081
48444	.129	.048

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FAX (604) 847-3005

**Assay Certificate**

**2V-1307-PA2**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-15-92**  
Copy 1. **ZAPPA RESOURCES LTD., VANCOUVER, B.C.**

*We hereby certify* the following Assay of 4 RC CORE samples submitted DEC-04-92 by **GEORGE NORMAN**.

Sample Number	CU TOTAL %	NON-SUL CU %
48445	.160	.091
48446	.061	.031
48447	.121	.061
48448	.106	.045

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TELEPHONE (604) 847-3004  
FAX (604) 847-3005

**Assay Certificate**

**2V-1327-RA1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-18-92**

Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 24 RC CORE samples submitted DEC-14-92 by GEORGE NORMAN.

Sample Number	NON-SUL CU %	TOTAL CU %
48464	.223	.228
48465	.582	.710
48466	.582	1.321
48467	.363	1.200
48468	.749	1.375
48469	.740	1.210
48470	.407	.513
48471	.167	.212
48472	.146	.175
48473	.206	.260
48474	.170	.206
48475	.096	.133
48476	.085	.147
48477	.101	.178
48478	.104	.240
48479	.086	.175
48480	.050	.097
48481	.080	.136
48482	.063	.102
48483	.069	.120
48484	.034	.052
48485	.163	.322
48486	.137	.387
48487	.096	.191

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**DEC 29 1992**



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FAX (604) 847-3005

**Assay Certificate**

**2V-1327-RA1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-18-92**

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Sample Number	NON-SUL CU %	TOTAL CU %
48464	.223	.228
48465	.582	.710
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48474	.170	.206
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48476	.085	.147
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48479	.086	.175
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48482	.063	.102
48483	.069	.120
48484	.034	.052
48485	.163	.322
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**Assay Certificate**

**2V-1327-RA2**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-18-92**  
Copy 1. **ZAPPA RESOURCES LTD., VANCOUVER, B.C.**

*We hereby certify* the following Assay of 13 RC CORE samples  
submitted DEC-14-92 by **GEORGE NORMAN**.

Sample Number	NCN-SUL CU %	TOTAL CU %
48488	.122	.242
48489	.118	.501
48490	.162	.563
48491	.137	.391
48492	.077	.160
48493	.076	.134
48494	.052	.102
48495	.119	.482
48496	.105	.764
48497	.134	.448
48498	.077	.204
48499	.112	.224
48500	.104	.618

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**SMITHERS LAB.:**

3176 TATLOW ROAD  
SMITHERS, B.C. CANADA V0J 2N0  
TELEPHONE (604) 847-3004  
FAX (604) 847-3005

**Assay Certificate**

**2V-1318-RA1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-15-92**

Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 24 RC CORE samples  
submitted DEC-10-92 by GEORGE NORMAN.

Sample Number	N-SULP CU%	TOTAL CU%
48507	.018	.023
48508	.013	.015
48509	.070	.076
48510	1.256	1.320
48511	.086	.093
48512	.204	.219
48513	.126	.138
48514	.117	.126
48515	.146	.157
48516	.211	.224
48517	.139	.156
48518	.161	.180
48519	.132	.148
48520	.157	.180
48521	.123	.136
48522	.284	.323
48523	.173	.195
48524	.119	.132
48525	.096	.108
48526	.092	.100
48527	.151	.165
48528	.207	.233
48529	.182	.201
48530	.173	.191

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FAX (604) 847-3005

**Assay Certificate**

**2V-1318-RA2**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-15-92**  
Copy 1. **ZAPPA RESOURCES LTD., VANCOUVER, B.C.**

*We hereby certify* the following Assay of 11 RC CORE samples submitted DEC-10-92 by **GEORGE NORMAN**.

Sample Number	N-SULP CU%	TOTAL CU%
48531	.283	.319
48532	.142	.169
48533	.034	.046
48534	.031	.041
48535	.027	.045
48536	.027	.043
48537	.038	.055
48538	.066	.187
48539	.095	.815
48540	.026	.109
48541	.041	.122

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**SMITHERS LAB.:**  
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FAX (604) 847-3005

**Geochemical Analysis Certificate**

**2V-1318-RG1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-15-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Geochemical Analysis of 19 RC CORE samples submitted DEC-10-92 by GEORGE NORMAN.

Sample Number	TOTAL-CU PPM
48542	424
48543	605
48544	343
48545	415
48546	241
48547	342
48548	166
48549	289
48550	149
48551	701
48552	332
48553	171
48554	199
48555	354
48556	563
48557	371
48558	148
48559	201
48560	147

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FAX (604) 847-3005

**Geochemical Analysis Certificate**

**2V-1319-RG1**

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-17-92**

Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Geochemical Analysis of 15 RC CORE samples submitted DEC-10-92 by GEORGE NORMAN.

Sample Number	TOTAL-CU PPM
48571	260
48572	370
48573	311
48574	181
48575	281
48576	139
48577	142
48578	168
48579	150
48603	586
48604	616
48605	638
48606	419
48607	407
48608	625

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TELEPHONE (604) 847-3004  
FAX (604) 847-3005

Assay Certificate

2V-1319-RA1

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

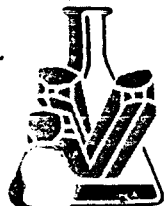
Date: **DEC-17-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 24 RC CORE samples submitted DEC-10-92 by GEORGE NORMAN.

Sample Number	NON-SUL CU %	TOTAL CU %
48561	.019	.022
48562	.079	.084
48563	.280	.289
48564	.151	.158
48565	.069	.073
48566	.045	.048
48567	.030	.032
48568	.034	.036
48569	.037	.041
48570	.028	.029
48580	.021	.021
48581	.019	.019
48582	.020	.020
48583	.050	.055
48584	.086	.092
48585	.043	.048
48586	.044	.050
48587	.140	.160
48588	.663	.867
48589	.177	.194
48590	.264	.285
48591	.358	.403
48592	.187	.259
48593	.088	.127

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Assay Certificate

2V-1319-RA2

Company: **ZAPPA RESOURCES LTD.**  
Project: **DOT**  
Attn: **WAYNE ROBERTS**

Date: **DEC-17-92**  
Copy 1. ZAPPA RESOURCES LTD., VANCOUVER, B.C.

We hereby certify the following Assay of 9 RC CORE samples submitted DEC-10-92 by GEORGE NORMAN.

Sample Number	NON-SUL CU %	TOTAL CU %
48594	.097	.157
48595	.106	.143
48596	.095	.183
48597	.056	.117
48598	.027	.061
48599	.032	.047
48600	.013	.022
48601	.039	.063
48602	.046	.089

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APPENDIX V

Min En Analytical Techniques

**MINERAL  
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Division of Assayers Corp. Ltd.

**AG, CU, PB, ZN, NI, AND CO ASSAY PROCEDURE**  
-----

Samples are dried @ 95 C and when dry are crushed on a jaw crusher. The -1/4 inch output of the jaw crusher is put through a secondary roll crusher to reduce it to -1/8 inch. The whole sample is then riffled on a Jones Riffle down to a statistically representative 300-400 gram sub-sample (in accordance with Gy's statistical rules.) This sub-sample is then pulverized in a ring pulverizer to 95% minus 120 mesh, rolled and bagged for analysis. The remaining reject from the Jones Riffle is bagged and stored.

A 2.200 - 3.000 gram sub-sample is weighed from the pulp bag for analysis. Each batch of 70 assays has a natural standard and a reagent blank included. The assays are digested using a HNO<sub>3</sub> - KClO<sub>4</sub> mixture and when reaction subsides, HCL is added to assay before it is placed on a hotplate to digest. After digestion is complete the assays are cooled, diluted to volume and mixed.

The assays are analyzed on atomic absorption spectrometers using the appropriate standard sets. The natural standard digested along with this set must be within 2 standard deviations of its known or the whole set is re-assayed. If any of the assays are >1% they are re-assayed at a lower weight.

OFFICE AND LABORATORIES:  
705 WEST FIFTEENTH STREET, NORTH VANCOUVER, B.C.  
CANADA V7M 1T2

PHONE: (604) 980-5814 (604) 988-4524  
TELEX: VIA USA 7601067  
FAX: (604) 980-9621



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FAX (604) 980-9621

SMITHERS LAB.:  
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SMITHERS, B.C. CANADA V0J 2N0  
TELEPHONE (604) 847-3004  
FAX (604) 847-3005

**ANALYTICAL PROCEDURE REPORT FOR ASSESSMENT WORK:**

**PROCEDURE FOR 31 ELEMENT TRACE ICP**

Ag, Al, As, B, Ba, Be, Bi, Ca, Cd, Co, Cu,  
Fe, K, Li, Mg, Mn, Mo, Na, Ni, P, Pb, Sb,  
Sr, Th, Ti, V, Zn, Ga, Sn, W, Cr and Geochemical Analysis

Samples are processed by Min-En Laboratories, at 705 West 15th Street, North Vancouver, employing the following procedures.

After drying the samples at 95 C, soil and stream sediment samples are screened by 80 mesh sieve to obtain the minus 80 mesh fraction for analysis. The rock samples are crushed by a jaw crusher and pulverized by ceramic plated pulverizer or ring mill pulverizer.

0.5 gram of the sample is digested for 2 hours with an aqua regia mixture.

