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**OVERBURDEN GEOCHEMISTRY AND  
POSSIBLE RELATIONS TO  
GOLD-BEARING ALLUVIAL DEPOSITS,  
SOUTHWEST GASPÉSIE, QUEBEC**

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**M.E.R.I. Project P85-15**

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## ABSTRACT

Possible sources of various elements and minerals, including gold and garnet, found in anomalous concentrations in stream sediments within the Matapedia River drainage basin in southwest Gaspésie were investigated in a study of the geomorphology and compositional nature of the upland overburden in an area of about 2400 km<sup>2</sup>.

Investigations on the geomorphology of the area show that Pleistocene events have had little influence in shaping the observed topography and drainage patterns. The project area is a dissected plateau with summit altitudes from 1300 to 1500 feet (396 to 457 m) above sea level, and floors of the major valleys from 150 to 750 feet (46 to 229 m) above sea level.

Upland overburden is thin and commonly not thicker than the soil profile. It varies from a diamict with a large proportion of the underlying bedrock and a lesser component of far-travelled material, to a rubble composed almost entirely of weathered underlying bedrock.

Major element analyses of the -60 mesh (<250  $\mu$ m) fraction of overburden and trace element analyses of heavy mineral concentrates show changing compositions across the area in approximate correspondence with changes in underlying rock types. The abundance of heavy minerals in overburden is related to the concentration of Shield-derived minerals within the heavy mineral fraction. The distribution of heavy mineral abundances in turn reflects the distribution of bedrock lithologies and suggests that lithological differences may have affected the pattern of

deposition and erosion of the Laurentide ice sheet as it flowed across the area.

Gold analyses within the -60 mesh (<250  $\mu\text{m}$ ) fraction of overburden show that gold is in very low concentrations. However, gold analyses within heavy mineral concentrates extracted from overburden provide a better estimate of gold variation across the area and outline a number of anomalous zones. High abundances of gold in heavy mineral concentrates from overburden samples collected north of the mouth of the Assemetquagan River and east of the Matapedia River (within a 36  $\text{km}^2$  area) support the hypothesis of a local source to the north-northwest for the alluvial gold at the mouth of the Assemetquagan River.

## 1.0 INTRODUCTION

In this project the geomorphology and compositional nature of upland overburden is examined in a 2400 km<sup>2</sup> area of southwestern Gaspésie (Fig. 1). The purpose of the investigation is to ascertain the role of the surficial cover as a source of specific elements and minerals, including gold and garnet, found in the stream sediments of tributaries of the Matapédia River. Stream sediments have been systematically collected across the entire report area by the Ministère des Ressources Naturelles du Québec (Tremblay and Choinière, 1978; Choinière, 1984) with more recent sampling in the Assemetquagan River basin by the Geological Survey of Canada (Maurice, 1986). This report is to provide data chiefly on the geochemical and heavy mineral composition of upland overburden which comprises thin glacial drift (till), alluvium of fluvioglacial and non-glacial origin, colluvium, rubble and other diamict materials of uncertain origin. Sampling for this project was carried out in 1985 and 1986. The 1986 sampling program work was conducted in a 36 km<sup>2</sup> area outlined by UTM coordinates 639000 and 645000 E and 5324000 and 5330000 N. This area encloses the Assemetquagan River alluvial gold occurrence (Maurice, 1986).

## 2.0 PREVIOUS WORK

The numerous bedrock geology maps of the Gaspésie Peninsula were compiled by McGerrigle and Skidmore (1967) and Skidmore (1977,1980), to provide a framework for the region under study.

Details of the lithostratigraphic characteristics of individual formations within the project area were obtained from Béland (1960), Stearn (1965), Lachance (1974,1977,1979) and Bourque and Lachambre (1980).

The surficial deposits of western Gaspésie have been studied for many years by J. Lebuis and P.P. David who summarized their findings in 1977 (Lebuis and David, 1977) and, following further work mainly by Lebuis, presented a revised theory of glaciation of the Gaspésie (David and Lebuis, 1985). Their studies are concentrated along the north coast of the peninsula where the glacial deposits are thicker and where there are stratigraphic successions, but the sparse drift of the southern part of the peninsula was also examined. Some of the findings of Prichonnet and Desmarais (1985) concerning the stratigraphy and composition of drift in the Lake Matapedia area may have some relevance to the compositional nature of drift in our area.

The Quebec Government conducted an extensive stream sediment trace element geochemical survey which covered the entire Gaspésie Peninsula (Choinière, 1982). Results of the 15-year survey were published as a series of 7 geochemical atlases. A set of 1:500 000 scale compilation maps summarizing the 75 000 samples analysed was presented by Choinière (1984). One of the main findings of the survey was that the concentration of trace elements in the stream sediments varies characteristically across the peninsula with changes in the underlying rock types. Choinière therefore suggested that Quaternary glaciations had very little influence in shaping the observed geochemical

patterns, except in the the north-central part of the peninsula where north-northeastward geochemical dispersion from the McGerrigle Mountains could be associated with northeastward glacial transport of drift as documented by Chauvin (1984).

Tremblay and Wilhelm (1978) evaluated the use of heavy mineral sampling of stream sediments as prospecting tools in a section of the peninsula which included a part of the present study area.

Girard (1985) investigated gold and associated minerals in heavy mineral and fine fractions of stream sediments of the Assemetquagan River. He proposed, from their physical appearance, that garnet and magnetite have a long transport history and that pyrite, hematite and ilmenite are of a more local origin.

Maurice (1986) made regional and detailed stream sediment surveys for gold in southwest Gaspésie. From an orientation survey in the auriferous lower segment of the Assemetquagan River he found that gold is preferentially concentrated in crevices and cleavage openings of fissile bedrock and also in moss on surfaces of boulders and bedrock. He suggested that the gold may have been eroded from Fortin Group lithologies by the Laurentide ice sheet during its southeastern movement across western Gaspésie. Deposition of gold and Shield-derived heavy minerals in the Assemetquagan valley would have been promoted by the retardation or obstruction of the debris-rich basal part of the ice sheet by the higher northwest-facing valley wall.

### 3.0 GEOLOGICAL FRAMEWORK

The stratigraphic and lithologic character of the sedimentary formations which underlie the report area (Fig. 2) are summarized in Table 1.

The Gaspésie peninsula forms the northwestern extremity of the Appalachian Physiographic Region on the continental mainland (Bostock, 1970). First order subdivisions include (1) the Notre Dame Mountains, an elevated upland surface which extends along the northern two-thirds of the peninsula and includes the Shickshock and McGerrigle mountains, and (2) the Chaleur Upland, a deeply dissected upland surface which borders Chaleur Bay to the south.

Within the western part of the Gaspésie peninsula, the Notre Dame Mountains and Chaleur Upland subregions consist of 3 major SW-NE trending, fault-controlled belts of tightly to gently folded Paleozoic rocks. The northern (Taconic) belt extends 40 km inland to the south margin of the Shickshock range and it is separated from the adjacent central belt to the south by the Shickshock fault. It consists of Cambro-Ordovician flysch sediments of the Quebec Group and includes the metavolcanic and metasedimentary rocks of the Shickshock Group, and the intrusive rocks associated with the McGerrigle plutonic complex. The adjacent central belt is a zone 65 km wide of gently folded, weakly recrystallized Silurian to Devonian strata. The northern 40 km consists of a basal Lower Silurian siltstone unit (St.-Léon Fm.) overlain by two units of Early to Middle Devonian limestone units (Cap Bon-Ami and Grande-Grève Fms.) and topped by a

sequence of Lower to Middle Devonian terrigenous clastic sediments (York Lake, York River, Lake Branch and Battery Point Fms.). The entire siltstone-limestone-sandstone succession is adjoined to the south by a trough of thick Lower Devonian siltstone and shale (Fortin Gp.) 25 km wide, with minor interbedded acid and intermediate volcanics and pyroclastics (Stearn, 1965). The southern belt borders Chaleur Bay to the south and is separated from the central belt by the Matapedia (Restigouche) fault. It contains a variety of rocks of different ages, ranging from Upper Ordovician to Lower Silurian flysch sediments (Honorat and Matapedia Gps.), through Upper Silurian to Lower Devonian shallow marine and terrestrial clastic sediments and associated mafic intrusions and andesitic volcanics (Chaleur Gp.), to Carboniferous terrestrial conglomerates and sandstones (Bonaventure Fm.).

The Taconic Orogeny (Upper Ordovician) caused most of the deformation recorded in the Cambro-Ordovician rocks (Williams and Hatcher, 1982). Later deformation during the Acadian Orogeny (Middle Devonian) affected the Siluro-Devonian strata and produced the major SW-NE trending structures observed across the southwestern part of the peninsula.

#### 4.0 GEOMORPHOLOGY

The project area is a dissected plateau with summit altitudes from 1300 to 1500 feet (396 to 457 m) above sea level, and floors of major valleys from 150 to 750 feet (46 to 229 m) above sea level. Projected profiles show that the upland is a rolling surface having about 300 feet (91 m) relief, and that the



present valleys are incised into broad valleys with floors now about 1000 feet (305 m) above sea level. The present streams occupy V-shaped valleys incised about 500 feet (152 m) below the ancient valley floors. The Assemetquagan and Milnikek Rivers have formed ingrown meanders in the younger inner valleys; however the present rivers are changing from a smoothly sinuous meander pattern to one of straight segments and sharp bends. This suggests that the river is actively cutting downward into bedrock, perhaps as a result of reduction in grain size of the load that it carried when the smoothly sinuous pattern formed.

Transverse profiles across the Matapedia valley suggest an ancient valley floor at about 1000 feet (305 m) above sea level, and one or two sets of lower terraces above an inner valley. North of Latitude  $48^{\circ} 15'$  the Matapedia valley is broad, with a floodplain in which the river meanders in an overfit fashion. South of this latitude the valley is a narrow, in places sinuous, canyon with little or no floodplain. The northern segment, which crosses relatively competent rocks (dominantly siltstones), is a typical glaciated valley, but the southern segment, deeply entrenched across weak shales of the Fortin Group, shows little evidence of glacial erosion.

A study of air photographs yielded no conclusive evidence of glacial erosion in the region: in two places small lakes occupy basins that may have been glacially formed in valleys that have broad U-shaped profiles, but such profiles may be produced by processes other than glacial activity.

The drainage pattern of the area, especially the first and

second order streams, is strongly controlled by structure. In the northwestern third (north of the Fortin Gp., Fig. 2) the main streams trend northwest across anticlinal and synclinal structures, and lesser subsequent streams are in valleys between resistant beds. In this region the drainage pattern is in part deranged, mainly by the glacial erosion of northwest-trending valleys and later deposition. In the central part of the area, from the northern boundary of the Fortin Group to the divide on the south side of the Assemetquagan valley, the streams have a modified dendritic pattern and are deeply incised into the shales. This is part of a larger basin trending southwest-northeast across the Matapedia valley, including the valleys of du Moulin River and the lower part of the Milnikék River in the west and the Assemetquagan River in the east. These rivers discharge southeast through the Matapedia valley. First and second order streams form a sharply rectangular pattern, with a tendency for northwest trends to be closely spaced and southwest trends to be farther apart. Some of the southwest-trending valleys form lineaments that can be traced for many kilometers. In some places the asymmetric, coarsely pinnate patterns of low order streams suggest a response to the dip directions of the rocks.

In the southeastern quarter of the area (south of the southern watershed of the Assemetquagan River) the streams have a parallel pattern flowing southeast, and the lower half to two-thirds of their courses lie in the relatively resistant rocks of the Matapedia and Chaleur Groups. These streams are eroding headward and capturing drainage areas within the Fortin Group.

The Matapedia is the main trunk of the drainage system. In the south half of the project area, below Ste. Florence, its valley is anomalously straight and narrow for the downstream segment of a valley that systematically broadens north of Ste. Florence. The southern part follows the regional southeast trend. North of the mouth of the Milnikek River, the valley appears offset to the east, and south of it, it is aligned with the lower reaches of the Milnikek River and a minor tributary. There are two occurrences of ice-contact stratified deposits plastered against the east side of the narrow Matapedia valley. The valley floor is gravel derived from fluvio-glacial outwash and younger alluvial fans.

The present drainage has evolved through a long interval, possibly extending back into late Paleozoic time. The major forms are little modified by Pleistocene events. The narrow, steep valleys incised into an older rolling surface suggest rejuvenation by uplift or obliteration of lithologic base levels rather than by climatic change. There are relatively few barbed tributaries and wind gaps, which suggests that much denudation has occurred since adjustment to structure was first achieved. The narrow valley of the lower Matapedia appears to be very young by comparison; however it must antedate the last glaciation. An arcuate valley as wide and deep as the Matapedia joins the head of a tributary of the Assemetquagan River to the Kempt-Ouest River and may have been the outlet of the Moulin-Assemetquagan basin before headward growth of a south-flowing stream captured the system and became the Matapedia River. It is also possible

that a diversion of the present Matapedia drainage through this valley was caused by a lobe of ice advancing down the Restigouche valley from the southwest, for which there is some other evidence, albeit sparse (Fig. 3).

If the Fortin Group formed a basin at some time in the past when the Matapedia and Restigouche Groups formed a range of hills on the north side of the Restigouche valley, the basin may have discharged around the west end of the hills via the Patapedia system or northward through the Matapedia valley (reversed) north of Ste. Florence. It can be speculated that the southern hills must have shed detritus that accumulated on the Fortin Group as a series of alluvial fans. The steep gradient of the southeast-flowing streams would have caused them to grow headward across the hills and eventually capture the drainage on the Fortin rocks.

## 5.0 SURFICIAL GEOLOGY

The surface deposits of the region were studied by Lebuis (1975), whose work is summarized in David and Lebuis (1985). In the northwest and northeast corners of the study area till is as thick as two meters, and eskers, other ice-contact stratified drift, and outwash are common. Colluvium on valley sides contains more than one lithology and includes a component derived from till. However in the southern part of the Matapedia valley, glacial deposits are represented by only two ice-contact deposits. The valley contains much sand and gravel on the floor and in terraces. Some of this is outwash, much of which is reworked by the present river. There are several alluvial fans

deposited by tributary streams, which are still active. These fans have a generous proportion of cobbles and boulders. Accumulations of reworked outwash sands reaching thicknesses of up to 90 meters are found on the valley sides of the Matapedia River below the mouth of the du Moulin River (Fig. 3). These thick sand deposits originated from the alluvial reworking of large outwash fans that were deposited in the narrow Matapedia valley at the mouths of du Moulin River and tributaries to the south when the last ice sheet was actively retreating to the north, releasing large volumes of detritus into the Matapedia River and its tributaries. The great thickness of the alluvium on the valley walls of the Matapedia valley from the mouth of the du Moulin River south to Chaleur Bay may have been associated with the presence of an ice lake in Chaleur Bay at the time of deglaciation. Damming of the Matapedia River at or near its mouth would have resulted in a wave of deposition upstream causing a rise in the elevation of the valley floor and deposition of large outwash fans at the mouth of tributaries possibly all the way up to the Assemetquagan River.

On the uplands of most of the area the drift is thin, commonly not thicker than the soil profile. It is generally a diamicton with a large proportion of underlying bedrock, but locally contains sparse to rare erratics. In places the upper part is fine sand or silt that may have been windblown. Exposures are poor, and it is not clear how important the role of cryoturbation is in mixing bedrock fragments with overlying material. The upland overburden over much of the south-central

section of the report area consists of a thin veneer of rubble derived from the physical weathering of the underlying rock, in which is mixed a very small component of far-travelled material. Sparse to rare erratics mantle the upland surface and provide evidence that the area was glaciated. Colluvium derived from the fissile shales and siltstones of the Fortin Group forms extensive and commonly thick sheets at the foot of valley slopes. It is clearly evident, from the often impressive recent accumulations of talus along road cuts within Fortin lithologies, that the colluvium-forming process provides a very effective pathway for moving materials from the upland areas to the valley floors and into the drainage systems.

David and Lebuis (1985) have strong evidence of a southward flow across our area in the distribution of erratics from the Precambrian Shield (gneisses, anorthosites), and the distinctive Val Brilliant quartzite. We are able to add to their data points, and to add another rock type, the red Lake Branch sandstone, from which erratics are distributed a few kilometers both to the north and south of the source exposure (Fig. 4). Some pink granitic erratics on the north side of the Restigouche estuary may have come from the west. A limited number of variations in northern New Brunswick reported by Rampton et al. (1984) support the tenuous evidence in our area for glacial flow at some time from the west or southwest.

## 6.0 METHODS

### 6.1 Field Methods

The 1985 field sampling program was designed to provide a

representative coverage of the study area. Overburden was sampled by collecting approximately 10 litres of diamicton from the C horizon of hand-excavated pits at approximately 1.5 to 3 km intervals along roads accessible by 2-wheel-drive vehicle. Samples with a high matrix (sand size and finer)-to-clast ratio were collected where possible, but where the matrix fraction was small, larger samples up to 20 litres were taken. In all, 295 samples were collected (Appendix 1).

During the summer of 1986, overburden samples were collected in a 36 km<sup>2</sup> area which encloses the mouth of the Assemetquagan River (Fig. 1). The sampling procedure was modified to take into account the very low abundance of far-travelled material in the upland drift in the general surrounding area. Overburden was sampled by collecting approximately 25 litres of material screened on site to -4 mesh (<4.75 mm). Care was taken to collect samples in the C horizon; however at most sites the overburden was thin and lacked distinct soil horizons. Sixty-nine samples were collected on the upland surface at a spacing of 500 to 700 m. A further 13 samples were collected on the valley slopes within the lower 1 km segment of the Assemetquagan River (Appendix 2).

Sample treatment is illustrated in Figure 5. Approximately 1 litre of material was separated from the 1985-series samples and then screened to -4 mesh (<4.75 mm) in preparation for geochemical analysis. The remaining fraction of the sample was then wet sieved to obtain the 63-250  $\mu$ m (very fine to fine sand) fraction from which heavy minerals were extracted and analysed. Sample treatment was slightly modified for the 1986-series

samples. These were wet sieved to a larger size fraction (63-450  $\mu\text{m}$ ) in an attempt to produce heavy mineral concentrates suitable in size for geochemical analysis.

Where observed, glacial erratics were identified and plotted on a map. Particular attention was given to the areal distribution of red sandstone erratics from the Lake Branch Formation which forms a narrow belt across the Matapedia River, 6.5 km south of the village of Causapsal (Fig. 2), and white orthoquartzite indicators of the Val Brillant Formation which outcrops west of Lake Matapedia. Specimens of each of the various sedimentary formations that outcrop in the study area were also collected for major and trace element analysis.

## 6.2 Geochemical Methods

The -4 mesh ( $<4.75$  mm) 1985-series samples were dry sieved to obtain 140-180 g of the -60 mesh ( $<250$   $\mu\text{m}$ ) fraction, for use in determining gold, major and trace element content. The samples were milled to -200 mesh ( $<75$   $\mu\text{m}$ ) and then sent to a commercial laboratory where they were analysed for gold by combined fire assay and plasma emission spectrometry (detection limit of 2 ppb). One hundred of these samples were also analysed for major and trace elements using a Philips PW 1400 X-ray spectrometer in the Department of Geological Sciences, McGill University. Loss on ignition was also measured on these samples.

Two hundred and forty (0.2 - 2.0 g) heavy mineral concentrates were sent to a second commercial laboratory where they were analysed for gold and 25 other elements by regular irradiation neutron activation. Ninety-six of these samples are



from the detailed 1986 study area.

### 6.3 Heavy Mineral Studies

Light and heavy preconcentrates were obtained from 165 1985-series samples and all of the 1986-series samples using a Mozely MK II mineral separator. Heavy minerals were then extracted from the preconcentrates by density-dependent settling in methylene iodide (3.3 s.g.). Magnetic minerals were removed from the heavy fraction using an automagnet. The resulting mineral fractions were weighed to provide weight percentages of heavy mineral in the total sample and magnetic minerals in the heavy mineral fraction. The loose non-magnetic heavy mineral fractions were examined under a binocular stereoscopic microscope in order to determine (1) the characteristic mineral assemblages, (2) the approximate proportion of local and far-travelled components and (3) the presence of visible gold in the samples.

## 7.0 RESULTS

### 7.1 Major and Trace Element Analyses

The results of the major and trace element analyses on the 100 -60 mesh (<250  $\mu\text{m}$ ) samples are given in Appendices 4 and 5. A legend for analytical Appendices 4 and 5, and Tables 2 and 3 is given in Appendix 3. Analyses are arranged by underlying rock formation or group and overburden type, and averages for the grouping are given in Tables 2 and 3.

As a generalization, overburden samples collected from the northern half of the project area have the highest concentrations of  $\text{SiO}_2$  (74.33-81.33% avg.) (underlain by York Lake, York River, York Branch, Cap Bon-Ami and Grande-Grève Formations). Samples

collected over Fortin Group lithologies in the central part of the area have intermediate  $\text{SiO}_2$  contents (70.9-74.6% avg.) and those underlain by the Matapedia Group show the lowest  $\text{SiO}_2$  concentrations (61.7-68.2% avg.) A cursory inspection of the lithological characteristics of the different formations (Table 1) and their respective approximate  $\text{SiO}_2$  contents (Table 4) shows that the north to south decrease in the concentration of  $\text{SiO}_2$  in overburden corresponds to a general trend of decreasing  $\text{SiO}_2$  across the different lithological units, from sandstones, siltstones and silty limestones in the northern units, shales and siltstones in the Fortin Group, and limestones and calcareous mudstones in the Matapedia Group. The parallelism of  $\text{SiO}_2$  variation in overburden with that of the underlying lithologies indicates that the composition of the former is greatly controlled by the composition of the latter, in other words, the overburden contains a significant proportion of locally-derived bedrock, a feature readily observed in the field.

The few bedrock specimens analysed do not permit estimation of compositional ranges within lithological units or comparison of  $\text{SiO}_2$  concentrations in overburden in relation to a specific formation or group. The bedrock samples roughly match their corresponding overburden equivalents. The principal exception is the rock specimen from the Grande-Grève Formation which is high in  $\text{CaO}$  and low in  $\text{SiO}_2$  in contrast to till and colluvium samples which are high in  $\text{SiO}_2$  and low in  $\text{CaO}$ . A second exception is the single Fortin Group bedrock sample with a  $\text{CaO}$  concentration (1.74%) well outside the range of  $\text{CaO}$  (0.07-0.43%) in the corresponding overburden samples.

A comparative summary of trace element averages for the major overburden types and the stream sediment data of Choinière (1982) is shown in Table 5. Observed concentration differences among upland overburden, stream sediments, and lithological formations and groups cannot be readily explained without taking into account variations associated with differing analytical procedures and detection limits, and the number of samples analysed. Considerable concentration overlap exists, and patterns like the ones observed with the major element distribution are not apparent.

#### 7.2 Heavy Mineral Concentrate Analyses

The results of the trace element analyses on the 221 heavy mineral concentrates are listed in Appendices 6 and 7. A further 19 analyses are from check samples inserted to verify the precision of the results (Appendix 7).

The geochemical data were subjected to factor analysis (Table 6). Selenium and Ag were excluded from all statistical treatment because of abundances below the detection limit in almost all cases. Because detection limits were highly variable for certain elements (a function of sample size and composition), analyses below the detection limit were considered as missing values. The missing values were treated by pair-wise deletion, that is, in the calculation of the correlation coefficient matrix, a sample would be deleted from the calculation of the coefficient for an element pair if one or both of the elements had missing values.

Geometric means and standard deviations of trace element

concentrations in the heavy mineral fraction of overburden are listed by lithological units in Table 7. The distribution of trace elements can be summarized as follows: (1) Heavy mineral concentrates from overburden samples collected over the Cap Bon-Ami and Grande-Grève Formations show the lowest contents of rare earth elements (REE), Mo, Ni, and Th. (2) Overburden from the Fortin Group shows the highest abundances of REE, Hf, Th, Sc, W, U, and Ni, with intermediate concentrations of As, Co, and Sb. (3) Arsenic, Co, Sb, Fe, and Zn are highest in abundance in heavy mineral concentrates of overburden from the Matapedia Group. Gold is also high but was detected in only two samples. (4) Finally, overburden collected from the York Lake and York River Formations shows the highest contents of Cr and Ta.

One of the more striking features of the trace element distribution is the high concentration of REE, Hf, Th, U, W, and Ni in the heavy mineral fraction of Fortin Group overburden. These elements also show a particularly strong loading on Factor 1 in the table of factor analysis (Table 6), the REE, Hf, Th, and U displaying the highest correlation coefficients among each other (Appendix 8). The strong correlation among the latter elements suggests an association with a specific mineral phase, the most likely being zircon. Qualitative X-ray fluorescence analysis of one concentrate from Fortin Group overburden which displayed a high Hf content (4700 ppm) also showed a high Zr content and the presence of zircon was confirmed by X-ray diffraction analysis.

If the concentration of REE, Hf, Th, and U in the heavy mineral samples is a function of the proportion of zircon, then

the proportion of zircon in the heavy mineral concentrates must be higher over the Fortin Group than over other formations. As expected, microscopic examination revealed higher proportions of zircon (>10%) in Fortin Group heavy mineral concentrates. Low abundances of zircon (<1-5%) are characteristic of heavy mineral concentrates from overburden both north of the Fortin Group (York Lake, York River, York Branch, Cap Bon-Ami and Grande-Grève Formations), where dilution by Shield-derived heavy minerals is widespread, and south of it (Matapedia Group), where pyrite, limonite and various other sulphides are the dominant heavy mineral constituents.

