ASSESSMENT REPORT OF THE

1984 GEOLOGICAL AND GEOCHEMICAL EXPLORATION PROGRAM

ON THE

SOUP 8-84 CLAIM GROUP

OMINECA MINING DIVISION NTS 94D/8

Located 15 km SSE of Johanson Lake, and approximately 200 km NNE of Smithers

Latitude 56°28' North; Longitude 126°03' West

OWNED BY VITAL RESOURCES LIMITED
OPTIONED BY BP RESOURCES CANADA LIMITED

GEOLOGICAL BRANCH ASSESSMENT REPORT

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SUMMARY

The 1984 exploration activities on the SOUP 8-84 Claim Group consisted of a comprehensive program of detailed geological mapping combined with talus fines and rock chip geochemical sampling. The program was carried out with the objectives of:

- To verify previously reported geological and geochemical data;
- To obtain a more detailed picture of known mineralized zones; and
- 3. To determine the potential for additional mineralization occurring within the volcanic stratigraphy.

Results indicate that highly anomalous gold values are intermittently associated with discontinuous magnetite-rich skarn horizons developed within Takla Group andesite tuffs and augite porphyry flows. Selective rock chip sampling of the skarn horizons returned a substantial number (32) of values >1000 ppb gold, of which 17 samples are in the 1000 - 3000 ppb range, 8 samples in the 3000 - 5000 ppb range and 6 samples >5000 ppb. The highest grade sample returned 62,300 ppb gold (1.82 oz/ton) followed by 23,550 ppb (0.69 oz/ton).

The anomalous values are widely and irregularly distributed throughout the skarn zones. However, several of the highest values are concentrated in the "Saddle Gully Zone", a fault-displaced and sheared section of skarn on the SOUP 4 claim.

Lithogeochemical sampling results indicate that relatively low background gold and copper occur in the Takla volcanic rock units on the property. Talus fines sampling substantiated anomalies reported from previous surveys and in particular, verified anomalies occurring uphill, above skarn mineralization in the "Saddle Gully Zone". These anomalies may result from small scattered gold-bearing quartz-sulphide veins to the extent that gold has been concentrated and possibly even enriched 5 to 10 times in talus fines by mechanical weathering processes. However, the alternative remains that an unknown gold-bearing source exists upslope from the anomaly as either an unexposed skarn horizon or a related hydrothermal vein system.

CONCLUSIONS

1. Gold mineralization on the SOUP claims is principally associated with conformable magnetite-rich skarn horizons developed within Takla Group volcanic rocks. Mineralization is irregular and at best, discontinuous however, mapping and sampling indicate that potential exists for a low to medium tonnage deposit of medium to high tenor.

- 2. Gold mineralization also occurs in minor volcanic-hosted sulphide + magnetite bearing quartz veins associated with dyking and hydrothermal alteration. Present information suggests that vein systems are poorly developed and do not offer economic potential.
- 3. Small disseminated and semi-massive pyritic zones developed adjacent to the skarn horizons contain weak gold and copper mineralization. Disseminated pyrite zones developed elsewhere within sheared and altered volcanic rocks are essentially barren of gold or copper mineralization.
- 4. Post-mineralization faulting and shearing has offset and deformed large sections of skarn bodies and may possibly have caused minor remobilization of gold mineralization within the fault zones. The "Saddle Gully Zone" displays such features and returned the highest rock chip gold values.
- 5. Gold enrichment in talus fines by mechanical weathering may be an important factor when considering the magnitude and extent of geochemical anomalies observed on the SOUP claims.

The stratigraphic assemablage on the SOUP claims lies on the northeast limb of a northwest-trending anticline straddling Kliyul Creek. In the opposite limb, exposed in a section southwest of the creek, volcanic derived limey tuffs and sediments are thought to be correlative with the SOUP assemblage. The skarn deposits are believed to have formed near the northeast pinchout, where the limey sediments project across the anticline and are transitional to the volcanic assemablage on the SOUP claims.

RECOMMENDATIONS

- 1. A program of hand trenching combined with further detailed mapping and sampling is warranted to define the limits of the magnetite-rich skarn zones. Priority should be given to the "Saddle Gully Zone" and other skarn horizons northwest of the rock glacier.
- 2. A detailed magnetometer survey should be completed on SOUP 1-11 claims with the objective of tracing unexposed extensions of the skarn zones and providing a guide for the detailed work mentioned above. Maximum spacings of 50 X 25 metres should be used over the known skarn horizons and 100 X 50 metres over the remainder of the claim block.

- 3. Diamond drilling is recommended, contingent upon continued positive results from surface work. Initial priority should be given to testing the subsurface extensions of the highest grade surface zones, however, it is emphasized that a substantial representative cross-section of the lithologies affected by skarn and hydrothermal alteration should be intersected.
- 4. The relatively high grade potential of gold-in-skarn mineralization as seen by results obtained on the SOUP claims, should encourage the search for similar mineralization elsewhere in the Takla Group, in areas where favourable geological and economic parameters exit.

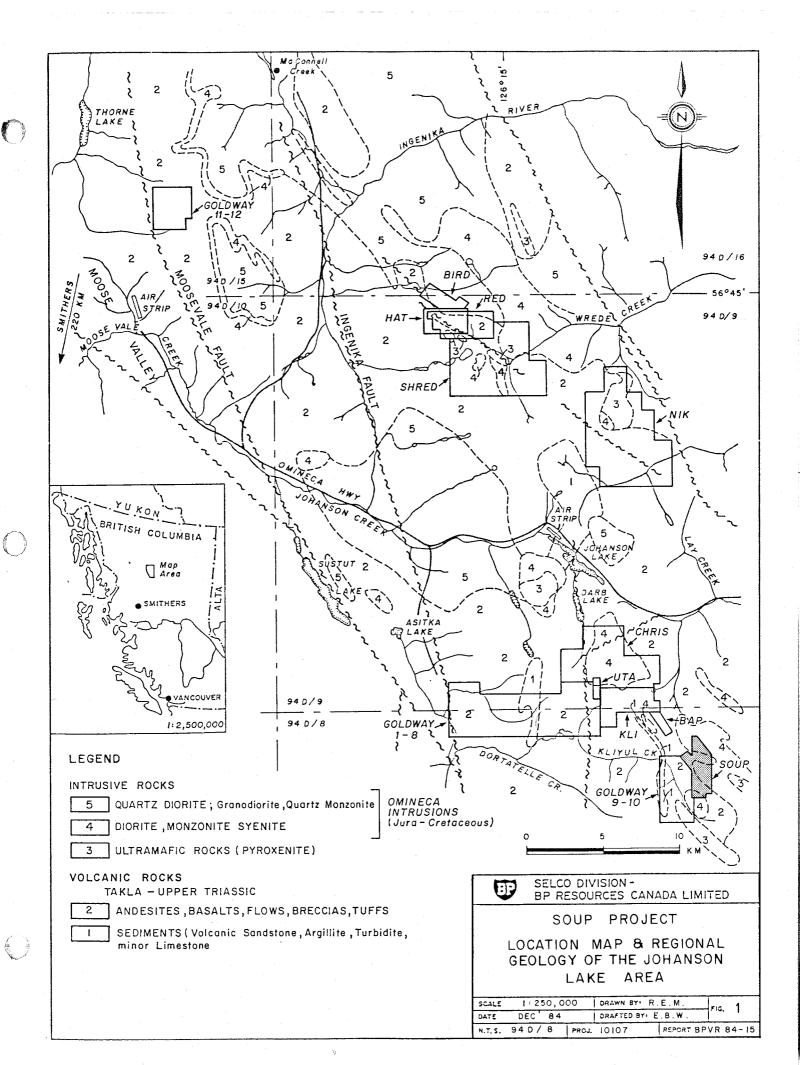
INTRODUCTION

1. Location, Access and Terrain

The SOUP claims (SOUP 8-84 Group) are located at 56^o28' north latitude and 126^o03' west longitude in the Omineca Mining Division approximately 200 km north-northwest of Smithers and 15 km southeast of Johanson Lake (NTS 94D/08, Figure 1).

Access to the property (Figure 2) is by helicopter from Johanson Lake which is reached by wheel or float equipped aircraft, or by the Omineca Highway, which is closed in winter. The road is reached from Fort St. James, north of Vanderhoof (430 km) or via Highway 97 north from Prince George (500 km). The Dease Lake extension of the British Columbia Railway is operational between Prince George and Driftwood, 65 km southeast of Johanson Lake.

The property is situated east of Kliyul Creek on a steep southwest-facing slope on which elevation ranges from 1300 to 2300 metres. Ubiquitous blocky talus, partially covered by alpine grasses and shrubs obscures much of the bedrock and the southwest side of the claim group is bisected by a prominent rock glacier.



2. Claim Status

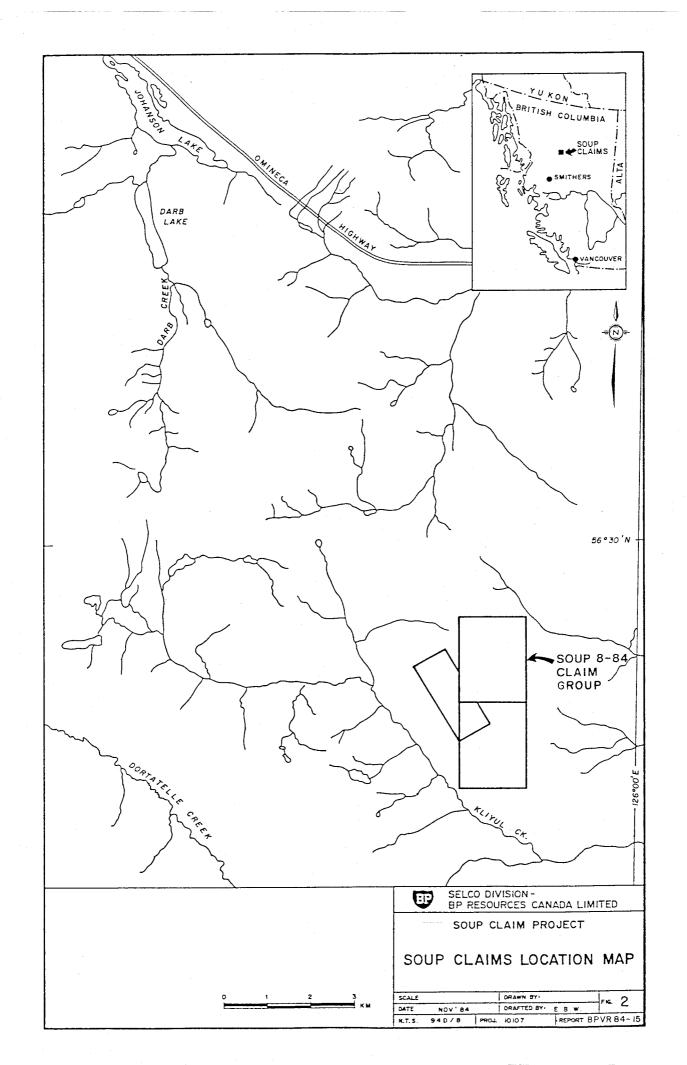
The SOUP 8-84 claim group (Figure 2) consists of ten, two-post claims (SOUP 1-10), SOUP 11 Fraction and MGS claims SOUP 12, 13, and 14, each consisting of 12 units, together totalling 37 units. They are 100% owned by Vital Resources Limtied and were optioned to BP Resources Canada Limited in 1984. With the filing of 1984 assessment work all claims in the group are in good standing unitl 1993.