After cooling samples are diluted to standard volume. The solutions are analysed by computer operated Jarrell Ash 9000 ICAP or Jobin Yvon 70 Type II Inductively Coupled Plasma Spectrometers. Reports are formatted and printed using a laser printer.

For Geochemical Analysis samples are tested by Atomic Absorption (A.A.) techniques rather than the above I.C.P. method.



**MINERAL  
• ENVIRONMENTS  
LABORATORIES**  
(DIVISION OF ASSAYERS CORP.)

**SPECIALISTS IN MINERAL ENVIRONMENTS**  
CHEMISTS • ASSAYERS • ANALYSTS • GEOCHEMISTS

**VANCOUVER OFFICE:**

705 WEST 16TH STREET  
NORTH VANCOUVER, B.C. CANADA V7M 1T2  
TELEPHONE (604) 980-5814 OR (604) 988-4524  
FAX (604) 980-9821

**SMITHERS LAB.:**

3178 TATLOW ROAD  
SMITHERS, B.C. CANADA V0J 2N0  
TELEPHONE (604) 847-3004  
FAX (604) 847-3005

**ASSESSMENT WORK FOR ZAPPA RESOURCES**  
-----

**NON-SUL CU %**

- 3% sulphuric acid
- boil 30 minutes
- then AA finish

**COPER OXIDE**

- 20% ammonium acetate hot leaching
- AA finish

**NATIVE COPER**

- 25% ammonium sulphate
- cold leaching for 3 hours
- then 20% ammonium acetate hot leaching
- AA finish

The result of the solution - Copper Oxide = Native Copper.

**SPECIAL CU LEACHING**

- using 3% and 10% of H2SO4
- leach for 3 hours
- AA finish



APPENDIX VI

Preliminary Evolution of the Leachability  
of Ores from the Dot Property  
of Zappa Resources

A. BRUYNESTEYN AND ASSOCIATES  
MINERAL LEACHING CONSULTANTS

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PHONE (604) 987-3183  
FAX (604) 987-5124

2175 GREYLYNN CRESCENT  
NORTH VANCOUVER, B.C.  
CANADA V7J 2X6

PRELIMINARY EVALUATION OF THE LEACHABILITY OF ORES FROM THE DOT  
PROPERTY OF ZAPPA RESOURCES.

Prepared for:

Zappa Resources Ltd.  
1500-675 West Hastings Street  
Vancouver, V6B 1N2

Prepared by:

A. Bruynesteyn & Associates  
2175 Greyllynn Crescent  
North Vancouver, V7J 2X6

January 19, 1993

## EXECUTIVE SUMMARY

1. The ore, when finely pulverized, is a relatively high acid consumer, but practical experiences shows that the ore, when dump/heap leached in -3" or -2" particle size, will probably consume only approximately 27 kg acid per tonne. Any sulphide in the ore can decrease this consumption at a rate of 30 kg/t for every 1% of sulphide present. With acid cost at \$20/t delivered at mine site, it is estimated that acid cost will amount to \$0.04-\$0.06 per pound of copper produced.
2. During the acid leach, 1.87 kg of copper was extracted per tonne of ore (4.1 lb/t)
3. This ore body is unique in that the sulphide ores consist mainly of chalcocite, a copper mineral that leaches hundreds of times faster and better than chalcopyrite.
4. It is anticipated that a properly designed heap leach operation will yield 70% extraction at operational costs of less than 37 cents per pound of cathode copper produced.
5. It is anticipated that a 2,800 t/d mining operation will produce 10 t/d cathode copper

## INTRODUCTION

Zappa Resources has engaged Westcoast Biotech to assess the chemical and biological leaching characteristics of a sample of ore from the DOT property. As a first stage in this assessment, the acid consuming characteristics of the ore were assessed. This report details the results of this assessment and disusses the potential viability of the proposed dump/heap leaching operation.

## RESULTS

### 1. ACID CONSUMPTION

To determine the acid consumption of this ore under typical dump/heap leach conditions, 10 gram of finely pulverized sample was suspended in 100 ml of neutral pH water. This suspension was stirred continuously while 1N sulphuric acid was added automatically. The addition of acid was stopped when the solution pH reached a value of 3.0 and the amount of acid added was noted. Subsequently, the acid addition was continued to pH 2.5 and 2.0, and the amounts of acid consumed noted.

## ACID CONSUMPTION AS A FUNCTION OF ACIDITY

Suspension pH	Acid consumed ml 1N H <sub>2</sub> SO <sub>4</sub>	Acid consumed kg/t
pH 3.0	11.25	67.3
pH 2.5	13.75	82.2
pH 2.0	16.20	96.9

## 2. EXPECTED COMMERCIAL ACID CONSUMPTION

It must be kept in mind that these tests were performed on finely pulverized material, providing optimum contact between ore and acid. In practise, acid consumption from -2" ore will be significantly less, probably only 27 kg/t when operating at pH 2.5. This amount will be further reduced by any acid produced biologically from the sulphides present. If the ore contains 0.5% copper as chalcocite, then this copper is associated with 0.13% sulphur. Any pyrite will add to this sulphur balance. When sulphur is oxidized, every 1% sulphur produces 30 kg of sulphuric acid per tonne. Therefore, the reported 0.5% sulphur content of the ore can produce up to 15 kg/t acid. Assuming that only 10 kg/t will be produced in practise, the acid consumption of the ore in a dump/heap leach will be 27-10 = 17 kg/t. Since acid can probably be supplied at mine site for \$20-\$30 per tonne, acid cost will be \$0.34-\$0.51 per tonne of ore treated.

The ore contains 0.5% copper, or 5 kg/t. Assuming that we can extract 75% of this, or 3.75 kg/t, the acid cost per kg of copper produced will be approximately 9.1-13.6 cents, equivalent to 4.1-6.2 cents per pound of copper.

## ANTICIPATED METALLURGICAL PERFORMANCE

## 1. CHEMISTRY AND BIOLOGY

When chalcocite is leached, there are two typical extraction phases. In the first phase, the chalcocite is converted into copper sulphate and covellite.



This reaction is mildly oxidative and involves the oxidation of the cuprous copper to cupric copper. Small amounts of ferric iron in the leach solution will readily accomplish this. The acid required for this reaction will be recovered in the solvent extraction plant.

The subsequent oxidation of the covelite into copper sulphate requires a much stronger oxidant and here is where the leaching bacteria play an important role. They readily oxidize the sulphide into sulphate, using oxygen from the air. Therefore, a properly designed heap leach operation, using DOT ores, should give 70-80% extraction in 34 months.

## 2. ANTICIPATED EXTRACTION

In the acid consumption test described, the acid added did extract 18.7 mg of copper from the 10.00 grams of ore tested. This is probably readily acid soluble copper, such as copper oxide, but it can also partly be from the chalcocite. Most B.C ores contain the primary mineral chalcocite, which leaches very slowly and, when treated in dumps, typically will give only 2-5% extraction per year. Chalcocite on the other hand leaches rapidly. At the Miami Globe mine of the Cyprus Copper Company, a chalcocite/chalcocopyrite ore is heap leached at a rate of 75,000 t/d in 150,000 tonnes heaps over a 120 day period. During that period, 60% copper extraction is obtained from the chalcocite, while a further 8% is obtained subsequently. The unleached residual copper is present mainly as chalcocopyrite and will not leach to any extent in this short time period. Since the DOT sulphide ore is mainly chalcocite, it is expected that 70-80% extraction should be possible over a 120 day period.

## 3. ANTICIPATED PLANT PERFORMANCE

It is anticipated that the DOT ore, containing 0.5% copper, will give at least 70% extraction in 4 months. This is equivalent to a production of 8,250,000 pounds of copper per million tonnes mined and treated.

A small mining operation of 1,400 t/d ore containing 0.5% copper, would produce 5 t/d cathode copper and a 2,800 t/d operation would produce 10 t/d. With 75% extraction, the size of the mining operation would be reduced to 2,615 t/d.

## 4. ANTICIPATED CAPITAL AND OPERATING COST

Based on the anticipated metallurgical performance of the DOT ore, the following capital and operating cost are estimated for a 10 t/d cathode copper plant with the following specifications:

Capacity 10 t/d cathode copper  
 Leach ore: 10,000,000 tonnes  
 Ore grade 0.6% copper as chalcocite & bornite  
 Extraction: 70% Acid requirement: \$0.06/lb Copper  
 Extractable copper: 84,000,000 lb.  
 Yearly production: 7,000,000 lb.  
 Life of leach plant: 12 years

Operating days: 350/year  
 Power cost: \$0.08/Kwh

Capital cost

SX-EW plant	\$3,500,000
Buildings	400,000
Site preparation	300,000
Pumps & Piping	200,000
Laboratory & Offices	100,000
	-----+
TOTAL	\$4,500,000

Operating costs:

1.	LABOUR		
		No.	
	Labour 2/shift	8	
	Maintenance	2	
	Warehouse	1	
	Supervisor	1	
	Accounting/sales	1	
	Manager	1	
		-----+	
	TOTAL	14	cents/lb
	Yearly cost @\$40,000 each = \$560,000		8.0
2.	Water \$40,000 per year		0.6
3.	Electricity 1.5 Kwh/lb copper		12.0
4.	Sulphuric acid		6.0
5.	Reagents \$160,000 per year		2.3
6.	Freight \$160,000 per year		2.3
7.	Laboratory \$100,000 per year		1.4
8.	Office & O/H \$150,000 per year		2.1
9.	Abandonment \$150,000 per year		2.1
			-----+
	Total operating cost		36.8 c/lb

If the \$4.5 million capital is to be repaid over 3 years, at 20% interest, the project must repay 7.7 million. Production over the three years will be 21 million pounds, so that the capital cost

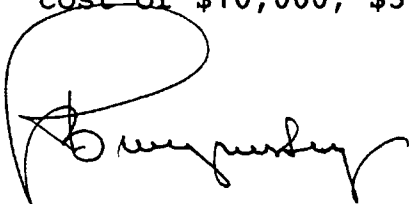
will be 36.6 c/lb. The total of operating and capital cost will be  $36.8 + 36.6 = 73.4$  c/lb. To this must be added the mining cost.