The other factors shown in Table 6 are less readily interpreted. Factor 2 (As, Co, Sb and Fe) could be an indicator of the presence of sulphides. The highest concentrations of Factor 2 elements are in heavy minerals of the Matapedia Group where a number of small sulphide mineralizations have been reported (Savard, 1985). The correlation of Au with Zn and Ta in Factor 4 and with -Fe, Ba and Mo in Factor 5 is accidental and reflects the effect of a very high outlier Au value (sample 272) on the calculation of the factors.

Choinière (1982,1984) has shown that the Fortin Group is overprinted by a large geochemical anomaly or "formational anomaly" of uranium in stream sediments. Choinière (1982, p. 129) concluded that

"the stream sediment samples from the Gaspé area reflect variations in the geochemical composition of the underlying bedrock."

Our data lend support to Choinière's findings in that the heavy

mineral fractions of Fortin Group overburden are enriched in uranium, and uranium is related to the presence of zircon, a significant mineral phase in the heavy fraction of the rubble and colluvium derived from the weathering of the underlying Fortin Group lithologies.

### 7.3 Gold Distribution

The regional distribution of gold in the -60 mesh and heavy mineral fractions of overburden are shown in Figures 6 and 7. Figure 8 shows the distribution of gold in overburden within the detailed study area.

The -60 mesh (<250  $\mu\text{m}$ ) results (Fig. 6) show that the overburden cover in general contains very low concentrations of gold. The highest concentrations observed, 30 ppb (sample 277) and 7 ppb (sample 67), occur in overburden samples collected over the Fortin and Matapedia groups respectively. An association of gold with sulphide minerals in sample 277 is suggested by the dominance of pyrite and limonite in the heavy mineral fraction. Sulphide minerals are present, but in lower abundances (<10%), in sample 67, collected on the upland surface which lies between the Matapedia and Milnikek Rivers.

A better estimate of the variation of gold in overburden on a regional basis is obtained from measurements within the heavy mineral concentrates (Fig. 7). The relative differences in gold abundance among the samples are more likely to be detected by this type of analysis because of the large concentration factor involved (100 to 1000) in separating the heavy mineral fraction from the original sample. The presence of gold is also more likely to be detected because of the large volume of original

material (25 1) that the heavy mineral sample represents.

Anomalous gold values in the heavy mineral concentrates analysed (Figs. 7 and 8) appear to be associated with 4 different geological settings: (1) in a number of colluvium and rubble samples collected over Fortin Group lithologies (including the highest observed gold concentration in a heavy mineral sample, 2 100 000 ppb in sample 272 collected off a west tributary of the Escuminac Nord River); (2) in 3 samples (4, 35 and 78) collected along the section of the faulted contact between the Fortin Group and York River formations where intermediate volcanic lavas outcrop; (3) in 5 overburden samples collected over Cap Bon-ami and St.-Léon lithologies on the west side of the Matapedia River; and (4) in 1 anomalous sample (280) collected over Matapedia Group rocks.

The distribution of gold concentrations in heavy mineral fractions of overburden within the 36 km<sup>2</sup> area surrounding the Assemetquagan alluvial gold occurrence is shown in Figure 8. The overburden cover in this area consists of a thin mantle of rubble on the upland surfaces, extensive colluvium on valley slopes, reworked outwash sands on the lower slopes of the Matapedia River, south of the du Moulin River, and more recent floodplain alluvium (Fig. 3). Heavy mineral abundances in the overburden (Fig. 9) reflect the distribution of overburden types, with alluvium containing five to ten times more heavy minerals than rubble and colluvium.

Figure 8 shows that, although the highest gold concentration in a heavy mineral sample is from overburden collected on a

small plateau south of du Moulin River, most high values occur within rubble and colluvium on slopes and upland areas north of the mouth of the Assemetquagan River and east of the Matapedia River.

Evidence put forward by Girard (1985), Maurice (1986) and Bergeron et al. (1986) indicates that the alluvial gold within the lower 2 km segment of the Assemetquagan River is derived from local Fortin Group lithologies. However, Maurice was the only one to outline a model explaining the presence of the gold. He tentatively proposed that:

"...gold was eroded from a bedrock source within Fortin Group sediments by the Laurentide ice sheet as it flowed southeasterly across western Gaspésie. It was then preferentially deposited in the Assemetquagan River valley, along with a large quantity of Shield-derived heavy minerals (almandite garnet and ilmenite), because the valley gorge constituted a major obstacle which impeded advancement of the debris-rich basal part of the ice sheet, thus concentrating the detritus in the river channel."

Our data point to a closed anomaly within the upland overburden north of the mouth of the Assemetquagan River. This supports Maurice's hypothesis of a local source to the north-northwest. However we find little evidence to support a model of glacial erosion within valleys to the northwest followed by transport and massive deposition of debris in the lower few kilometers of the Assemetquagan River. The valleys and upland areas within Fortin lithologies south of Ste. Florence show little if any evidence of large-scale glacial erosion and transport. The upland drift is very thin and consists largely of underlying bedrock, with which is mixed a very small component of englacially transported material. Other evidence (see



Section 7.4) suggests that lithological differences across the area may have locally affected ice flow conditions, and hence the pattern of erosion, transport and deposition of debris.

Although conclusive evidence to support Maurice's model was not found, glacial erosion and transport is not precluded from the area in question. Small ice tongues may have occupied individual valleys and caused localized erosion and deposition of material. However, it appears unlikely that any of these would have been large enough to cross upland divides and deposit material in adjacent valleys.

#### 7.4 Heavy Mineral Studies

The heavy mineral concentrates separated from the overburden samples were examined under a stereoscopic microscope and the relative proportions of individual mineral phases were noted. The heavy mineral assemblages in the sand fraction of till and colluvium samples collected in the northeastern third of the study area (within the Causapsca drainage basin including the area which forms the upper section of the Assemetquagan drainage area) consist predominantly of magnetite, garnet and ilmenite. When combined, these form an estimated 80 to 95% of the total heavy mineral fraction. The remaining 5-20% is epidote and clinopyroxene, with minor orthopyroxene, hematite, zircon, pyrite, leucxene, rutile, chromite, amphibole and kyanite. Three common varieties of garnet are identified: a subangular, dull orange variety; a subangular, transparent, bright orange variety which locally contains oxide inclusions; and a subangular to subrounded, transparent, light pink variety. These garnet

almandites with various proportions of pyrope and grossularite molecules. Ilmenite, hematite and magnetic minerals occur as subangular to subrounded grains which characteristically display surfaces reminiscent of euhedral habits. Electron microprobe analyses of ilmenites found in stream sediments of the region (Maurice, 1986) indicated that these are exsolved hemo-ilmenite varieties. At least 5 different types of magnetic minerals are noted: (1) highly magnetic true magnetite grains, (2) layered magnetite-hematite grains, (3) magnetic hematite grains, (4) weakly magnetic ilmenite grains, and (5) magnetic chromite grains. The garnet and ilmenite grains along with a large portion of the magnetite minerals are thought to be derived from the Canadian Shield. These form the bulk of the far-travelled component within the heavy minerals. Other far-travelled components include elongate pink and fluorescent zircons, rutile, amphibole, and kyanite.

Four varieties of zircon grains are noted within the samples: (1) a subhedral to euhedral, large (>100  $\mu\text{m}$ ), elongate, light pink and transparent variety, (2) a subhedral to euhedral, small (<100  $\mu\text{m}$ ), cloudy white variety, (3) a less common subhedral to anhedral, fluorescent to non-fluorescent, frosted tabular to rounded variety, and (4) a rare subhedral to anhedral, non-frosted, colourless to purple variety.

The heavy minerals in these samples are almost entirely angular and appear fresh, and few grains show evidence of pre- or post- erosional mechanical rounding or chemical alteration. Mineral grains which display "mature" appearances

show either surface frosting, chemical replacements, or rounded habits. Features observed include limonite replacements of pyrite cubes, leucoxene alteration of ilmenites, frosted garnets, pyroxene and epidotes, and frosted round zircon grains. Locally-derived mineral phases include zircon, epidote, pyrite, hematite, barite and chromite. Zircon, hematite and chromite are accessory phases within the arenaceous units in the area.

In contrast to overburden samples examined above, the heavy fraction of till and colluvium samples collected in the northwest third of the report area show variations in both mineral assemblages and mineral proportions. Magnetic minerals continue to be ubiquitous components of the samples, but the relative proportion of far-travelled and locally-derived minerals varies. Oxidized sulphide (oxidized pyrite, limonite, botryoidal sulphides) and irregular oxide minerals in several samples become the dominant phase, with subordinate garnet, pyroxene and true ilmenites. Some of the samples with high sulphide content show anomalous concentrations of gold.

Samples examined from the south third of the study area (below Ste. Florence) show the lowest proportions of far-travelled minerals. Garnet and ilmenite on average form less than 10% of the total heavy mineral fraction. Pyrite, zircon and epidote along with barite and fluorescent minerals form the major components. Zircon in Fortin Group rubble samples reaches proportions of up to 80%. Pyrite and associated sulphides reach highest proportions in samples within Fortin Group overburden samples in the southeast corner of the area and in Matapedia

Group samples.

Figure 10 shows the distribution of heavy mineral abundances in the overburden within the regional area. The heavy mineral concentration in the samples examined varies from <0.1% to 1.5% (see also Appendix 9).

On the basis of the relative abundance of total heavy minerals, the study area can be divided into three broad zones. This division is approximately the same as the division into zones on the basis of mineral proportions that were defined earlier. The zone of highest proportion of Shield-derived heavy minerals (northeastern third) corresponds to the zone of highest abundances of heavy minerals (>0.5%), which in turn corresponds to the zone of thickest till in the area. As expected, similar heavy mineral concentrations were observed in the adjacent stream sediments by Maurice (1986), who defines the area as a zone of high abundance of heavy minerals (0.1 to 1.5%).

The northwest third of the report area corresponds to a zone of both intermediate heavy mineral concentrations (0.1-0.5%) and intermediate proportions of Shield-derived heavy minerals. The samples within this zone overlie lithologies of the Cap Bon-Ami, Grande-Grève and St.-Léon Formations. Higher concentrations to the immediate south appear to be associated with the underlying sandstone units (York River and Lake Branch Formations). The southern third of the project area (south of Ste. Florence) lies within Zone III of David and Lebluis (1985). This zone is characterized by a lack of glacial deposits, sparse erratics, deeply weathered bedrock and extensive colluvium containing a high content of locally-derived bedrock. Overburden samples

collected within this area show the lowest heavy mineral abundances ( $<0.1\%$ ) along with the lowest proportions of Shield-derived components. Low abundances of heavy minerals were also recorded in stream sediments of the upper central Assemetquagan drainage basin by Maurice (1986). However, the lower central section, which includes a segment of the Assemetquagan River and the lower part of the Creux River, is shown as a zone of high abundance of heavy minerals. The lack of available data on the composition of overburden in this area does not allow the nature of the contribution of heavy minerals from overburden to the stream sediment to be evaluated.

The lower section of the Assemetquagan drainage area can be divided into two segments on the basis of the abundance of heavy minerals in the stream sediments (Maurice, 1986): a lower 900 m segment rich in heavy minerals and gold where the Assemetquagan River is narrow, rapid and turbulent; and an upper segment where heavy minerals show intermediate abundances. Overburden samples collected within the detailed study area surrounding the mouth of the Assemetquagan River (Fig. 9) show very low abundances of heavy minerals ( $<0.1\%$  for rubble and colluvium). This suggests that the overburden in the lower section of the river basin is not a significant contributor of heavy minerals to the stream sediments. A possible explanation for the high abundance of Shield-derived heavy minerals within this section of the river is that they are derived from an overburden source within the central and/or upper sections of the drainage basin, and that they have been transported downstream and concentrated in the lower segment by stream processes. The Assemetquagan drainage

basin contains important sources of Shield-derived heavy minerals in its upper reaches. High discharges associated with northward-retreating ice during deglaciation could have transported and deposited outwash rich in Shield-derived minerals in the Assemetquagan River and its tributaries. Post-glacial hydraulic action (most effective under seasonal high meltwater discharges) would erode, redistribute and concentrate this material within different reaches of the river as well as erode tills in the upper part of the drainage basin and colluvium in the central and lower parts. Shield-derived heavy minerals at the mouth of the Assemetquagan River owe their presence to a combination of all these processes. Which process(es) account(s) for the bulk of the heavy minerals present could not be resolved with the evidence at hand.

Perhaps equally as interesting is the apparent relation between heavy mineral abundances in overburden and underlying lithological units. The highest abundances of heavy minerals are in tills and colluvium underlain by the more resistant sandstone units, the intermediate abundances in tills and colluvium underlain by siltstones and limestones and the lowest abundances in colluvium and rubble of soft Fortin shales and Matapedia Group limestones. One may speculate that competency differences between the various lithological groups affected the local resistance to basal flow and hence the pattern of erosion and deposition of the Laurentide ice sheet as it moved south-eastward across the area. Debris derived from competent sandstone units may have provided the most resistance to flow and

hence localized accumulations of materials within the basal debris layer of the ice. How this may have affected the pattern of deposition of Shield-derived materials or the pattern of ice flow across the less competent Fortin Group lithologies is unclear, and more detailed investigation will be required to explain all the observed features.

## 8.0 CONCLUSIONS

1. The project area is a dissected plateau with summit altitudes from 1300 to 1500 feet (396 to 457 m) above sea level, and floors of the major valleys from 150 to 750 feet (46 to 229 m) above sea level. Investigations on the geomorphology of the area show that Pleistocene events have had little influence in shaping the observed topography and drainage patterns. The upland is a rolling surface having about 300 feet (91 m) relief. The present rivers and streams occupy V-shaped valleys incised about 500 feet (152 m) below floors of ancient valleys now approximately 1000 feet (305 m) above sea level. The drainage pattern, especially the first and second order streams, is strongly controlled by structure. The Matapedia River is the main trunk of the drainage system. It is incised 1000 feet (305 m) below the floor of a large cross-axial valley. North of Ste. Florence the present Matapedia valley is broad and typical of glaciated valleys. South of Ste. Florence the Matapedia River is entrenched in a narrow straight canyon.

2. Upland overburden is thin and commonly not much thicker than the soil profile. It consists of glacial drift (till), alluvium of fluvio-glacial and non-glacial origin, colluvium, rubble and other diamict materials of uncertain origins. The overburden is composed largely of material derived locally with a small component of far-travelled materials. Lake Branch sandstone erratics are transported south in the drift and alluvium near and in the Matapedia valley; erratics of this formation are also found in drift and alluvium up to 20 km north of their source. Evidence of southward flow across the area is also provided by the distribution of Val Brilliant erratics. Some eastern transport, as suggested by meagre erosional features, is not precluded.
3. Major element analyses of the -60 mesh (<250  $\mu\text{m}$ ) fraction of overburden support a model of changing composition from north to south in approximate correspondence with changes in bedrock composition. This agrees with the observation that the overburden contains a significant component of locally-derived bedrock. Trace element analyses of the same material are less reflective of this trend. However, analyses of heavy mineral concentrates show significant trace element variations between overburden from different lithological units. The relative enrichment of REE, Hf, Th and U in heavy mineral samples extracted from Fortin Group rubble and colluvium reflects the higher proportion of locally-derived zircon within the heavy mineral fraction. Low abundances of zircon are characteristic of heavy mineral concentrates from overburden both north of the Fortin, where dilution by



Shield-derived heavy minerals is common, and south of it, where sulphide minerals form the dominant heavy mineral constituent.

4. The distribution of heavy mineral abundances and the proportions of individual mineral phases are related to the distribution of overburden types. High heavy mineral abundances are associated with high proportions of Shield-derived heavy minerals and correspond to till and colluvium from the northeast third of the area. Low heavy mineral abundances reflect low proportions of far-travelled mineral components and correspond to rubble and colluvium from the Fortin and Matapedia Groups.

Heavy mineral abundances in overburden reflect heavy mineral abundances in adjacent stream sediments for the upper and central parts of the Assemetquagan drainage basin. Low concentrations of heavy minerals in overburden adjacent to the lower 1000 m of the Assemetquagan River do not explain the high abundance of Shield-derived minerals in the stream sediments. A possible explanation is that these were transported from upstream sources and concentrated in the lower section of the river by hydraulic activity. Possible sources include outwash deposited within the valley and till and colluvium rich in Shield-derived heavy minerals in the upper reaches of the drainage area.

5. The distribution of overburden types can be related to the distribution of underlying rock types. Tills rich in Shield-derived minerals occur over sandstone lithologies, with some

overlap onto Fortin shales and siltstones to the south. Overburden with intermediate heavy mineral abundances and proportions of Shield-derived minerals overlies siltstone and limestone units in the northwest third of the area. Overburden with the lowest abundances of heavy minerals and proportions of Shield-derived heavy minerals occur on soft Fortin Group shales and siltstones and Matapedia Group limestones. This suggests that lithological differences may have affected the local resistance to basal flow and hence the pattern of erosion and deposition of the Laurentide ice sheet as it moved southeastward across the area.

6. Analyses of -60 mesh ( $<250 \mu\text{m}$ ) show that gold is in apparently very low concentrations with the exception of two samples containing 30 ppb and 7 ppb gold. A better estimate of the variation of gold in overburden is obtained from measurements within heavy mineral concentrates. Although sample density is low, heavy mineral gold analyses outline a number of anomalous areas. High abundances of gold in heavy mineral concentrates from overburden samples collected north of the mouth of the Assemetquagan River and east of the Matapedia River (1986 study area) support Maurice's (1986) hypothesis of a local source to the north-northwest for the alluvial gold at the mouth of the Assemetquagan River.

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Table 1. Lower Paleozoic formations and groups within the study area and their chief lithological characteristics. After Stearn (1965), Lachance (1977, 1979), Skidmore (1977) and Bourque and Lachambre (1980).

Period or Group (Gp.)	Formation (Fm.)	Lithology	Tectono-Stratigraphic Classification
Middle Devonian	Battery Point Fm.	Green and red arkosic sandstone, siltstone; shale and conglomerate; minor intermediate volcanics.	
	Lake Branch Fm.	Red arkosic sandstone, siltstone; minor shale.	
Lower Devonian	York River Fm.	Grey-green argillaceous sandstone, siltstone.	Central Belt*
	York Lake Fm.	Grey-green arkosic sandstone, calcareous siltstone; minor silty limestone, shale, volcanics.	Gaspé-Connecticut Valley Synclinorium**
	Fortin Gp.	Dark grey to brown, calcareous to non-calcareous shale and siltstone; arkosic sandstone, conglomerate; minor andesitic flows and pyroclastic volcanics.	
	Grande-Grève Fm.	Medium grey to light brown silty limestone, calcareous siltstone; minor mudstone.	
	Cap Bon-Ami Fm.	Medium grey argillaceous to silty limestone, shale, mudstone; minor sandstone, andesitic volcanics.	
Lower Silurian to Lower Devonian	St.-Léon Fm.	Greenish-grey calcareous siltstone, limestone, calcareous mudstone, shale; minor sandstone.	
	Chaleur Gp.	Grey siltstone interbedded with calcareous mudstone; minor volcanics.	Southern Belt*
Upper Ordovician to Lower Silurian	Matapedia Gp.	Grey banded argillaceous and silty limestone and shale; locally hydrothermally altered and intruded by porphyritic dykes of intermediate compositions.	Aroostook - Percé Anticlinorium**

\* After McGerrigle (1950)

\*\* Poole and Rodgers (1972)



Table 2. Summary of major element averages (%) by soil type and underlying rock  
(recalculated to exclude LOI in summation)

SOIL	ROCK	n	EAST	NORTH	SiO2	TiO2	Al2O3	Fe2O3	MnO	MgO	CaO	Na2O	K2O	LOI
C	B	1	2555.	4795.	81.33	.760	9.52	3.54	.040	1.00	.20	1.46	1.71	3.58
T	B	1	3755.	5300.	74.87	.749	13.55	6.70	.034	.97	.29	1.18	1.51	15.79
T	BY	1	3555.	5400.	67.72	.992	16.89	8.24	.044	2.08	.12	.97	2.79	8.25
T	C	2	3940.	6205.	76.92	.746	12.04	4.26	.040	1.39	.55	1.56	2.00	4.83
A	F	1	5100.	2820.	70.09	.905	15.46	7.97	.035	1.88	.20	1.50	2.10	9.06
C	F	27	5450.	4248.	70.53	.875	16.42	6.79	.041	1.66	.19	1.41	2.10	10.88
T	F	10	4995.	4835.	74.60	.825	13.91	5.50	.043	1.51	.19	1.34	2.02	7.08
U	F	8	5281.	4211.	71.35	.886	15.73	6.95	.042	1.58	.24	1.24	2.02	14.21
U	FM	1	5180.	2665.	67.38	.929	17.59	8.59	.036	1.64	.12	1.44	2.00	14.12
C	G	3	3255.	6092.	74.95	.635	12.94	5.60	.070	1.66	.27	.87	2.37	5.82
T	G	14	2566.	6497.	77.27	.662	11.76	5.11	.049	1.37	.34	1.11	1.90	7.85
U	G	1	2405.	6170.	81.06	.655	9.35	3.69	.180	.85	.71	1.10	1.56	13.90
C	H	4	6153.	3085.	61.70	.886	16.16	6.94	.069	1.74	8.19	1.45	2.80	11.63
T	H	5	6829.	3296.	66.59	.898	17.14	8.10	.084	2.75	.32	1.58	2.80	9.11
U	H	3	7028.	3510.	68.02	.891	17.03	7.24	.087	2.33	.21	1.43	2.97	7.28
C	Y	3	4385.	5755.	74.33	.906	12.93	6.72	.073	1.16	.32	1.44	1.96	8.21
T	Y	13	4242.	6282.	76.06	.852	12.62	5.16	.048	1.35	.28	1.55	1.93	5.40
C	YG	1	3975.	7090.	79.09	.724	10.51	4.26	.036	1.13	.33	1.52	1.63	5.07
T	YG	1	3745.	5700.	75.04	.811	13.62	5.86	.031	1.18	.25	1.23	1.67	8.31

Table 3. Summary of trace element averages (ppm) by soil type and underlying rock

SOIL	ROCK	n	EAST	NORTH	Ba	Co	Cu	Ni	Zn	Nb	Zr	Y	Sr	Rb	Pb	Th	U
C	B	1	2555.	4795.	288.	5.	28.	36.	40.	12.	411.	14.	80.	67.	15.	10.	11.
T	B	1	3755.	5300.	250.	9.	34.	24.	33.	15.	296.	13.	75.	64.	18.	12.	14.
T	BY	1	3555.	5400.	328.	23.	14.	75.	69.	18.	249.	18.	67.	117.	18.	13.	9.
T	C	2	3940.	6205.	394.	5.	28.	33.	55.	9.	389.	22.	104.	86.	20.	12.	10.
A	F	1	5100.	2820.	374.	31.	28.	59.	69.	19.	310.	22.	63.	98.	19.	12.	11.
C	F	27	5450.	4248.	323.	19.	33.	58.	69.	16.	264.	20.	63.	98.	21.	13.	10.
T	F	10	4995.	4835.	316.	11.	31.	65.	73.	13.	276.	17.	68.	89.	21.	12.	10.
U	F	8	5281.	4211.	299.	17.	32.	56.	68.	17.	272.	21.	58.	97.	22.	12.	10.
U	FM	1	5180.	2665.	270.	22.	24.	55.	115.	19.	239.	19.	71.	105.	20.	12.	9.
C	G	3	3255.	6092.	435.	11.	42.	54.	78.	15.	179.	26.	66.	114.	24.	11.	8.
T	G	14	2566.	6497.	404.	8.	40.	37.	70.	13.	232.	19.	79.	90.	22.	11.	11.
U	G	1	2405.	6170.	485.	5.	26.	16.	96.	6.	232.	11.	84.	79.	26.	7.	11.
C	H	4	6153.	3085.	302.	21.	23.	70.	70.	19.	222.	27.	186.	114.	22.	14.	10.
T	H	5	6829.	3296.	320.	29.	29.	80.	87.	20.	203.	23.	57.	123.	23.	14.	9.
U	H	3	7028.	3510.	361.	28.	26.	72.	78.	22.	231.	20.	58.	131.	26.	16.	7.
C	Y	3	4385.	5755.	337.	15.	27.	47.	53.	14.	370.	21.	91.	94.	18.	13.	10.
T	Y	13	4242.	6282.	329.	12.	33.	49.	56.	15.	314.	17.	90.	83.	21.	11.	11.
C	YG	1	3975.	7090.	331.	5.	38.	36.	44.	10.	303.	13.	95.	69.	21.	12.	13.
T	YG	1	3745.	5700.	288.	13.	48.	41.	103.	14.	270.	14.	80.	84.	19.	12.	10.