CLAIM NAME	RECORD NUMBER	NO. OF UNITS	RECORDING DATE	EXPIRY DATE
SOUP 1	26941		Aug. 7/64	Aug. 7/93
SOUP 2	26942		Aug. 7/64	Aug. 7/93
SOUP 3	26943		Aug. 7/64	Aug. 7/93
SOUP 4	26944		Aug. 7/64	Aug. 7/93
SOUP 5	26945		Aug. 7/64	Aug. 7/93
SOUP 6	26946		Aug. 7/64	Aug. 7/93
SOUP 7	26947		Aug. 7/64	Aug. 7/93
SOUP 8	26948		Aug. 7/64	Aug. 7/93
SOUP 9	26949		Aug. 7/64	Aug. 7/93
SOUP 10	26950		Aug. 7/64	Aug. 7/93
SOUP 11 F	r. 4206		Aug. 15/81	Aug. 15/93
SOUP 12*	5805	12	Oct. 5/83	Oct. 5/93
SOUP 13	5806	12	Oct. 5/83	Oct. 5/93
SOUP 14	6491	12	Aug. 13/84	Aug. 13/93

^{*}SOUP 12 overstaked by SOUP 14 - August 9, 1984 (Tags had been filled out incorrectly on SOUP 12)

3. History

Work in the Kliyul Creek area dates back to the 1940's, shortly after placer gold was discovered at McConnell Creek, about 50 km northwest. More recently, in the 1960's and



1970's, exploration was directed towards porphyry copper-molybdenum and volcanic-hosted stratabound copper deposits. The most impressive find has been the Sustut Copper deposit discovered by Falconbridge Nickel Limited, 40 km west of Johanson Lake. Other drilling ventures were undertaken in 1982 by Lornex Mines Limited on upper Lay Creek and by Getty Mines Limited on Porphyry Creek, northwest of the SOUP claims (Carter, 1983).

The original SOUP claims (SOUP 1-10) were located in 1964, SOUP 11 Fraction was staked in 1981, SOUP 12 and 13 were staked in 1982 and SOUP 14 in 1984. Exploration over the last 20 years consisted of detailed mapping, surface sampling, petrographic studies, magnetic profiles and the drilling of three short x-ray holes. Vital Resources Limited acquired the claims in 1980 and completed additional geochemical surveys in 1981. In 1982 Noranda Exploration Company Limited carried out detailed soil and rock chip sampling, but later dropped their option. The claims were optioned to BP Resources Canada Limited in 1984.

4. 1984 Exploration Activities

The 1984 program consisted of detailed geological mapping, lithogeochemical sampling (rock chip), talus fines sampling

and selective rock chip sampling of mineralized showings.

Detailed mapping was carried out on an orthophoto base map at 1: 5000 scale prepared by McElhanney Surveying and Engineering Limited. The mapping was carried out to verify previous geological data and to obtain a more detailed outline of mineralized zones within the stratigraphy. Talus fines sampling was carried out to substantiate the results of an earlier sampling program and a continuous lithogeochemical sampling line was completed to obtain representative metal values for the various units and to establish the presence or absence of pathfinder elements within the volcanic sequence as a guide to exploration.

REGIONAL GEOLOGY

The SOUP claims (Figure 1) are situated in the northern extension of the Quesnel Trough within the Intermontane Tectonic belt of the Canadian Cordillera. The Quesnel Trough assemblage consists principally of the Upper Triassic to Lower Jurassic Takla Group volcanic and sedimentary rocks, correlatives of which extend from south of the U.S. border to north of the Stikine River (Monger, 1977). The volcanic rocks consist of island-arc type calc-alkaline to alkaline flows and volcaniclastic rocks of predominantly submarine origin, although subaerial volcanic complexes also occur in the region.

The extrusive rocks are interlayered with volcanogenic sandstones that grade into laminated siltstones, argillites, with minor conglomerates, tuffaceous limestone and limestone breccia.

The stratigraphy is intruded by granitic to intermediate plutons Jura-Cretaceous in age which are satellitic to the Hogem Batholith and by remnants of ultramafic intrusions, a few of which are locally related to deep-seated faults transecting the region. In the Kliyul Creek area on the SOUP claims, the principal rock types are augite porphyry flows and andesitic tuffs with numerous comagnatic diorite intrusions of similar composition. Southwest of the claims, across a north-northwest-trending antiformal structure which straddles Kliyul Creek, there is a lithologic transition from volcanic to sedimentary facies where tuffaceous sediments are interlayered with calcareous sediments and gritty limestones which can be traced to the north, parallel to the antiformal axis.

Mineralization in the area occurs in a variety of forms and associations (Bradley, 1984). Copper and molybdenum porphyrytype occurrences are associated with monzonitic intrusions and copper occurs as minor fracture fillings and disseminations in the Takla volcanic rocks. Numerous gold-bearing quartz-carbonate veins with semi-massive sulphides, including

chalcopyrite and pyrite are commonly and spatially associated with subvolcanic hornblende diorite sills and dykes within the Takla stratigraphy.

One of the most impressive mineral occurrences is on the SOUP claims where gold and minor chalcopyrite are closely associated with magnetite-rich skarn horizons, also within the Takla volcanics. Similar skarn mineralization occurs on the Kli claims and is described in another report.

PROPERTY GEOLOGY

Most of the rocks within the SOUP 1-11 claims (Figure 3) are andesitic to basaltic volcanic rocks of the Upper Triassic to Lower Jurassic Takla Group that have undergone greenschist grade metamorphism. They are composed of a lower unit of feldspar-rich andesitic flows and tuffs overlain by andesitic to basaltic augite porphyry flows. These rocks have been intruded by diorite stocks, dykes and sills and by a small quartz monzonite intrusive, as well as by small dykes of feldspar porphyry.

The stratigraphy dips about 30° to the northeast and is transected by northerly and easterly striking faults. Shearing is confined along the faults and in a few other zones but generally the rocks are not highly deformed.

Massive conformable magnetite skarn bands (1-5 metres thick) occurring near the base of the augite porphyry are exposed discontinuously along strike (northwest-southeast) and contain appreciable gold and copper. Similar mineralization is found along a few fault zones cross-cutting the skarn horizon. Other minor gold-copper mineralization is found in small sulphide-bearing quartz veins. Disseminated pyrite zones occur in sheared and altered flows and tuffs but are poorly mineralized. Property geology, displayed in Figure 3, has been expanded from 1:5 000 to 1:2 000 to show more detail.

1. LITHOLOGIES

(a) Volcanic Rocks

(i) Andesite: (Unit 1)

The lower half of the large slope to the east above Kliyul Creek is composed of fine to medium grained, grey-green andesitic feldspar crystal tuffs and massive flows. Minor lithic fragments are present in some flow sections and the flows and crystal tuffs are grossly similar in appearance. However, the presence of <1% lapilli (< 2.0 cm) of similar composition to the host rock, indicates that much of the unit is tuffaceous.

Variable amounts of augite and minor hornblende occur but are never abundant. The unit is generally massive, weakly foliated in places and weathers to a homogeneous light to medium grey.

Mafic minerals are chloritized and most feldspars are sericitic pale green.

Rare amygdaloidal andesitic flows are found which have an aphanitic groundmass with epidote and feldspar amygdales up to 5 mm in diameter.

(ii) Augite Porphyry: (Unit 2)

Augite porphyry overlies the andesite (Unit 1) and in places interfingers at the contact, notably in the southeast. The augite porphyry is andesitic to basaltic, consisting mainly of flows and minor tuffs. The rocks vary from fine to coarse grained, medium to dark green with 10% to 70% augite phenocrysts in an aphanitic to fine grained, often feldspar-rich matrix. The generally stubby augite phenocrysts are usually 1 to 2 mm but may be up to 10 mm long. Minor flow breccias and laminated tuffs occur within the unit. Mafic minerals (augite and minor hornblende) are highly chloritized and feldspar crystals (<1 mm) in the matrix are saussuritized.

(b) INTRUSIVE ROCKS

Dykes and sills of dioritic composition occurring within the Takla volcanic rocks probably represent subvolcanic feeder dykes. Other intrusive types differing in composition and age also occur. A small monzonite plug intrudes diorite in the southeast and is likely an Omineca offshoot. In the Kliyul Creek valley an ultramafic plug (not mapped) is possibly older and may be fault related.

(i) Diorite: (Unit 3)

Diorite outcrops in the headwall and northern rim of the rock glacier. The diorite is part of an intrusion that is approximately two kilometres wide lying mainly northeast of the rock glacier. McTaggert (1965) describes the unit as a typical plagioclase-hornblende diorite, in places displaying flow orientation. Some phases are augite-bearing while others are monzonitic rather than dioritic. At its margins the diorite forms narrow dykes, varying in width from a few centimetres up to two metres. Some dykes contain abundant angular inclusions of the Takla host rocks. The diorite is highly altered with chloritic to sericitic hornblende and saussuritic plagioclase.

In the area northwest of the rock glacier there is a 20 to 100 metre thick diorite sill that is intruded along the contact between the augite porphyry and andesite units. It is believed to be an offshoot of the main diorite pluton but has a fine grained "microdiorite" texture (McTaggert, 1965). The rock is fine to medium grained, grey-green, equigranular and highly altered. It is more resistant than the surrounding volcanic rocks and characteristically forms blockier talus.

No contact metamorphic effects are visible at the diorite/andesite contacts, possibly due to the compositional and temporal association between the intrusive and its host rocks.

(ii) Feldspar Porphyry Dykes: (Unit 4)

Several feldspar porphyry dykes 2 to 5 metres in width, intrude the augite porphyry volcanic rocks northwest of the rock glacier. They contain 20 to 60% sub-rounded to euhedral phenocrysts (2-10 mm) in a fine grained, quartz-poor, grey-green matrix. The dykes dip down-slope, resulting in outcrops much wider than true thickness. Contacts with the augite porphyry are slightly pyritic and iron-stained along the dyke margins.

(iii) Quartz Monzonite: (Unit 5)

Intruding the Takla rocks in the southwest corner of the property, not within the mapped area, is a quartz monzonite body which is probably a satellitic offshoot of the Omineca Intrusions.

McTaggert (1965) describes the rock as a biotite quartz monzonite that is somewhat heterogeneous at its contacts. Contact effects include recrystallization and weak localized pyritization of the volcanic rocks.

(iv) Other Dykes:

Several minor dykes occur including a fine grained diabase, a pale green felsite dyke and a hornblende porphyry dyke were observed in "Cross-Section Creek". A 10 metre wide fine to medium greenish-grey diorite dyke occurs southeast of the rock glacier above the major fault.

McTaggert (1965) also notes the occurrence of minor mica lamprophyre dykes.

2. STRUCTURE

Attitudes in the volcanic rocks were difficult to measure but generally the volcanic stratigraphy strikes northwest-southeast and dips about 30° to the northeast (into the

hill). Some minor warping was observed but no major folding occurs.

A number of faults cross-cut and offset lithologies at several localities. Southeast of the rock glacier a major fault that follows the augite porphyry/andesite contact trends 100° to 120° and has a steep to vertical dip with left lateral displacement resulting in about 350 metres offset in the skarn band (Figure 3). Northwest of the rock glacier several faults within the augite porphyry trend 020° to 030° and also appear to have steep to vertical dips. One such fault has uplifted the skarn band about 60 metres on its southeast side. However, just northeast of the property, displacements on dykes show a downdrop of more than 20 metres on the southeast side so that no consistent sense of movement can be determined.

The volcanic rocks along the faults are moderately to highly sheared, resulting in fault zones up to 30 metres wide. Shearing is parallel to nearly perpendicular to fault orientations within the shear zones. Such variations in shear direction are probably due to drag effects along the faulted blocks.

Other fault zones occur elsewhere with various orientations. Most are within the lower andesite unit where steep to vertical east-west striking shears are the most common. Many shear zones are pyritic, leached and iron-stained, however, most do not carry significant gold values.

3. ALTERATION AND MINERALIZATION

(a) Modes of Occurrence

The rocks, of the Soup claims have undergone regional greenschist metamorphism, resulting in chloritized mafic minerals, sausseritized and sericitized plagioclase with actinolite and/or tremolite coatings on fracture surfaces. Weak propylitic alteration is common with epidote, quartz and calcite stringers and veins, especially in the augite porphyry unit.