#### RECOMMENDATIONS

It is my opinion that the ore from the DOT property will be highly amenable to the dump/heap leach process. The only possible severe restriction may be associated with the ore's acid consumption. Practical experience has shown that such acid consumption from commercial sized material is often only a fraction of the consumption of pulverized ore. It is therefore recommended that a simple, inexpensive column test be carried out to determine the acid consumption of -2" ore. To do this, 50-100 pounds of ore is charged to a column and treated continuously with a solution of pH 2.0. When the acid consumption stabilizes, probably after 1-2 months, the ore will be inoculated with an active culture of the leaching bacteria and the progress of the subsequent biological copper extraction will be followed closely until maximum extraction has been reached. In the unlikely case that the acid consumption turns out to be unrealistically high, the test will be terminated before the biological phase is introduced.

We will be pleased to carry out this testing program for you. The cost of the first phase, representing the acid leach, will be \$3,500. TAP, the Technical Assistance Program of the Government, will pay \$1,250 of this cost, so Zappa's cost will be only \$2,250.

If the second biological phase is justified, we will do this at a cost of \$10,000, \$3,750 of which will again be paid by TAP.



A. Bruynesteyn  
President

**A. BRUYNESTEYN AND ASSOCIATES**  
MINERAL LEACHING CONSULTANTS

---

PHONE (604) 987-3183  
FAX (604) 987-5124

2175 GREYLYNN CRESCENT  
NORTH VANCOUVER, B.C.  
CANADA V7J 2X6

**C E R T I F I C A T E**

I, Albert Bruynesteyn do hereby certify that I am an independent Consultant in Mineral Leaching, with offices at 2175 Greyllynn Crescent, North Vancouver, B.C. V7J 2X6.

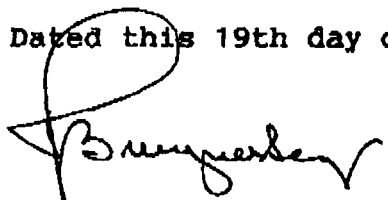
**I FURTHER CERTIFY THAT**

1. I am a chemical engineer with 25 years of experience in mineral biotechnology, 17 of which were with the B.C. Research Corporation where I was Head of the Division of Extractive Metallurgy. I have three patents on biological treatment processes for sulphide ores and have been an independent consultant since 1984 specializing in the commercial application of mineral biotechnology. I have consulted on most major biological leaching operations world wide.

2. I have based this report on the results of test work carried out on the DOT ore and on my experience with mining operations where ores with similar mineralogy are being treated.

3. I have no interest in the DOT property, nor do I expect to receive any sort of interest.

Dated this 19th day of January 1993 in Vancouver, B.C.



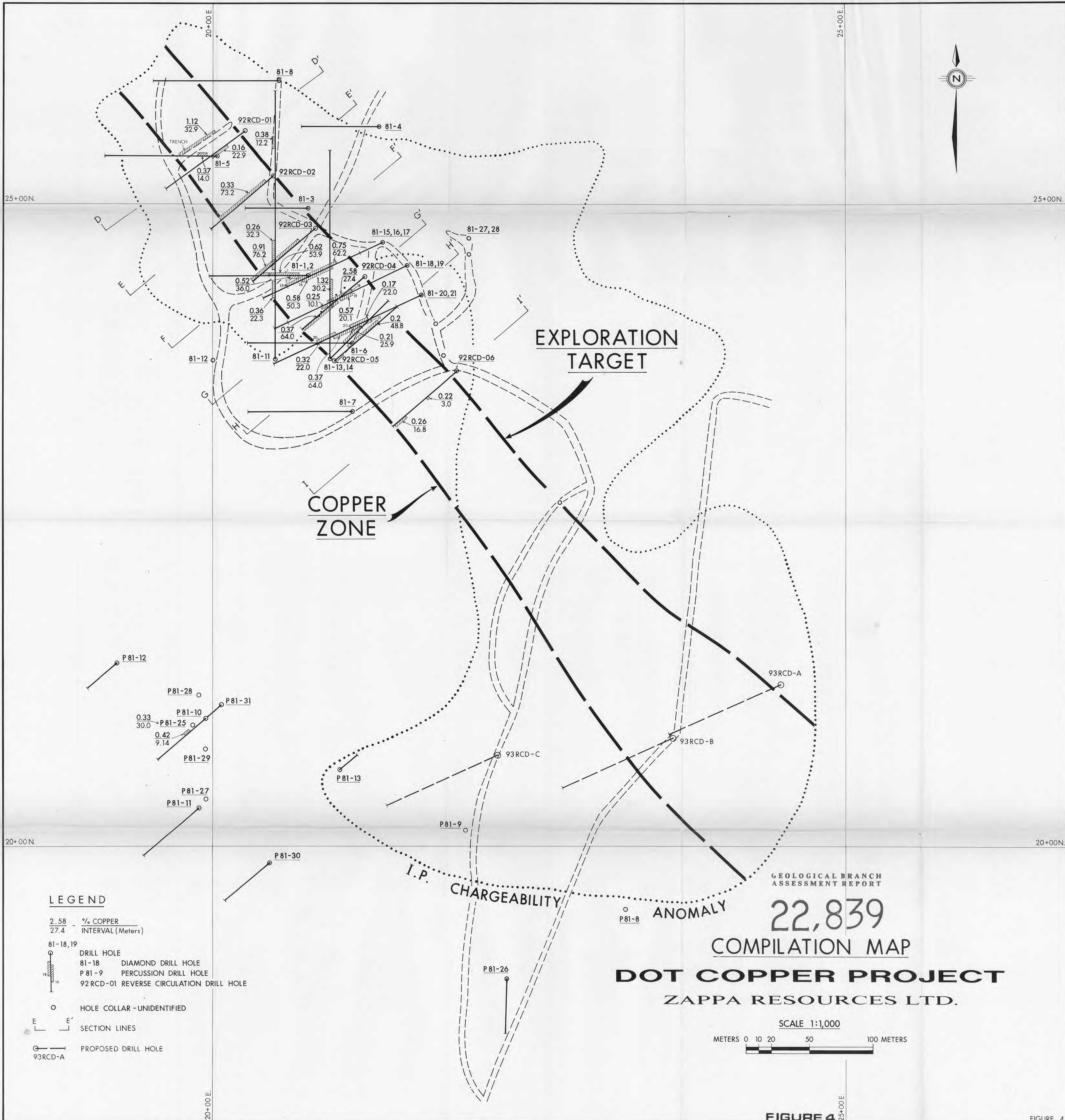
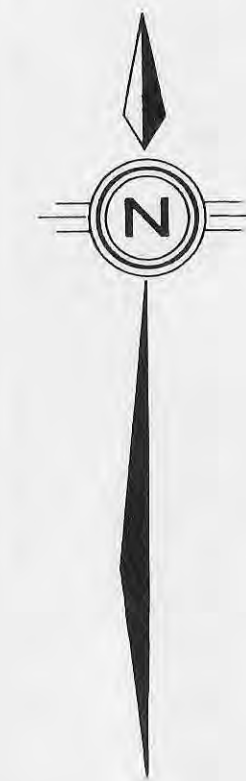
A. Bruynesteyn, Ph.D.