Table 4. Chemical analyses of rocks in the Matapedia area

Sample	Val Brilliant sandstone	St.-Leon Formation	Battery Point sandstone	York River sandstone	Fortin Group	Ste. Marguerite volcanics	Matapedia Group	Grande-Grève Formation
	85-8	85-7	85-2	85-3	85-4	85-1	85-5	85-6
Major Elements (%) recalculated to exclude LOI in summation								
SiO <sub>2</sub>	98.21	81.23	79.68	76.99	66.67	63.35	61.85	50.25
TiO <sub>2</sub>	.05	.43	.55	.49	.87	1.25	.74	.19
Al <sub>2</sub> O <sub>3</sub>	.52	8.15	9.54	10.40	15.96	16.99	13.71	4.04
Fe <sub>2</sub> O <sub>3</sub>	.11	3.16	3.19	3.17	6.72	5.34	5.49	1.84
MgO	n.d.	1.77	1.23	2.05	2.68	2.14	3.03	4.08
CaO	.02	1.27	.42	1.14	1.74	1.74	9.74	36.97
Na <sub>2</sub> O	.03	.90	2.31	1.83	1.91	6.83	2.21	.34
K <sub>2</sub> O	.17	1.94	2.13	2.69	2.99	1.09	2.78	1.01
LOI	.24	2.82	1.89	3.19	4.60	2.93	10.36	25.43
Trace Elements (ppm)								
Ba	n.d.	124	192	228	248	537	192	226
S	n.d.	n.d.	n.d.	40	n.d.	n.d.	n.d.	n.d.
Cr	n.d.	196	111	102	94	n.d.	57	n.d.
Co	47	13	18	21	15	5	4	n.d.
Ni	n.d.	31	27	202	78	n.d.	37	n.d.
Cu	41	124	45	106	78	6	n.d.	n.d.
Zn	8	60	37	33	84	111	72	23
Nb	<10	12	15	14	18	29	17	10
Zr	184	386	230	184	221	417	185	63
Y	<10	23	21	22	27	45	29	14
Sr	13	58	74	71	135	589	199	387
Rb	16	70	66	72	115	34	101	41
Pb	12	18	12	16	22	24	21	10
Th	<10	<10	<10	<10	<10	13	11	<10
U	12	10	<10	<10	<10	16	10	16

n.d. = not detected

Table 5: Comparison of trace element analyses with alluvium (Choinière, 1982)

Sample Formation Type or Group	No. of Samples	Cu (ppm)		Zn (ppm)		Pb (ppm)		Ni (ppm)		Co (ppm)		U (ppm)		MnO (%)		
		Arith.	Std. Dev.	Arith.	Std. Dev.	Arith.	Std. Dev.	Arith.	Std. Dev.	Arith.	Std. Dev.	Arith.	Std. Dev.	Arith.	Std. Dev.	
Colluv. Fortia	27	33	12	69	17	21	21	3	58	18	19	10	10	1	.041	.022
Till Fortia	10	31	9	73	19	21	21	1	55	23	11	6	10	2	.043	.013
Alluv.† Fortia	1988	12.6	5.2	80.5	32.9	19.1	19.1	10.2	34.6	11.3	11.2	4.1	4.1	4.6	.1006	.0995
Rock Fortia	1	78	-	84	-	22	-	-	78	-	15	-	<10	-	n.d.	-
Till Gr.-Grève	14	40	24	70	22	22	22	4	37	14	8	5	11	2	.019	.027
Cap Bon-Ami																
Alluv.† Gr.-Grève	2730	30.6	29.5	99.6	41.7	21.9	21.9	8.2	38.5	15.2	20.1	9.7	2.4	2.2	.1157	.1031
Cap Bon-Ami																
Rock Gr.-Grève	1	n.d.	-	23	-	10	-	-	n.d.	-	n.d.	-	15	-	n.d.	-
Till York River	13	33	12	56	17	21	21	3	49	17	12	9	11	2	.048	.028
York Lake																
Alluv.† York River	1102	8.7	4.6	75.6	32.0	19.0	19.0	7.2	40.2	11.5	14.3	5.2	3.1	3.6	.1003	.0909
York Lake																
Batt. Point																
Rock York River	1	106	-	33	-	16	-	-	202	-	21	-	<10	-	n.d.	-

† Data from Choinière (1982)

Table 6: Factor analysis, Varimax-rotated loadings on log transformed data, H.M.C. analyses.

Element	Factors				
	1	2	3	4	5
Hf	0.953	-0.078	-0.0129	-0.163	0.093
Lu	0.950	-0.132	-0.171	-0.080	0.119
Yb	0.949	-0.147	-0.152	-0.079	0.138
Th	0.920	-0.173	-0.217	-0.116	0.196
U	0.903	-0.157	-0.160	-0.227	0.180
Ce	0.885	-0.232	-0.219	-0.015	0.240
Sm	0.878	-0.269	-0.236	-0.045	0.251
La	0.859	-0.295	-0.242	-0.058	0.256
Eu	0.822	-0.314	-0.313	-0.054	0.235
Cr	0.756	0.025	0.176	-0.336	-0.089
Sc	0.743	-0.076	0.068	-0.298	-0.246
W	0.598	-0.227	-0.464	-0.260	0.031
As	0.200	-0.913	-0.050	0.045	0.123
Co	0.267	-0.850	-0.160	-0.162	-0.045
Sb	0.341	-0.779	-0.122	-0.124	0.066
Fe	-0.094	-0.594	0.049	0.075	-0.626
Ni	0.234	-0.397	-0.864	-0.331	-0.006
Na	0.239	0.025	-0.814	0.010	0.123
Ta	0.431	0.189	0.008	-0.762	-0.006
Zn	0.080	-0.444	-0.146	-0.756	0.080
Au	0.155	-0.095	-0.275	-0.633	0.695
Ba	0.142	-0.393	0.130	-0.226	0.695
Mo	0.326	0.057	-0.268	0.004	0.629

Variance explained by rotated components:

9.513                  3.542                  2.294                  2.145                  1.931

Percent of total variance explained:

41.363                  15.401                  9.974                  9.326                  8.394

Table 7: Characteristics of trace element concentrations in the heavy mineral fraction of overburden in relation to the underlying bedrock formations or groups.

Element	Fortin			Br.Grève-Cap Bon-Ami			York Lake-York River			Matapedia			St-Léon			Lake Branch		
	n.	Geom. Mean	log Std. Dev.*	n.	Geom. Mean	log Std. Dev.*	n.	Geom. Mean	log Std. Dev.*	n.	Geom. Mean	log Std. Dev.*	n.	Geom. Mean	log Std. Dev.*	n.	Geom. Mean	log Std. Dev.*
As	122	116.95	2.42	38	64.42	2.90	27	37.24	2.32	9	242.66	2.37	8	36.98	2.88	4	43.45	1.00
Au	48	368.13	13.09	13	389.94	37.15	7	46.67	1.74	2	1534.62	60.81	3	29.38	1.29	0		
Ba	83	1177.61	2.12	20	1253.14	3.98	13	668.34	1.74	4	1737.80	2.01	3	1940.89	10.45	2	489.78	1.00
Ca	4	7.82	1.96	6	2.75	1.31	0			2	11.22	1.37	2	3.87	1.44	1	3.00	
Co	124	100.69	1.88	38	76.03	2.10	27	57.54	1.63	9	179.89	1.78	8	60.95	1.95	4	34.12	1.00
Cr	124	7244.36	1.56	38	5035.01	2.84	27	10351.42	1.92	9	3953.67	2.25	8	5069.91	2.04	4	7961.78	1.00
Fe	124	20.89	1.53	38	22.23	1.70	27	20.99	1.55	9	35.24	1.32	8	21.33	1.67	4	22.44	1.00
Hf	124	1073.99	1.86	38	401.79	2.39	27	674.53	1.75	9	344.35	2.49	8	528.45	1.77	4	494.31	1.00
Mo	35	23.88	2.13	13	12.16	1.74	6	14.93	2.06	3	50.58	2.96	3	27.67	1.67	2	14.42	1.00
Na	103	.10	1.60	33	.09	1.35	20	.11	1.69	5	.11	1.34	7	.11	1.31	3	.07	1.00
Ni	16	1253.14	1.97	3	543.25	1.43	1	899.50	0	1	1300.17	0	0			0		
Sb	123	7.40	2.16	36	4.53	2.70	26	3.47	1.97	9	12.82	2.72	8	3.50	2.26	4	4.58	1.00
Sc	124	53.83	1.30	38	44.06	1.77	27	49.20	1.56	9	29.04	1.61	8	46.77	1.62	4	37.58	1.00
Ta	117	14.00	1.69	37	14.16	2.08	27	16.83	1.65	6	11.09	2.07	8	12.19	2.10	4	13.87	1.00
Th	124	128.82	2.19	38	37.76	2.43	27	68.71	1.73	9	55.34	1.92	8	42.56	1.98	4	46.56	1.00
U	124	56.10	2.02	37	19.91	2.49	27	36.39	1.90	9	26.67	2.14	8	22.86	1.99	4	29.24	1.00
W	73	43.95	2.00	13	20.99	2.34	8	25.23	1.37	2	15.89	1.49	1	26.00	0	3	18.11	1.00
Zn	93	205.12	1.75	33	250.03	1.94	24	240.99	1.71	7	486.41	2.92	7	208.93	1.79	3	157.04	1.00
La	124	597.04	2.32	38	150.31	2.30	27	260.62	1.76	9	391.74	1.49	8	166.34	1.85	4	171.79	1.00
Ce	124	924.70	2.20	38	295.12	2.16	27	496.59	1.75	9	550.81	1.33	8	286.42	1.66	3	423.64	1.00
Sm	124	63.24	2.25	38	20.42	2.07	27	30.90	1.71	9	35.89	1.52	8	23.01	1.61	4	21.38	1.00
Eu	123	13.55	2.31	38	4.31	2.19	27	5.58	1.84	9	7.21	1.64	8	4.32	1.68	4	4.92	1.00
Yb	124	78.16	1.84	38	27.86	2.11	27	40.46	1.69	9	25.64	2.14	8	34.04	1.64	4	28.18	1.00
Lu	124	15.67	1.85	38	5.40	2.13	27	8.18	1.67	9	4.89	2.27	8	6.67	1.69	4	5.62	1.00

Geom. = Geometric

n. = Number of cases

\* log Std. Dev. = Antilog of the standard deviation of the log data

ppm = As, Ba, Co, Cr, Hf, Mo, Ni, Sb, Sc, Se, Ta, Th, U, W, Zn, La, Ce, Sm, Eu, Yb, Lu

µg = Ca, Fe, Na, Pb = Au



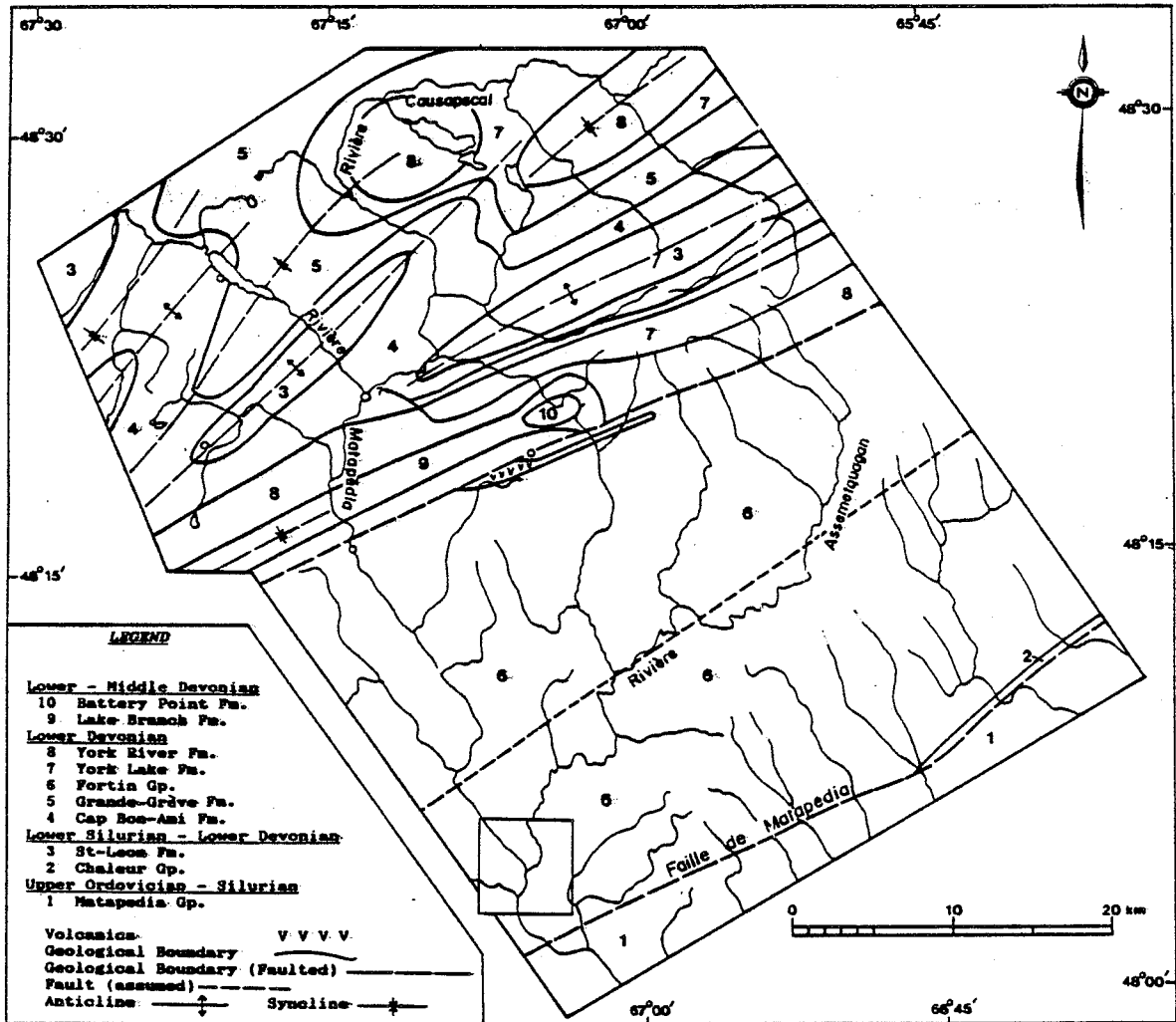


Fig. 2. Geological map of the project area



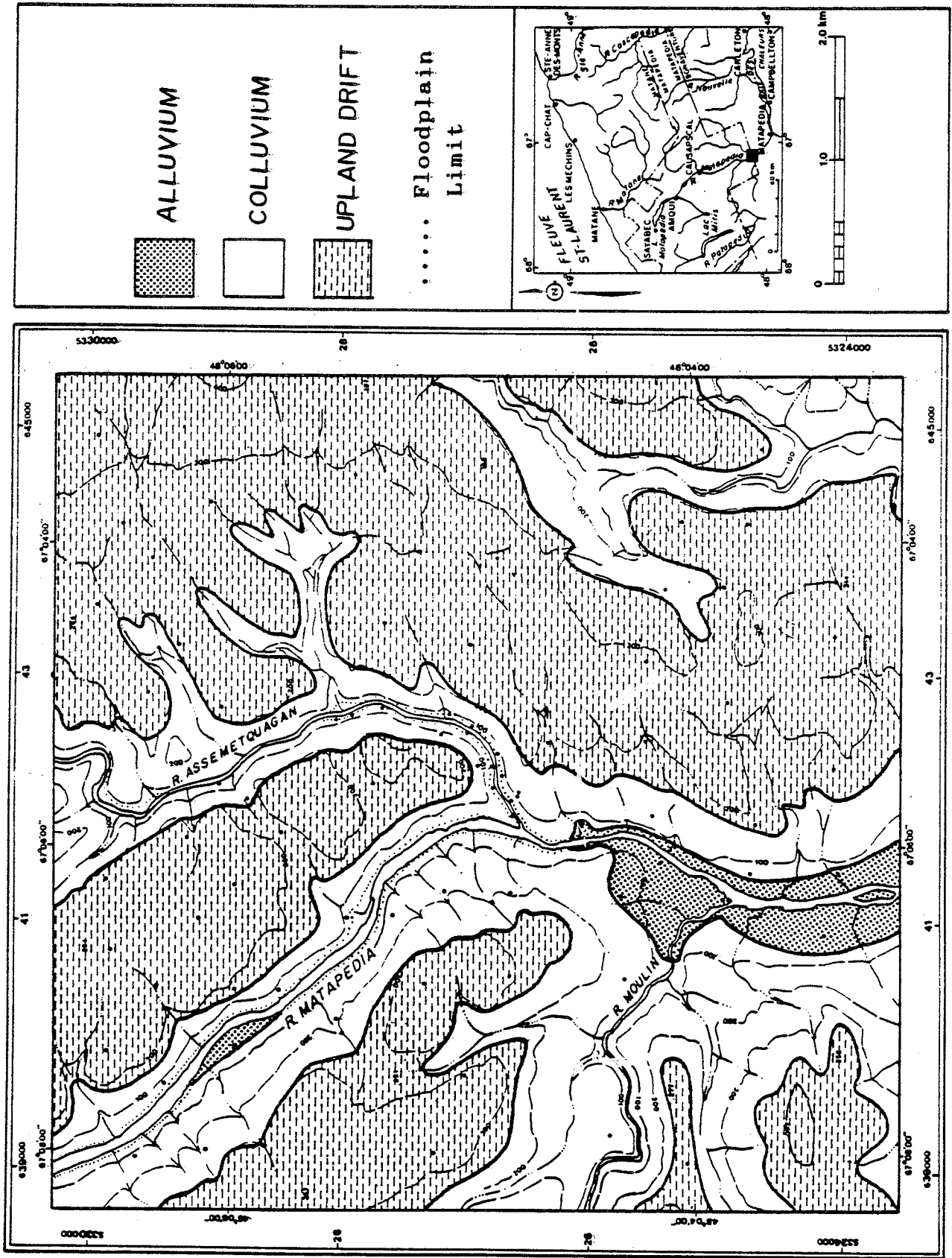


Fig. 3 Distribution of overburden types within the 1986 study area.

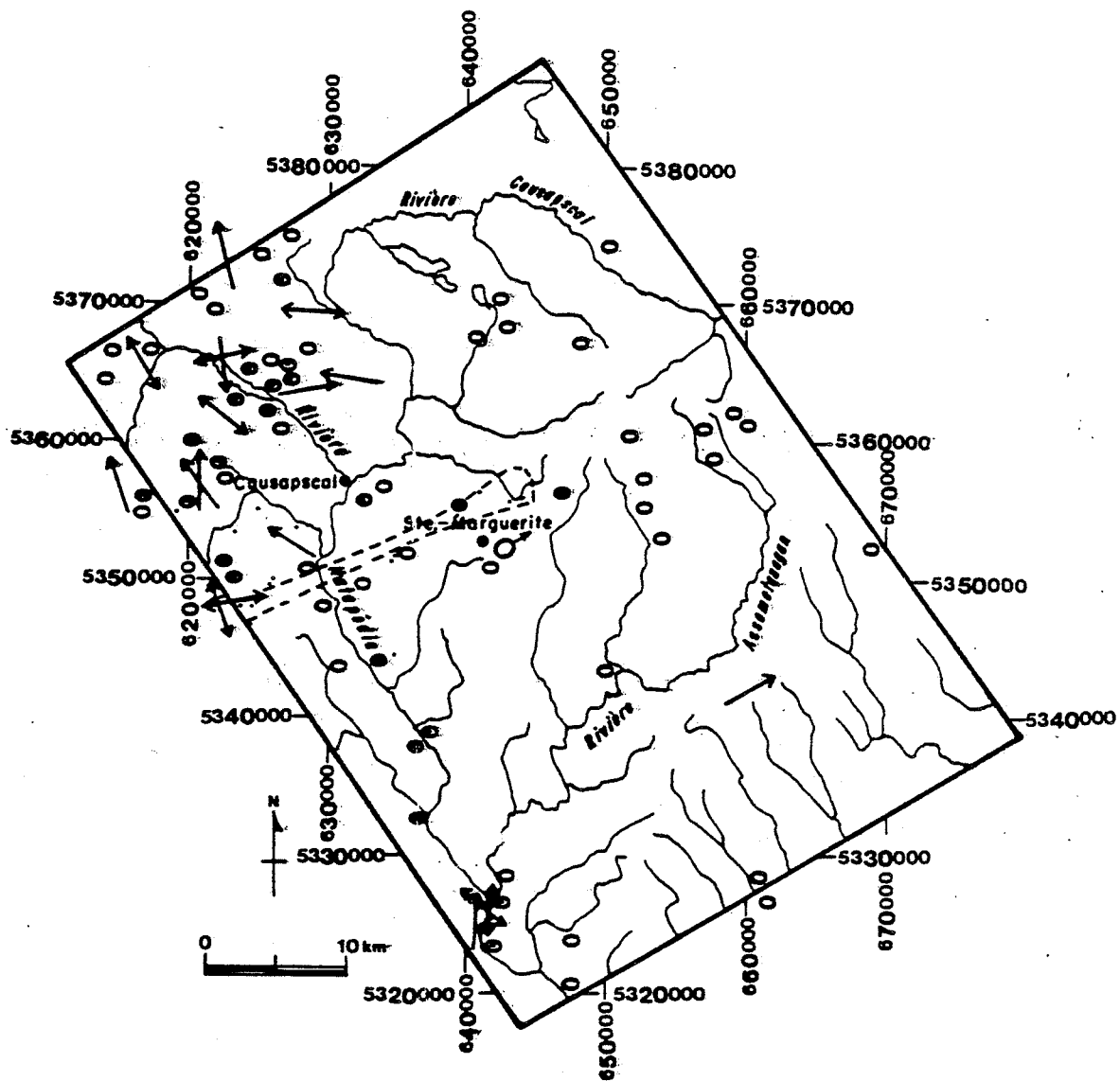


Fig. 4 Map of ice flow direction indicators. (Modified after Leblais, 1975, David and Leblais, 1985)

Glacial striae

↗ direction known  
↘ trend only

--- York Branch Fm.

○ rock boss indicating ice flow direction

Erratics

○ Val Brilliant  
● York Branch

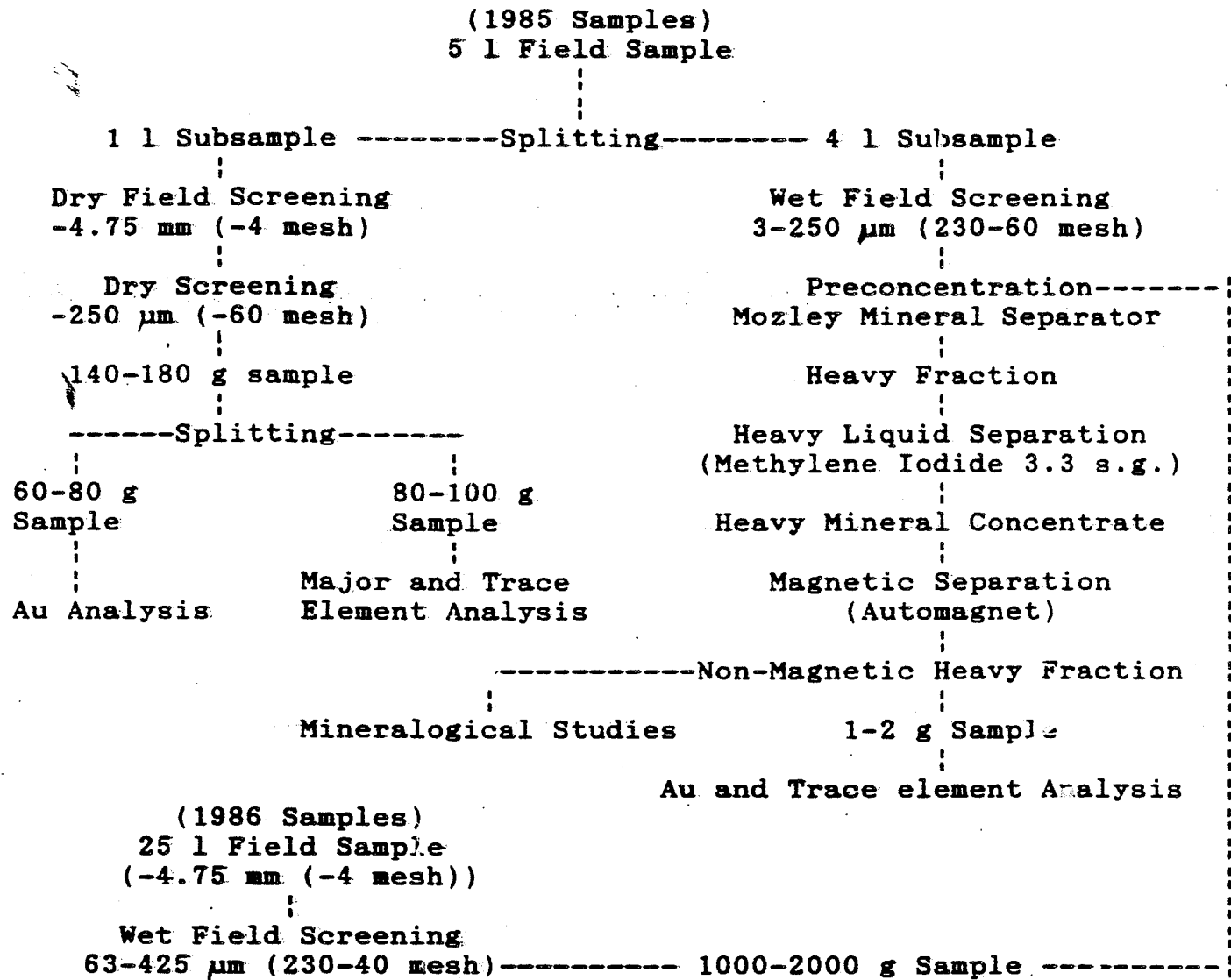


Fig. 5: Sample Treatment Chart

Appendix 7. Trace element concentrations in 19 heavy mineral concentrates and their respective duplicates.