Mineralization occurs in three ways on the property:

- (1) associated with magnetite-rich skarn zones;
- (2) in silicified, pyritic andesite tuffs; and
- (3) in small minor sulphide bearing quartz veins. Gold and copper sulphide in magnetite skarns are the only economic minerals of any significant abundance on the claims.

(b) Magnetite Skarn Zones

(i) General Features

The magnetite-rich skarns are more or less massive and concordant. They vary along strike in thickness from 1 to 5 metres and in magnetite content from 60 to 100%. Most magnetite is concentrated near the top of the horizon. Peripheral disseminated zones of 5 to 20 metres thick underlie most massive horizons usually with a magnetite content of 2 to 10%, minor pyrite, chalcopyrite and appreciable malachite staining. At some localities iron and minor copper sulphides form massive lenses up to 1 metre thick, commonly above the top of the magnetite zone. In most occurrences the sulphides and magnetite appear to be somewhat mutually exclusive, where magnetite-rich zones are sulphide-poor and sulphide-rich zones are magnetite-poor.

Magnetite zones are predominantly composed of fine grained granular magnetite with coarse disseminated crystalline pyrite. Chalcopyrite occurs as small inclusions in pyrite or gangue, but is rarely enclosed in magnetite (Sinclair, 1975). There is a general correlation between high gold and high copper values and Sinclair believes that gold occurs in the chalcopyrite. No visible gold was observed.

Outcrops of the magnetite skarn horizons are highly oxidized, forming bright orange-brown stain zones commonly characterized by epidote alteration with minor actinolite and rare fine grained garnet, however, no major massive or zonal concentration of calc-silicate minerals was found in any of the skarn zones mapped.

The skarn assemblage is mainly concentrated along diorite/augite porphyry or andesite/augite porphyry contacts. As with the iron minerals, alteration is peripherally concentrated below the skarn horizons but only rarely above. In some areas the mineralization follows shear zones cutting upwards into overlying augite porphyry.

(ii) Skarn Zone Occurrences

Southeast of the rock glacier there is one band which is offset 350 metres by a left lateral fault displacement and then continues at the northwest edge of Soup 10 claim. It occurs near the base of the augite porphyry unit and has a total exposed length of about 600 metres. Northwest of the rock glacier there is one band at the base of the augite porphyry unit which extends along its contact with the underlying diorite sill. The band has

a mapped length of 800 metres. Another band in the northwest corner is 40 to 50 metres higher and was mapped over a 350 metre length.

The most encouraging rock geochemcial results were obtained from the "Saddle Gully Zone", a magnetite zone which continues up one of the fault zones (020° to 030°) that crosscuts the main skarn band in the "Saddle Gully" northwest of the rock glacier (Figure 3). Abundant disseminated and massive magnetite is found within a 50 x 100 metre area at 2100 metres elevation where mineralized shears up to 1 metre wide occur along with discontinuous but stratigraphically conformable skarn beds which are enclosed within the fault zone. Gold values up to 60 grams per tonne were obtained from this zone. Values in excess of 1000 ppb gold were obtained from a 300 metre long zone between 1900 metres and 2100 metres elevation. Width of the latter zone is generally 25 metres or less and it appears that only about one quarter of the rock within the zone is mineralized. There is generally a sharp contact between mineralized and non-mineralized rock in the zone. It is possible that faulting either occured penecontemporaneously with skarn mineralization or that the skarn horizon has been off-set and perhpas remobilized by fault deformation.

(c) Pyrite Zones

Adjoining the magnetiferous bands in the Saddle Gully zone are a few irregular silicified, disseminated pyritic zones up to 10 metres thick within the augite porphyry. Southeast of the rock glacier a few large zones occur in shears adjacent to the skarn bands. Several disseminated pyrite zones up to 150 metres wide occur elsewhere within the andesite and augite porphyry units. The zones are usually silicified, iron-stained, highly sheared and contain 2-5% pyrite. They have a strongly bleached appearance with sericitized feldspars and chloritized mafic minerals.

Northwest of the claims a similar pyritic alteration zone continues for several hundred metres along strike. This zone is far more extensive than the smaller zones on the Soup claims and is overlain by augite porphyry which is intruded by diorite. The same bleached pyritic zone is traced to the northwest as a prominent gossan further along the northeast side of Kliyul Creek for about 5 km through the Bap and Kli claims.

Minor iron-stained pyrite zones also occur at the contacts of feldspar porphyry dykes and are likely the result of contact metamorphism. Notably, none of the pyritic zones carry appreciable gold or copper values.

(d) Quartz Veins

Small to medium sized quartz veins (<25 cm wide) occur throughout the property, particularly in the augite porphyry. The veins do not occur in great numbers, but some contain up to 20% chalcopyrite and minor magnetite. They are most common adjacent to faults and shear zones and along contacts with the feldspar porphyry dykes.

GEOCHEMISTRY

1. Objectives

Comprehensive geochemical sampling was carried out on the Soup claims with the following objectives:

- 1) Talus fines were collected to substantiate results from an earlier (1982) sampling program by Vital Resources Limited and to relocate anomalies delineated by that program.
- 2) Detailed rock chip sampling was carried out to determine the source of the anomalies.
- 3) Additional detailed rock samples were taken from the skarn bands to further delineate associated gold mineralization.
- 4) A continuous cross section lithogeochemical sampling line was completed at "Cross-section Creek" to determine representative gold values for the various lithologic units.

5) The lithogeochemical sampling was also to determine pathfinder elements which would serve as an exploration guide for copper and gold zones.

2. Talus Fines Samples

Talus fines samples were collected at 25 metre spacing along six lines (Figure 4A). Three lines were placed on each side of the rock glacier, beginning at the top of the ridge and continuing downhill to well below the skarn horizons. The results from these lines reconfirms that anomalously high gold values occur in talus throughout the Soup 1 through 10 claims. As well, the occurrence of some exceptionally high values in the 5,000 to 10,000 ppb range was reconfirmed and a zone of highly anomalous values above the main skarn band in Saddle Gully was substantiated. The total area, having values in excess of 500 ppb, appears to be the same or smaller than that previously outlined. However, this may have been influenced by line spacing.

Northwest of the rock glacier almost all the samples had gold values greater than 50 ppb and 60% were in excess of 100 ppb, illustrating the unusually high gold enrichment on that portion of the property. Southeast of the rock glacier gold values tend to be lower and only 40% of the samples have values in excess of 100 ppb, however, almost all still have values well above the average crustal abundance for gold.

Samples that are anomalous above the high background fall into three groups.

- The first lie at and slightly below the skarn bands and can be directly attributed to gold from the skarn bands. The talus fines samples do not, however, always show high values directly coincident with the skarn bands.
- 2) The second group of high values involves four samples, (853075 through 853078), with values in the 1700 ppb to 1800 ppb range which are located at and below 2100 metres elevation, coincident with the Saddle Gully zone. The Vital Resources grid had an anomaly of similar magnitude at this location. Both anomalies can be attributed to faulted and sheared skarn-type mineralization found in the zone. The very high values do not continue above or below known mineralized zones.
- 3) The third group of anomalous values includes a few sacttered, very high values (up to 8000 ppb Au) along with moderately high, more extensive zones above the skarn bands in the augite porphyry unit. Very high values were also found above the skarn band in Vital's soil grid, but the locations do not everywhere coincide

with those in the present work. This, combined with the spotty nature of the high values and general lack of continuity of anomalies, indicates that the high values are perhaps due to sources of very limited extent, such as small chalcopyrite and gold bearing veins or talus blocks shedding from them. Alternatively, it remains possible that an unknown gold-mineralized source exists upslope from the anomaly.

Samples 853007 through 853014 are all greater than 200 ppb gold. Most samples in the next line to the southeast (853068 through 853072) are also above 200 ppb gold at the same elevation (2100 to 2200 metres) suggesting that a possible conformable zone of higher gold values occurs. This zone is in the same area as a >500 ppb gold zone, outlined on Vital's grid. However, the third line to the southeast does not show the extension of this anomaly.

Southeast of the rock glacier only samples 862125 through 853127 are anomalous (2250 to 2300 metres elevations). Other highs are spotty and directly attributable to the skarn bands.

It is important to note that all of the above anomalies occur within the augite porphyry unit or near its lower contact and

most are directly related to the magnetite-rich skarn horizons. No notable anomalies of any consequence occur in the underlying andesite unit.

3. Rock Chips Samples

A rock chip sample line of 125 continuous 10-metre samples (854001 through 854125) was placed from the base of outcrop to the top of the ridge along Cross-Section Creek (Soup 1 and 2 claims) to identify additional gold mineralization and to correlate rock values with talus fines results (Figure 4B). Gold values obtained from the rock chips fall predominantly in the 5 to 20 ppb range (Figure 4C). Only four samples had greater than 100 ppb gold.

An arithmetic average of 19.8 ppb was obtained from 73 samples taken from the andesite (Unit 1), whereas 50 samples from the augite porphyry (Unit 2) averaged 55.3 ppb gold. However, of those from the augite porphyry, two high samples (860 ppb) contained quartz-epidote vein material and are probably biased. Thus, neither unit appears to have exceptionally high background gold values. In contrast, samples from three short sample lines across mineralized zones in lower skarn horizon averaged 149.3, 169.6 and 281.3 ppb respectively. Each sample line includes massive magnetite, underlying disseminated material and overlying barren material.

Rock chip samples taken along a second line (Figure 4C) across the upper part of the augite porphyry unit, above and below the Saddle Gully Zone returned similar low background values for the augite porphyry (e.g. 853501-853509) and equally low values (853532-853534 and 852525-852528) representing the microdiorite (Unit 3).

The highest rock chip gold values were obtained from fault-displaced magnetite-bearing sections in the Saddle Gully zone. The fault trends 020° to 030° with steep to vertical dips. Values greater than 1000 ppb gold occur over a zone 300 metres long which include values up to 60,000 ppb. The zone is about 25 metres wide, but only magnetite-rich shears and some stratigraphically conformable magnetite-rich beds within the zone are mineralized. Actual widths of mineralization are generally between 0.5 and 5 metres. Rock adjacent to the mineralized zones have only very weak gold values. As much of the zone is talus-covered, the total extent of mineralization is unknown.

The magnetite-rich skarn bands were chip sampled at several other locations (lines 1-5, Figure 4A, 4B). At most localities the gold values averaged considerably less than

the higher grade Saddle Gully zone. Although a few values (10%) are greater than 1000 ppb, about 30% fall between 100 and 1000 ppb and more than 60% are less than 100 ppb. The results indicate that although high gold values occur within the magnetite-rich skarn zones, not all of the skarns are gold-rich and mineralization is sporadic, at best.

Elsewhere, isolated pyritic zones (described earlier) occurring in both the augite porphyry and andesite units have no indicated gold enrichment whatsoever and thus, have no apparent economic potential. However, some malachite-stained copper-bearing zones which underlie the skarn at several localities appear to have some degree of gold enrichment and perhaps reflect a genetic relationship between gold and copper.

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- Bradley, M.D., 1984. Geological Compliation Report on the Kliyul Creek, B.C. Cu-Au Play. Unpublished BP-Selco Report.
- Carter, N.C., 1983. Report on Soup and Klisum Mineral Claim Groups of Vital Resources Limited, Omineca Mining Division, British Columbia, Unpublished company report.
- McTaggert, K.C., 1965. Geology of the Soup Mineral Claims, Nos. 1 to 10 and Soup Fraction, B.C.D.M. Assessment Report No. 675.
- Monger, J.W.H., 1977. The Triassic Takla Group in McConnell Creek Map-Area, North Central British Columbia, G.S.C. paper 76-29.
- Sinclair, A.J., 1975. A Mineralographic Study of Surface and Drill Core Specimens from the Soup Group of Claims and its Importance to Beneficiation. B.C.D.M. Assessment Report No 5562.