APPENDIX VII  
Composite Sample  
Listing of Component Samples

LIST OF SAMPLES FOR COMPOSITE SAMPLE

<u>Drill Hole</u>	<u>Assay No.</u>	<u>Assay % Cu</u>	<u>Average % Cu</u>
<u>92RCD-02</u>	48330	.424	
	48341	.143	
	48351	1.096	
	48362	0.184	
	48372	0.438	.46
<u>92RCD-03</u>	48405	0.385	
	48414	1.105	
	48424	0.923	
	48432	0.341	
	48443	0.164	.5066
<u>92RCD-04</u>	48466	1.321	
	48474	0.206	
	48481	0.136	
	48489	0.501	
	48497	0.448	.5224
<u>92RCD-05</u>	48510	1.32	
	48516	0.224	
	48522	0.323	
	48528	0.233	
	48534	0.041	.4282
		<hr/>	
		Average of Total	.48



**LEGEND**

- 2.58 - % COPPER
- 27.4 - INTERVAL (Meters)
- 81-18,19 - DRILL HOLE
- 81-18 - DIAMOND DRILL HOLE
- P 81-9 - PERCUSSION DRILL HOLE
- 92 RCD-01 - REVERSE CIRCULATION DRILL HOLE
- - HOLE COLLAR - UNIDENTIFIED
- E - E' - SECTION LINES
- - PROPOSED DRILL HOLE
- 93RCD-A

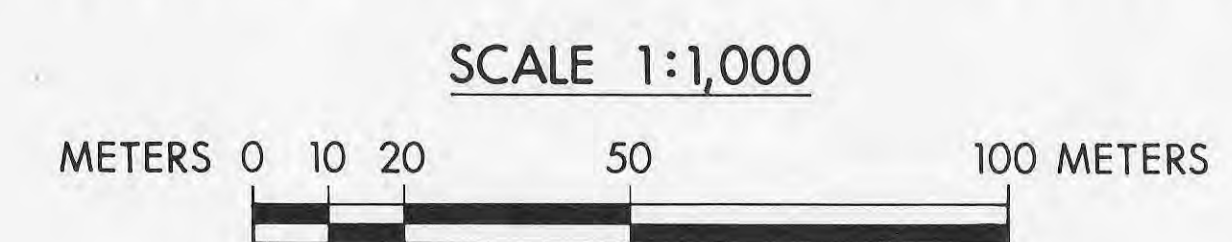
GEOLOGICAL BRANCH  
ASSESSMENT REPORT

**22,839**

**COMPILATION MAP**

**DOT COPPER PROJECT**

**ZAPPA RESOURCES LTD.**



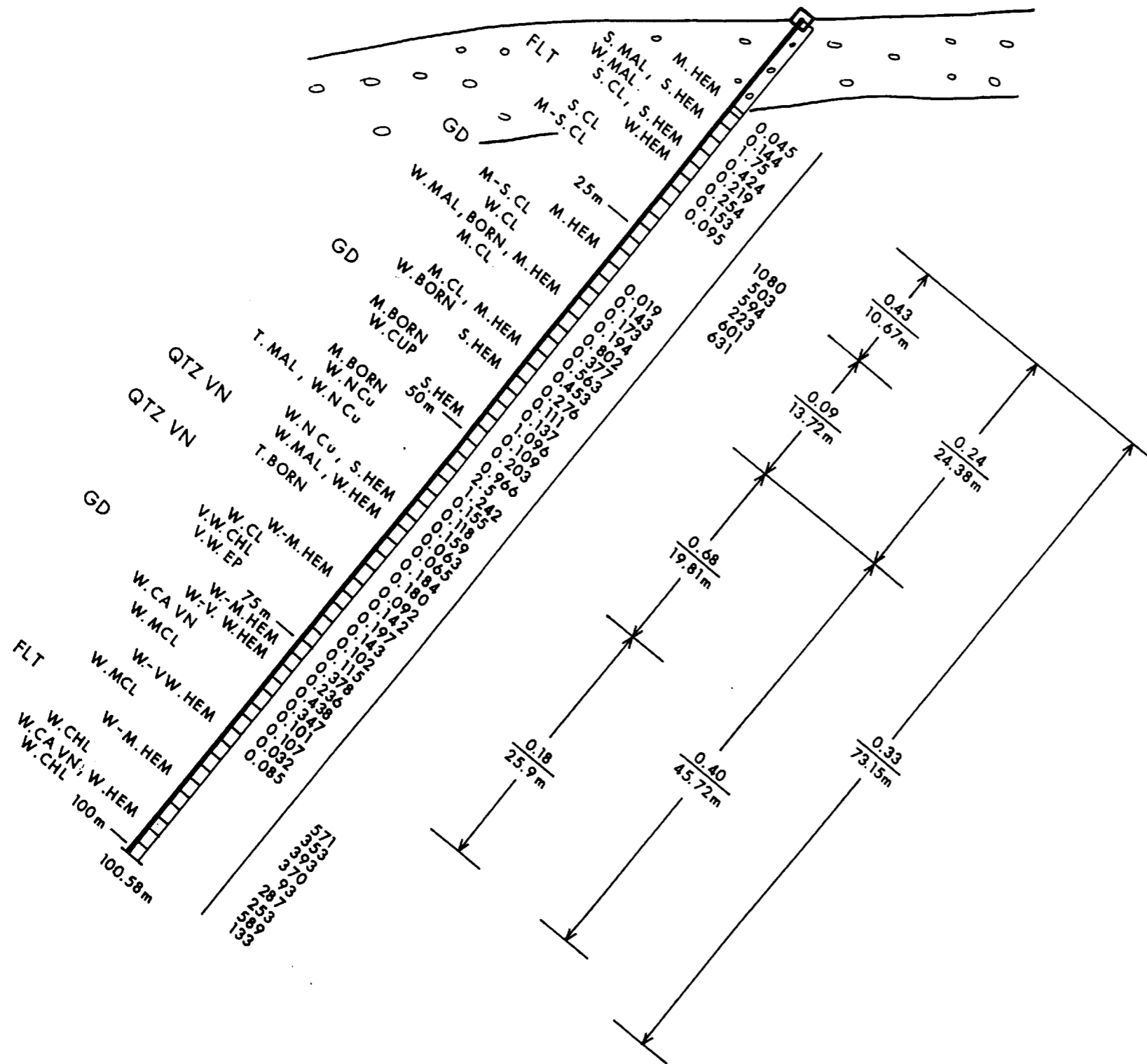
**FIGURE 4**





### 92 RCD - 02

DIP -50°  
AZIMUTH 230°  
LENGTH 100.58m



## LEGEND

### LITHOLOGIC UNITS

- Overburden
- LOWER JURASSIC**
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

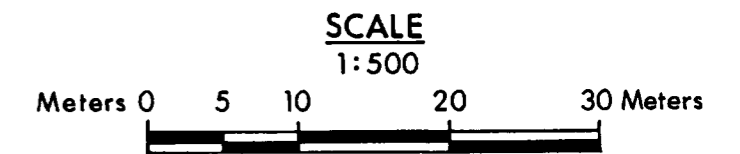
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

Total Copper	
0.141	1330
% Cu	ppm Cu
↑	
AVERAGE GRADE	
0.91	
↓	
76.2 m	
Total Length metres	

### MISCELLANEOUS

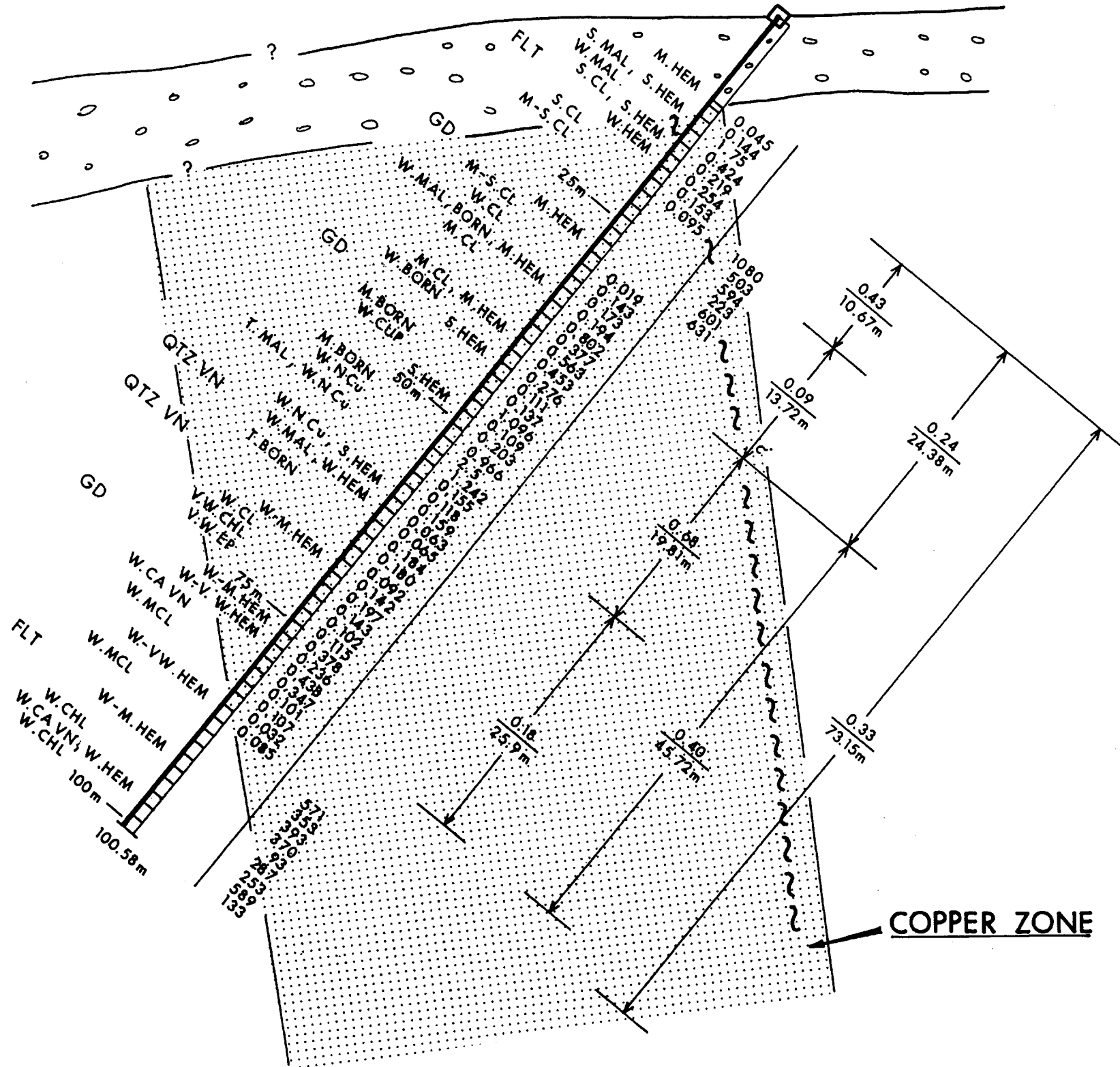
- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION E-E'</b>		
<b>92RCD-02-GEOLOGY-ASSAYS</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 921/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	7
DATE: Jan./1993	NICOLA M.D.	