Element	Sample No.													
	001*	005	002*	053	003*	074	310*	089	307*	222	008*	231	306*	226
Ag (ppm)	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5
As (ppm)	49	63	29	31	28	30	5	8	<2	28	16	23	7	11
Au (ppb)	200	420	<9	13	<20	<22	<15	<13	<22	85	<21	30	<19	<19
Ba (ppm)	600	<500	400	300	<500	<600	<300	<800	900	<1300	<500	<400	700	<400
Ca (%)	<3	<3	2	<1	<3	<3	2	<1	<3	<6	<2	<3	4	<2
Co (ppm)	65	74	41	43	52	59	37	35	39	110	45	42	42	38
Cr (ppm)	9500	8900	1500	1500	15000	15000	700	790	6200	15000	6900	7400	4000	41000
Fe (%)	21.4	23.8	32.4	31.5	21.5	18.1	18.9	18.6	16.8	43.8	20.4	25.3	18.4	18.4
Hf (ppm)	720	620	200	180	740	730	140	150	620	1300	590	620	440	460
Mo (ppm)	21	<5	<5	<5	<5	<5	<5	7	<5	<14	<5	<5	<5	<5
Na (%)	.09	.07	.13	.13	<.05	<.05	.08	.06	<.05	.13	.17	.06	.06	.07
Ni (ppm)	400	<400	<200	<200	<400	<400	<300	<300	<400	<400	<400	<300	<400	<400
Sb (ppm)	5.7	8	1.1	1.2	2.1	3.1	<0.2	<.04	.9	3.3	.80	1	<.6	<.6
Sc (ppm)	46.4	49.10	44.7	45.4	45.8	52.4	54.9	52.5	60.7	145	52.5	46.2	59.4	59.5
Se (ppm)	<16	<17	<5	<6	<16	<20	<10	<9	<18	<43	<18	<13	<15	<15
Ta (ppm)	13	13	8	9	15	25	19	13	17	16	23	13	18	19
Th (ppm)	54	49	16	19	65	66	13	17	65	130	59	59	44	42
U (ppm)	32.7	31.4	9.1	8.1	35.1	42	5.3	7.4	36.2	77	29.6	29.9	16.6	12.4
W (ppm)	19	<16	<7	<8	18	<12	<9	<8	14	<32	24	<15	<11	<11
Zn (ppm)	260	270	110	140	220	170	140	140	110	500	120	250	190	130
La (ppm)	189	194	75	78	255	254	55	58	241	476	194	219	157	155
Ce (ppm)	384	367	164	270	474	461	147	170	728	1520	367	365	514	551
Sm (ppm)	25	23.1	14.3	14.5	31.1	27	9	8.9	26.2	56.4	20.5	25.1	18.6	18.6
Ku (ppm)	3.4	3.4	2.5	2.5	5.9	64	1.4	1.9	6.5	11.3	2.70	3.4	4.1	22
Yb (ppm)	39.8	35.9	19.9	19.4	40.5	39.9	20	19	40.8	90.2	33.3	34.9	33.4	33.2
Lu (ppm)	7.79	6.95	3.36	3.23	8.32	8.16	3.62	3.50	7.96	17.5	6.96	6.89	6.48	6.13

\* Duplicate

Appendix 7. (cont.)

Element	Sample No.													
	398*	361	397*	368	396*	367	395*	338	394*	364	393*	343	391*	316
Ag (ppm)	<5	<5	<15	<5	<5	<5	<5	<9	<5	<5	<8	<5	<5	<5
As (ppm)	17	19	150	190	81	98	160	190	47	40	83	74	70	100
Au (ppb)	<31	26	86	67	<27	<24	<28	14	<20	2100	<14	<16	940	460
Ba (ppm)	700	400	500	<500	600	<1200	400	500	<300	400	<200	600	600	<600
Ca (%)	<4	<2	<3	<3	<4	<3	<3	<1	<2	<3	<1	<2	<4	<5
Co (ppm)	43	48	95	110	70	74	100	110	49	55	56	55	74	82
Cr (ppm)	5600	6700	2600	2400	3800	3700	4200	4600	4000	5000	1900	2300	5900	6000
Fe (%)	22.4	26.3	29.7	32.2	24.6	27.5	26.9	31.4	25.7	25.6	25.1	25.8	21	23.4
Hf (ppm)	780	760	230	210	520	440	450	390	430	600	140	190	770	720
Mo (ppm)	<5	<5	15	12	13	9	<8	<5	13	23	7	<5	<5	31
Na (%)	.14	.06	.08	.06	<.05	.06	.08	.07	.05	.05	.05	.06	.12	.12
Ni (ppm)	<500	<300	<500	<500	<400	<400	<400	<200	500	<400	<200	<300	600	1000
Sb (ppm)	1.6	1.7	5.4	7.1	2.6	5	6.2	7.4	2	2.6	2.8	3	2.7	3.8
Sc (ppm)	45.2	47.4	41	44.8	38.5	40.8	39	40.5	38.8	40.6	43.3	44.1	44.3	48.8
Se (ppm)	<34	<8	<21	<24	<27	<25	<12	<12	<28	<29	<12	<14	<13	<42
Ta (ppm)	12	11	8	7	9	11	7	7	9	11	5	5	9	10
Th (ppm)	67	67	20	16	49	41	37	34	31	44	10	15	70	67
W (ppm)	31	32.7	7.5	7.8	26.8	19.8	22	16.1	16.1	22.8	5.1	9	34.9	29.1
U (ppm)	28	44	20	<15	20	<12	<15	<6	<12	<11	<9	11	<17	<12
Zn (ppm)	100	60	160	<60	80	120	170	190	<210	130	130	140	140	190
La (ppm)	291	254	86	74	232	199	173	126	152	215	45	57	322	302
Ce (ppm)	347	444	135	176	329	322	296	216	260	321	95	116	538	487
Sm (ppm)	27.5	30.4	9.9	8.8	19.9	18.3	16.9	15.8	13	18.6	6.1	5.7	31.4	30
Ru (ppm)	5.9	4.8	2.2	2.4	4.6	5.9	5.8	3.6	3.3	4.1	1.4	1.8	8	7.7
Yb (ppm)	45.2	50.4	21	24.8	33.2	29.5	31.4	31.9	27.5	35.4	20.2	21.4	45.2	45.8
Lu (ppm)	9.68	10	4.61	4.69	7.3	6.24	6.02	6.07	5.51	7.72	3.99	4.30	9.74	9.42

\* Duplicate

Appendix 7. (cont.)

Element	Sample No.									
	390*	372	389*	341	388*	317	309*	157	392*	383
Ag (ppm)	<5	<5	<7	<7	<6	<5	<5	<5	<5	<5
As (ppm)	65	73	150	170	62	79	44	37	70	67
Au (ppb)	<19	<19	<11	27	<11	<24	<16	<21	19	<15
Ba (ppm)	<300	<1000	200	200	400	<1300	1000	<400	<700	400
Cu (%)	<2	<2	<1	<1	3	<3	<3	<3	1	<1
Co (ppm)	66	69	85	93	62	73	39	39	56	51
Cr (ppm)	4900	5000	1900	2100	3400	3300	10000	11000	2300	2400
Fe (%)	24.9	26.2	31.6	32.7	22	22.8	25.5	26.3	25.6	25.6
Hf (ppm)	410	450	150	140	320	300	470	530	190	190
Ho (ppm)	12	<5	<5	<5	<5	<5	<5	<5	<5	7
Na (%)	.07	.06	<.05	.05	.12	.13	.08	.13	<.05	.06
Ni (ppm)	400	<300	<200	<200	300	<500	<300	<400	<300	<300
Sb (ppm)	2.9	2.9	4.9	5.7	2.4	3.3	6.4	7	2.4	2.6
Sc (ppm)	41.7	44	40	41.4	44.2	49.3	35.3	36.8	44.3	42.8
Se (ppm)	<9	<21	<5	<5	<5	<23	<12	<15	<12	<12
Ta (ppm)	8	10	7	5	7	6	11	10	5	7
Th (ppm)	32	37	13	10	26	23	58	66	13	14
U (ppm)	15.6	16	4.6	6.7	13.3	12.5	28.7	35.3	7.8	8.1
V (ppm)	<12	<10	8	7	47	<10	<12	21	<10	21
Zn (ppm)	340	110	140	140	110	200	310	370	170	150
La (ppm)	144	161	39	41	109	125	205	204	54	58
Ce (ppm)	253	279	79	80	197	195	356	397	99	94
Sm (ppm)	14.9	16.6	6	6.2	15.3	14.8	23.5	25.9	6.3	6.5
Ru (ppm)	5	6.6	1.6	1.7	3.7	5.9	4.5	5.8	1.5	1.3
Yb (ppm)	28.2	29.1	21.8	19.8	27.4	26.9	28.7	31.2	23.3	22
Lu (ppm)	5.71	6.32	3.79	3.64	5.1	5.44	6.03	6.09	4.78	4.65

\* Duplicate

Appendix B: Correlation coefficients for log transformed data, H.M.C. analyses

	As	Au	Ba	Ca	Co	Cr	Fe	Zn	Hf	Mo	Nb	Ni	Sb	Sc	Ta	Tb	U	V	La	Ce	Sm	Eu	Yb	Lu
As	1.00																							
Au	.201	1.00																						
Ba	.361	.425	1.00																					
Ca	.820	.271	.344	1.00																				
Co	.113	.213	.145	.245	1.00																			
Cr	.348	-.252	-.129	.447	-.043	1.00																		
Fe	.374	.437	.497	.513	.294	.113	1.00																	
Zn	.279	.354	.270	.355	.761	-.130	.254	1.00																
Hf	.091	.0303	.333	.056	.175	-.291	.121	.352	1.00															
Mo	.088	.148	.150	.223	.145	-.192	.202	.338	.380	1.00														
Nb	.408	.647	.141	.522	.078	.247	.525	.437	.225	.720	1.00													
Ni	.812	.133	.318	.724	.251	.249	.506	.414	.180	.148	.532	1.00												
Sb	.130	.221	.180	.338	.588	.206	.269	.709	.128	.222	.169	.265	1.00											
Sc	-.086	.635	.075	.075	.615	-.224	.485	.485	.242	.068	.279	.197	.502	1.00										
Ta	.380	.383	.317	.425	.675	-.140	.295	.960	.432	.392	.524	.497	.623	.446	1.00									
Tb	.348	.435	.315	.433	.745	-.136	.374	.963	.415	.383	.450	.498	.624	.513	.969	1.00								
U	.347	.355	.287	.437	.481	-.125	.405	.782	.340	.369	.789	.519	.477	.345	.725	.711	1.00							
V	.477	.405	.355	.517	.573	-.066	.312	.898	.477	.389	.554	.553	.553	.359	.968	.925	.683	1.00						
La	.413	.360	.357	.456	.582	-.074	.223	.899	.174	.390	.507	.505	.597	.365	.957	.912	.672	.969	1.00					
Ce	.552	.368	.354	.493	.571	-.074	.301	.905	.452	.412	.512	.546	.591	.371	.970	.924	.706	.985	.970	1.00				
Sm	.419	.373	.375	.524	.519	-.073	.285	.866	.406	.391	.687	.579	.570	.360	.925	.887	.689	.953	.932	.954	1.00			
Eu	.344	.332	.287	.503	.644	-.110	.240	.968	.401	.354	.435	.466	.726	.427	.964	.938	.705	.930	.931	.946	.913	1.00		
Yb	.334	.326	.281	.387	.655	-.115	.223	.974	.384	.358	.468	.458	.724	.429	.963	.940	.706	.923	.922	.938	.906	.996	1.00	

Appendix 9: Heavy mineral separation data sorted by decreasing heavy mineral content, 1986 samples.

Sample No.	Comp.	Weight (g)	Wt. H.M. (g)	Wt. M.M. (g)	%H.M. x 100	%M.M.
343	A	618.40	9.012	.8010	145.73	8.89
383	A	1000	12.238	1.4421	122.38	11.78
341	A	569.80	3.683	.6489	64.64	17.62
364	C	1151.97	5.712	.7296	49.58	12.77
372	A	1177.80	5.393	.6311	45.79	11.70
317	A	1000	3.703	.4653	37.03	12.57
338	A	1092.40	3.694	.5408	33.82	14.64
368	A	824.90	2.686	.4232	32.56	15.76
316	A	1175.40	3.468	.4325	29.50	12.47
318	D	816	2.121	.2586	25.99	12.20
361	A	1001.70	2.523	.1457	25.19	5.78
362	C	1000	2.510	.4490	25.10	17.89
353	A	1000	2.123	.3760	21.23	17.71
367	A	1861.62	3.890	.6264	20.90	16.10
321	R	1000	1.611	.0839	16.11	5.21
354	C	1000	1.381	.1229	13.81	8.90
355	A	1000	1.110	.1205	11.10	10.86
312	C	1071	1.161	.1196	10.84	10.30
353	C	1000	1.072	.1068	10.72	9.96
350	C	2079.79	2.167	.1552	10.46	7.16
315	C	1151.90	1.204	.0050	10.45	.42
386	R	1000	1.038	.0671	10.38	6.46
374	A	1000	1.035	.1591	10.35	15.37
369	A	1145.57	1.131	.0954	9.87	8.44
365	C	1040.50	1.020	.1346	9.80	13.29
313	C	977.20	.854	.0895	8.74	10.48
352	C	1000	.865	.0685	8.65	7.92
387	R	1161.72	1.002	.0296	8.63	2.95
329	C	1057.60	.906	.0617	8.57	6.81
331	R	1017.50	.850	.0745	8.35	8.65
308	C	1000	.830	.0950	8.30	11.45
351	C	1047.10	.799	.0678	7.63	8.49
319	R	1000	.752	.0605	7.52	8.05
314	A	1249.22	.903	.0954	7.23	10.56
332	R	1000	.706	.0531	7.06	7.52
322	R	1000	.700	.0335	7.00	4.79
307	R	1031.20	.707	.0353	6.86	12.07
328	D	1000	.679	.0545	6.79	8.03
347	C	1000	.670	.0551	6.70	8.22
346	R	1931.77	1.293	.0989	6.69	7.58
306	R	1269.33	.816	.1041	6.43	12.76
330	R	1000	.643	.0655	6.43	10.19
326	R	1007.20	.636	.0287	6.31	4.51
370	C	1000	.628	.0915	6.28	14.57



Appendix 9. (cont.)

384	R	1091.50	.610	.0109	5.59	1.79
349	D	1660.84	.909	.0687	5.47	7.56
357	D	1000.50	.506	.0346	5.06	6.84
325	R	1000	.504	.0372	5.04	7.38
324	C	861.80	.421	.0381	4.89	9.05
340	C	1010.20	.487	.0401	4.82	8.23
339	C	982.60	.445	.0424	4.53	9.53
382	C	1000	.434	.0038	4.34	.88
385	R	1081.30	.467	.0320	4.32	6.85
327	C	1006.60	.427	.0463	4.24	10.84
358	C	1101.50	.459	.0432	4.17	9.41
366	C	1000	.415	.0541	4.15	13.04
336	R	1005.60	.408	.0146	4.06	3.59
337	R	1319.29	.518	.0417	3.93	8.05
344	R	1079.30	.424	.0202	3.93	4.76
311	C	1000	.368	.0300	3.68	8.15
323	R	1000	.368	.0301	3.68	8.18
333	D	1000	.366	.0171	3.66	4.67
345	R	1000	.362	.0202	3.62	5.58
342	A	639.30	.222	.0161	3.47	7.25
356	R	1000	.325	.0317	3.25	9.75
375	C	1000	.325	.0250	3.25	7.69
334	D	1024.40	.328	.0320	3.20	9.76
376	C	1000	.303	.0202	3.03	6.67
379	R	728.80	.220	.0007	3.02	.32
309	R	1000	.296	.0438	2.96	14.80
359	R	1000	.292	.0284	2.92	9.73
320	R	1261.83	.329	.0269	2.61	8.18
381	C	1082.90	.280	.0243	2.59	8.68
373	C	1000	.241	.0216	2.41	8.96
360	R	1000	.239	.0135	2.39	5.65
380	R	1000	.229	.0251	2.29	10.96
335	R	1002.10	.207	.0132	2.03	6.38
378	R	927	.184	.0350	1.98	19.02
310	C	1000	.190	.0148	1.90	7.79
348	C	1094.10	.195	.0058	1.78	2.97
371	C	1011.40	.130	.0025	1.29	1.92
377	R	1000	.057	.0025	.57	4.39

A: Alluvium  
D: Diamicton

C: Colluvium  
R: Rubble

Wt. : Weight

H.M.: Heavy Mineral Fraction

M.M.: Magnetic Mineral Fraction

%H.M.: Heavy Mineral abundance: (H.M. / Wt. of Sample)100

%M.M.: Magnetic Mineral abundance: (M.M./H.M.)100



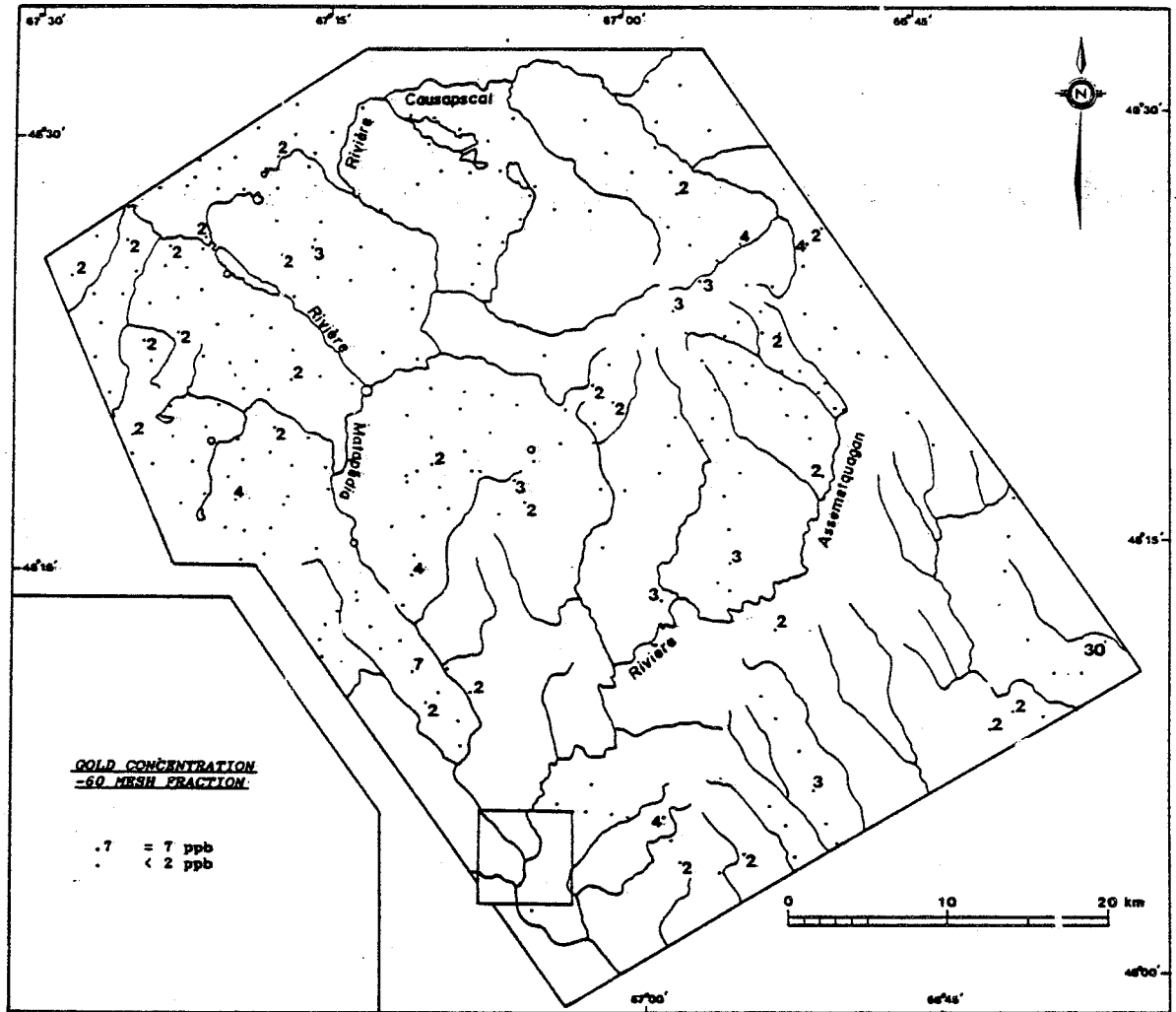


Fig. 6. Gold -60 mesh samples

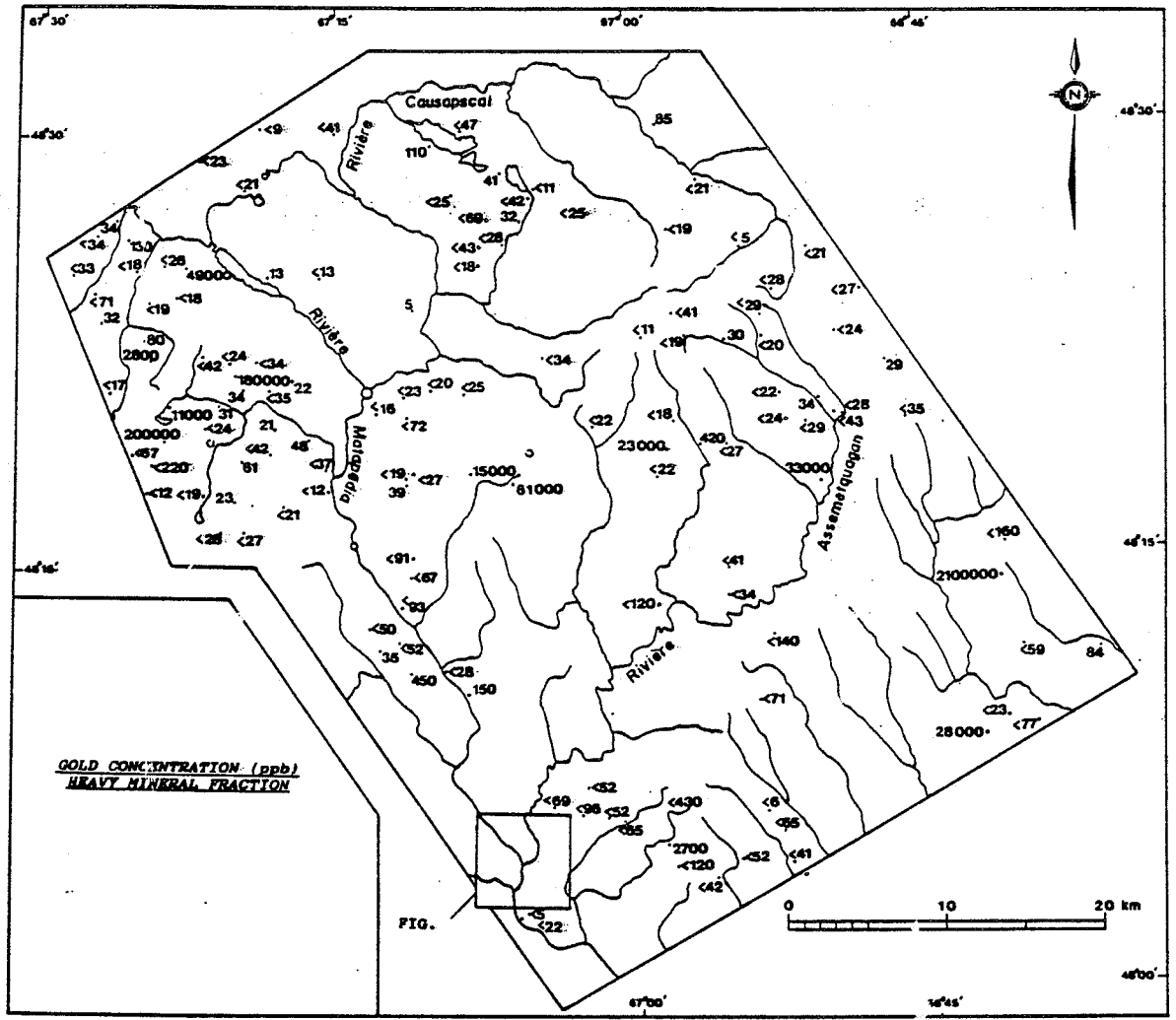


Fig. 7. Gold in heavy mineral concentrates

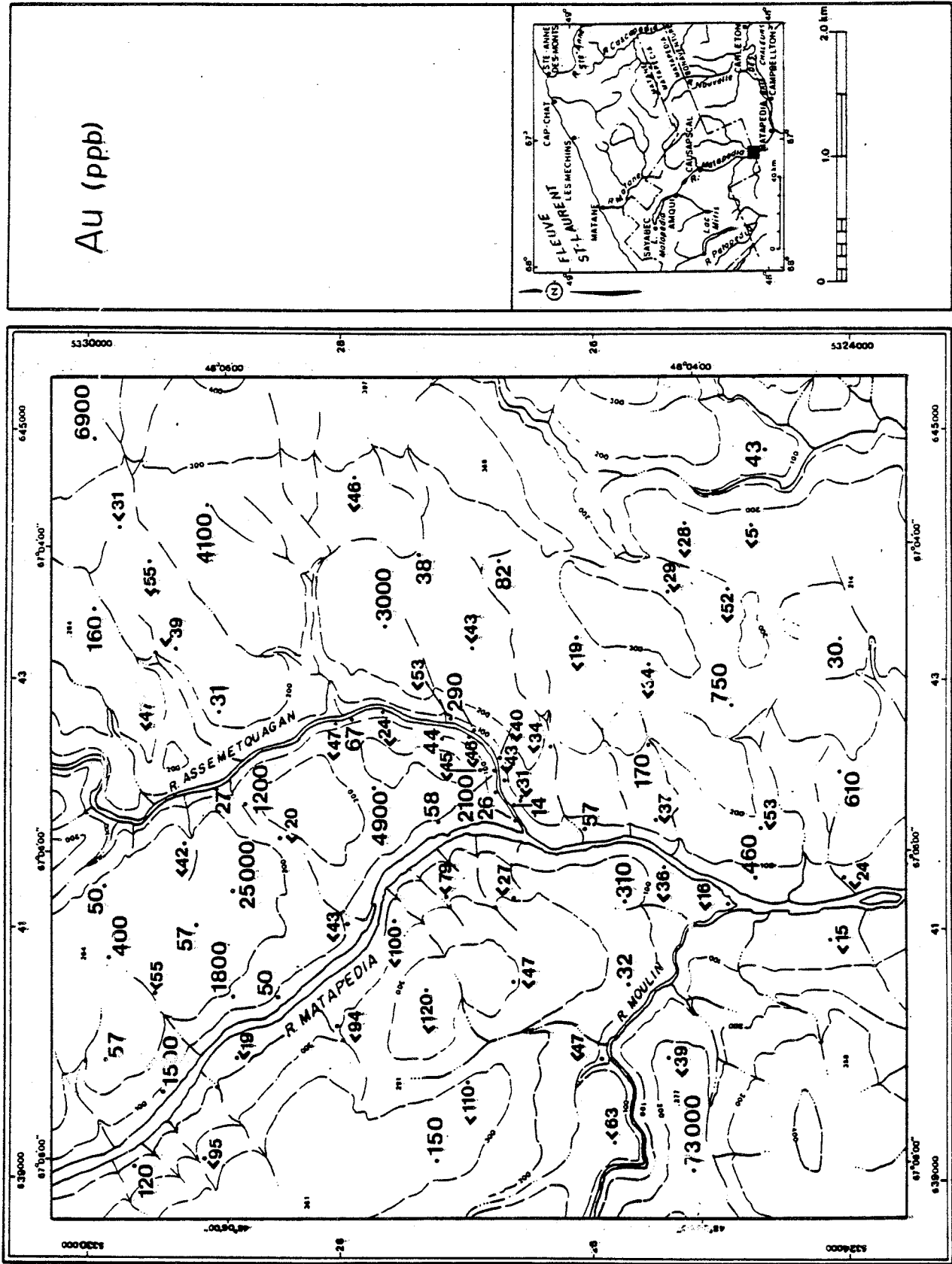


Fig. 8 Gold concentration in the heavy mineral fraction of overburden; 1986 study area.