APPENDIX 1 GEOCHEMICAL PREPARATION AND ANALYTICAL PROCEDURES

VANGEOCHEM LAB LTD. 1521 Pemberton Ave. North Vancouvelr, B.C. V7P 2S

RECEIVED

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SELCO - BP EXPLORATION VANCOUVER, B.C.

TO:

Dr. Stan Hoffman BP - Selco Mining

Suite 700 - 890 West Pender Street

Vancouver, B.C. V6C 1K5

FROM:

Vangeochem Lab Ltd. 1521 Pemberton Ave.

North Vancouver, B.C. V7P 2S3

SUBJECT:

Analytical procedure used to determine elements in hot acid soluble by Induction Couple Plasma Spectrometer

(ICP) analysis.

1. Method of Sample Preparation

- (a) Geochemical soil, silt or rock samples were received in the laboratory in wet-strength 4" x 6" Kraft paper bags or rock samples sometimes in 8" x 12" plastic bags.
- (b) The dried soil and silt samples were sifted by hand using a 8" diameter 80-mesh stainless steel sieve. The plus 80-mesh fraction was rejected and the minus 80-mesh fraction was transferred into a new bag for analysis later.
- (c) The dried rock samples were crushed by using a jaw crusher and pulverized to 100-mesh or finer by using a disc mill. The pulverized samples were then put in a new bag for later analysis.

2. Method of Digestion

- (a) 0.500 gram of -80 mesh sample was used.
- (b) Samples were digested in a hot water bath with conc. HNO3 and conc. HCl acids.
- (c) The digested samples were diluted to a fixed volume and shaken well.

3. Method_of_Analysis

The ICP analyses elements were determined by using Jarrel Ash, model 885. Direct reading emission spectrograph of a inductive coupled plasma excitation source. All major matrix and trace elements are interelement corrected to trace elements. All data is entered into Apple II plus, stored on floppy disks, and printed by Epson 100.

4. The analyses were supervised by Mr. Dean Toye and Mr. Conway Chun of Vangeochem Lab Ltd. and their staff.

Conway Chun

VANGEOCHEM LAB LTD.

VANGEOCHEM LAB LTD. 1521 Pemberton Ave. North Vanvouver, B.C. V7P 2S3

TO:

Dr. Stan Hoffman BP - Selco Mining

Suite 700 - 890 West Pender Street

Vancouver, B.C. V6C 1K5

FROM:

Vangeochem Lab Ltd. 1521 Pemberton Ave.

North Vancouver, B.C. V7P 2S3

SUBJECT:

Analytical procedure used to determine gold by fireassay method and detected by atomic absorption spec. in geological samples.

1. Method of Sample Preparation

- (a) Geochemical soil, silt or rock samples were received in the laboratory in wet-strength 4" x 6" Kraft paper bags or rock samples sometimes in 8" x 12" plastic bags.
- (b) The dried soil and silt samples were sifted by hand using a 8" diameter 80-mesh stainless steel sieve. The plus 80-mesh fraction was rejected and the minus 80mesh fraction was transferred into a new bag for analysis later.
- (c) The dried rock samples were crushed by using a jaw crusher and pulverized to 100-mesh or finer by using a disc mill. The pulverized samples were then put in a new bag for later analysis.

2. Method of Extraction

- (a) 20.0 30.0 grams of the pulp samples were used. Samples were weighed out by using a top-loading balance into fusion pot.
- (b) A Flux of litharge, soda ash, silica, borax, flour, or potassium nitrite is added, then fused at 1900 degrees F and a lead button is formed.

- (c) The gold is extract by cupellation and part with diluted nitric acid.
- (d) The gold bead is saved for measurement later.

3. Method of Detection

- (a) The gold bead is disolved by boiling with sodium cyanide, hydrogen peroxide and amonium hydroxide.
- (b) The gold analyses were detected by using a Techtron model AA5 Atomic Absorption Spectrophotometer with a gold hollow cathode lamp. The results were read out on a strip chart recorder. The gold values in parts per billion were calculated by comparing them with a set of gold standards.
- The analyses were supervised or determined by Mr. Conway Chun or Mr. David Chiu and his laboratory staff.

David Chiu

VANGEOCHEM LAB LTD.

VANGEOCHEM LAB LTD. 1521 Pemberton Ave. North Vancouver, B.C. V7P 2S3

TO:

Dr. Stan Hoffman BP - Selco Mining

Suite 700 - 890 West Pender Street

Vancouver, B.C. V6C 1K5

FROM:

Vangoechem Lab Ltd. 1521 Pemberton Ave.

North Vancouver, B.C. V7P 2S3

SUBJECT: Analytical procedure used to determine Aqua Regia soluble gold in geochemical samples

1. Method of Sample Preparation

- (a) Geochemical soil, silt or rock samples were received in the laboratory in wet-strength 4" x 6" Kraft paper bags or rock samples sometimes in 8" x 12" plastic bags.
- (b) The dried soil and silt samples were sifted by hand using a 8" diameter 80-mesh stainless steel sieve. The plus 80-mesh fraction was rejected and the minus 80mesh fraction was transferred into a new bag for analysis later.
- (c) The dried rock samples were crushed by using a jaw crusher and pulverized to 100-mesh or finer by using a disc mill. The pulverized samples were then put in a new bag for later analysis.

2. Method of Digestion

- (a) 5.00 (10.00) grams of the minus 80-mesh samples were used. Samples were weighed out by using a top-loading balance into beakers.
- (b) 20 ml of Aqua Regia (3:1 HCl : HNO3) were used to digest the samples over a hot plate vigorously.
- (c) The digested samples were filtered and the washed pulps were discarded and the filtrate was reduced to about 5 ml.

- (d) The Au complex ions were extracted into diisobutyl ketone and thiourea medium. (Anion exchange liquids "Aliquot 336").
- (e) Separate Funnels were used to separate the organic layer.

3. Method of Detection

The gold analyses were detected by using a Techtron model AA5 Atomic Absorption Spectrophotometer with a gold hollow cathode lamp. The results were read out on a strip chart recorder. A hydrogen lamp was used to correct any background interferences. The gold values in parts per billion were calculated by comparing them with a set of gold standards.

4. The analyses were supervised or determined by Mr. Conway Chun or Mr. Eddie Tang and his laboratory staff.

Eddie Tang

VANGEOCHEM LAB LTD.

APPENDIX 2
LIST OF ANALYTICAL DATA

SOUP CLAIMS

REANALYSIS OF ROCK CHIP SAMPLES >1000 ppb Au
(By Fire Assay - Atomic Absorption Finish)

Sample #	Length (m)	ANALYSIS 20 gm Geochem Sample FA-AA ANALYSIS	REANALYSIS Average of two 20 gm Samples FA-AA ASSAY
852002	0.4	2359	2550
852004	0.2	4700	5600
852006	3.0	3600	3950
852009	2.0	1050	1050
852010	1.5	7340	8500
852505	2.0	3250	3900
852507	5.0	4100	4600
852515	5.0	1360	1550
852516	3.5	15560	13500
852517	1.0	20640	23550
852542	3.0	2025	1850
852543	3.0	1750	1800
852544	3.0	1700	1650
853517	4.0	3750	4250
853524	3.0	2750	3100
853525	5.0	1750	1900
853526	5.0	3450	2900
853529	5.0	1675	1300
853530	1.5	60850	62300
853539	2.5	2950	3050
853547	3.0	3515	3650
853549	3.0	1975	2100
853564	3.0	1900	2200
853565	3.0	1400	1600

		ANALYSIS	REANALYSIS
	Length	20 gm Geochem Sample	Average of two 20 gm Samples
Sample #	(m)	FA-AA ANALYSIS	FA-AA ASSAY
854153	5.0		
854159	2.5	7800	8500
854163	5.0	1270	500
854176	3.0	2400	2300
854177	3.0	1450	1400
854179	3.0	1500	1550
854182	3.0	3475	3400
854183	3.0	1220	1450
854184	3.0	2325	2300

ROCK SAMPLE ANALYSES

SELECTION # 1

SAMPLE TYPE(S)				80/81	/82/83													
BEDROCK TYPE (3)			ALL														
SOIL HORIZON(S	j)			ALL														
SAMPLE TEXTURE	E(S)			ALL														
OVERBURDEN ORT	(GIN(S)			ALL														
LABORATORY-SI	ZE FRACT	ION-EXTRACT	ION(S)	ALL														
PAIR STATUS				ALL														
REC#	SHPL#	UTH-E UTH-	-#								HO	CU	FB	711	NI	U	ĦN	FE AG
1 8184555	852001			94D08 3	1 129DGY	42	MGYT222		40	LC	3 .	6753	Ģ	38	33	5	428	3.97 1.3
2 8184555	852002		Ū	94008 3	1-136L0B	4	WH TII				17	97	7 :	12	11	E.A.		4.88 1.4
3 8384555	852003		0	94008	7 136LOB	41	WH T15				3	51	3	20	5	5	120	4.55 .5
4 8184555	852004		Ħ	94008 3	1 179 GR	22	LGRT222	CU			32	9978	4	35	20	5	653	8.31 5.2
5 8184555	852005		Ğ.	94009	1 136L08	3	WH T11	CU			4	4477	5.	24	10	-5		2.11 1.9
8 8184555	852006		I	94D0B 3	1-136MOB	31	MGYT123		45		6	559	15	5 0 .	36	5	542	15.5 1.1
7 8184555	852007			94008	11 139HGR	32	MGGT123		45	CL.	. 2	2038	11	66	65	Į.		6.84.5
8 8184555				94D08-3	1 129MGR	32	MGGT123			CL	1	1995	5	32	43	5	1. 1. 1. 1.	4.62 .7
9 8184555	852009				1 139WOB			CA	45	Ļ ·	1	2187		43	28	5	566	7.35 2
10 8184555	852010			94008 3	1 136MOB	34	MGYT123		45		4	742	11	54	-39	11		13.866.4
11 8184555	852011				1 13690B				45	1	79	620	12	80	62	5		12.99.7
	852012		.1	74009 3	1 136000	31	LGRT123				1	216	4	13	5	5		7.03 .4
184555	852501				12 631MRB						29	705	5	65	35	5		14.571.2
14 8184555	852502				2 52 HBR					EC	-		1	52	32	5.		5.1 .5
15 8184555				94008	1 52 MGR	22	MGRT212	MAAI	7	ECL	1	168	i	27	w.	5		2.88 .6
16 8184555				94D08 3	1 52 MGR	22	MGRT212	MAAZ	9			148		31	-32	5		4.28 .5
17 8184555					1 321MOB			FE				2007		24	4			23.347.7
18 8184555					1 531MRB			FE		CEL	2	108		46	20	5		5.09.2
19 8184555					1 531MBR				7	EL	-,-	566		- 61	28	5		17.031.4
70 8184555					I 531MBR					CE		364		48	35	5		7.74 .8
21 8184555					1 531MBR				7.			539		28	28	5		3.78 .2
22 8184555					1 531MBR			CPRYFE	9	CQE		-	5	45	-	5		5.9 .5
23 8184555					11 526MBR		MGRT222				3		7	51	14	5		7.15 .5
24 8184555					11 526HBR						1		2	38	11	5		2.59 ,2:
25 8184555					1 526MOB						4		5	26	18	5		3.16.4
26 8184555					1 526HOB		MGRT224	<u> </u>			į	76	3	29	9	5		3.43 .1
27 8184555					72 445MBR:			FE	9.		4		á	44	21	5		8.46 .4
28 8184555					2 431M8B			FE	9		13		8	37	14	5		12.151.3
29 8184555					32 431MOB			FE	ņ	QE .	99	719		23	5	5		18.837.8
	852518				12 43 MGR		MGR	FE			2	271		32	ó.	5		2.51 .1
31 8184555	852519				52 42 MBR		MGY 123	CV			5	505	2	13	11	5		4.27 .1
32 8184555	852520				12 42 MDR	33	MGY 123	CA			4		2	23	8	5		4.41 .2 1.71 .1
33 8184555	852521				51 52 MBR		LGY 123				4		1 7	11		5 5		4.74 .2
34 8184555	852522				61 52 MBR		L6Y 123			400	j. L	228	3	20 23	6	5 - 5		5.58 .3
35 8184555	852523				31 52 MBR		DGR 222	rć ·			i	206	5	23 23	14 7	5 5		7.9 .3
38 8184555	852524				1 42 MBR		DGR	FE	7		i .	214 70	1.2	23 37	72 72	5		2.55 .1
37-9184555	852525				11 2 MBR		DGRT222	cc			i	70 107	1	30	11	5		3.2 .1
3 84555	852526				1 26MBR		LBRT223	FE	9.	c	1	7	2 4	36 26	31	J 5		2.58 .1
39 8184555	852527			74000	SI 2 MBR	11	DGRT22	FE	7	Ε	\$	1		40	U.S.		494	LIUU el