# 92 RCD-02

DIP -50°  
 AZIMUTH 230°  
 LENGTH 100.58m



## LEGEND

### LITHOLOGIC UNITS

- Overburden
- GD Grandodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

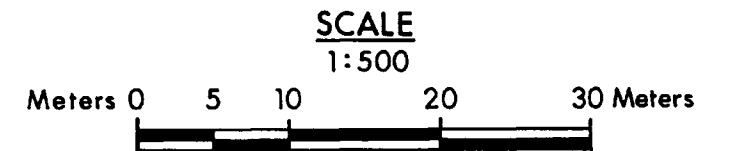
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

Total Copper		% Cu	ppm Cu
0.141	1330		
AVERAGE GRADE		% Total Copper	Total Length metres
0.91			76.2 m

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



ZAPPA RESOURCES LTD.  
 DOT PROPERTY

## CROSS SECTION E-E' COPPER ZONE

VIEW AZIMUTH 320°

GEOLOGY BY: George Norman	N.T.S. 921/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	8
DATE: Jan./1993	NICOLA M.D.	





**92 RCD-03**

DIP -50°  
 AZIMUTH 230°  
 LENGTH 100.58m

**LEGEND**

**LITHOLOGIC UNITS**

- Overburden
- LOWER JURASSIC**
- GD Granodiorite  
Guichon Batholith

**ALTERATION**

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

**DEGREE OF ALTERATION**

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

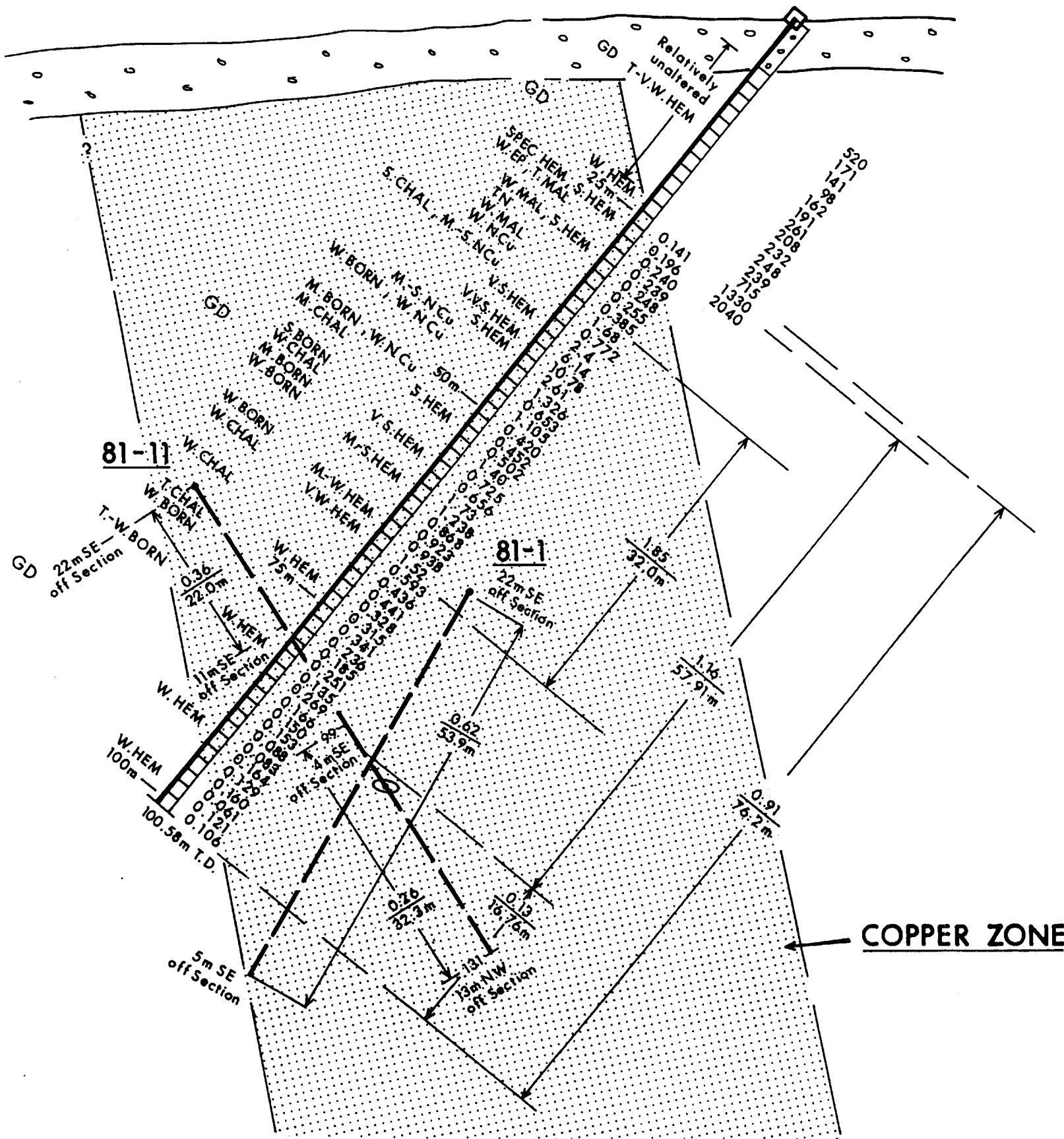
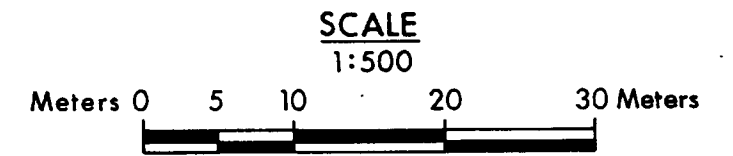
**MINERALIZATION**

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

0.141	1330	Total Copper
		%Cu   ppmCu
↑		<b>AVERAGE GRADE</b>
0.91		% Total Copper
↓		Total Length metres
76.2 m		

**MISCELLANEOUS**

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



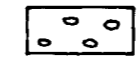
<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION F-F'</b>		
<b>COPPER ZONE</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 92I/7 W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	<b>10</b>
DATE: Jan./1993	NICOLA M.D.	

# 92RCD-04

DIP -50°  
AZIMUTH 230°  
LENGTH 100.58m

## LEGEND

### LITHOLOGIC UNITS

-  Overburden
- LOWER JURASSIC**
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

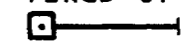
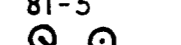

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

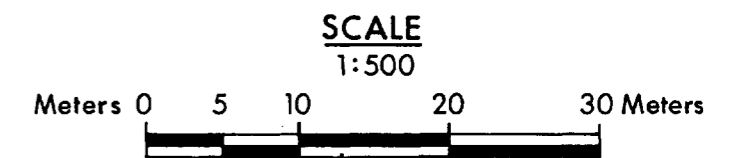
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

Total Copper	
%Cu	ppmCu
0.141	1330
AVERAGE GRADE	
% Total Copper	Total Length metres
0.91	76.2 m

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01  Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5  Diamond Drill Hole - On Section,  
Off Section
-  Pierce Point Of Section