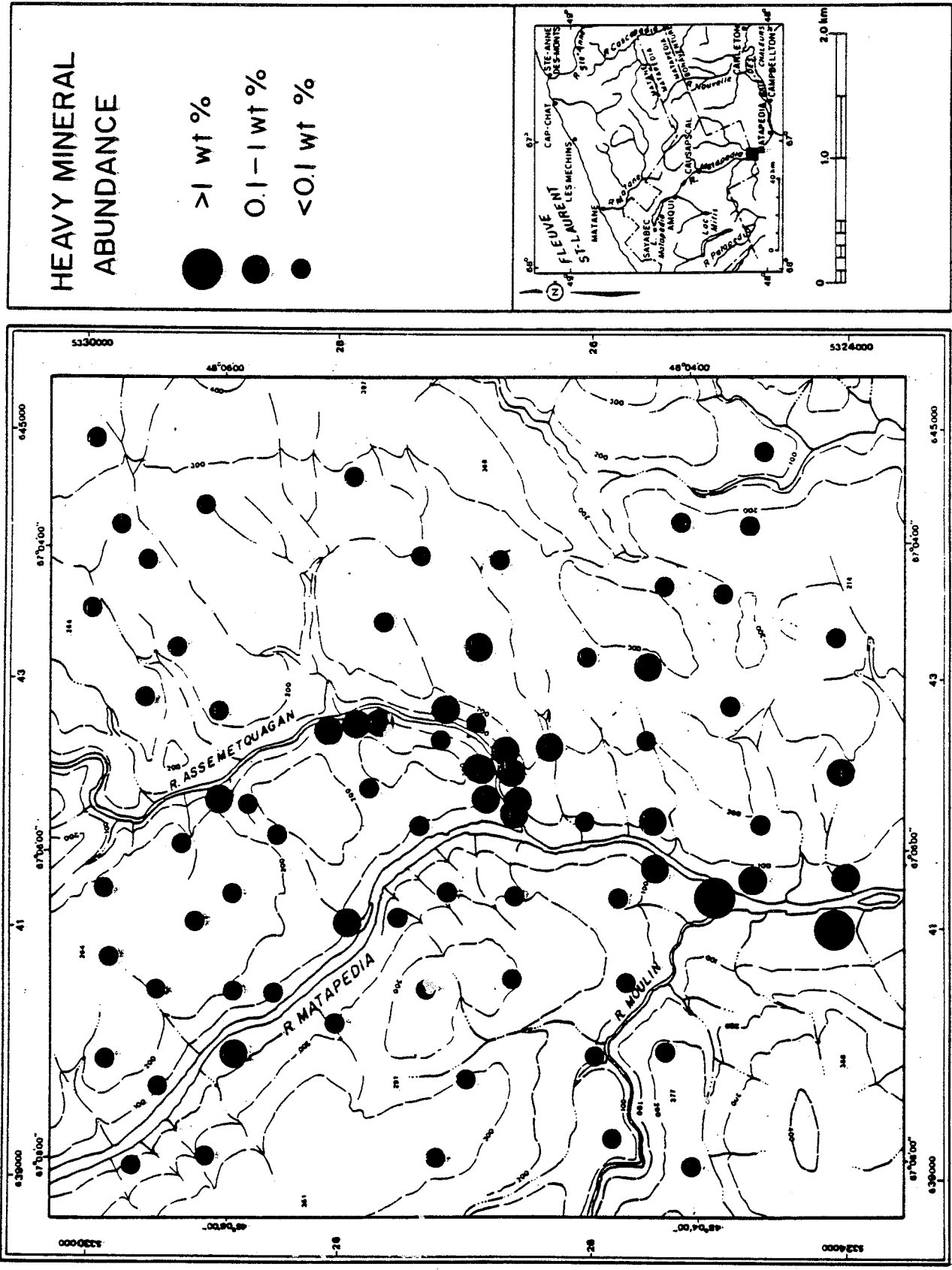


Fig. 9 Heavy mineral abundance in overburden; 1986 study area.

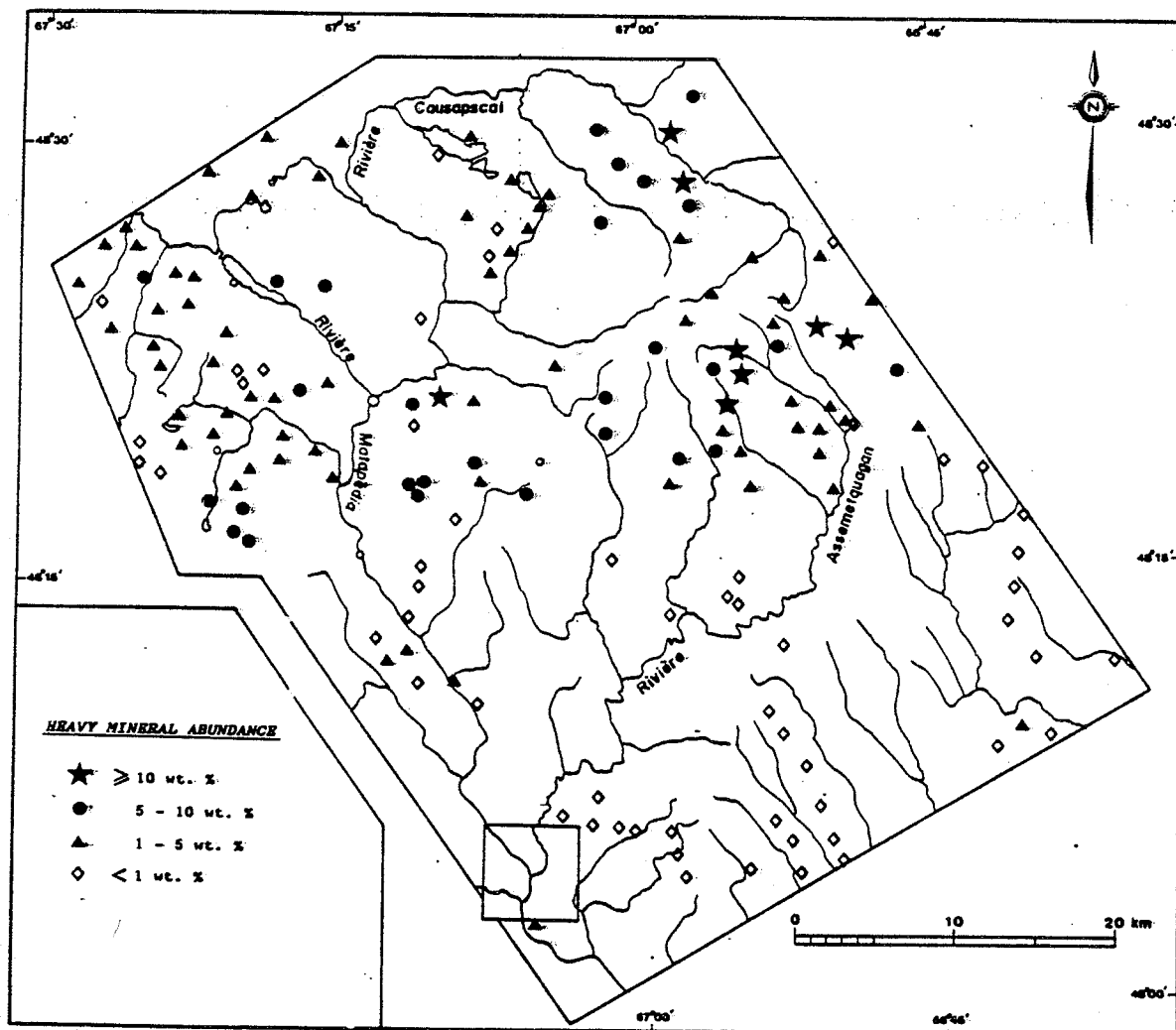
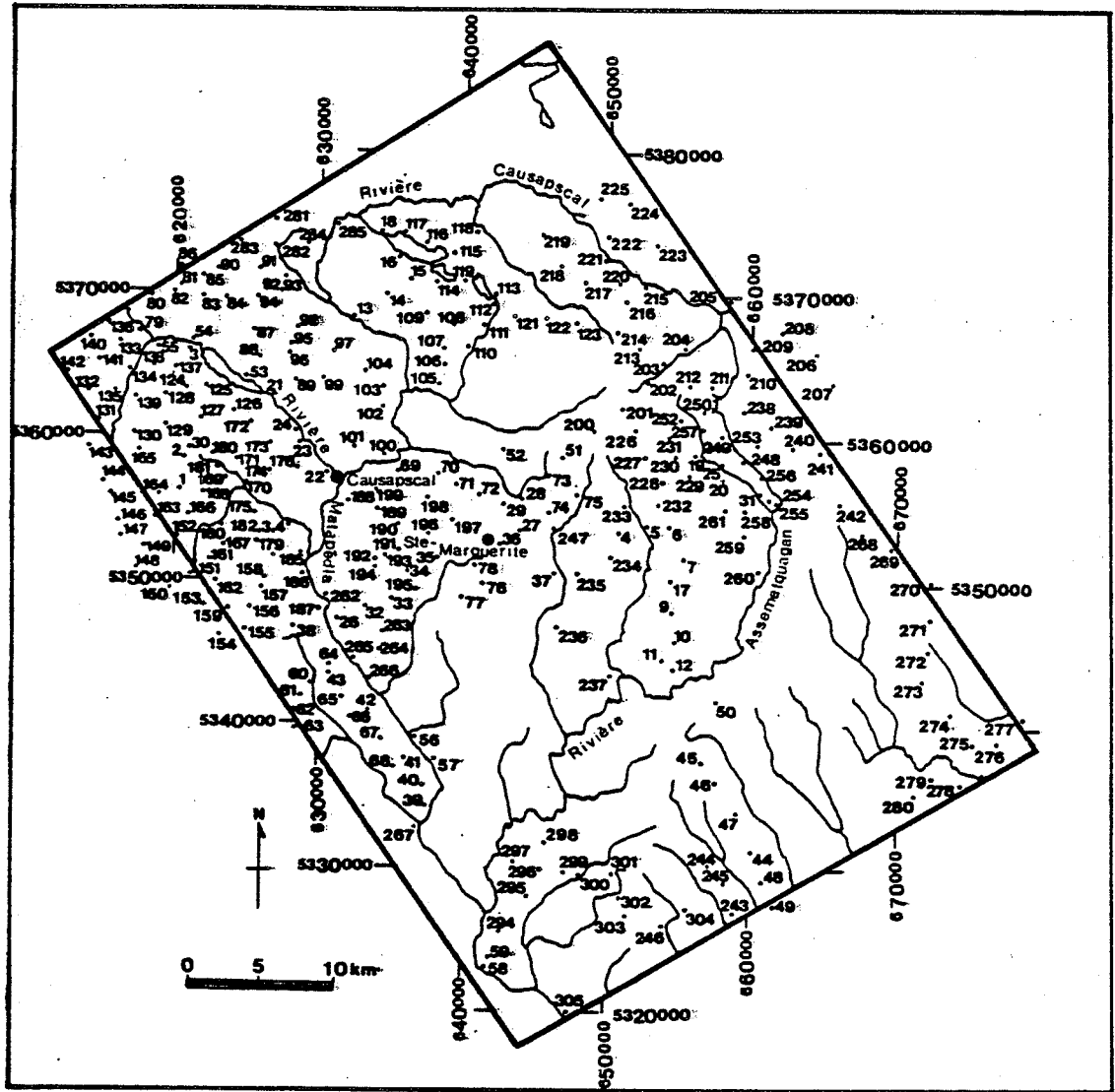
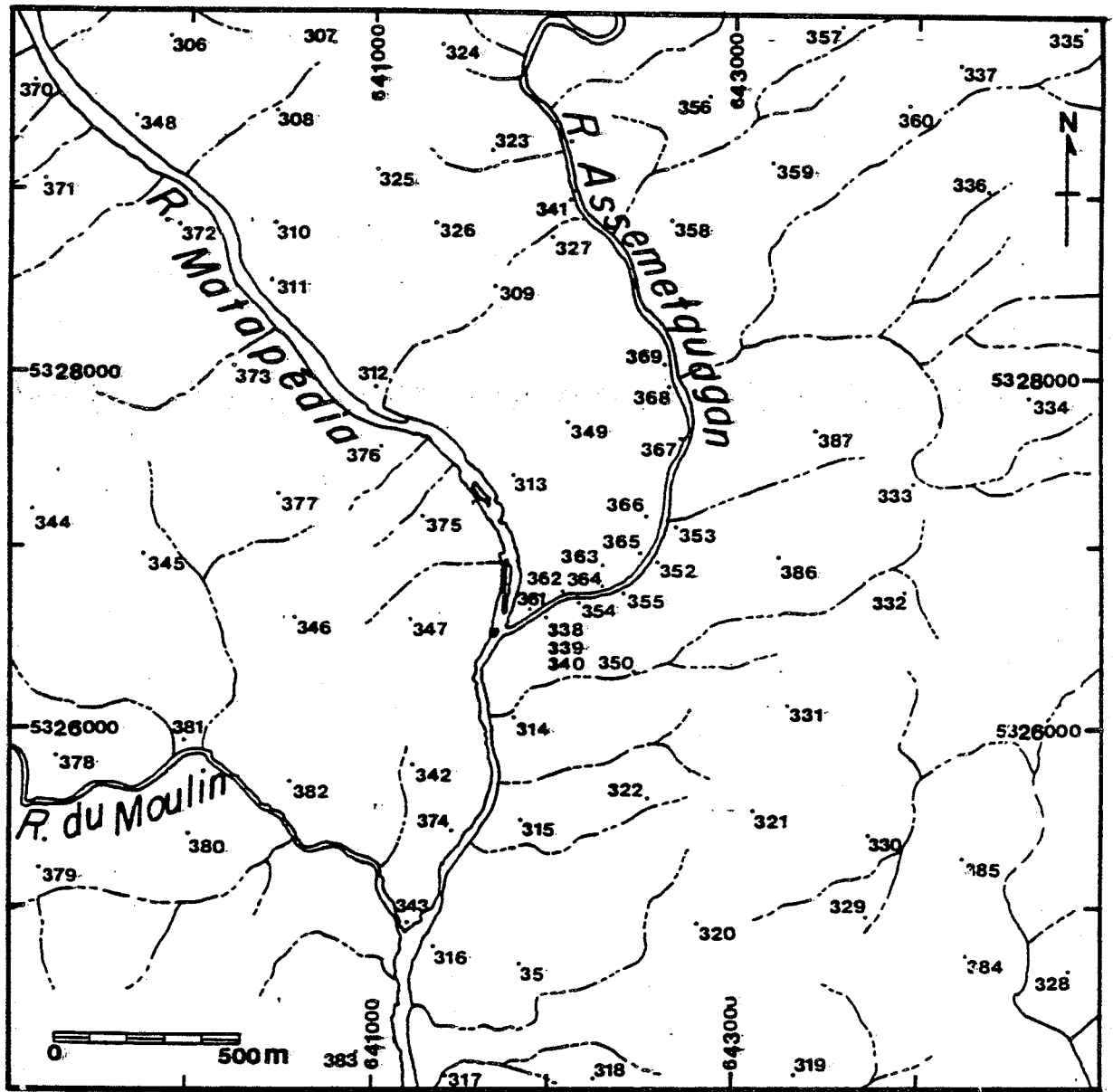


Fig. 1) Heavy mineral abundance in overburden



Appendix I. Sample location map, 1985-series samples





Appendix 2. Sample location map, 1986-series samples

Appendix 3. Legend for Appendices 4 and 5 and tables 2 and 3

Symbols for "SOIL" (overburden)

C - Colluvium

T - Till

A - Alluvium

U - Undifferentiated diamicton

Symbols for "Rock" (lithological units)

B - York Branch Fm.

C - St.-Léon Fm.

F - Fortin Gp.

G - Cap Bon-Ami and Grande-Grève Fms.

M - Matapedia Gp.

Double letters indicate samples at or close to lithological boundaries

n - Number of samples in a given category

LOI - Loss on ignition

North - Abbreviated northing

East - Abbreviated easting

e.g. East 2555 and North 4795 represent  
UTM designation 19UFD25554795

Appendix 4: Overburden Samples (1985-series)- Major element (%) recalculated to exclude LOI in summation.

SAMPLE	SiO2	TiO2	Al2O3	Fe2O3	MnO	MgO	CaO	Na2O	K2O	LOI	EAST	NORTH	SOIL	ROCK
156	81.33	.760	9.52	3.54	.040	1.00	.20	1.46	1.71	3.58	2555	4795	C	B
196	74.87	.749	13.55	6.70	.034	.97	.29	1.18	1.51	15.79	3755	5300	T	B
190	67.72	.992	16.89	8.24	.044	2.08	.12	.97	2.79	8.25	3555	5400	T	BY
174	78.15	.660	11.53	4.19	.028	1.46	.23	1.31	1.94	4.66	2675	5765	T	C
213	75.70	.833	12.55	4.33	.052	1.33	.87	1.81	2.06	4.99	5205	6645	T	C
302	70.09	.905	15.46	7.97	.035	1.88	.20	1.50	2.10	9.06	5100	2820	A	F
266	73.84	.807	13.65	6.90	.016	1.43	.17	1.27	1.81	12.01	3425	4325	C	F
56	76.82	.809	12.21	4.48	.060	1.58	.24	1.81	2.25	4.84	3650	3930	C	F
57	68.95	.878	16.68	7.29	.025	2.19	.13	1.67	2.69	7.85	3810	3755	C	F
12	68.56	.954	17.26	6.90	.073	2.22	.16	1.21	2.62	6.63	5430	4450	C	F
272	70.81	1.023	16.54	6.84	.024	1.18	.16	1.42	2.37	13.60	7190	4545	C	F
11	69.84	.872	16.63	7.25	.032	1.88	.14	1.46	2.00	9.38	5310	4505	C	F
301	56.27	.880	24.81	13.72	.123	1.28	.11	.89	1.62	29.46	5055	2955	C	F
9	71.25	.915	15.51	6.25	.057	2.03	.26	1.38	2.43	5.29	5460	4795	C	F
300	73.42	.856	14.15	5.43	.031	1.85	.14	1.82	2.03	4.37	4865	2960	C	F
296	68.67	.858	18.01	6.55	.032	1.92	.13	1.54	2.23	12.10	4550	2990	C	F
297	69.18	1.008	16.76	7.57	.021	1.56	.11	1.34	2.25	11.51	4360	3030	C	F
270	59.38	.785	24.16	11.08	.050	1.65	.08	1.18	1.97	23.46	7240	5020	C	F
271	72.65	1.066	15.55	6.42	.024	.98	.07	1.07	1.99	13.69	7200	4750	C	F
261	73.82	.834	14.69	5.95	.033	1.22	.27	1.29	1.68	12.45	5790	5545	C	F
242	76.15	.561	14.35	5.86	.031	.93	.16	.94	1.25	18.24	6595	5575	C	F
269	72.18	.898	15.32	5.92	.060	1.88	.28	1.10	2.45	5.13	6975	5310	C	F
294	74.94	.834	13.26	5.04	.025	1.67	.22	1.88	1.84	5.40	4285	2575	C	F
10	67.60	.928	18.51	6.88	.040	1.83	.43	1.37	2.36	10.77	5455	4605	C	F
237	66.47	.986	18.47	7.21	.042	2.38	.11	1.53	2.85	7.37	5030	4350	C	F
274	72.16	.882	15.30	6.39	.043	1.51	.13	1.50	1.96	8.47	7325	4095	C	F
236	72.70	.803	14.89	5.87	.025	1.89	.17	1.74	2.07	5.64	4665	4645	C	F
273	68.62	.946	18.12	6.81	.050	1.71	.28	1.48	2.22	11.77	7145	4315	C	F
234	70.97	.765	17.02	5.91	.030	1.50	.19	1.29	1.78	15.28	5010	5170	C	F
260	72.38	.844	15.63	6.19	.021	1.64	.18	1.42	2.13	8.94	6050	5120	C	F
295	70.31	.869	15.87	6.66	.039	1.96	.21	1.64	2.32	8.02	4450	2810	C	F
233	74.00	.809	14.91	5.49	.069	1.32	.36	1.14	1.76	12.16	5120	5535	C	F
244	72.44	.959	15.14	6.42	.031	1.56	.18	1.63	1.79	9.97	5720	3030	C	F
240	79.32	.872	11.14	3.78	.041	.96	.32	1.36	1.84	3.83	6275	5905	T	F
44	74.08	.910	13.80	5.82	.039	1.37	.19	1.63	1.79	7.32	6000	3150	T	F
6	76.20	.821	13.19	4.84	.041	1.41	.26	1.30	1.94	5.36	5450	5370	T	F
47	71.51	1.038	15.58	6.38	.038	1.46	.15	1.49	2.06	9.37	5925	3380	T	F
37	74.33	.740	13.45	5.52	.023	1.70	.13	1.43	1.78	7.75	4610	5080	T	F
258	71.12	.925	16.36	6.57	.067	1.49	.20	1.11	2.01	11.32	5980	5470	T	F
77	77.25	.593	12.60	5.35	.060	1.31	.10	1.13	1.56	10.72	3975	4895	T	F
38	72.12	.838	14.94	5.88	.054	2.14	.24	1.67	2.62	3.48	2850	4690	T	F
33	75.97	.726	13.26	5.10	.033	1.78	.16	1.55	2.05	4.40	3530	4885	T	F

## Appendix 4 (cont.)

SAMPLE	SiO <sub>2</sub>	TiO <sub>2</sub>	Al <sub>2</sub> O <sub>3</sub>	Fe <sub>2</sub> O <sub>3</sub>	MnO	HgO	CaO	Na <sub>2</sub> O	K <sub>2</sub> O	LOI	EAST	NORTH	SOIL	ROCK
232	74.06	.787	14.83	5.76	.036	1.50	.10	.69	2.50	7.26	5350	5525	T	F
45	67.25	.930	17.48	7.93	.059	1.98	.12	1.19	2.53	8.92	5650	3750	U	F
4	76.14	.807	13.40	5.67	.030	1.28	.17	1.11	1.73	10.66	5060	5355	U	F
46	68.83	.945	17.69	7.39	.060	1.67	.13	1.21	2.19	14.31	5755	3615	U	F
17	70.41	.930	16.52	6.27	.031	1.79	.83	1.25	2.18	32.25	5430	5010	U	F
59	74.97	.829	13.73	5.25	.028	1.56	.18	1.61	1.80	6.31	4230	2365	U	F
7	71.86	.872	16.18	6.12	.048	1.71	.24	1.14	2.23	8.27	5520	5165	U	F
298	69.85	.934	16.82	6.69	.047	1.82	.16	1.50	2.10	8.39	4615	3155	U	F
259	71.51	.839	14.05	10.27	.033	.83	.09	.93	1.38	24.54	5985	5275	U	F
303	67.38	.929	17.59	8.59	.036	1.64	.12	1.44	2.00	14.12	5180	2665	U	FM
100	78.82	.509	11.53	4.52	.027	.96	.28	.65	1.80	7.02	3450	5915	C	G
175	76.13	.645	11.57	5.07	.134	1.82	.37	1.21	2.70	3.62	2575	5485	C	G
109	69.91	.750	15.73	7.22	.048	2.21	.16	.74	2.62	6.82	3740	6875	C	G
107	77.71	.652	11.37	5.66	.043	1.06	.19	.85	1.35	10.65	3855	6645	T	G
98	78.53	.601	11.85	4.38	.042	1.09	.50	1.20	1.40	12.20	2825	6805	T	G
166	74.64	.782	12.57	5.57	.035	1.65	.65	1.46	2.26	7.39	2115	5430	T	G
92	78.67	.677	10.87	4.16	.048	1.57	.29	1.50	1.88	3.69	2775	7120	T	G
139	80.84	.632	10.08	3.65	.043	1.13	.25	1.13	1.80	4.68	1765	6245	T	G
90	79.16	.569	10.72	6.00	.031	.94	.20	.97	1.31	11.57	2295	7180	T	G
124	80.54	.515	9.83	3.90	.077	1.28	.32	.91	2.01	3.06	2060	6350	T	G
89	75.62	.794	13.07	5.55	.047	1.38	.54	1.22	1.51	11.39	2840	6410	T	G
103	72.94	.733	14.12	5.21	.027	2.19	.09	1.02	2.95	5.10	3445	6370	T	G
88	73.44	.608	14.30	6.10	.051	1.39	.78	1.23	2.01	10.10	2585	6565	T	G
133	70.72	.725	14.28	6.94	.130	2.60	.19	.87	3.24	4.63	1600	6650	T	G
165	81.46	.623	9.92	3.21	.029	.96	.23	1.14	1.89	4.64	1765	5845	T	G
105	76.89	.734	11.40	5.99	.028	1.05	.28	1.25	1.69	9.85	3815	6380	T	G
83	80.55	.618	10.22	5.17	.057	.83	.25	.89	1.34	10.95	2180	6965	T	G
126	81.06	.655	9.35	3.69	.180	.85	.71	1.10	1.56	13.90	2405	6170	U	G
277	60.66	.877	16.68	6.64	.075	2.76	7.85	1.25	3.30	10.86	7825	4130	C	M
246	68.87	.785	16.76	7.10	.099	1.50	.19	2.12	2.87	5.09	5380	2600	C	M
245	71.41	.973	15.31	7.31	.031	1.19	.16	1.58	1.81	10.69	5835	2895	C	M
304	45.88	.910	15.87	6.70	.069	1.52	24.57	.86	3.23	19.86	5570	2715	C	M
280	63.45	.890	19.44	8.71	.095	2.90	.14	1.47	3.47	9.25	7110	3535	T	M
48	72.34	1.002	14.69	6.79	.055	1.21	.31	1.53	1.93	12.55	6075	2930	T	M
49	64.60	.923	17.40	7.49	.069	4.54	.17	1.60	3.13	6.38	6190	2730	T	M
279	64.64	.773	17.74	9.77	.155	1.92	.48	1.67	3.17	8.32	7295	3650	T	M
278	67.93	.903	16.45	7.76	.048	3.19	.47	1.63	2.28	9.07	7475	3635	T	M
276	65.27	.923	18.43	7.76	.073	3.27	.12	1.19	3.39	7.87	7700	3900	U	M

Appendix 4 (cont.)