290 8184555	853512	94D08 33 126DGY	32 MGGT212	4	IC EQ 9	414 5 46	18 5	676 7.24 .9
291 8184555		94D08 33 12 DGY			C E 1		18 5	648 4.07.5
292 8184555	853514	94008 33 125MGB	32 MG6T212	0	EQ 2	188 1 20		341 2.52 .2
293 8184555	853515	94D08 33 126HGY	42 MGYT212	Ą	C3 CEQ 2	423 5 41	20 5	687 3.19 .1
294 8184555		94D08 33 126M6Y			4 EQC 1	131 2 43		672 3.96 .1
295 8184555	85 3517	94D08 33 126MBL		C	4 87	473 8 42		517 12.39.9
296 8184555	853518	94D08 33 121MGY			4 EQC 22	364 8 50		847 10.45.2
297 8184555	853519	94008 33 121DGY		C	EQC 1	74 4 28	29 5	493 3.17 .1
298 8184555		94D08 33 121M8R			4 EC			
299 8184555		94D08 33 12 MGY			0CE 1	351 6 37	25 5	756 5.85 .1
300 8184555	853522	94D08 33 121MBR			5	335 2 25		377 4.57 .1
301 8184555	853523	94008 33 121MGY	31 MGYT224		3	158 3 24		317 3.26 .1
302 8184555	853524	94D08 33 121DGY	31 MGYT212 F	FE 4	6	579 8 64		935 11.19.3
303 8184555	853525	94008 33 121067	31 MGYT212 F	FE	3	690 6 5 3	20 5	785 9.78 .3
304 8184555		94D08 33 121D6Y	31 MGYT212 F	FE 9	4 E 8	1257 5 53	24 5	825 9.96 .5
305 8184555	853527	94D08 33 121D6Y	32 MGYT212	į.	5	561 2 - 35	26 5	553 4.86 .1
304 8184555	853529	94D08 33 121D6Y	42 MGYT212	4	6	699 6 45	25 5	773 6.46 .2
307 8184555	853529	94B08 33 121DGY	32 MGYT212		11	1098 3 51	14 5	758 7.06 1.3
308 8184555	853530	O 94008 33 1260GY	41 DGYT212	4	QC 57	2274 22 31	7 5	294 24.825.5
309 8184555	853531	94008 33 12 DBY	32 MGGT212	C	Ε 1	176 1 15	13 5	193 1.71 .1
310 8184555	853532	94D08 33 12 DGY	32 MGGT212	C	E 1	26 1 15	11 5	177 1.3 .1
311 9184555	853533	94D08 33 12 MGY	32 MG6T212	C	EQ 1	79 3 29	31 5	361 2.91 .1
312 8184555	853534	94D08 33-126MGY	32 M00T212	0	EI	41 3 48	12 5	519 3.07 .1
317-3194555	853535	94D08 3 6D8Y	31 MGYT212		E 1	112 1 31	16 5	429 4.14 .1
3 1 84555	853536	94D08 33 12 DGY	32 MGYT212		Ε 1	15 5 28	12 5	307 3.43 .1
315 8184555	853537	94008 33 12 DGY	32 MGYT212	C	E 1	570 3 29	12 5	267 3.05 .3
316 8184555	853538	94008 33 12 06 Y	21 MGYT212		EQ 1	23 4 31	3 5	314 3.18 .3
317 8184555	853539	94D08 33 126MGY	MGYT123	4	2	3780 18 49	29 5	364 27 3.4
318 8184555	853540	94D08 33 126MBR	MGYT123		1	90 2 43	25 5	748 4.36 .1
319 8184555	853541	94D08 33 126MBR	32 MGYT123	C	E 1	301 2 12	4 5	132 2.09 .5
320 8184555	853542	94D08 33 121MBR	32 MGYT123	0	4 E 1	184 4 29	43 5	597 4.52 .4
321 8184555	853543	94D08 33 121MBR	32 MGYT123	C	EQ 1	87 3 42	59 5	797 4.24 .3
322 8184555	853544	94D08 33 121MBR	32 MGYT123	C	ED 1	41 1 24	38 5	329 2.11 .2
323 8184555	853545	94D08 33 12 MBR	32 MGYT123 F	FE C	E 1	1268 2 16	4 5	519 2.23 .2
324 8184555	853544	94008 33 121MBR	32 MGYT123 F	FE C	1	2605 2 32	9 5	521 7.3 .7
325 8184555	853547	94D08 33 121DRE	32 DRET123 F	FE	QC 1	1807 12 766	5 5	574 30.2 5
326 8184555	853548	94D08 33 121DRE		FE	QC 1	4337 5 48		586 10.491.4
327 8184555				FE	1	979 27 51		558 30.853.1
328, 8184555		94D08 33 12 LGY	32 MGYT22?	7	0 1	50 8 50		731 3.77 .2
329 8184555				T	0 2	88 13 10		331 2.68 .4
330 8184555	853552	94008 33 126MBR		FE 9		z 78 _. c. 2 -75		661 3.27 .6
331 8184555		94D08 33 126MGY		FE 9		82 53 18		126 2.7 1.4
332 8184555		94008 33 12 MGY	31 MGRT22		PC OC 1	22 4 70		803 2.54 .1
333 8184555		94D08 33 12 MGY		7	C 1	504 2 16		158 1.3 .1
334 9184555		94D08 33 12 MGY			1	367 3 24		215 1.84 .2
335 8184555		74D08 33 121MGY			_ 1	324 6 12		115 .92 .2
334 8184555		94D08 33 126MRB			C 1	498 3 15		134 1.31 .7
337 8184555		94D08 33 126MRB			EC 1	638 5 25		186 2.33 .2
338-9184555		94008 33 126MRB	31 T227		E 2	449 5 10		111 1.85 .4
33 84555		94D08 3			11	645 3 7	1 5	78 1.83 .6
340 8184555	853562	9400 8 3			i	79 3 24	7 5	267 2.52 .3

341 8184555	853543			94D08 32	131DRB	Ą	HOKT	FE	· a ·	G	20	707	71.	64	21	5	319	29.091.	4
342 8184555				94D08 32				FE	· p	0	99	1346	22	76	38	5	307	30.663.	5
	853565			94D08 31			MBKT	FE	94	0	11	589	26	74	20	5	251	30.423.	5
	853566			94008 31	131088	4	M8KT223	FE		0	1	513	17	39	11	5	308	30.681	
345 8194555				94008 32			MBKT223	FECU	7		1		11	44	14	5		15,53,5	
	853568			94008 38			MBRT223	FECU	ş.	E0	7	2436		71	14	5		11.971.	
347 8184555				94D08 35			MGRT223	FECU	Q.	ΕO	Á	5111	_	93 83	12	5		6.09 .3	
348 8184555				74D08 33			MGRTZ23	FELG	ģ.	EL	3	2320		58	10	্র -			
										-								14.89.8	
349 8184555				94D08 33					ņ	EL		2820		48	12	5		2.32 .1	
	853572			94D08 33		-	HGRT22?	FE	9	EL	1	1763		53	11	5		4.02 .1	
351 8184555				94008 33			MGRT227		ÿ Ā	EL	1	152		40	· •	5		2.39 .2	
352 8184555				74008 33					9	EL			1	48	6	5		2.53 ,2	
353 8184555				94008 32			MGRT223		Ģ.	EL	1	24	4	44	5 .	5		2.29 .2	
	853576			94D08 32			HGRT223		9	EL	1	30	2	35	5	5		2.78 .1	
	853577			94D08_32					Ģ	EI.	1	8	3	43	8	5		2.31 .2	
	853578			94D08 36					· Ģ	EL	1		1	55	61	5		4.05 .2	
357 8184555	853579			94008 32	13 MBR	2 -	DGRT22		Ŗ.	EI_	1		3	39	28	5	372	2.33 .1	
389 8184555	853931			94D08 33	126DRE	42	LGGT223		94	@LC	3	LV	ā	95	Ģ:	5	473	5,22 .3	
390 8184555	853932			94D08 33	126MRB	32	MGYT223		9	QL.	2 -	29	. 7	36	7	5	379	3.68 .2	
391 8184555	853933			94008:33	126MR9	32	MGGT223		9 :	BLC	2	106	5	26	4	5	284	3.79 .3	
392 8184555	853934			94008 33	126MGY	34	LGYT123PF		P	0	<u> </u>	64	3	48	7	<u> </u>	677	3.35 .1	
393 8184555A	854001			94D08 33	12 LGY	32	MGYT222		C		1	27	5.,	άl	13	<u> </u>	738	2.73 .1	
394 8184555	854002			94008:33	12 LGY	31	HGYT222		C		1	24	7	65	9 .	5	572	2.5 ,1	
395 9184555	854003			94008-33	12 LGY	32	MGYT222		C4		1	17	5	65	9	5	769	2.51 .1	
39 84555	854004			94D08:33	12 LGY	32	MGYT223		C		1	19	3	77	6	5	827	2.55 .1	
397 8184555	854005			94008 33	12 LGY	32	MGYT223		C		1	37	3	63	8	5	744	2.66 .1	
398 8184555	854006			94008 33	12 LGY	32	MGYT223		Г		. 1 1	20	4	73,	8	5	. 796	2.68 .1	
399 8184555	954007			94008 33	12 LGY	32	MGYT223		C		1	36	2	57	25	5	665	2.65 .1	
400 8184555	854008			94008 33	12 LGY	31	MGYT223		04		1	75	6	72 -	32	15	1047	4.12 .1	
401 8184555	854009			94D08 33	12 LGY	- 32	MGYT223		C4 -		2	38	6	62	32	15		3.7 .1	
402 8184555	854010			94008 33	12 LGY	31	MGYT223		C		1	43	á	71	15	5		3.28 .1	
403 8184555	854011			74D08 33	12 LGY	32	MGYT223		$\mathbb{C}_{\mathbb{R}^{n}}$	Ε	1	30 .	5	óć	16	5	825	3.12 .1	
404 8184555	854012			74008 33	12 LGY	32	MGYT223		04		1	43	5	83	15	5	1063	3.54 .1	
405 8184555	854013			94008 33	12 LGY	32	MGYT223		C		1	32	8	115	15	5	1018	3.24 .3	
406 8184555	854014			94D08 33	12 LGY	31	MGYTZ23		C		1	32	- 1	146	7	5	868	2.52 .1	
407 8184555	854015			94008 33	. 12 LGY	37	MGYT223		Ç .		2	33	7	180	3	5	755	2.61 .1	
408 8184555	854016	·		94008 33	12 LGY	31	HGYT223		0.		1	51	5	115	5	5	754	2.91 .1	
	854017			94D08 33	12 LGY	-32	MGYT223		C		1	55	5	ĠΪ	ā	5	690	3.07.1	
410 8184555				94008 33	12 L6R	42	HGYTZZ3		CA		1	42	3	32	é	5	866	3.01 .1	
111 8184555				94008 33	12. LGR	32	MGYT223		C4		1	35	2	77 -	40	5	765	2.88 .1	
412 3184555	854020			94008 33	12 LDR	47	L0YT223		C1		2	23	7	40	7	5.	474	3.38 .1	
	854021			94000 33			MGYT223		C4	0	1	24	1	48	10	5	591	4.07 .1	
414 9184555	854072			94D08 33			H66T223		C4		1	E t	. 6	55	91	11		3.56 .1	
	854023			94D08 33			MGYT223		C4:		1	35	5	57	58	5		3.83 .1	
416 8184555	854024			94D08 33			MGYT223		C		1	37	4	62	37	5		4.18 .1	
	854025			94008 33			MGYT223		C		1	39	5	75	32	5 .		3.74 .1	
418 8184555				94D08 33			MGYTZZZ		Ç		2	56	6	90	7	5		3.8 .1	
417 8184555				94D08 33			MGYT223		C		1	57	Ę	72	4	5		3.83 .1	
420 0184555	854028			94008 33			LGYT223		C		1	25	ó	52	7.	.		3.48 .1	
42 84555	854029			74008 33			LSYT223		C4		2	55	8	73	14	5		4.28 .2	
422 8184555				94D08-33					C		1	57	1		Ģ	5		3.73 .1	
	- 1																		