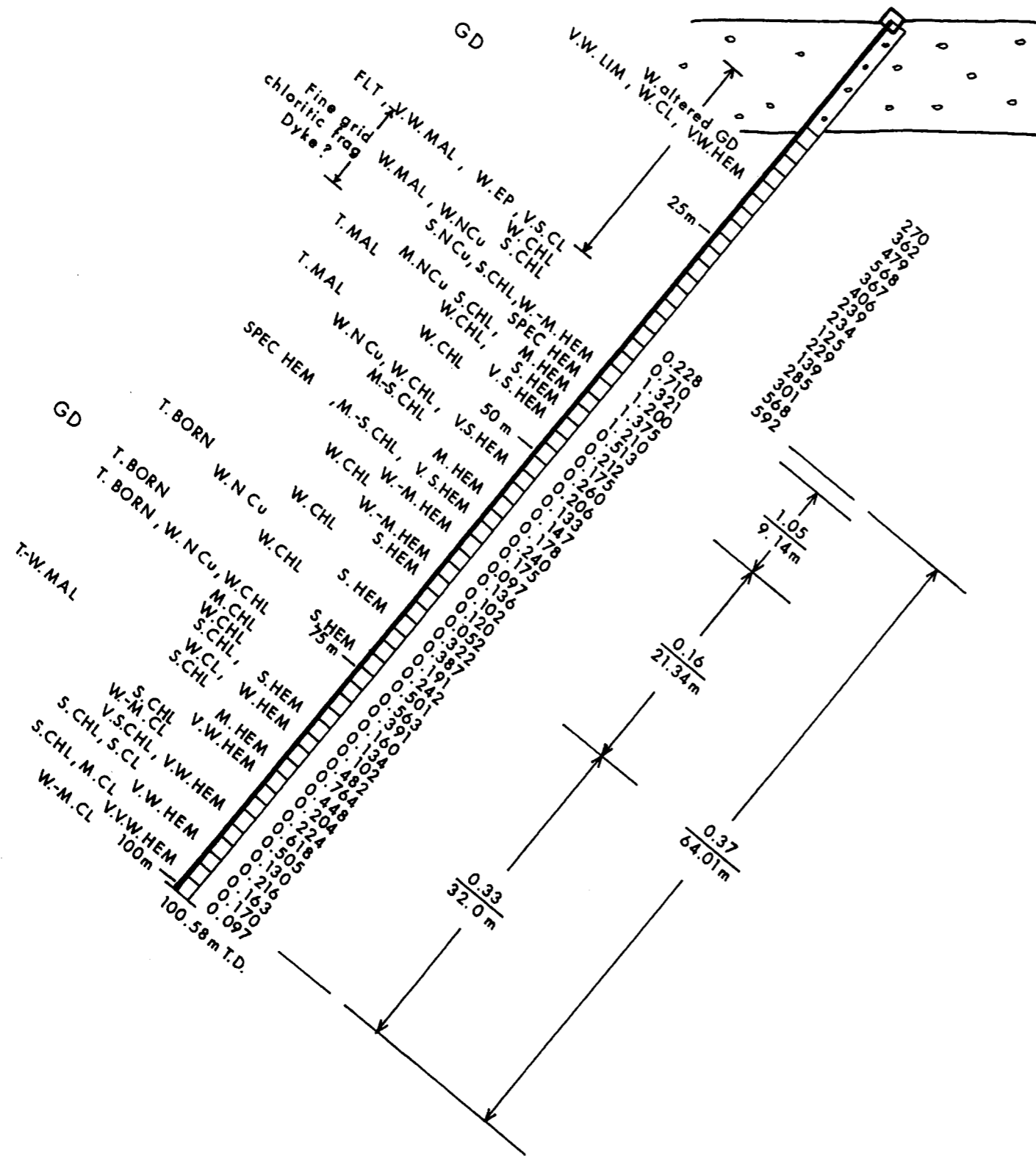
**ZAPPA RESOURCES LTD.**  
**DOT PROPERTY**

## CROSS SECTION G-G'

### 92RCD-04-GEOLOGY-ASSAYS

VIEW AZIMUTH 320°

GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	11
DATE: Jan./1993	NICOLA M.D.	



92RCD-04

DIP -50°  
AZIMUTH 230°  
LENGTH 100.58m

81-18

81-19  
14m SE  
off Section

# LEGEND

## LITHOLOGIC UNITS

Overburden

## LOWER JURASSIC

GD Granodiorite  
Guichon Batholith

## ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

## DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

## MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

Total Copper	%Cu	ppmCu
0.141	1330	
AVERAGE GRADE		
0.91	% Total Copper	Total Length metres
76.2 m		

## MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section

## SCALE

1:500



ZAPPA RESOURCES LTD.

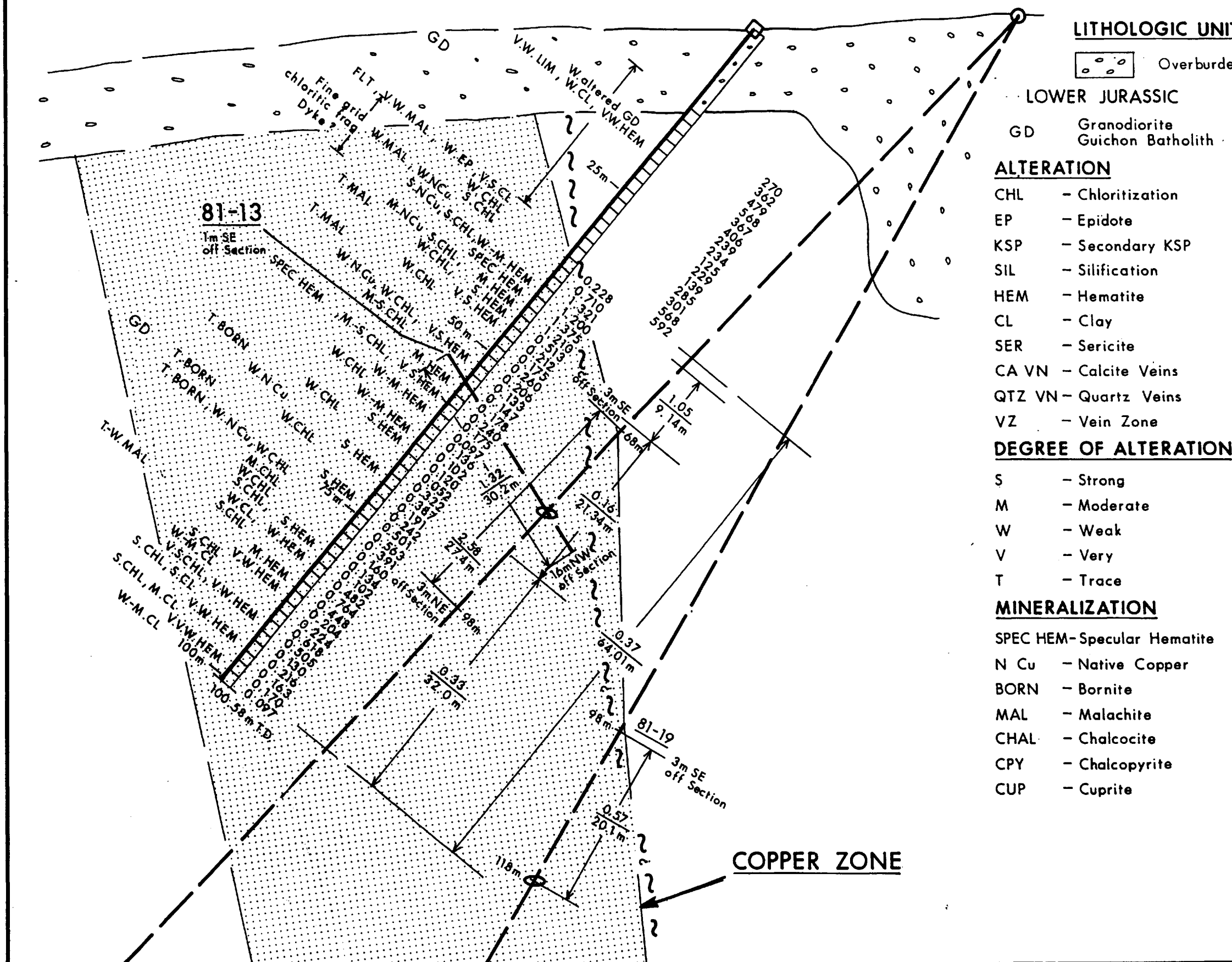
DOT PROPERTY

CROSS SECTION G-G'