SAMPLE	SiO2	TiO2	Al2O3	Fe2O3	MnO	HgO	CaO	Na2O	K2O	LOI	EAST	NORTH	SOIL	ROCK
243	68.81	.953	16.14	7.19	.103	2.22	.32	1.55	2.95	7.11	5895	2715	U	M
275	69.99	.797	16.53	6.77	.084	1.50	.19	1.54	2.59	6.86	7490	3915	U	M
158	75.24	.890	12.55	5.30	.028	1.42	.18	1.70	2.14	4.32	2665	5020	C	Y
201	70.24	.995	14.68	9.87	.057	.80	.32	1.00	1.89	15.78	5090	6195	C	Y
231	77.49	.833	11.55	5.01	.133	1.28	.46	1.61	1.84	4.53	5400	6050	C	Y
185	74.78	.797	13.51	5.25	.032	1.53	.19	1.60	2.07	5.42	2885	5210	T	Y
120	77.76	.838	11.44	4.44	.043	1.25	.18	1.46	1.98	4.06	4025	6365	T	Y
249	75.57	.962	13.46	4.96	.042	1.12	.18	1.49	2.19	4.08	5760	6035	T	Y
223	78.00	.721	12.25	4.44	.026	1.34	.21	1.31	1.62	5.32	5320	7360	T	Y
220	78.67	.770	11.30	4.07	.039	1.05	.41	1.76	1.69	4.11	5045	7110	T	Y
218	81.69	.764	9.64	2.87	.041	.91	.41	1.78	1.74	2.31	4650	7205	T	Y
74	77.14	.876	12.04	4.50	.038	1.33	.30	1.51	2.03	5.26	4595	5490	T	Y
194	79.34	.794	11.17	4.27	.032	1.28	.19	1.39	1.87	4.27	3395	5105	T	Y
13	77.98	.814	11.71	4.39	.028	1.29	.34	1.57	1.57	6.66	3200	6835	T	Y
15	68.63	.825	16.14	9.93	.037	1.45	.35	1.29	1.48	11.51	3600	7075	T	Y
161	72.69	.928	13.49	5.93	.116	1.75	.28	1.73	2.82	4.36	2235	5145	T	Y
211	72.91	.976	14.69	6.30	.056	1.42	.28	1.67	1.57	9.19	5740	6355	T	Y
51	73.66	1.005	13.19	5.79	.101	1.82	.27	1.54	2.47	3.63	4690	5875	T	Y
119	79.09	.724	10.51	4.26	.036	1.13	.33	1.52	1.63	5.07	3975	7090	C	YG
70	75.04	.811	13.62	5.86	.031	1.18	.25	1.23	1.67	8.31	3745	5700	T	YG

Appendix 5. Overburden samples - Trace elements (ppm) except Au (ppb)

SAMPLE	Ba	Co	Cu	Ni	Zn	Nb	Zr	Y	Sr	Rb	Pb	Th	U	Au	EAST	NORTH	SOIL	ROCK
156	288	<5	28	36	40	12	411	14	80	67	15	10	11		2555	4795	C	B
196	250	9	34	24	33	15	296	13	75	64	18	12	14		3755	5300	T	B
190	328	23	14	75	69	18	249	18	67	117	18	13	9		3155	5400	T	BY
174	388	<5	29	34	48	8	239	21	79	93	17	10	9	2	2675	5765	T	C
213	400	5	26	32	62	9	538	22	129	79	22	13	10		5205	6645	T	C
302	374	31	28	59	69	19	310	22	63	98	19	12	11		5100	2820	A	F
266	233	20	19	38	47	17	208	17	58	99	20	12	12		3425	4325	C	F
56	279	9	31	54	46	21	270	22	75	107	24	12	10		3650	3930	C	F
57	397	31	29	70	71	20	214	21	58	119	23	15	9	2	3810	3755	C	F
12	391	32	33	70	80	21	265	25	61	118	22	11	8		5430	4450	C	F
272	354	8	12	25	29	13	264	20	59	109	19	13	7		7190	4545	C	F
11	309	19	33	60	73	16	234	21	56	101	16	12	10		5310	4505	C	F
301	198	44	30	40	95	17	191	18	46	83	22	14	12	4	5055	2955	C	F
9	442	21	42	67	72	24	289	24	71	107	25	15	10		5460	4795	C	F
300	314	9	24	67	71	14	302	23	67	94	19	13	10		4865	2960	C	F
296	318	13	58	62	63	17	283	21	59	101	21	14	9		4550	2990	C	F
297	355	21	19	48	83	19	230	21	53	117	18	13	9		4360	3030	C	F
270	251	30	21	41	81	17	189	16	40	94	19	12	9		7240	5020	C	F
271	325	11	30	27	38	12	273	22	47	97	20	14	9		7200	4750	C	F
261	259	20	29	43	65	10	283	17	73	80	20	12	11		5790	5545	C	F
242	153	8	38	30	48	9	175	10	50	69	21	13	11		6595	5575	C	F
269	425	21	28	69	66	21	308	22	72	102	20	11	10		6975	5310	C	F
294	316	<5	33	48	60	11	318	20	76	84	16	13	9		4285	2575	C	F
10	440	31	50	78	90	22	294	24	83	103	28	13	8	3	5455	4605	C	F
237	468	34	45	95	101	19	246	24	57	127	26	13	7	3	5030	4350	C	F
274	264	13	27	54	73	11	291	19	70	96	20	13	11		7325	4095	C	F
236	292	18	65	60	81	20	285	21	63	95	24	15	10		4665	4645	C	F
273	317	13	36	87	69	15	307	22	70	96	21	12	8		7145	4315	C	F
234	238	10	26	57	61	8	235	16	58	86	20	12	10		5010	5170	C	F
260	329	22	28	70	59	17	296	18	61	101	21	11	9	2	6050	5120	C	F
295	380	20	42	61	76	18	277	22	68	102	22	13	13		4450	2810	C	F
233	348	10	28	51	72	10	281	15	86	76	18	9	11		5120	5535	C	F
244	316	15	29	87	89	15	307	21	71	88	15	11	11		5720	3030	C	F
240	395	<5	28	31	39	12	373	16	94	75	20	12	10		6275	5905	T	F
44	266	8	21	52	60	14	292	19	74	87	19	13	11	3	6000	3150	T	F
6	339	10	26	75	89	11	307	17	75	86	21	12	10		5450	5370	T	F
47	309	20	25	120	93	17	314	22	70	100	23	11	7		5925	3380	T	F
37	202	11	35	62	65	12	224	16	59	87	22	13	11		4610	5080	T	F
258	333	14	23	58	97	12	268	19	64	95	22	11	9		5980	5470	T	F
77	213	<5	49	50	55	8	198	13	46	77	22	13	14		3975	4895	T	F
38	383	17	42	71	64	14	311	18	86	82	19	11	11		2850	4690	T	F
33	325	11	25	66	77	9	246	19	68	93	21	14	10		3530	4885	T	F
232	391	8	34	68	90	19	224	15	48	108	23	12	9		5350	5525	T	F

Appendix 5. (cont.)

SAMPLE	Ba	Co	Cu	Ni	Zn	Nb	Zr	Y	Sr	Rb	Pb	Th	U	Au	EAST	NORTH	SOIL	ROCK
45	384	31	31	66	91	21	249	23	49	118	21	11	5		5650	3750	U	F
4	253	10	21	44	51	15	266	15	58	83	18	11	12		5060	5355	U	F
46	304	17	76	61	76	11	288	23	49	98	26	12	10	2	5755	3615	U	F
17	288	9	22	50	64	19	297	22	65	100	23	13	9		5430	5010	U	F
59	326	10	27	55	54	12	296	27	73	94	20	12	11		4230	2365	U	F
7	337	18	22	70	81	19	283	21	65	100	24	14	11		5520	5165	U	F
298	313	18	28	87	87	20	293	24	63	100	23	14	11		4615	3155	U	F
259	187	24	28	14	43	19	203	12	40	79	21	11	12		5985	5275	U	F
303	270	22	24	55	115	19	239	19	71	105	20	12	9	2	5180	2665	U	FM
100	348	5	32	38	68	10	135	11	55	100	22	11	9		3450	5915	C	G
175	518	15	57	57	88	18	241	50	78	109	27	11	9	2	2575	5485	C	G
109	441	12	38	66	78	17	160	18	66	134	23	11	7		3740	6875	C	G
107	203	5	46	36	100	16	196	9	76	80	20	10	11		3855	6645	T	G
98	246	5	15	32	57	8	258	11	90	64	19	10	12		2825	6805	T	G
166	348	12	39	43	79	19	352	16	81	97	21	10	11		2115	5430	T	G
92	279	9	49	49	42	6	322	18	93	79	23	13	12		2775	7120	T	G
139	595	5	57	24	44	8	226	22	71	90	19	11	10		1765	6245	T	G
90	190	5	105	21	112	17	179	9	70	73	27	10	14		2295	7180	T	G
124	729	5	28	33	67	10	193	28	73	92	22	13	11		2060	6350	T	G
89	310	5	23	32	55	10	275	14	91	73	22	11	13		2840	6410	T	G
103	399	8	58	31	105	17	175	14	63	131	24	16	9		3445	6370	T	G
88	509	12	18	55	50	16	206	46	105	90	19	9	8	2	2585	6565	T	G
133	806	17	52	61	78	19	157	49	52	145	32	14	7	2	1600	6650	T	G
165	534	5	32	12	57	7	201	9	77	94	20	10	13		1765	5845	T	G
105	286	19	9	34	63	17	293	13	81	77	19	8	11		3815	6380	T	G
83	226	5	31	54	73	8	214	10	79	75	19	8	10		3180	6965	T	G
126	485	5	26	16	96	6	232	11	84	79	26	7	11		2405	6170	U	G
277	266	25	24	70	72	18	193	26	236	132	20	13	8	30	7825	4130	C	M
246	359	19	29	94	66	25	201	43	57	119	25	15	8		5380	2600	C	M
245	271	18	15	39	66	15	317	21	78	91	21	14	12		5835	2895	C	M
304	312	20	25	76	74	18	177	17	374	113	23	12	12	2	5570	2715	C	M
280	424	36	41	76	125	21	177	16	53	146	27	15	6	2	7110	3535	T	M
48	236	19	34	38	64	17	265	19	71	99	20	11	8		6075	2930	T	M
49	411	28	17	137	56	22	207	14	40	122	23	16	9		6190	2730	T	M
279	284	39	29	60	86	21	158	53	59	136	25	18	9	2	7295	3650	T	M
278	246	25	24	90	91	20	207	11	62	111	20	12	12		7475	3635	T	M
276	353	32	30	80	107	20	218	21	46	139	27	17	7		7700	3900	U	M
243	385	27	26	78	69	27	240	17	51	139	24	14	6		5895	2715	U	M
275	345	24	22	59	59	19	234	21	76	115	26	18	8		7490	3915	U	M
158	323	10	14	47	35	12	311	16	87	95	15	13	10		2665	5020	C	Y

Appendix 5. (cont.)

SAMPLE	Ba	Co	Cu	Ni	Zn	Nb	Zr	Y	Sr	Rb	Pb	Th	U	Au	EAST	NORTH	SOIL	ROC
201	321	25	25	41	74	18	328	19	84	109	20	14	9	3	5090	6195	C	Y
231	368	9	41	53	50	12	471	28	101	78	20	11	11		5400	6050	C	Y
185	300	10	66	44	68	16	279	17	85	100	17	12	10		2885	5210	T	Y
120	289	<5	31	45	50	9	297	16	91	85	20	12	12		4025	6865	T	Y
249	365	11	21	54	58	16	396	21	88	90	19	12	10	2	5760	6035	T	Y
223	202	6	34	58	58	15	264	13	82	84	21	13	10		5320	7360	T	Y
220	349	<5	27	32	39	12	318	15	104	80	17	9	11		5045	7110	T	Y
218	335	<5	26	19	25	7	351	16	107	71	18	10	13		4650	7205	T	Y
74	378	13	36	49	57	14	336	17	85	78	19	11	12		4595	5490	T	Y
194	343	<5	26	44	42	16	336	15	81	79	20	11	13		3395	5105	T	Y
13	258	<5	32	35	37	15	340	15	89	76	23	12	12		3200	6835	T	Y
15	305	33	28	43	69	16	239	14	83	73	21	11	12		3600	7075	T	Y
161	455	14	35	75	62	19	288	24	91	95	26	12	8		2235	5145	T	Y
211	270	22	36	65	86	17	295	18	96	74	22	11	15		5740	6355	T	Y
51	429	20	25	78	74	20	340	26	94	96	28	13	10		4690	5875	T	Y
119	331	<5	38	36	44	10	303	13	95	69	21	12	13		3975	7090	C	Y
70	288	13	48	41	103	14	270	14	80	84	19	12	10		3745	5700	T	Y



NUCLEAR ACTIVATION SERVICES LIMITED

1280 MAIN STREET WEST, HAMILTON, ONTARIO, L8S 4K1  
 PHONE (416) 522-5666 TELEFX 06-986947

CERTIFICATE OF ANALYSIS

TO: MC GILL UNIVERSITY  
 ATTN: M.A. BERNIER  
 DEPT. OF GEOLOGICAL SCIENCES  
 3450 UNIVERSITY STREET  
 MONTREAL, QUEBEC

CUSTOMER NO. 133/01/07  
 DATE SUBMITTED  
 04-FEB-87

REPORT: 7661

FILE NUMBER: 9356

100 UNPREPARED SAMPLES

PO# C 500431R

WERE ANALYZED AS FOLLOWS:

ELEMENTS	DETECTION LIMIT	UNITS	METHOD	ELEMENTS	DETECTION LIMIT	UNITS	METHOD
AG	5.0000	PPM	INAA	SC	0.1000	PPM	INAA
AS	2.0000	PPM	INAA	SE	5.0000	PPM	INAA
AU	5.0000	PPB	INAA	TA	1.0000	PPM	INAA
BA	100.0000	PPM	INAA	TH	0.5000	PPM	INAA
CA	1.0000	%	INAA	U	0.5000	PPM	INAA
CO	5.0000	PPM	INAA	W	4.0000	PPM	INAA
CR	10.0000	PPM	INAA	ZN	50.0000	PPM	INAA
FC	0.0200	%	INAA	LA	1.0000	PPM	INAA
HF	1.0000	PPM	INAA	CE	3.0000	PPM	INAA
MC	5.0000	PPM	INAA	SM	0.1000	PPM	INAA
NA	0.0500	%	INAA	EU	0.2000	PPM	INAA
VI	200.0000	PPM	INAA	YB	0.2000	PPM	INAA
SB	0.2000	PPM	INAA	LU	0.0500	PPM	INAA

COMMENTS:

NOTE: DETECTION LIMIT RAISED DUE TO SAMPLE SIZE & COMPOSITION.

DATE 23-FEB-87

NUCLEAR ACTIVATION SERVICES LIMITED  
 CERTIFIED BY: *R. Blakely*

\*\*\* UNLESS INSTRUCTED OTHERWISE WE WILL DISCARD ALL SAMPLES \*\*\*  
 IRRADIATED SAMPLES AFTER 20 DAYS. ANY OTHER MATERIAL AFTER 120 DAYS

Appendix 6. Trace element results, heavy mineral concentrate analyses.

NUCLEAR ACTIVATION SERVICES LIMITED

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SAMPLE NUMBERS

85-0296-H\*\* 85-0207-H\*\* 85-0299-H\*\* 85-0300-H\*\*

<35	<61	<5	<79	<31	<5
250	310	290	140	320	100
<61	<96	<52	<52	<65	<65
3100	<1800	<1400	2000	1200	2600
<11	<21	<14	<10	<11	<1
150	170	100	91	200	90
9400	9300	5600	8000	7000	12000
28.2	26.9	31.1	18.7	26.0	15.8
1400	1300	1400	2400	1700	3200
<5	<5	<5	23	<5	59
0.07	0.16	0.12	0.07	0.07	0.13
1000	<1600	<1300	<800	<900	<1000
13	21	12	9.9	15	7.9
64.4	62.7	49.7	70.1	58.7	80.5
<59	<87	<59	<21	<61	<120
16	<13	13	17	15	27
200	180	200	310	230	380
72.9	67.8	75.0	106	89.2	142
28	61	<74	90	44	46
290	390	<190	270	150	280
567	1170	984	1350	1100	1660
1480	1840	1630	2270	1840	2580
102	118	106	153	120	175
17.3	18.9	31.0	29.4	29.0	43.4
108	103	182	178	135	207
21.9	19.2	20.9	35.0	26.0	43.3

NUCLEAR ACTIVATION SERVICES LIMITED

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SAMPLE NUMBERS

86-0301-H\*\* 86-0300-H\*\* 86-0337-H\*\* 86-0346-H\*\* 86-0349-H\*\* 86-0350-H\*\*

AG	PPM	<5	<34	<20	<24	<5	<21
AS	PPM	240	280	310	120	180	150
AU	PPB	<630	57	<11	<67	4900	<34
9A	PPM	<9100	1500	1800	1700	2200	700
CA	%	<70	<11	<6	<9	<1	<7
CO	PPM	250	140	180	120	100	100
CR	PPM	11000	6200	7100	8500	6700	6900
FE	%	32.0	29.9	30.0	28.2	19.9	21.5
HF	PPM	1400	790	1000	2100	1500	1200
MO	PPM	<290	5	<5	<5	<5	20
NA	%	0.42	0.07	0.09	0.12	<0.05	0.09
NI	PPM	9100	<900	<600	<800	<200	<500
SB	PPM	49	9.7	11	8.5	9.1	8.2
SC	PPM	43.0	47.9	52.5	79.7	56.3	50.0
SE	PPM	<520	<27	<34	<60	<5	<51
TA	PPM	<45	19	10	12	13	17
TH	PPM	260	99	170	230	190	140
U	PPM	89.4	34.0	53.8	87.5	70.3	56.6
W	PPM	530	<21	36	<16	33	34
ZN	PPM	<1000	<150	<90	190	90	150
LA	PPM	845	774	849	1140	906	714
CE	PPM	1370	1160	1390	1650	1390	1070
SM	PPM	80.6	66.0	96.3	115	93.9	63.3
EU	PPM	<35.1	24.4	22.7	25.2	22.5	14.3
YB	PPM	98.2	69.6	90.8	137	108	78.7
LU	PPM	22.7	13.2	16.9	29.4	23.0	16.3



NUCLAR ACTIVATION SERVICES LIMITED

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SAMPLE NUMBERS

SAMPLE NUMBERS

318-H**	319-H**	320-H**	321-H**	322-H**	323-H**	ELEMENT	324-H**	325-H**	326-H**	327-H**	328-H**	329-H**
**	**	**	**	**	**	UNITS	**	**	**	**	**	**
<5	<13	<5	<5	<14	<5	AG	<5	<15	<5	<17	<13	<5
66	150	45	73	76	190	AS	160	190	58	160	220	120
910	30	750	<14	170	<42	AU	50	57	25000	1200	43	<52
900	900	1200	800	600	2200	BA	1200	1200	700	700	400	1300
<5	<4	<8	<7	<5	<3	CA	<8	<6	<14	<6	<4	<9
63	98	54	74	74	110	CO	99	98	49	100	130	96
6000	6900	11000	9800	14000	7900	CR	7900	6500	6400	7500	7100	8700
20.5	25.1	20.2	16.9	18.1	25.0	FE	20.9	26.7	16.1	25.9	29.8	17.7
980	840	2200	960	1400	1400	HF	1600	880	1600	1400	800	1300
<5	<5	2.6	<5	<5	<5	MO	18	<5	49	16	<5	33
0.12	0.14	0.09	0.19	0.08	0.11	NA	0.13	0.12	0.19	0.08	0.12	0.11
<50	500	<700	<600	<400	<700	NI	<700	<500	<1100	<600	<400	900
3.3	20	4.6	4.5	5.1	7.3	SB	8.1	6.9	5.0	6.6	52	11
52.7	50.5	64.6	52.6	54.3	49.5	SC	57.6	51.2	61.5	53.4	43.6	59.6
<35	<11	<25	<41	<13	<50	SE	<16	<31	<94	<42	<10	<25
11	9	16	14	15	12	TA	13	11	16	8	8	13
98	110	220	170	160	210	TH	240	130	240	150	93	160
43.4	44.4	93.2	50.6	64.1	62.8	U	77.6	50.6	172	69.6	40.5	68.2
24	24	61	16	24	50	W	59	30	84	84	97	22
170	120	250	170	140	170	ZN	220	170	370	200	90	<120
474	453	957	635	644	1200	LA	1220	675	1140	712	377	745
754	772	1550	894	1060	1940	CE	1900	1070	1660	1230	529	1160
45.2	52.9	106	60.5	71.9	123	SM	129	70.3	112	80.2	45.5	74.0
11.6	11.0	20.6	15.7	17.6	24.9	EU	25.9	12.7	24.4	18.3	10.3	17.3
62.3	65.8	147	71.0	96.0	113	YB	128	73.4	116	102	59.1	91.7
13.3	13.5	30.3	14.1	18.8	22.5	LU	25.3	14.1	24.9	20.6	11.8	19.2