				A -															
	423 8184555	954031	94008 33	17AL GY	32	MRYT273		C4		1	70	7	112	10	5	1110	4.49	.3	
	424 8184555		94DOB 31		-	MGYT223		Ĉ.		2	33	8	123		5		5.21		
	425 8184555		94008 33			MGYT223		C4		•	52	8	139	8	5		4.35		
	426 8184555		94D08 31			MGYTZ23		Ĉ	Ε	1	47	5	89	7	5		4.85		
	127 8184555		94008 33			MGYT223		C4	•	•	41	7	37	6	5		3.68		
	428 8184555		.94D08 31			MGYT223		C4	Ε	1	36	7		11	5		3.31		
	429 8184555		94D08 33			MGYT223		C4	E	1	39	2	67	5	÷		3.81		
	430 8184555		94008 31			LGYT223		C4	Ε	1	35	5	69 69	á			3,89		
	431 8184555		94D08 31			LGYT223		.01 .01	1.	1	35 -	3	69 69	ė.	5		3.14		
	432 8184555		9400B 31			MGYT223		ſ		i.	41	a E	67 . 67	7	5		3.06		
	433 8184555		94D08 33			M671223		C		1	35.	5	.a., 76		J E3		3.22		
	434 8184555		94D08 33			MBYT223		, E		f.	27	5	7 u 5 t	17 .	5		3.83		
	435 8184555		94008 33			MGYT223		Ē.	E	1	33·	7	56	á	5		3,5		
	436 8184555							ς.	E	2	26	4	52	Ω 3	5		3.85 .		
			94D08 33			MGYT223		C		4	41	2		о 7	ر 5		3,30 		
	437 8184555		94008 33			MGYT223		C		i	52	2	65 65	9	5 5		3.04		
	438 8184555		94B08 33			MGYT223		C	ε'	. <u>3</u> . #		3		ี 13	5		3.71		
	439 8184555		94D08 33		44	MGYT223			E	i	49 47	3 3			5 5		3.15 .		
	440 8184555		94D08_33			T223		C .	Ε	2 .	43		64	8					
	441 8184555		94008 33			1223		C	E	1	42	1	52	3	<u>.</u>		2.71		
	442 8184555		94D08 33					€.	EL	1	40 40	1	50	6	5	522 -		. [
	443 8184555		94008 33					C		1	40	3	54 .	8	5		2.77		
	444 8184555		74008 33					:C	EL	1	36	4	52	7	5		2.92		
	445 8184555		94008 33					°C .	EL.	1		1 .	47.	14	5		2.67		
	44 \$ 84555		94008 33				-	0	Ε	1	20	2	41.	28	5	501	2,33		
	44 184555	854055	94D08 33					C4	E	4	40 .	3.	48		10	959	3.55		
	448 8184555		94008 33			LGGT222		C45		3	<u> 3</u> 0 ·	4	37	48	Ģ E		1.84		
	449 8184555		94008 33			MGGT223		C	E	<u> </u>	5	2	69	5	5		3.29		
	450 8184555		94008 33					C	E	I	10	3	42.	1.7	5	399	3.48		
	451 8184555		94D08 33					C	Ε	1	10	i	40	<u>.</u>	5		3.06		
	452 8184555		94008 33			1223		U.		1	41	1	77	6	5		3,58		
	453 9184555		74008 33			T223		C -		1	27	1	62	5	5		3.08		
	454 8184555		94D08 33			T223		C		1	47	2 -	38	6	5		3.15		
	455 8184555		94D08 33					_		1	116	2	53	64	5		4.17		
	456 8184555		94D08 33					C		1	39	2	19	24	5	619	4.15		
	457 8184555		94D08 33						EL	1	17	1	35	ត់	5		2.95		
	458 8184555		94008 33							1		4	38	ά -	₩		3,27		
	459 8184555		94008 33							1	27	3	41	5	5	444	3.24		
	460 8184555		94008 33						EL.	ŀ	33	4	32	3a	5		3.06		
	461 8184555.		94D08 33							1	22	3	35	7	5		3.58		
	462 8184555		94008 33					C		1		.1 .	28.	18	5		2.13		
	463 8184555		94D08 33			L00T123		. C -		1		2.	30	15	5 -		2,27		
	464 8184555		94008 33			L66T123		C	E	1		2	29	14	5	298	2,65		
	465 8184555		 94008 33			M6YT123		С	F	1	79	2.	28	14	5	352	3,02		
	466 8184555		94D08 33					 E	Ε	1	87	3	33 .	lá	5 5		2,0		
	467 8184555	854075	94008 33			MGYT123		C		. 1	211	4	47	29	5	720	3.52		
	468 8184555	854076	94D08 33			HGYT123		C .		. 1	79	3.	32	16	5		2.46		
	467 8184555		94009 33			MGYT123		C		1	109	3	34	10	5		2.75	1.0	
•	470 8184555		94D08 33			MGYT123		C		1	230	7	42	30	5 //	621	6.12		
	471 9184555	954079	74008 33			MGYT123.		C		1	361	8	49	60	5 .		12.99		
	47 84555	954080	94008 33			LGYTZZZ			1.	3	127	ģ	40	17	5		7:92		
	473 8184555	854081	94D08 33	12 LGY	21	M6Y T222			EL	1	129	7	35	15	5	457	3.68	* 4	

474	8184555	R54082			94D08 33 1	2 MGN	12	H6YT222		EL	4	115	1	43	25	5	664	4.28	. 3
	8184555				94008 33 1						1		3	34	30	11		3.35	
	8184555				94008 33 1			MGYT222			1		i	43.	12	5		3.45	
	8184555				94008 33 1				4	Ē.	1		1	40	13	5		4.28	
	8184555				94D08 33 1			MGYT222	C	_	ţ		1	41	42	5		4.38	
	8184555				94008 33 1			GY0T222	462	•	7	217		30	19	5		2.69	
	8184555				94D08 33 1			LGYT222	46		4		1	-38	150	5		4.36	
	8184555				94008 33 1				64		16		3	11	11	5		9.28	
	8184555				94D08 33 1				C		3		1	21	79	5		1.54	
	8184555				94008 33 1						1	137		34	50	5	575	3	
	8184555				94D08 33 1				C		1	113		379	31	5		3.59	
	8184555				94D08 33 1			MGYT222	4C		1	118		38	41	5		2.94	
	8184555				94D08 33 1				C .		1		5	47	111	5		4.96	
	8184555				94008 33 1				C.		1		1	32	18	5	195	4.07	
	8184555				94D08 33 1				C		1		2	34	17	5		4.01	
	8184555				94D08 33 1				64.	c	1		1	24	21	.u 5	384	2.32	
	8184555				94D0B 33 1				64		1		1	41	23	, 4 5		4.19	
								MGNT222	46		1		7	60	57	5		5.4	
	8184555				94008 33 1	100				E.	-		12	46	97 33	5 5			
	8194555				74008 33 1			DGGT222	ó		1		12 6					4.9	
	8184555				94008 33 1				₹.	E	1		9 .	43	29	5		5.26	
	8184555				94D08 33 1			LGYT222	48	15.	1			66	15	5		3.83	
	8184555		1.		94008 33 1			DGRT222		LE			7	44	40 10	5		4.42	
	8194555				94D08 33 1	,		DERT222		EL.		26	11	62	103	្រ		4.94	
-	30	854105			94008 33 1				, C4		1	93	10.	60	45	5		4,76	
	84555	854106			94D08 33 1				C4		1	83	10	á7 	92	5		5.2	
		854107.			94008 33 1			MGYT222	C4 -		5		13	32	177	5		5.15	
	8184555				94008 33 1			H6YT222	4C		1	85.	5	45	47	5		3,88	
	8184555				94008 33 4			MGYT222	40		1	82	9	87	40 .	. <u>박</u>	1211		.5
	8184555				94D08 33 1				6	. EL			7	38	20	5		1.51	
	8184555				94008 33 1				ó.		1		1	35	38	5		3.12	
	8184555				94D08-33 I			MGNT222	ò	_	1	55	3	21	15	.5		1.63	
	8184555				94008:33				6	L.	1		8	24	30	5		2.24	
	8184555				94009 33 1				- 6 -	EL.	2	88	3	31	26	5		2.8	
	8194555				94008 33 1				6	E	1		1	28	43	5	389	2.56	
	8184555				94D08 33 1			MGRT222	6		1	107		37	12	5	1000	3.82	
	8184555				94008 33 1				6	Ε	7	174		47	30	5		3.8	
	8184555				94008 33 1				á	E	2	154		41	20	5		4.25	
	9134555				94008 33				. 6	E	1		3	53	22	5	708	4.17	
	8184555				9400B 33 I			MGYT222			1	75	1	32	14	5		4.13	
	8184555			١.	94D08 33 1			MGYT222			1		1	40.	14	5	522	4,67	
	8184555				94D08 -3 1					Ü	2	79.	7	75		5	293	2.34	
	8184555				94008 33			MGYT222		Œ	1	72.	7	34	44	5	797	4.97	
	8184555				94D08 33 1			MGYTZZZ			1.	79	1 - 1	28	25	5	37,8	7,72	
517	8184555	854125			94008 33	12 DGY -	23	MGYT222	5	ELC	1	75	7	42	27	5	860	4.77	
	8184555				94008 42 6				6		10	336	10	26	á	5		11.72	
	8184555				94008 42: 8	426L0B	33	ORBT222	6 -		10		16	21	3	5- :-,	189	22.3	
520	8184555	854128			94008-42 8	526L0B	33	ORBT222	å		13		11	30::	11	5	271	21.21	
521	8184555	854129			94008 42	126L08	3,1	LGRT123	6		1 .	330 -	12	55	26	5	457	13.25	
522	~184555	854130			94008 42 8	526LBR	33	WH6T123	ó		5	507	17	32	26	5		14.71	
52	84555	851131			94008 42	626LBR	22	WHST123	5		9	474	10.	24	13	5		11.27	
	8184555				94008 42 1	12 1.BR	21	LGYT123		٤	1	213	5	38	15	E.	. 532	3.13	, đ