COPPER ZONE

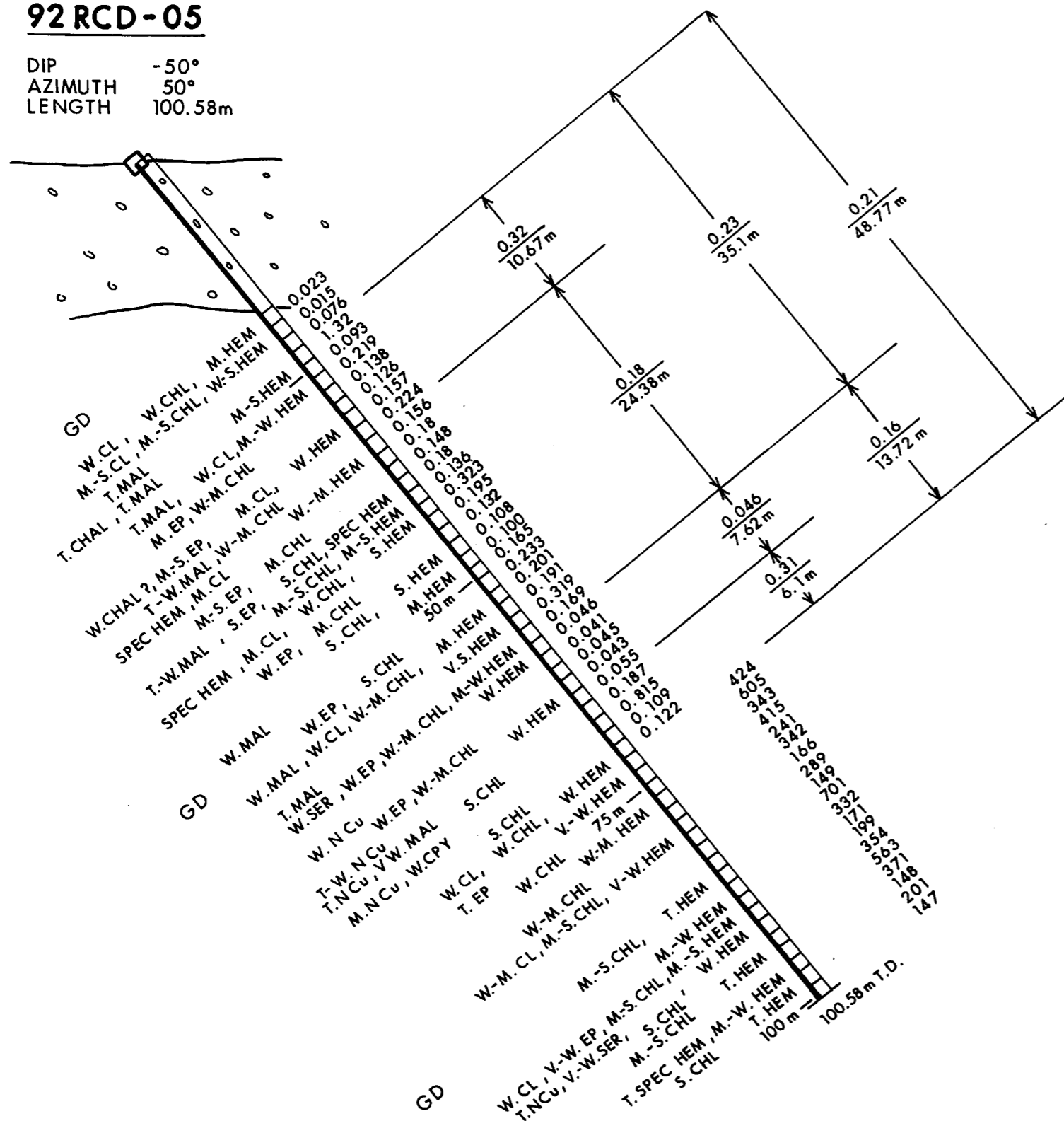
VIEW AZIMUTH 320°

GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	12
DATE: Jan./1993	NICOLA M.D.	



# 92 RCD-05

DIP -50°  
 AZIMUTH 50°  
 LENGTH 100.58m



## LEGEND

### LITHOLOGIC UNITS

- Overburden
- LOWER JURASSIC
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

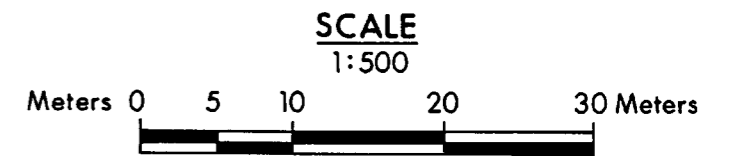
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

Total Copper	% Cu	ppm Cu
0.141	1330	
AVERAGE GRADE		
0.91		
76.2 m	Total Length metres	

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section

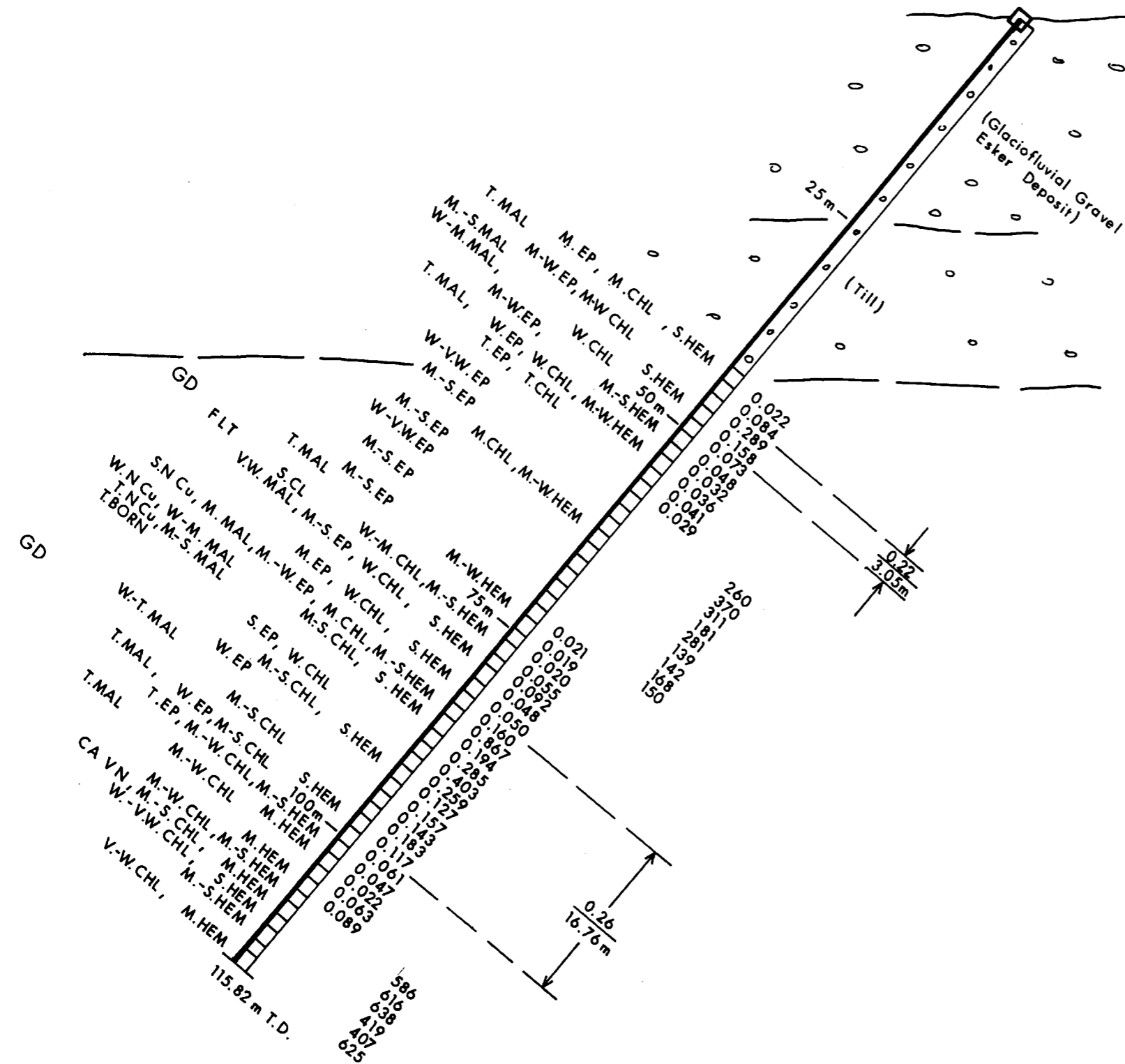


<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION H-H'</b>		
<b>92RCD-05-GEOLOGY-ASSAYS</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	<b>13</b>
DATE: Jan./1993	NICOLA M.D.	



# 92RCD-06

DIP -50°  
AZIMUTH 230°  
LENGTH 115.82



## LEGEND

### LITHOLOGIC UNITS

- Overburden
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

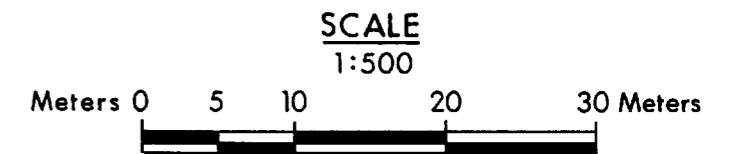
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

Total Copper	
0.141	1330 %Cu   ppmCu
↑	
AVERAGE GRADE	
0.91	% Total Copper
↓	
76.2 m	
Total Length metres	

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section



<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION I-I'</b>		
<b>92RCD-06-GEOLOGY-ASSAYS</b>		
VIEW AZIMUTH 320°		
GEOLOGY BY: George Norman	N.T.S. 921/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	<b>15</b>
DATE: Jan./1993	NICOLA M.D.	

# 92RCD-06

DIP -50°  
AZIMUTH 230°  
LENGTH 115.82

## LEGEND

### LITHOLOGIC UNITS

- Overburden
- LOWER JURASSIC**
- GD Granodiorite  
Guichon Batholith

### ALTERATION

- CHL - Chloritization
- EP - Epidote
- KSP - Secondary KSP
- SIL - Silification
- HEM - Hematite
- CL - Clay
- SER - Sericite
- CA VN - Calcite Veins
- QTZ VN - Quartz Veins
- VZ - Vein Zone