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**	**	**	**	**	**	**	**	**	**	**	**	**	**	**	**	**	**
330-H**	331-H**	332-H**	333-H**	334-H**	335-H**	ELEMENT & UNITS	336-H**	338-H**	339-H**	340-H**	341-H**	342-H**	343-H**	344-H**	345-H**	346-H**	347-H**
<15	<5	<5	<20	<5	<5	AG PPM	<5	<9	<17	<17	<7	<26	<12	<12	<12	<12	<12
150	85	150	110	180	120	AS PPM	130	190	260	180	170	560	120	120	120	120	120
<29	<19	32	38	<46	6900	AU PPS	4100	14	<30	<31	27	310	1900	1900	1900	1900	1900
300	500	2300	1100	1400	1900	BA PPM	2100	500	1100	1100	200	1100	<8	<8	<8	<8	<8
<5	<3	<11	<7	<8	<12	CA %	<8	<1	<5	<6	<1	<8	<5	<5	<5	<5	<5
94	62	110	83	140	79	CD PPM	90	110	120	96	93	190	8300	8300	8300	8300	8300
3000	4700	6800	11000	8500	8300	CR PPM	8200	4600	6300	6600	2100	5000	1400	1400	1400	1400	1400
20.6	23.2	21.3	18.6	14.6	14.7	FE %	17.0	31.4	26.6	24.7	32.7	30.3	2200	2200	2200	2200	2200
1500	750	1200	1400	1600	2200	HF PPM	1900	390	770	1000	140	860	<5	<5	<5	<5	<5
<5	<5	37	<5	<5	<5	MO PPM	36	<5	<5	<5	<5	<5	0.12	0.12	0.12	0.12	0.12
0.12	0.12	0.11	0.23	0.12	<0.07	NA %	0.12	0.07	0.12	0.08	0.05	0.09	<600	<600	<600	<600	<600
<400	<300	<1000	<600	<700	<1000	NI PPM	<600	<200	<400	<500	<200	<300	7.4	7.4	7.4	7.4	7.4
9.4	4.4	8.3	6.1	15	7.3	SB PPM	5.5	7.4	16	7.9	5.7	21	68.7	68.7	68.7	68.7	68.7
15.5	49.2	54.3	59.3	68.4	68.7	SC PPM	63.6	40.5	48.8	50.6	41.4	40.2	<24	<24	<24	<24	<24
<12	<20	<56	<38	<23	<24	SE PPM	<12	<12	<14	<30	<5	<17	15	15	15	15	15
12	10	16	17	25	15	TA PPM	10	7	11	14	5	6	320	320	320	320	320
180	79	160	200	270	320	TH PPM	250	34	100	150	10	170	107	107	107	107	107
13.5	34.9	63.6	69.8	95.0	107	U PPM	87.3	16.1	37.9	54.3	6.7	49.8	52	52	52	52	52
51	38	56	48	100	52	W PPM	42	<6	17	19	7	75	<140	<140	<140	<140	<140
170	140	270	150	210	<140	ZN PPM	170	190	140	<80	140	190	1010	1010	1010	1010	1010
711	366	795	775	1010	1430	LA PPM	1150	126	473	773	41	589	1810	1810	1810	1810	1810
1200	611	1060	1380	1810	2370	CE PPM	1960	216	799	1160	80	959	133	133	133	133	133
2.1	40.4	76.9	94.6	133	171	SM PPM	129	15.8	54.4	79.2	6.2	68.1	29.2	29.2	29.2	29.2	29.2
16.8	9.7	18.7	16.2	29.2	32.9	EU PPM	25.6	3.6	15.2	16.5	1.7	12.5	103	103	103	103	103
111	57.8	83.6	103	136	165	YB PPM	141	31.9	63.0	82.9	19.8	73.2	11.1	11.1	11.1	11.1	11.1
21.9	11.1	17.5	20.2	27.7	33.5	LU PPM	28.4	6.07	11.8	15.5	3.64	14.5	11.1	11.1	11.1	11.1	11.1

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343-H**	344-H**	345-H**	347-H**	348-H**	351-H**	ELEMENT & UNITS	352-H**	353-H**	354-H**	355-H**	356-H**
<5	<5	<5	<14	<5	<5	AG PPM	<5	<5	<5	<5	<18
74	160	110	170	120	71	AS PPM	140	150	170	210	270
<16	150	<110	<27	1500	<53	AU PPR	290	<30	<40	<47	160
600	900	2100	800	3300	1100	BA PPM	2200	500	<800	<700	600
<2	<7	<22	<5	<16	<9	CA %	<12	<5	<7	<8	3
55	97	95	110	40	61	CO PPM	130	100	110	110	110
3300	4900	11000	7100	8000	10000	CR PPM	11000	5500	6200	6900	6700
25.6	18.1	8.66	25.1	9.74	19.8	FE %	29.6	23.0	23.6	23.8	22.2
100	1300	2500	1000	3300	1400	HF PPM	1900	950	700	1100	1200
<5	<5	<5	<5	90	3	MO PPM	61	<5	<5	<5	<5
0.06	0.12	0.16	0.06	0.13	0.07	NA %	0.12	0.08	0.10	0.13	0.17
<300	<500	<1600	<400	1700	1000	NI PPM	<1100	<500	<700	<200	<500
3.0	10	18	7.9	8.6	2.5	SB PPM	5.7	4.6	7.6	10	13
4.1	55.1	73.3	46.1	67.9	57.2	SC PPM	76.4	45.6	49.2	53.0	48.7
<14	<16	<170	<10	<87	<63	SE PPM	<77	<42	<49	<14	<12
5	18	37	11	15	13	TA PPM	28	9	10	16	9
15	220	400	130	570	140	TH PPM	210	83	77	180	170
9.0	72.8	130	49.1	157	71.1	U PPM	85.3	39.4	39.2	66.6	67.2
11	33	66	40	79	35	W PPM	42	30	<19	35	56
140	190	<250	170	270	<110	ZN PPM	380	100	170	170	130
57	966	1630	646	2450	607	LA PPM	946	305	377	1000	304
116	1580	2510	1050	4110	911	CE PPM	1500	571	598	1600	1400
6.7	113	176	68.0	288	59.0	SM PPM	97.3	36.4	36.9	110	93.9
1.8	24.4	38.3	15.6	67.7	10.0	EU PPM	21.6	10.2	8.4	22.7	16.4
71.4	112	190	75.8	300	89.8	YB PPM	124	99.5	55.3	98.6	95.3
4.30	21.8	39.8	14.5	56.2	19.0	LU PPM	25.9	11.5	11.7	18.4	17.9

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**	**	**	**	**	**	**	**	**	**	**	**	**	**	**	**	**
358-H**	359-H**	350-H**	361-H**	362-H**	363-H**	ELEMENT	364-H**	365-H**	366-H**	367-H**	368-H**	370-H**				
		UNITS :	UNITS :	UNITS :	UNITS :		UNITS :	UNITS :	UNITS :	UNITS :	UNITS :	UNITS :				
<5	<23	<5	<5	<20	<5	AG	<5	<25	<16	<5	<5	<5				
210	310	120	19	96	58	AS	40	130	280	98	100	110				
41	<39	<55	26	<26	<45	AU	2100	<46	44	<24	57	120				
1500	800	<2400	400	<1300	1100	BA	400	<900	1100	<1200	<500	1100				
<6	<7	<11	<2	<4	<8	CA	<3	<7	<4	<3	<3	<13				
110	140	88	48	85	65	CO	55	99	130	74	110	80				
6700	6100	9500	6700	5700	7300	CR	5000	8000	7500	3700	2400	6100				
25.4	30.3	19.0	26.3	23.1	21.2	FE	25.6	21.8	29.1	27.5	32.2	24.5				
1200	930	1500	760	770	1000	HF	600	1100	910	440	210	920				
<5	<5	<5	<5	<5	18	MO	23	55	<5	9	12	<5				
0.07	0.11	0.11	0.06	0.06	<0.05	NA	0.05	0.08	0.11	0.06	0.06	0.14				
<900	<600	<800	<300	<400	<700	NI	<400	<700	<500	<400	<500	1500				
9.2	11	7.4	1.7	4.1	4.1	SB	2.6	4.8	13	5.0	7.1	5.9				
52.5	53.0	60.6	47.4	42.5	48.8	SC	40.6	49.3	47.0	40.8	44.8	46.0				
<17	<39	<26	<8	<31	<48	SE	<29	<23	<26	<25	<24	<25				
8	11	25	11	11	12	TA	11	7	9	11	7	11				
170	140	260	67	66	110	TH	44	100	100	41	16	100				
63.6	54.1	87.0	32.7	28.2	43.3	U	22.8	45.3	43.8	19.8	7.8	4.2.9				
57	20	98	44	<13	<21	W	<11	32	40	<12	<15	<26				
130	160	<120	60	140	200	ZN	130	220	150	120	<60	210				
72c	672	1250	254	342	400	LA	215	439	428	199	74	803				
1310	1110	2130	444	534	846	CE	321	710	727	322	176	944				
89.6	76.6	143	30.4	31.1	48.3	SM	18.6	45.8	51.9	16.3	8.8	67.0				
20.3	12.7	27.5	4.8	13.3	13.3	EU	4.1	9.1	11.0	5.9	2.4	18.9				
93.2	79.3	122	50.4	46.6	66.4	YB	35.4	67.5	66.2	29.5	24.8	64.7				
13.4	14.1	23.2	10.0	0.85	13.3	LU	7.72	14.5	13.6	6.24	4.69	12.3				

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371-H**	372-H**	373-H**	374-H**	375-H**	376-H**	ELEMENT & UNITS	377-H**	378-H**	379-H**	380-H**	381-H**	392-H**
<5	<5	<5	<5	<5	<5	AG PPM	<5	<79	<5	<20	<5	<5
01	73	350	210	230	55	AS PPM	120	300	99	100	120	11
<5	<19	<74	<36	<79	<100	AU PPB	<120	663	73000	<39	<47	32
300	<1000	<1600	<600	1800	2600	BA PPM	2300	1700	2100	<600	2800	700
<17	<2	<19	<5	<14	<18	CA %	<24	<10	13	<7	<9	<5
93	69	170	130	120	48	CO PPM	120	150	50	66	99	36
0000	5000	4800	4300	7100	13000	CR PPM	9900	8000	11000	5400	8500	11000
3.77	26.2	25.8	25.4	20.8	13.0	FE %	10.0	23.6	6.33	22.0	13.7	13.3
6700	450	1000	500	1200	2000	HF PPM	2300	1100	2600	1000	2000	1000
<5	<5	31	16	<5	<5	MO PPM	<5	73	55	<5	<5	<5
0.12	0.06	0.15	0.08	<0.10	0.14	NA %	<0.17	0.13	0.08	0.10	0.09	0.19
1200	<300	<1500	<600	1800	<1400	NI PPM	<1700	<200	<800	<600	<700	<500
5.3	2.9	15	7.3	9.9	8.0	SB PPM	14	17	17	6.3	8.3	3.5
28.4	44.0	48.2	48.0	52.9	68.9	SC PPM	73.3	55.9	77.1	52.6	64.9	57.2
<140	<21	<93	<33	<80	<130	SE PPM	<120	<47	<29	<39	<47	<14
21	10	<11	11	13	<11	TA PPM	28	17	27	10	21	14
640	37	150	51	150	310	TH PPM	450	200	390	140	320	200
224	16.0	69.2	20.9	73.2	168	U PPM	195	65.5	144	60.8	114	73.2
440	<10	<42	<17	<37	<43	W PPM	110	33	100	27	67	64
<190	110	<220	200	260	<220	ZN PPM	<270	260	290	<90	170	130
7470	161	707	285	822	1120	LA PPM	1960	1020	1420	732	1350	736
4050	279	1190	427	1140	1800	CE PPM	3420	1740	2510	1130	2220	1280
290	16.6	79.7	28.9	79.8	127	SM PPM	265	114	192	74.6	159	94.1
59.9	4.6	15.6	11.1	26.1	27.1	EU PPM	50.1	18.6	36.0	16.0	29.0	15.4
367	29.1	77.8	39.2	85.9	193	YB PPM	197	105	206	80.1	167	115
71.6	6.32	17.4	8.05	18.5	41.7	LU PPM	39.2	19.5	42.6	16.3	34.7	24.7



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393-H**	394-H**	395-H**	386-H**	387-H**	388-H**	ELEMENT & UNITS	389-H**	390-H**	391-H**	392-H**	393-H**	
<5	<21	<5	<5	<5	<6	AG	<7	<5	<5	<5	<8	<5
67	69	110	69	72	62	AS	150	65	70	70	83	47
<15	<5	<28	<43	3000	<11	AU	<11	<19	940	19	<14	<20
420	3400	1000	<2100	2400	400	BA	200	<300	600	<700	<200	<300
<1	<1	<5	<7	12	3	CA	<1	<2	<4	1	<1	<2
51	69	96	62	64	62	CO	85	66	74	56	56	49
2400	35000	9700	6300	7200	3400	CR	1900	4900	5900	2300	1900	4000
25.6	3.93	21.3	16.1	13.9	22.0	FE	31.6	24.9	21.0	25.6	25.1	75.7
190	2900	1100	1200	1500	320	HF	150	410	770	100	140	430
7	50	<5	<5	<5	<5	MO	<5	12	<5	<5	7	13
0.06	<0.05	0.08	0.13	0.12	0.12	NA	<0.05	0.07	0.12	<0.05	0.05	0.05
<300	600	<400	<600	<600	300	NI	<200	400	600	<300	<200	<200
2.6	6.7	10	4.1	3.8	2.4	SB	4.9	2.9	2.7	2.4	2.8	2.0
42.8	94.7	53.9	52.6	57.6	44.2	SC	40.0	41.7	44.3	44.3	43.3	38.8
<12	<5	<13	<45	<75	<5	SE	<5	<9	<13	<12	<12	<28
7	27	13	13	14	7	TA	7	8	9	5	5	9
14	380	150	150	190	26	TH	13	32	31	13	10	31
9.1	153	58.8	68.2	75.3	13.3	U	4.6	15.6	34.9	7.8	5.1	16.1
21	63	5.0	<22	<24	47	W	8	<12	<17	<10	<9	<12
150	230	520	200	130	110	ZN	140	340	140	170	130	<210
58	1180	613	725	828	109	LA	39	144	322	54	45	152
29	1900	1020	1040	1270	197	CE	79	253	578	99	95	260
6.5	124	71.2	69.5	85.7	15.3	SM	6.0	14.9	31.4	6.3	6.1	13.0
1.3	26.7	14.3	13.8	17.2	3.7	FU	1.6	5.0	8.0	1.5	1.4	3.3
22.0	109	83.6	81.8	100	27.4	YB	21.8	23.2	45.2	23.3	20.2	27.5
4.65	41.6	16.6	17.8	21.6	5.10	LU	3.79	5.71	9.74	4.78	3.99	5.51

NUCLEAR ACTIVATION SERVICES LIMITED

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ELEMENT UNITS	395-H**	396-H**	397-H**	398-H**
AG	PPM	<5	<15	<5
AS	PPM	81	150	17
AU	PPM	<27	96	<31
BA	PPM	400	500	700
CA	%	<3	<3	<4
CO	PPM	100	70	43
CR	PPM	4200	3800	2600
FE	%	26.9	24.6	22.2
HF	PPM	450	520	780
MO	PPM	<8	13	<5
NA	%	0.08	<0.05	0.14
NI	PPM	<400	<400	<500
SB	PPM	6.2	2.6	5.4
SC	PPM	39.0	38.5	41.0
SE	PPM	<12	<27	<14
TA	PPM	7	9	12
TH	PPM	37	49	67
U	PPM	22.0	26.8	31.0
W	PPM	<15	20	28
ZN	PPM	170	80	100
LA	PPM	173	232	291
CE	PPM	296	329	347
SM	PPM	16.9	19.9	27.5
EU	PPM	5.8	4.6	5.9
YB	PPM	31.4	33.2	45.2
LU	PPM	6.02	7.30	9.68

EXPLANATION OF CODES

VARIABLE DETECTION LIMITS DUE TO SAMPLE COMPOSITION

NUCLEAR ACTIVATION SERVICES LIMITED

1250 MAIN STREET WEST, HAMILTON, ONTARIO, L8S 4K1

PHONE (416) 522-5666 TELEX 06-986947

CERTIFICATE OF ANALYSIS

TO: MC GILL UNIVERSITY  
 ATTN: M.A. BERNIER  
 DEPT. OF GEOLOGICAL SCIENCES  
 3450 UNIVERSITY STREET  
 MONTREAL, QUEBEC

CUSTOMER NO. 133/01/02

DATE SUBMITTED  
 26-MAR-87

REPORT: 7945

FILE NUMBER: 9638

140 UNPREPARED SAMPLES

PO# C-50063-IR

WERE ANALYZED AS FOLLOWS:

ELEMENTS	DETECTION LIMIT	UNITS	METHOD	FLEMENTS	DETECTION LIMIT	UNITS	METHOD
AG	5.0000	PPM	INAA	SC	0.1000	PPM	INAA
AS	2.0000	PPM	INAA	SE	5.0000	PPM	INAA
AU	5.0000	PPB	INAA	TA	1.0000	PPM	INAA
BA	100.0000	PPM	INAA	TH	0.5000	PPM	INAA
CA	1.0000	%	INAA	U	0.5000	PPM	INAA
CO	5.0000	PPM	INAA	W	4.0000	PPM	INAA
CR	10.0000	PPM	INAA	ZN	50.0000	PPM	INAA
FE	0.0200	%	INAA	LA	1.0000	PPM	INAA
HF	1.0000	PPM	INAA	CE	3.0000	PPM	INAA
MO	5.0000	PPM	INAA	SM	0.1000	PPM	INAA
NA	0.0500	%	INAA	EU	0.2000	PPM	INAA
NI	200.0000	PPM	INAA	YB	0.2000	PPM	INAA
SD	0.2000	PPM	INAA	LU	0.0500	PPM	INAA

COMMENTS:

NOTE: DETECTION LIMIT RAISED DUE TO SMALL SAMPLE SIZE & HIGH ACTIVITY.

DATE 16-APR-87

NUCLEAR ACTIVATION SERVICES LIMITED  
 CERTIFIED BY *[Signature]*

\*\*\* UNLESS INSTRUCTED OTHERWISE WE WILL DISCARD ALL SAMPLES \*\*\*  
 IRRADIATED SAMPLES AFTER 30 DAYS. ANY OTHER MATERIAL AFTER 120 DAYS.

NUCLEAR ACTIVATION SERVICES LIMITED

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NUCLEAR ACTIVATION SERVICES LIMITED

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0001-H**		0002-H**		0003-H**		0008-H**		0057-H**		0283-H**		0306-H**		0307-H**		0309-H**		0310-H**		85-5**		85-6*	
***	***	***	***	***	***	***	***	***	***	***	***	***	***	***	***	***	***	***	***	***	***	***	***
S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S		S A M P L E N U M B E R S	
ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS	ELFMENT	% UNITS
<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5
49	29	28	16	28	18	240	27	240	27	27	27	42	39	39	39	37	74	74	110	110	110	110	110
200	<9	<70	<21	150	<9	150	<9	150	<9	<9	<9	<19	<22	<22	<15	<15	420	420	420	420	420	420	420
600	400	<500	<500	<4400	300	<4400	300	<4400	300	300	300	700	900	1000	<300	<300	<500	<500	500	500	500	500	500
<3	2	<3	<2	<37	<1	<37	<1	<37	<1	<1	<1	4	<3	<3	2	<3	<3	<3	<4	<4	<4	<4	<4
65	41	52	45	210	27	210	27	210	27	27	27	42	39	39	37	74	74	74	110	110	110	110	110
9500	1500	15000	6900	6000	2000	6000	2000	6000	2000	2000	2000	4000	6200	10000	700	8900	8900	8900	7900	7900	7900	7900	7900
21.4	32.4	21.5	20.4	20.8	36.8	20.8	36.8	20.8	36.8	36.8	36.8	18.4	16.8	25.5	18.9	23.8	23.8	23.8	25.2	25.2	25.2	25.2	25.2
720	200	740	590	3100	240	3100	240	3100	240	240	240	440	620	470	140	620	620	620	600	600	600	600	600
21	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	20	20	20	20	20
0.09	0.13	<0.05	0.17	<0.22	0.10	<0.22	0.10	<0.22	0.10	0.10	0.10	0.06	<0.05	0.08	0.08	0.07	0.07	0.07	0.07	0.07	0.07	0.07	0.07
400	<200	<400	<400	<3100	<200	<3100	<200	<3100	<200	<200	<200	<400	<400	<300	<300	<400	<400	<400	<500	<500	<500	<500	<500
5.7	1.1	2.1	0.8	56	1.3	56	1.3	56	1.3	1.3	1.3	<0.6	0.9	6.4	6.4	8.0	8.0	8.0	3.8	3.8	3.8	3.8	3.8
46.4	44.7	45.8	52.5	76.2	29.0	76.2	29.0	76.2	29.0	29.0	29.0	59.4	60.7	35.3	54.9	49.1	49.1	49.1	48.8	48.8	48.8	48.8	48.8
<16	<5	<16	<18	<77	<6	<77	<6	<77	<6	<6	<6	<15	<18	<12	<10	<17	<17	<17	<36	<36	<36	<36	<36
13	8	15	23	88	7	88	7	88	7	7	7	18	17	11	19	13	13	13	19	19	19	19	19
54	16	65	59	360	19	360	19	360	19	19	19	44	65	58	13	49	49	49	65	65	65	65	65
32.7	9.1	35.1	29.6	270	9.5	270	9.5	270	9.5	9.5	9.5	18.6	36.2	28.7	5.3	31.4	31.4	31.4	30.1	30.1	30.1	30.1	30.1
19	<7	18	24	<80	<7	<80	<7	<80	<7	<7	<7	<11	14	<12	<9	<16	<16	<16	<20	<20	<20	<20	<20
250	110	220	120	1200	230	1200	230	1200	230	230	230	190	110	310	140	270	270	270	<110	<110	<110	<110	<110
189	75	255	194	1740	69	1740	69	1740	69	69	69	157	241	205	55	194	194	194	236	236	236	236	236
344	164	474	367	2020	151	2020	151	2020	151	151	151	514	778	356	147	367	367	367	452	452	452	452	452
25.0	14.3	31.1	20.5	163	13.3	163	13.3	163	13.3	13.3	13.3	18.6	26.2	23.5	9.0	23.1	23.1	23.1	31.0	31.0	31.0	31.0	31.0
3.4	2.5	5.9	2.7	23.1	2.7	23.1	2.7	23.1	2.7	2.7	2.7	4.1	6.5	4.5	1.4	3.4	3.4	3.4	5.1	5.1	5.1	5.1	5.1
39.8	19.9	40.5	33.3	223	16.1	223	16.1	223	16.1	16.1	16.1	33.4	40.8	28.7	20.0	35.9	35.9	35.9	33.7	33.7	33.7	33.7	33.7
7.79	3.36	8.32	6.96	46.5	2.97	46.5	2.97	46.5	2.97	2.97	2.97	6.48	7.96	6.03	3.62	6.95	6.95	6.95	6.89	6.89	6.89	6.89	6.89

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85-10\*\* 85-12\*\* 85-31\*\* 85-45\*\* 85-50\*\* 85-53\*\*  
 86-3-11\*\* -H 86-5-7\*\* -H 86-6-10\*\* -H 86-5-9\*\* -H 86-5-7\*\* -H 86-6-10\*\*

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85-56\*\* 85-58\*\* 85-59\*\* 85-65\*\* 85-66\*\*  
 -H 86-3-17\*\* -H 86-6-10\*\* -H 86-2-28\*\* -H 86-5-7\*\* -H 66-5-4\*\* -H 86-5-9

ELEMENT & UNITS	85-10**	85-12**	85-31**	85-45**	85-50**	85-53**	85-56**	85-58**	85-59**	85-65**	85-66**
AG PPM	<5	<18	<5	<17	<5	<5	<5	<5	<5	<5	<15
AS PPM	130	140	110	300	220	31	38	47	22	120	91
SA PPM	<41	<34	34	<71	<140	13	<28	<22	<5	<50	35
BA PPM	<900	900	600	2600	<3500	300	1500	1300	700	<3100	900
CA %	<8	<7	<3	<16	<27	<1	<3	<4	<5	<8	<6
CO PPM	110	120	110	330	250	43	50	65	48	190	110
CR PPM	4400	6300	14000	9100	9600	1500	5800	7300	7800	12000	14000
FE %	23.7	24.4	27.1	23.7	23.5	31.5	21.8	22.5	23.9	23.3	24.2
HF PPM	830	750	1000	1500	1400	180	490	700	700	650	860
MO PPM	<5	<5	<5	<5	<93	<5	<5	<5	<5	18	<5
NA %	0.07	<0.05	0.14	<0.09	<0.17	0.13	0.16	0.08	0.08	0.05	0.07
NI PPM	<700	<600	<400	1500	<2600	<200	<500	<400	<500	<1000	<500
SB PPM	14	8.6	6.7	24	25	1.2	2.5	3.2	1.7	15	6.5
SC PPM	58.2	55.5	54.4	53.6	49.7	45.4	54.6	50.8	50.6	52.8	55.8
SE PPM	<32	<27	<15	<68	<140	<6	17	<19	20	<41	31
TA PPM	20	18	12	8	86	9	10	12	12	43	11
TH PPM	120	130	110	210	170	19	48	72	67	97	86
U PPM	58.9	51.8	49.2	115	62.9	8.1	21.0	40.6	37.1	45.2	43.5
W PPM	<20	<17	66	<33	<64	<8	<19	<15	25	34	<15
ZN PPM	260	180	190	<350	1100	140	260	140	310	390	<23
LA PPM	555	524	371	1080	791	78	200	277	364	594	489
CE PPM	691	663	737	1320	952	270	361	529	591	1000	555
SM PPM	63.9	60.2	52.1	120	74.5	14.5	28.4	36.2	38.8	58.0	46.4
EU PPM	11.3	13.8	5.7	23.5	19.0	2.5	5.5	7.0	8.7	13.4	9.5
YB PPM	61.5	56.1	61.1	137	89.0	19.4	36.2	45.3	47.0	53.9	53.7
LU PPM	12.5	11.2	11.2	24.9	16.2	3.23	7.08	9.15	8.65	9.62	10.0

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ELEMENT & UNITS	85-70**	85-74**	85-84**	85-86**	85-106**	85-107**
AG	<5	<14	<5	<5	<5	<5
AS	30	52	86	85	43	43
AU	<25	<21	<23	<18	<43	<43
BA	700	<600	500	500	<1000	<1000
CA	<4	<3	2	2	<6	<6
CO	62	59	75	81	78	78
CR	3800	15000	2300	3000	6300	6300
FE	28.1	18.1	27.5	24.3	29.6	29.6
HF	630	730	340	290	480	480
MO	<5	<5	<5	<5	<24	<24
NA	0.05	<0.05	0.05	0.05	0.09	0.09
NI	<500	<400	800	<400	<900	<900
SB	2.0	3.1	23	1.6	3.5	3.5
SC	55.7	52.4	32.9	46.3	57.6	57.6
SE	<22	<20	<19	<13	<35	<35
TA	16	25	12	19	35	35
TH	52	66	22	30	49	49
U	33.8	42.0	11.9	12.9	13.7	13.7
W	<14	<12	<11	<9	<25	<25
ZN	130	170	370	180	350	350
LA	222	254	113	108	166	166
CE	306	461	178	202	241	241
SM	25.3	27.0	14.3	14.0	19.1	19.1
EU	7.0	6.4	3.6	4.1	7.5	7.5
YB	43.7	39.9	28.1	20.4	28.5	28.5
LJ	8.36	8.16	5.51	4.52	5.83	5.83