525 8184555	854133	94008 42 12 LBR	21 LGYT123	1 3	56 2	32 10	5 437	2.81.3
526 8184555		94D08 4Z 12 LGG		Ēi	56 2	30 14		1.97 .4
527 8184555		94008 42 12 LGG	22 LGYT123	6 E 1	149 2	37 11		2,59 .5
528 8184555		94008 42 12 MGG		5 E 2	38 3	35 15		2.65 .3
529 8184555		94D08 42 12 NGG		2	13 4	38 15		3.05 .4
530 8184555		94D08 34 626M08		ó 55	602 11	28 7		26.51.8
531 9184555		94D08 34 626M00		6 6	620 19	28 6		37.461.5
532 8184555		94D08 34 626M08		6 2	376 11	23 5		18.8.3
533 8184555		94008 34 626L08		5 2	352 10	20 4		18.78.7
534 8184555		94D08 34 126L08		6 E 4	746 10	33 20		10.4 .5
535 8184555		94D08 34 12 LBR		6 E 2	70 1	33 14		3.04 .3
536 8184555		94008 34 12 HR6		6 E 1	101 1	37 15		2.67 .4
537 8184555		74008 41 12 MGY		C4 . 1	127 6	35 41		4.36 .5
538 8184555		94008 41 12 MGY		C42 E 1	64 3	42 74	•	4.87 .
537 8184555		94D08 41 12 MGY		6 E 1	121 2	33 14		4.43 .5
540 8184555		94008 41 12 GYB		64 E 1	58 6	37 7.6		3.2 .5
541 8184555		94D08 41 12 GRB		6 E 2		19 13		1.7 .2
542 8184555		74D08 41 12 GRB		5 3	117 1	7 7		4.37 .5
543 8184555		94008 41 126L08		6 5	228 - 3	14 13		3.88 .8
544 8184555		94D08 41 126L08		ó 16		ນ ນ 1ນ		5.47 .9
545 8184555		94008 41 126L08		6 E 33		12 5		8.7 1.3
546 8184555		74D08 41 126H08		6 E 2	2009 12	71 14		19.84.2
547 8184555		94D08 41 12 MO8		0 C 4	495 13	27 10		24.36.5
					471 9			
549 84555 54 84555		94008 41 12 1.08		5 E 4	123 . 3	25 13 26 23		16.92.7
		94D08 41 12 MG6			153 3	20 20 38 27		2.39 .5
550 8184555 551 8184555		94008 41 12 NGG		69 Q 13'	T 17 T T 1 T 1 T 1 T 1 T 1 T 1 T 1 T 1 T	27 8		3.69 .6
551 8184555		94D08 31 631M0B			115 3		~	18.074.8 J
552 8184555		94D08 31 12 NGS		4 1		49 32 31 31		5.33.1
553 8184555		94008 31 12 MGG		4 1	1283			6.28 .1
554 8184555		74008 31 12 MGG		6 EC 1	121 4			4.78 .1
555 8184555		94009 31 12 LGG		6 E0 2 6 EC 2	2515 7	35 26 37 29		5.5 1.2
556 8184555		94008 31 12 LGG 94008 31 12 LGG			433 3 267 1	37 29 20 13		5.41 .4
557 8184555				E 1	124 3	23 30	5 320	2.65 .1
558 8184555 FFO 0104FFF		94D08 31 12 BRG		E 1	124 6	23 30 32 15		4.37 .2
559 8184555		94008 31 12 BRG		6 1	97 3	36 17	7 77	3.67 .1
560 8184555		94008 42 12 MGY		E 1				
561 8184555		94008 42 12 MBY		E 1	90 3	33 16	5 419	2.94 .1 2.18 .2
562 8184555 563 8184555		94008 42 12 MGY 94008 42 12 MGY		E 1	64 3 ·	27 28 33 20		3.08 .1
				E 1	109 4			4.21.1
564 8184555 565 8184555		94D08 42 12 LGY		6 E 1	93 2 58 7	46 22 45 19		3.08 .2
		94008 42 12 LGY				10 15 27 15		1.76 .2
566 8184555		94008 42 124GYB			126 4		5 279	8.75 .7
567 8184555		94808 42 126L08		5 2	375 9 542 15	22 40 27 15	5 210 5 228	22.4 1.9
568 8184555 548 0104555		94008 47 525NOB		33		34 80	5 228 5 319	26.231.4
569 8184555 E7A 0103EEE		94008 42 126MOB		18 514 OC 5		51 27	5 545	7.2 .4
570 9184555 571 9194555		94D08 11 126ORG		546 QC 5 546 QC 4		12 6	5 109	6.1 2.3
571 8194555		94008 31 1260RG	43 LGYT223					4.48 .9
572 8184555		94D08 31 126NOR	33 MGYT127	56 QC 4	69 6	58 30 50 16	5 437 5 476	
573 9184555			33 L0YT22?	56 2	31 á		5 40 5	8.73 .4
57 84555		74D08 31 126L0B		C 25	24 7			8.47 3.1 / 25.981.5
575 8184555	5341 8 3	91008 11 626M08	34 UNBIZZ/	6 18	908 18	. 3á 11 .	J 310	4.1.701.3

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576	8184555	854194				74D08	11 526	H08	34 G	RBT227			6		21	353 :	5 á(33	5	4	53	25.76	3	
	8184555						11 626			GYT227			6			511 7			5	- 1	61	7.07	.2	
1.	8184555					94008	11 125	GYB	22 1.	GYT222					1 1	161 8	2	7 22	5	2	95	2.27	2	
579	8184555	854187				94008	11 12	MGY .	22 L	97T223			6	E	1	232 8	4	7 20	5	6	17	4.25	7	
REC#	SMPL# CO	AU	AU7. AS	3 : HG	SD	SH	lik 1	F	TH	CD	BI	Ų.	BA	SR.	- 51	AL (TA 1	1G : N	A A	K :	ZR?	CE?	11	_
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298 853520																					
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319 853541		20 1	5	7 -	2	2	2	1	2	36	14	39	.01			.21		.02	7	4	. 14
329 353542		30 1	2	2	2 :	7	4	1	2	92	11	76	.01	-2.01	5.88	2.47	.01	60	6	2	.11
32 3543	17	25 1	2	2	2	2	3	"- 1	2	123	11	55	.01	2.88	4.99	3.8	.01	.09	3	2	.08
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323 853545	24	100 1	3	3	2	2	7	- 1	2	35	8	31	.01	.74	1.59	.18	.02	.02	3	3	.07
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329 853551		20 1	24	2	2	2	2	1	3	47	114			1.15				. 34		2	,11
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335 853557		10 1	2	2	2	2	2	1	2	41	31			.87						2 2	.17 .2
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340 853562 341 853563			17		2	2	£ .	1	2	75	25	11		1.15				.11		7	.12
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345 953567		135 1	17	4	2	2	4	1				21		1.47						2	.22
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532 854140 40	1 11	2 2	2 7	1 2	62 25	35 .01	.99 1.27 .13 .01	.03 4	2 "07
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534 854142 21 70	1 4	2 2	2 5	1 7	77 78		1.2 .3 .8 .04	.28 5	2 .16
535 854143 9 50	1 4	2 2	2 4	1 2	53 81	47 .01	1.49 .47 1.05 .04	.11 3	2 .12
536 854144 11 20	1 3	2 2	2 3	1 7	18 66	8101	1.92 .65 1.51 .05	.09 2	2 .15
537 854145 14 100	1 2	2 2	2 6	1 - 2	75 203	18 .01	1.7 .58 1.71 .04	.22 2	2 .13
538 854146 16 30	1 2	2 2	2 4	1 2	75 202	22 .01	1.96 1.06 2.08 .03	.27 2	7 .13
537 854147 3 10	1 2 2	3 2	2 4	1 2	64 176	53 ,01	1.06 .69 .62 .05	.14 3	2 .12
540 854148 10 15	1 2	2 2	2 5	1 2	91 96	45 .01	2.28 1.53 2.71 .03	.24 2	2 .15
541 854149 4 35	1 3 3	2 2	2 4	1 2	50 60	52 .01	1.31 .76 .74 .05	.13 4	2 .17
542 854150 1 30	1 4	3 2	2 3	1 2	37 14	48 .01	.82 .79 .23 .05	.05 6	2 .17
543 854151 6 65	1 3	2 2	2 4	1 2	62 -60	65 .01	1.11 .85 .5 .03	.16 5	2 .23
544 854152 1 180	1 5	2 2	2 5	1 2	95 34	88 .01	1.02 1.04 .16 .03	.04 5	7 , 25
545 854153 1 1500	1 10	2 2	2 4	1 2	71 43	55 .01	.94 1.72 .09 .02	.03 9	2 .22
546 854154 1 120	1 7,	2 2	8 8	1, 2,	57 10	47 .01	.66 1.27 .13 .01	.03 4	2 .09
517 854155 1 90	1 5	2 2	ਵ ਹ	1 2	70 26	55 .01	.8 .79 .24 .03	.04 6	2 .12
548 854156 7 160	1 9	2 7	2 8	1 2	77 28	45 .01	.98 .45 .46 .05	.07 8	2 .24
549 854157 7 10	1 5	2 2	2 5	1 2	74 88	54 .01	1.77 .81 -1.4 .04	.13 3	2 .25
550 854158 7 30	1 2		2 4	1 2	109 74	36 .01	2.34 .94 2.05 .04	.17 3	2 .29
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553 954141 18 20	1 2		12 2	1 2			2.17 1.06 2.27 .02	.32 4	4 .14
554 854162 22 165	1 2		2 2	1 2	149 67		2.78 3.88 3.38 .02	.15 3	4 .11
555 854163 21 1270		2 2	2 2	1 2	145 107		2.5 1.71 2.73 .02	.28 3	3 .13
55 4164 21 140	1 3		2 2	1 2	131 79			.17 1	4 .15
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560 854168 7 0	1 2	2 2	2 2	1 2			2.04 .58 1.92 .04	.17 2	2 .15
561 854169 6 5	1 3		2 2	1 6	97 42			.12 4	2 .15
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554 854172 10 10	1 2		2 3	1 2			2.34 .5 2.41 .03	.45 2	7 .18
565 854173 16 10	1 2		2 2	1 3	97 63		2.41 .64 2.39 .03	.13 3	2 .18
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567 854175 16 140			2 4	1 6			1 .42 .88 .03		5 .16
548 854176 9 2400.			2 1	1 2			.53 .24 .08 .03		7
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572 854180 35 900 573 854181 23 430			2	1 2			2.24 .11 2.51 .04		7 .01
573 854181 23 430 574 854182 47 3475			7 7	1 2			1.86 .47 1.98 .04		2 .02
575 854183 20 1220			7 3 2 5	1 4			.13 .01 .08 .04		2 .01
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TALUS FINES SAMPLES ANALYSES