### DEGREE OF ALTERATION

- S - Strong
- M - Moderate
- W - Weak
- V - Very
- T - Trace

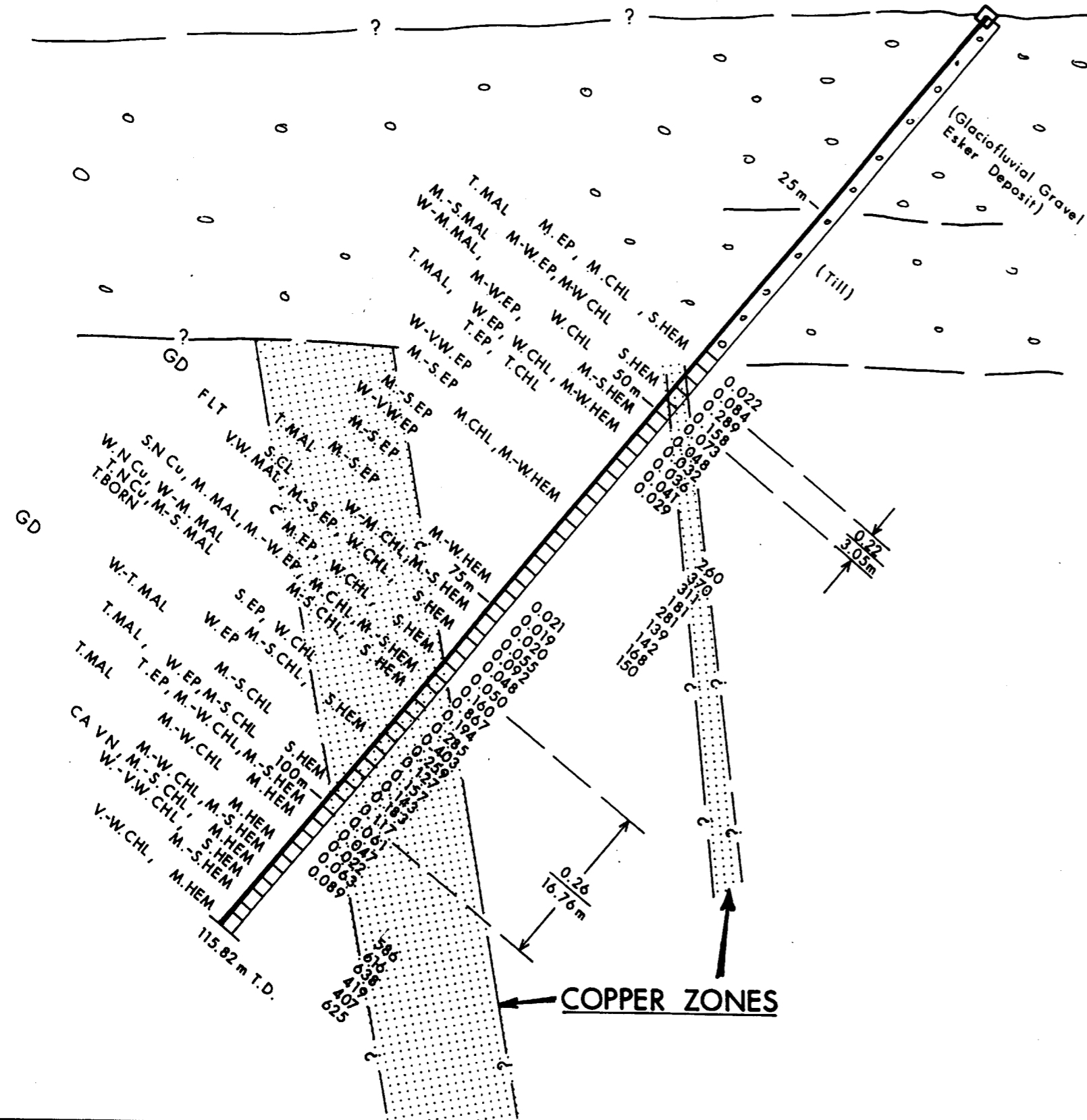
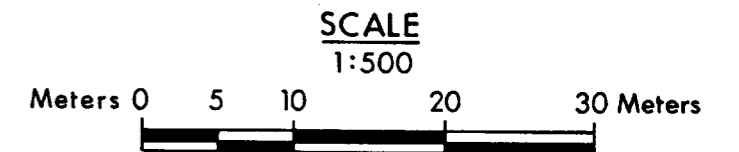
### MINERALIZATION

- SPEC HEM - Specular Hematite
- N Cu - Native Copper
- BORN - Bornite
- MAL - Malachite
- CHAL - Chalcocite
- CPY - Chalcopyrite
- CUP - Cuprite

0.141	1330	Total Copper %Cu	ppmCu
↓		<b>AVERAGE GRADE</b>	
0.91		% Total Copper	
76.2 m		Total Length metres	
↓			

### MISCELLANEOUS

- FLT ~ ~ Fault Zone
- 92RCD-01 Reverse Circulation Drill Hole ;  
Year, Hole Number
- 81-5 Diamond Drill Hole - On Section,  
Off Section
- Pierce Point Of Section

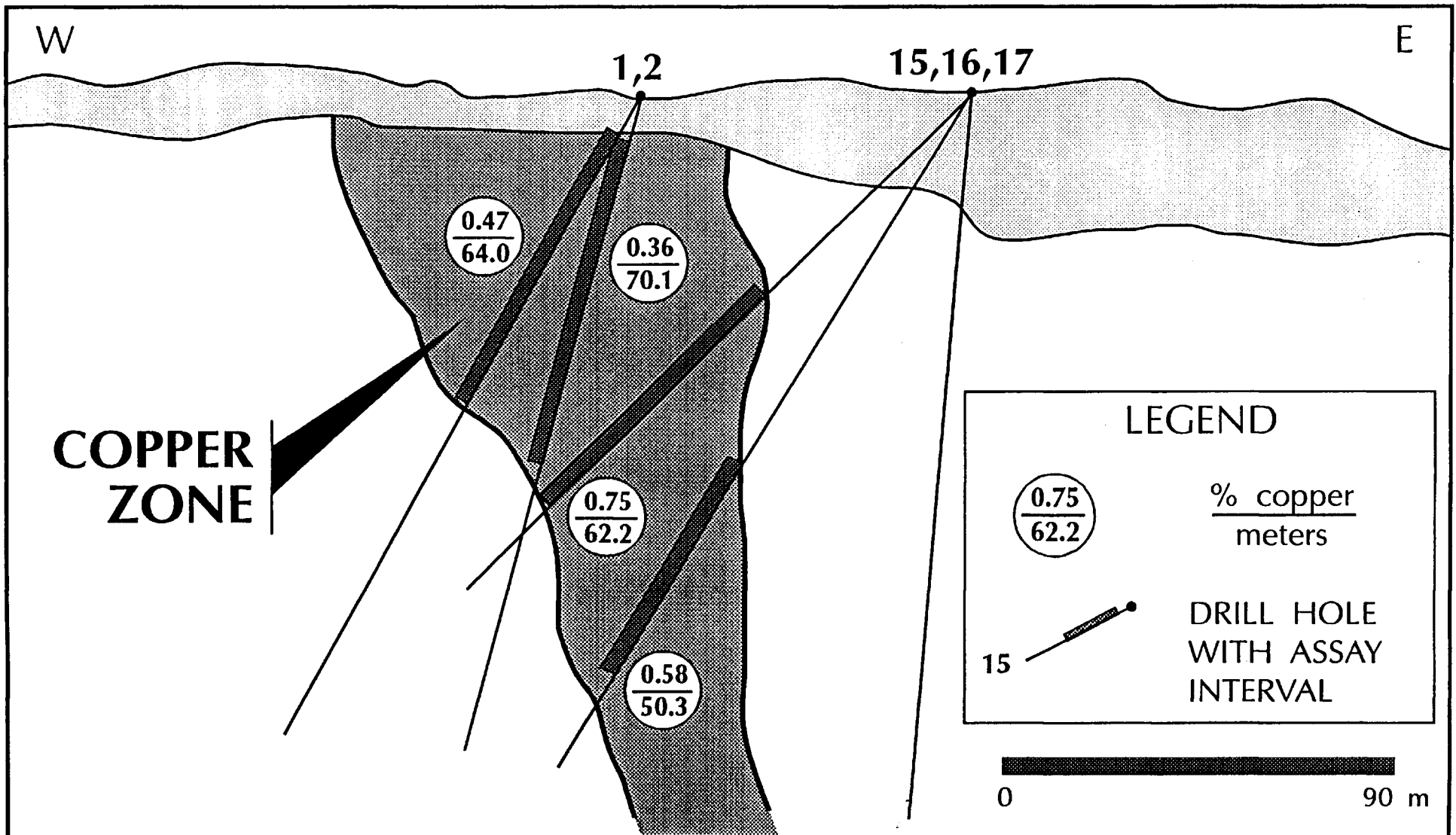


<b>ZAPPA RESOURCES LTD.</b>		
<b>DOT PROPERTY</b>		
<b>CROSS SECTION I-I'</b>		
<b>COPPER ZONE</b>		
<b>VIEW AZIMUTH 320°</b>		
GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:500	<b>16</b>
DATE: Jan./1993	NICOLA M.D.	

# ZAPPA RESOURCES LTD.

## DOT PROPERTY CROSS SECTION COPPER ZONE

Figure 17







**LEGEND**

2.58 - % COPPER  
27.4 - INTERVAL (Meters)

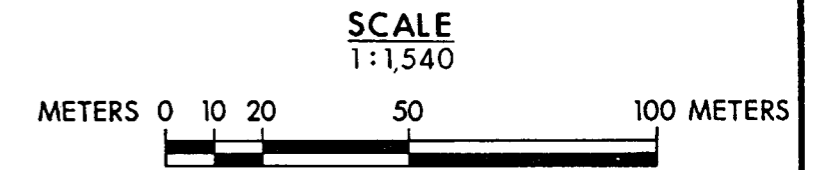
- 81-18,19  

 DRILL HOLE  
 81-18 DIAMOND DRILL HOLE  
 P 81-9 PERCUSSION DRILL HOLE  
 92RCD-01 REVERSE CIRCULATION DRILL HOLE

HOLE COLLAR - UNIDENTIFIED

SECTION LINES

TONNAGE BLOCK



ZAPPA RESOURCES LTD.  
DOT PROPERTY  
**TONNAGE BLOCKS**

GEOLOGY BY: George Norman	N.T.S. 92I/7W	FIGURE No.
DRAWN BY: J. Serwin	SCALE: 1:1540	18
DATE: Jan./1993	NICOLA M.D.	