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ELEMENT & UNITS	85-108**	85-119**	85-122A**	85-131**	85-138**
AG	<5	<5	<5	<5	<16
AS	22	52	15	84	190
AU	<69	41	<25	32	<33
BA	<1600	1500	800	<200	1900
CA	<11	<6	<3	3	<4
CO	76	78	44	96	150
CR	6500	37000	5200	4600	6400
FE	29.9	12.9	26.2	32.2	30.4
HF	770	1400	450	380	410
MO	22	<5	7	<5	7
NA	0.10	0.07	0.09	0.11	0.11
NI	<1400	<600	<500	<200	<700
SB	7.1	2.7	2.0	4.1	5.8
SC	64.8	58.0	49.3	47.2	50.6
SE	<57	<31	30	<7	<24
TA	18	24	17	9	5
TH	54	170	40	27	38
U	24.2	68.3	17.2	13.7	12.6
W	<37	<24	26	73	17
ZN	350	330	350	280	180
LA	235	554	152	95	139
CE	480	847	334	397	272
SM	27.2	63.9	19.6	16.8	18.0
EU	7.7	11.0	2.4	2.0	4.0
YB	41.2	82.2	27.3	26.7	27.6
LJ	8.84	15.8	5.86	4.94	5.78

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 86-5-27\*\* -H 86-6-10\*\* -H 86-3-11\*\* -H 86-6-10\*\* -H 86-6-10\*\* -H 86-3-11\*\*

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85-194\*\* 85-199\*\* 85-200\*\* 85-207\*\* 85-222  
 86-5-7\*\* -H 86-4-24\*\* -H 86-6-10\*\* -H 86-3-17\*\* -H 86-5-27

ELEMENT & UNITS	85-144**	85-157**	95-159**	85-167**	85-174**	85-180**	85-194**	85-199**	85-200**	85-207**	85-222
AG PPM	<5	<7	<5	<5	<5	<14	<5	<5	<5	<5	<5
AS PPM	48	160	70	49	200	200	34	51	11	18	28
AU PPB	<17	81	<27	22	<24	<24	48	<23	<11	<27	85
BA PPM	500	<200	<600	300	<1400	<4	700	61000	300	1100	<1300
CA %	4	<3	3	<2	<4	<4	<3	<2	<1	<4	<6
CD PPM	63	39	46	72	160	160	45	61	24	61	110
CR PPM	3800	11000	10000	9700	5800	12000	13000	4200	3500	20000	15000
FE %	25.1	26.3	21.7	36.2	30.1	26.8	20.5	19.7	6.88	19.3	43.8
HF PPM	470	530	540	460	600	800	760	280	250	1000	1300
MO PPM	5	<5	<5	<5	<5	<5	<5	11	<5	<5	<14
NA %	0.11	0.13	0.05	0.07	0.08	0.08	<0.05	0.08	<0.05	<0.05	0.13
NI PPM	<400	<400	<500	400	<300	<500	<500	<500	<200	<400	<1100
SB PPM	2.3	7.0	6.5	7.3	9.4	9.4	7.1	2.2	1.0	3.1	3.3
SC PPM	45.2	36.8	41.8	26.3	47.5	47.5	47.8	49.3	24.9	51.0	145
SE PPM	<13	<15	<22	<9	<13	<13	<22	<18	<9	<19	<43
TA PPM	11	10	22	8	7	7	17	10	21	17	16
TH PPM	34	66	52	51	55	55	80	24	21	85	130
U PPM	19.0	35.3	37.2	25.4	27.3	27.3	51.1	13.7	13.3	50.7	77.0
W PPM	<14	21	18	74	<13	<13	21	13	<6	40	<32
ZN PPM	290	370	<170	140	370	370	170	140	<50	230	500
LA PPM	150	204	210	181	242	242	289	126	88	360	476
CE PPM	267	397	425	361	324	324	518	335	247	578	1520
SM PPM	19.7	25.9	19.9	24.8	28.4	28.4	32.3	15.6	9.2	43.5	56.4
EU PPM	3.0	5.8	5.7	3.9	5.8	5.8	7.4	4.2	1.9	8.3	11.3
YB PPM	28.4	31.2	29.9	27.5	35.0	35.0	41.8	21.6	15.9	57.4	90.2
LU PPM	5.56	6.09	5.64	5.07	7.01	7.01	8.39	4.06	3.12	11.8	17.5





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86-5-27\*\* -H 86-5-1\*\* -H 86-2-28\*\* -H 86-5-1\*\* -H 86-3-11\*\* -H 86-3-24\*\* -H

85-293\*\* 85-298\*\* 85-302\*\* 85-303\*\* 85-304\*\*  
86-4-24\*\* -H 86-5-1\*\* -H 86-3-11\*\* -H 86-3-24\*\* -H 86-3-24\*\* -H

86-5-27\*\* 86-5-27\*\* 86-5-27\*\* 86-5-27\*\* 86-5-27\*\* 86-5-27\*\*  
86-5-27\*\* 86-5-27\*\* 86-5-27\*\* 86-5-27\*\* 86-5-27\*\* 86-5-27\*\*

<5 36 <17 <13 <10 <5 <5  
240 260 290 150 47  
<160 <59 84 <23 <41 <5  
<800 <1400 <600 9 <500 <4  
<41 <11 9 <3 <4 <6  
840 230 340 160 97 66  
11000 10000 3800 1400 1200 19000  
156 260 427 345 473 171  
2500 1400 330 130 94 1500  
<5 <5 <5 <5 <5 <5  
<0.28 0.11 0.08 <0.05 <0.05 <0.05  
<2600 1200 <600 <500 <400 <700  
99 23 14 55 34 45  
71.2 50.5 28.4 23.7 11.8 55.5  
210 <86 <21 <14 <39 <39  
<1 17 5 6 <1 44  
460 170 52 28 21 160  
169 97.7 25.4 11.2 10.7 99.1  
<76 65 21 <11 12 <18  
<660 <260 270 220 <100 <200  
4030 810 342 251 314 663  
4540 884 521 350 558 1160  
351 78.1 35.6 23.9 26.2 62.3  
90.9 14.6 6.8 7.1 3.2 12.9  
212 94.3 22.9 10.6 9.5 90.9  
37.1 18.6 4.61 2.11 1.61 18.3

AG PPM <5 <5 <5 <5 <5 <5  
AS PPM <14 <14 170 170 170 <36  
AU PPB <52 <52 <270 <270 <270 1100  
BA PPM 1800 <3100 <3100 <3100 <3100 <52  
CA % <9 <9 <6 <6 <6 <10  
CO PPM 56 92 170 140 290 <36  
CR PPM 31000 11000 9900 25000 2700 86  
FE % 12.8 25.9 21.3 23.2 42.9 21.3  
HF PPM 2500 1100 1600 2500 170 540  
MO PPM <5 <5 <5 <5 <5 <5  
NA % 0.10 <0.05 0.12 <0.15 <0.15 0.06  
NI PPM <700 <670 <900 <1700 <1100 <400  
SB PPM 3.7 8.3 12 31 37 4.6  
SC PPM 70.3 60.3 66.7 26.1 54.0  
SE PPM <41 <29 <44 <100 <31 <17  
TA PPM 16 27 10 38 <1 22  
TH PPM 290 120 200 240 44 39  
U PPM 138 52.0 104 217 17.5 27.3  
W PPM <32 <16 39 <56 <26 <11  
ZN PPM 220 <140 <220 1400 1700 170  
LA PPM 1090 438 1020 885 451 161  
CE PPM 1710 593 1180 1100 505 303  
SM PPM 118 50.2 104 97.0 26.0 20.4  
EU PPM 16.8 7.9 12.8 12.8 4.4 3.7  
YB PPM 151 64.2 106 147 17.0 34.1  
LU PPM 30.5 12.6 20.5 29.7 2.65 7.22

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30\*\* 69\*\* 93\*\* 116\*\* 137\*\* 139\*\*  
 86-11-19\*\*H 86-10-30\*\*H 86-11-19\*\*H 86-11-19\*\*H 86-10-30\*\*H

<5	<5	<5	<13	<11
170	16	15	140	130
<42	<20	<47	4900	<19
<1000	<400	800	4200	<500
<7	<2	<5	<2	<2
130	39	55	110	100
13000	1300	14000	3900	8600
33.2	25.6	17.9	24.1	22.0
690	230	910	410	550
21	18	<5	<38	16
0.08	0.14	0.12	0.07	0.08
<90	<40	<50	<40	<40
9.6	1.1	0.8	3.5	9.0
52.6	54.2	58.7	48.2	49.2
<38	<12	<25	<17	<16
13	12	22	9	13
71	22	99	33	42
48.9	9.8	54.2	17.8	25.4
<25	<18	<15	<10	11
440	210	180	230	150
233	100	398	103	173
569	178	471	232	661
31.3	16.7	43.5	16.3	20.9
5.2	2.5	6.5	2.5	5.6
40.3	25.4	53.7	25.9	29.8
7.59	4.43	11.5	5.18	5.79

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AG	PPM	<5	<5	<12	<5	<14	<19
AS	PPM	51	38	85	67	150	200
AU	PPB	<19	<28	23	<24	31	<34
BA	PPM	500	<700	800	<1400	800	1300
CA	%	<3	<3	<3	<3	<3	<4
CO	PPM	37	47	67	66	120	170
CR	PPM	13000	12000	17000	14000	13000	18000
FE	%	29.3	18.9	20.6	21.3	24.5	33.1
HF	PPM	630	730	730	790	650	890
MO	PPM	8	<5	<5	<5	<5	<5
NA	%	0.07	0.07	0.05	0.06	0.05	0.07
NI	PPM	<400	<500	<400	<400	<500	<700
SB	PPM	6.3	4.6	7.5	6.1	7.6	10
SC	PPM	35.7	43.1	47.3	42.2	47.8	65.3
SE	PPM	<17	<25	<20	<21	<21	<28
TA	PPM	12	34	15	20	17	19
TH	PPM	53	72	80	73	51	65
U	PPM	36.4	43.9	47.4	45.2	33.8	47.3
W	PPM	<14	17	18	23	<13	<17
ZN	PPM	220	220	190	180	250	340
LA	PPM	220	252	300	287	216	297
CE	PPM	425	411	912	562	429	616
SM	PPM	27.1	24.9	32.6	31.6	25.5	34.7
EU	PPM	4.0	5.3	7.9	6.1	5.7	10.2
YB	PPM	30.9	36.0	39.8	39.8	34.3	48.2
LU	PPM	6.22	7.91	8.18	8.72	6.87	9.66

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175\*\* 184\*\* 185\*\* 192\*\* 209\*\*  
 86-11-19\*\*H 86-10-30\*\*H 86-11-14\*\*H 86-11-19\*\*H 86-11-19\*\*H

<5	<5	<5	<5	<11
100	160	74	38	97
21	40	<72	<19	<21
<900	900	<1700	400	19000
<2	<5	<14	<3	<3
97	120	110	27	87
3700	12000	29000	10000	13000
14.9	28.7	27.0	29.2	16.3
240	700	1600	520	740
<5	11	37	13	<5
0.08	0.14	<0.09	0.07	0.09
500	<600	<200	<400	500
5.0	6.5	10	5.8	6.6
27.7	41.5	63.4	34.9	41.4
<13	<20	<61	<13	20
12	9	33	12	9
26	70	150	51	71
19.5	31.4	65.2	32.0	43.5
<8	<23	<37	<14	<10
180	400	530	230	260
89	280	495	183	269
172	491	1240	369	503
10.9	38.1	59.8	25.6	30.5
2.7	8.0	10.1	4.7	5.7
15.7	39.9	68.9	27.9	40.6
2.96	8.18	14.0	5.94	8.71

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211\*\* 214\*\* 215\*\* 231\*\* 251\*\*  
 86-11-14\*\*H 86-11-12\*\*H 86-11-19\*\*H 86-3-17\*\*H 86-11-10\*\*H

ELEMENT	211**	214**	215**	231**	251**
% UNITS	86-11-14**H	86-11-12**H	86-11-19**H	86-3-17**H	86-11-10**H
AG	PPM	<5	<5	<5	<5
AS	PPM	55	10	45	12
AU	PPB	<28	<19	30	<20
BA	PPM	<600	<500	<400	400
CA	%	<1	3	<3	<2600
CD	PPM	88	49	42	74
CR	PPM	6100	5900	7400	15000
FE	%	30.0	15.0	25.3	21.9
HF	PPM	450	500	620	1200
MO	PPM	6	<5	<5	<5
NA	%	0.15	0.10	0.09	0.19
NI	PPM	<500	<400	<300	<500
SB	PPM	2.4	1.4	<0.7	1.8
SC	PPM	43.5	65.5	59.3	34.5
SE	PPM	17	<17	<15	<15
TA	PPM	11	21	24	12
TH	PPM	47	48	30	26
U	PPM	26.0	27.5	21.0	14.3
W	PPM	<20	<11	15	31
ZN	PPM	340	280	120	150
LA	PPM	171	190	123	87
CE	PPM	319	347	432	158
SM	PPM	22.9	25.0	26.9	10.6
EU	PPM	4.6	4.9	9.4	2.1
YB	PPM	28.1	37.4	38.9	16.9
LU	PPM	5.45	7.18	6.79	3.54

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257\*\* 278\*\* 96-134\*\* 4\*\* 16\*\* 35\*\* 42\*\* 52\*\* 78\*\* 89\*\* 102\*\* 109\*\*  
 86-11-11\*\* -H 86-5-1\*\* -H 86-2-28\*\* -H 87-2-25\*\* -H 87-2-13\*\* -H 87-2-25\*\* -H 87-2-23\*\* -H 87-2-19\*\* -H 87-2-23\*\* -H 87-2-23\*\* -H 87-2-19\*\*

Element	Units	Sample 1	Sample 2	Sample 3	Sample 4	Sample 5	Sample 6	Sample 7	Sample 8	Sample 9	Sample 10	Sample 11	Sample 12
AG	PPM	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5
AS	PPM	26	52	240	50	140	75	310	610	400	550	780	0.12
AU	PPB	<20	<77	110	23000	110	240	15000	<52	6100	10	160	<0.08
BA	PPM	<500	<1600	<700	700	<700	<600	<1200	1200	600	<800	1700	<25
CA	%	<3	<12	<6	<3	<6	<4	<4	<12	<2	<1	<1	<3
CO	PPM	47	310	56	17	320	27	590	1700	38	35	430	170
CR	PPM	24.6	42.2	27.6	26.5	25.5	24.7	20.8	17000	4000	18.6	2.53	33.8
FE	%	630	410	280	470	320	590	770	21.1	19.5	150	27	300
HF	PPM	<5	<41	23	17	27	5	<5	1200	460	7	<5	11
MD	PPM	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5	<5
NA	%	0.12	0.16	0.12	0.05	0.20	0.10	0.23	0.07	0.06	0.06	<0.05	0.08
NI	PPM	<400	<1500	<470	<400	<800	<400	<900	<500	<300	<300	<200	<600
SB	PPM	1.5	15	3.0	2.1	6.4	3.0	15	3.4	0.9	<0.4	0.7	7.2
SC	PPM	48.0	23.5	46.8	47.1	38.5	44.5	49.8	58.2	53.7	52.5	2.9	40.6
SE	PPM	<14	<51	<18	<17	<26	<19	<38	<24	<13	<9	<5	<21
TA	PPM	12	19	9	13	15	15	2	16	19	13	<1	6
TH	PPM	70	47	24	47	46	73	120	130	40	17	3.2	25
U	PPM	35.7	33.0	12.8	21.0	17.1	33.6	57.7	61.1	24.8	7.4	1.7	13.5
W	PPM	<15	<39	<15	24	<22	<19	<28	<22	<10	<8	4	17
ZN	PPM	230	<390	320	180	690	280	620	250	140	140	80	360
LA	PPM	242	400	130	169	177	259	910	477	153	58	13	91
CE	PPM	493	505	233	294	402	490	854	838	288	170	26	170
SM	PPM	29.1	38.8	18.5	19.4	30.0	33.9	70.9	56.8	16.9	8.9	2.0	12.5
FU	PPM	3.6	7.6	2.5	1.7	4.1	4.6	17.3	9.7	4.2	1.9	0.4	2.8
YB	PPM	41.0	26.0	23.9	30.8	23.9	34.9	57.1	71.6	31.7	19.0	1.9	19.3
LU	PPM	7.60	5.86	4.48	5.84	4.93	7.63	11.3	14.0	5.98	3.50	0.39	3.53

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110**	111**	112**	113**	128**	130**	ELEMENT	132**	133**	136H**	140H**	147H**	148H
87-2-13**	87-2-25**	87-2-19**	87-2-19**	87-2-19**	87-2-19**	C UNITS	I-H	87-2-13**	87-2-19**	87-2-19**	87-2-13**	87-2-13
<5	<5	<5	<5	<5	<11	AG	<5	<15	<5	<5	<34	<5
17	15	22	11	32	170	AS	33	210	89	31	330	40
<28	32	<42	<18	80	80	AU	<71	130	34	<34	<67	<12
<1600	<500	<900	200	600	5200	BA	1700	600	27000	900	<3500	20000
<4	<4	<7	<2	<2	<2	CA	<10	<3	<3	<5	<8	<1
60	53	68	14	50	180	CO	78	260	90	53	170	19
10000	16000	14000	3900	3500	7800	CR	6100	1400	2900	5400	2800	1700
20.2	16.4	19.2	5.40	16.7	25.6	FE	25.1	39.2	27.1	26.9	36.1	10.2
1000	860	890	170	370	510	HF	670	120	310	900	170	100
<5	<5	<12	<5	<5	<5	MO	18	25	49	18	<38	<5
0.11	0.18	0.19	<0.05	0.09	0.06	NA	0.12	0.09	0.08	0.15	0.12	<0.05
<500	<400	<700	<200	<400	<400	NI	1700	<600	<500	<600	<1400	<200
2.2	2.0	2.8	1.8	17	9.2	SB	7.2	29	6.3	2.5	13	1.1
22.8	59.2	74.1	13.4	52.5	39.4	SC	64.2	39.6	48.2	55.3	29.0	16.5
<25	25	<33	<7	<14	<17	SE	<60	73	<15	<23	<44	<9
28	17	18	4	17	20	TA	53	8	6	14	22	3
100	110	97	18	37	46	TH	62	12	22	78	29	8.4
54.6	45.7	53.2	7.7	20.5	27.4	U	34.7	7.4	10.1	35.5	<9.7	4.3
25	<21	<24	<8	<10	<10	W	<41	<16	<23	<27	<36	11
<130	190	220	60	<50	190	ZN	<310	300	240	<50	790	<50
383	386	354	67	133	177	LA	223	98	100	287	119	38
665	650	759	125	269	329	CE	423	101	221	527	122	71
41.4	45.3	41.9	8.8	16.8	22.4	SM	30.1	22.6	16.5	39.3	15.1	6.0
7.8	8.0	8.2	1.5	3.4	4.9	EU	9.0	5.7	2.9	6.3	7.7	0.8
59.5	54.9	54.8	11.0	25.6	27.8	Y8	44.0	16.1	26.4	46.4	14.3	8.5
12.2	10.4	10.9	2.41	5.22	5.56	LU	8.54	3.06	4.87	9.29	2.83	1.49

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143H\*\* 163H\*\* 164H\*\* 165H\*\* 169H\*\* 170H\*\*  
 87-2-23\*\* 87-2-19\*\* 87-2-19\*\* 87-2-19\*\* 87-2-19\*\* 87-2-25\*\*

AG	<5	20	<19	<15	<5
AS	95	120	370	170	41
AU	<220	11070	2800	34	<35
BA	<5700	1100	1000	<700	<2200
CA	<41	<5	<3	<5	<7
CO	100	110	250	140	69
CR	32000	19000	11000	11000	15000
FE	9.24	23.2	29.4	29.8	20.9
HF	2500	1100	760	760	1300
MO	<5	22	<5	<5	<5
NA	<0.29	0.11	0.09	0.13	0.15
NI	<3900	<800	<500	<600	<600
SB	30	5.9	13	9.1	7.7
SC	113	50.6	54.0	47.7	62.6
SE	<170	<54	<23	<23	<29
TA	180	25	15	14	30
TH	350	100	72	55	130
U	184	65.8	35.8	36.0	75.4
W	<110	<4	<13	<17	<19
ZN	2600	<190	<130	310	380
LA	1070	466	259	216	471
CE	1400	538	461	298	571
SH	118	54.8	32.2	26.0	46.2
EU	29.0	9.5	7.2	5.7	10.1
YB	177	62.6	42.5	39.7	75.6
LU	34.3	11.5	8.67	8.51	15.4

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FLEMENT ! 179\*\* 181\*\* 186\*\* 188\*\* 193\*\* 20  
 87-2-23\*\* 87-2-25\*\* 87-2-11\*\* 87-2-23\*\* 87-2-25\*\* 87-2-1

AG	PPM	<5	59	<5	<5	<5	<5	<5	<5
AS	PPM	93	130	29	35	22	46	22	22
AU	PPB	<42	180000	<12	<16	<41	<27	<41	<41
BA	PPM	<1100	<5200	<300	900	600	600	1000	1000
CA	%	<5	<16	<1	3	<4	<4	<3	<3
CO	PPM	85	170	33	57	33	33	62	62
CR	PPM	17000	51000	4000	1700	10000	10000	15000	15000
FE	%	19.6	6.58	13.6	20.7	29.4	29.4	20.9	20.9
HF	PPM	1000	3000	260	210	820	820	1300	1300
MO	PPM	<5	<78	<5	<5	16	<5	<5	<5
NA	%	0.39	<0.12	<0.05	0.13	0.09	0.09	0.09	0.09
NI	PPM	<700	<1200	<200	<400	<400	<400	<100	<100
SB	PPM	9.5	4.4	2.2	1.9	4.8	4.8	4.8	4.8
SC	PPM	56.4	81.1	33.2	50.0	41.3	41.3	72.1	72.1
SE	PPM	<35	<93	20	<12	<19	<19	<33	<33
TA	PPM	29	62	10	11	14	14	22	22
TH	PPM	91	310	25	18	71	71	140	140
U	PPM	60.5	145	14.4	8.5	42.7	42.7	68.5	68.5
W	PPM	<20	70	11	<10	30	30	<23	<23
ZN	PPM	220	440	80	80	210	210	<170	<170
LA	PPM	356	1060	87	93	260	260	536	536
CE	PPM	891	1260	<3	279	484	484	692	692
SH	PPM	39.6	109	12.6	13.7	32.6	32.6	59.5	59.5
EU	PPM	11.2	18.5	3.7	4.2	5.9	5.9	10.5	10.5
YB	PPM	53.7	153	18.2	20.8	41.6	41.6	80.2	80.2
LU	PPM	11.7	31.1	3.39	3.64	8.80	8.80	15.9	15.9

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204\*\* 226\*\* 228\*\* 244\*\* 254\*\* 260\*\*  
 87-2-13\*\*-H 87-2-25\*\*-H 87-2-19\*\*B-H 87-1-6\*\*-H 87-2-19\*\*-H 87-2-25\*\*-H

272\*\*  
 87-2-23\*\*-H 87-2-25\*\*-H

ELEMENT :

C UNITS

ELEMENT	204**	226**	228**	244**	254**	260**	272**
AG	<5	<5	<5	<5	<5	<5	<5
AS	7	11	25	6	20	110	1900
AU	<5	<19	<18	<6	<28	33000	2100000
RA	<200	<400	400	200	<700	<4000	35000
CA	<2	<2	<2	<1	<4	<9	<63
CO	15	38	57	6	52	220	1200
CR	1900	4100	5900	1200	15000	15000	5800
FE	4.93	18.4	19.3	1.06	16.6	13.7	12.0
HF	230	460	370	200	1200	2000	1500
MO	<5	<5	<5	<5	<5	<5	<460
NA	<0.05	0.07	0.08	<0.05	0.23	0.58	<0.44
NI	<200	<400	500	<200	<500	<1100	<6000
SB	1.0	<0.6	2.4	0.8	3.0	6.3	61
SC	14.8	59.5	55.3	5.9	60.1	68.4	44.1
SE	<7	<15	<14	<5	79	<72	<400
TA	3	19	16	16	25	34	<1
TH	23	42	33	22	120	220	290
U	13.0	12.4	18.3	11.2	73.2	126	234
W	<9	<11	<11	<4	<14	38	270
ZN	70	130	90	<50	170	340	3700
LA	87	155	118	77	376	712	1850
CE	156	551	417	90	673	1230	2070
SM	11.5	18.6	14.7	8.4	41.0	76.3	140
EU	2.0	2.2	3.8	1.6	9.3	21.8	71.7
YB	13.5	33.2	26.3	11.7	69.4	107	102
LU	2.58	6.13	4.52	2.26	14.2	22.1	16.5

EXPLANATION OF CODES

VARIABLE DETECTION LIMITS DUE TO SAMPLE COMPOSITION