SELECTION # 1

SAMPLE TYPE(S)			50/51/6	Λ/ Δ 1												
BEDROCK TYPE(S			ALL	MOL												
SOIL HORIZON(S			ALL													
SAMPLE TEXTURE			ALL											19.75		
OVERBURDEN ORI			ALL													
		ON-EXTRACTION(S														
	CC LUHCIT	UNTEXIRHUI (UNI)	ALL													
PAIR STATUS	ewer #	UTH-E UTH-N	HLL						MO	CD	PB	7N	ΗI	U	MN FE A	G
REC# 80 6084555	SMPL#	6262616253304	94D08 1D2	. 50	Ato to	tcoananao	40A		365W2		2	100	23	5	1068 6.67	
	853001		94D08 3E2			TFR222DBR	30A		348M4	1045		35	61	5	1341 7.52	
	353002	6262436253297	94008 3E2			IFRZZZDON DM8222NOLS			368W12	998		28	46	5	1110 7.81 .	
82 5094555	853003 057004	6262266253291 · 6262096253285	94D08 3E2			TFR222HOLE			365W17	1043		20	. 55	5	981 7.37	
83 6084555 84 6084555	853004 853005	6261916253276	74000 3E2 94D08 3E2			TFR222MOLE			36889		4	37	47	5	1073 4.54	
	853006	6261766253266	94008 3E2			BMB222MOLB			36888		4	37	25	5	638 5.22	
85 5084555						DNB2ZZNOLD TFR2ZZMOLD			21000 21000	-	7 3	38	74 74	9. 5	756 6.61 .	
86 6084555	853007	6261566253252	94008 3E3					- 1. j	375 4 9	105á		36	36	5	774 d.71	
	853008	6261496253238	94D08 2E3		410 151					1120		36 36	32 ·	5 5	1018 7.38 4	
88 5084555	853009	6261416253222	94008 3E2		410 20				375 13		_			. ජ වි	1727 7.39 1	
89 5084555	853010	6261366253206	94D08 3E2		410 151				328 8		ם ח	48 43	47 65	5 5	1470 6.53	
90 5084555	853011	6261296253187	94D08 3E2			BMB22 MOLE			378 5		7			_		
	853012	6261236253165	94D08 3E2			TFR22 MOLE			37544	635	4	35	81	5	1248 6.11 .	
084555	853013	6261256253145	94D08 3E2		410 101				345W4	556 	å .	58	80	5	2160 6.3 1	
75 3084555	853014	6261226253123	94008 3E2		510, 201				37W 8	737	8	34	51	5	1054 6.23 1	
94 4084555	853015	6261156253106	94D08 3E2		210 10				379 1		15	39	45	5	605 4.91 .	
95 6084555	853016	6261026253096	94D08 3E2			TFR222MOLE			35W 4	634	9	37	50 50	5	1419 7.43	
96 6084555	853017	6260856253087	94D08 3E2			TFR22 MOLE			35MW1		2	39	56	5 .	774 5.3 ,	
97 5084555	853018	<i>\$</i> 260866253077	94008 3EZ		410 20				364 5	472	ė.	3á	51	5	1314 6.33 .	
78 5084555	853019	6260456253069	.94008 3E2		510 35				36W 4	425	8	57	48	5	1620 6.43 .	
99 6084555	853020	6260256253061	94D08 3EZ		310 15		LB 50A		360 2		4	45	199	5	2091 6.13	
100 5084555	853021	8260056253051	94008 3E2		410 30				380 1		5	48	92	5.	1197 5.11 .	
101 5084555	853022	6259846253045	94D08 3E2		410 15				388 1	288	8	10 .	43	5	1120 4.55 .	
102 5084555	853023	6259646253038	94008 3E2		410 15				384 1		5	47	44	5	1141 5.04 .	
103 5084555	853024	6259456253031	94D08_3E2		410 15		20A		3814 52	1702		-37	58	5	778 8.43 .	
104 6084555	853025	6259256253023	94D08 3E2		310,20				38W 10	1034		48	51	5	1271 7.43).	
105,6084555	853026	6259056253012	94D08 3E2		210 10				35% 25	.846		45	25	5	958 17.23.	
106,6084555	853027	6258866253902	94D08 3E2	L 8P	310 15				35W 8	407	.7	48	46	5	2417 6.42 .	
107 5084555	853028	6258856252992	94D08 3E2		410 15				35W 7	639	δ,	19	47	5	1864 6.32 .	
108 6084555	853029	6258486252981	94D08 3E2						35W 13	912	12	35	- 25	5	987 15.94.	
109 6084555		6258486252981											12		743 17.37.	
110 6084555		6258326252971					3R 25A		35M (9	856			32			
111 5084555	853032	6258116252967							35W 11			12	33	5	949 8.35 .	
112 6084555		8258008252958							35# 14	1166		40	46	5		
113 6084555	953034	6257816252954							35W 22	1807		35	32	5	1134 11.73.	
114 6084555	853035	6257646252944					3R 90A		354 5	662		35	30	5	464 7.68.	
115 5094555	853036	6257466252938	94D08 3E2	L ap	410 15		IR 15A		35W 8	962		37	28	-5	1114 8.22 .	
116 5084555	853037	6257276252935	-94008-3E2	L-89	410 25		9R25M			256		42	27	5	865 6.02 .	
11 184555		6257086252928					1011			139		11		.		
113 084555	853039	6256916252922	94D08 3E3	L BP	510 20	CIR MGY	3N: 25H		35W 3	885	. 5	46	56	5	1265 7.67 .	. 4

272 6084555	853193	6273036251173	94D08 3E2L 8P	310 20TFR	MRORBR	10A	3.5	ISW22	62 17	171	8 25	3951	8.76	.1
273 6094555	853194	6272906251155	94D08 2E2L 8B	2 5 15TFR -	MRD	50A			41 5	51	3 5	539	1,96	.2
274 6084555	853175	6272636251151	94008 3EZL 8B	4 5 10TFR	MRB	20A -			213 22	141	34 5		6.03	
275 6084555	853196	6272406251142	94D08 3E2L 8B	4 5 10TFR	DBR	.30A			217 18	110	12 5		5.41	
276 6084555	853197	6272196251143	94008 3E2L 8B	4 5 10TFR	Red	20A			183 15	118	16 5		5.45	.6
277 6084555	853198	6272086251124	94B08 3E2L 8B	4 5 10TFR	DBR	15A			270 11	115	31 5	4156		3
278 6084555	853199	6271996251102	94D08 3E2L 8B	1.2 STFR	DBR	40%		55 41	20d 14	205	26 5		7.88	
358 6084555	853700		94D08 321L P	310 10BTL2	DBR -	90A			755 9	45	24 5		4.08	
357 6084555	853901		94D08 321L P	310 15BTL2	MBR	80A		39 5	463 8	41	14 5	931	5.01	
340 4084555	853902		94D08 321L P	310 10BFF	HOB	70A			334 12	47	.10 :5	612	5.12	
-361 6084555	853903		94008 321L P	311 10BFP	MOB	60A			512 13	49	11 5	497	3.73	
362 6084555	853704		94D08 321L P	310 10BFP	MOB	á0A			347 10	43	7 5	497	6.05	
363 6081555	853905		94D08 3 IL 1P	310 10TFP	MOB.	60A		55 5	296 . 11	45	6 5	462	6.78	
364 6084555	853906		94D08 - IL 19	310 10TFP	MOB .	50A		15 á	430 16	65	7 5		6.85	
365 6084555	853907		74D08 - IL 1P	310 (OTFP	MOB	50A		55 6	370 12	51	7 5	918	6.53	
366 6084555	853908		94D08 - 11 1P	310 10TFP	MOB	60A.			148 22	40	19 5	659	8.64	
367 6084555	853909		94008 - 1L 1P	310 10TFP	MOB	50A		55 14	285 15	65	21 5		6.36	
368 6084555	853910		94D08 3 IL F	310 10TFP	HOB	40A		56 6	231 30	50	15 5		9.35	
369 6084555	853911		94D08 3 1L P	310 10TFP	HOB	60A		55 8	121, 12	25	7 5	346		.2
370 6084555	853912		94D08 3 IL P	310 10TFP	HOB	80A		15 5	73 61	23	3 5	17.4	. 6. 76	
371 6084555	853913		94008 3 1L P	310 10TFL	MGYBR	80A		5S 6	335 6	33	7 5	640	5.78	
372 6084555			94D08 3 1L P	310 10TFL	MGYBR	70A		59 4 · ·	307 9	55	19 5		4.63	
373 6084555			94D08 3 1L 1P	310 10TFL	MBR	70A		55 20	521 7	64	11 5		5.12	
374 6084555			94D08-3-1L 1P	310 10TFP	MORBR	60A		59 7	202 12	44	11 5		7.55	
37 84555	853917		94D08 3 1L 1P	310 10TFP	DOB	40A		55 9	130 13	41	18 5	781	6.11	
378 8084555	953918		94D08 3 1L 1P	310 10TFF	HGYBR	70A		55 5	229 . 13 :	44	19 5	822	6.54	
377 6084555			94008 3 IL IP	310 10TFL	MBR.	808		55 10	307 8	50	16 5		4.34	
378 6084555			94D08_321L_1P	310 10TFP	MBR	70A		55 6	400 5	39	36 5	697	4.23	
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380 6084555			94D08 3211 1P	310 10TFL	DBR	95A		59 5	340 7	215	22 5		4.59	
381, 6084555	34 707 17 54 3		74D08 321L 1F	310 15TFP	MOB	70A		58 8	762 5	54	48 5		6,44	
382 6084555			94D08 321L 1P	310 10TFP	HOR	70		55 2	267 4	84	. 29 . 5		5.12	
383 6084555			94008 321L 18	310 10TFL	DBR	70A		55 3	313 4	71	25 5		5.12	
384 6084555			94D08 321L 18	310 10TFL	MBR	75A		38 4	208 . 7	54	14 .5		5.6	
385 6084555			94D08_321L_1P	310 LOTEP	MOB	60A		56 5	114 8	49	10 5		5,39	-
386 6084555			94D08 3Z1L 1P		MOB	80A		55 B	271 13		15 5		5.55	
387 6084555			94008 321L 1P		MO9	80A		55 6	167 17		21 5		8.73	
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359	853711	.2	8	4	11
370	853912	.15	10	3	6
371	853913	.7	11	2	11
37	53714	.15	9	3	24
375	653915	.16	11	5	7
374	853916	.23	10	2	16
375	853917	.17	11	2	25
376	853918	.21	9	7	27
377	853919	.16	7	2	25
378	853720	.14	7	3	51
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382	853924	.19	2	3	61
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384	853926	.19	4	1	21
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388	853930	.17	2	7	25

APPENDIX 3
STATEMENT OF COSTS

SOUP 8-84 CLAIM GROUP 1984 EXPLORATION PROGRAM

STATEMENT OF COSTS

ANALYTICAL COSTS:

ANALYTICAL COSTS:					
200 soil samples @ \$12.13 (Au geochem, +ICP)	\$2,426.00				
345 Rock samples @ \$15.77 (Au geochem, +ICP)	5,440.65				
545 Samples Shipping Charges Computer processing @ \$2.00/sample	7,866.65 200.00				
x 545	1,090.00				
Interpretaion - 1 day - Dr. S.J Hoffman @ \$300.00	300.00				
TOTAL ANALYTICAL COSTS		\$ 9,456.65			
FIELD LABOUR COSTS:					
Project Geologist - 15 days @\$141.13/day	\$2,116.95				
Geologist - 20 days @\$89.24	1,784.80				
Geological Assistant - 15 days @ \$68.78	1,031.70				
Geological Assistant - 24 days @ \$55.00	1,320.00				
Field Assistant - 24 days @ \$55.17	1,324.08				
Field Assistant - 20 days @ \$65.63	1,312.60				
Supervisory visits - 2 days @\$300	600.00				
TOTAL FIELD LABOUR		\$9,490.13			

CAMP COSTS:

5	men, 10 days @	\$50/day	\$2,500.00
	men, 6 days @		1,800.00

TOTAL (All inclusive of equipment, \$4,300.00 room & board, pilot)

AIR CHARTERS:

Helicopter - 5 hours @ \$450/hr. \$2,250.00 -24 hours @ \$486/hr. \$11,664.00 \$13,914.00

Fixed Wing (Total) 1,000.00

TOTAL AIR CHARGES \$14,914.00

TRAVEL EXPENSES:

Hotel, meals - 2 men, 2 nights @\$50/night 200.00
TRUCK RENTAL 500.00

MAPS AND REPORT PREPARATION:

Project Geologist
- 5 days @ \$141.13 \$ 705.65

Drafting - 20 hours @ \$17.00/hr. 340.00

Materials 100.00

Orthophoto preparation
(McElhanney Ltd.) 5,820.00

TOTAL COSTS \$45,826.43

APPENDIX 4 STATEMENT OF QUALIFICATIONS

H.Q. SMIT R.E. MEYERS

STATEMENT OF QUALIFICATIONS - H.Q. SMIT

B.Sc. (Hons.) Geology, 1984 - University of British Columbia, Vancouver

H.Q. Smit was employed as an exploration geologist with BP Resources Canada Limited from May to November, 1984. Prior to this he was employed for two field seasons with the Geological Survey of Canada and has held field assistant positions with various mining and exploration companies in Western Canada.

STATEMENT OF QUALIFICATIONS - R.E. MEYERS

B.Sc. (Hons.) Geology 1974 - Carleton University, Ottawa
M.Sc. Economic Geology 1980 - McGill University, Montreal
Associate Member of the Geological Association of Canada (1974)
Member of the Canadain Institute of Mining and Metallurgy

I have practised my profession continuously since graduation in 1974, as a Mine Geologist (1974-1977); in Economic Geology research (1977-1979); and in mineral exploration (1979-present).

