Gmelin Handbook of Inorganic Chemistry

Ath Faition

Supplement Volume A 1

Technology of Platinum-Group Metals

Gmelin Handbook of Inorganic Chemistry

8th Edition

Pt Platinum

Supplement Volume A 1

Technology of Platinum-Group Metals

With 37 illustrations

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Platinum A 3 (Preparation of Platinumum Metals) - 1339

. Platinum A 4 (Detection and Determination of the Platinumum Metals) - (940

* Platinum A.S. (Alloys of Platinumum Metals: Ru, Rh, Pd) - 184

Platenum, A 6 (Alloys of Platenumum Metals) Os, Ir. Pt) - 1951

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Planeum C r (Compounds up to Platinumum and Bismuth) - 193

Plannum C 2 (Compounds up to Platinumum and Caesium) - 1940

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Preface

Like most supplement volumes of the platinum-group metal series, Platinum Suppl. Vol. A 1 has been written by an international team of specialists. It comprises technological data of all six platinum-group metals and their technically relevant alloys and compounds.

The volume starts with a review on the recovery of the platinum-group metals (23 pages); the next 42 pages are devoted to processes for separating and refining the PGM in order to obtain metals of high purity. The electrodeposition of the PGM and their alloys is treated on 26 pages. The by far most extensive section deals with PGM and their alloys and compounds in catalysis. After a historical survey and a list of important reviews on PGM catalysis, the catalytic properties of the metals are treated in a general way, followed by unsupported metals and alloys including preparation of catalysts and their reactions in various industrial processes. The role of supported metals and alloys is described in a similar manner. This is followed by an extensive description of the preparation and the reactions of PGM compounds with various nonmetals and their catalytically active role in a number of industrial processes (226 pages).

The last chapter (21 pages) is a compilation of data on the medical use of cytostatic platinum compounds.

Gelnhausen, December 1985

Kurt Swars

Table of Contents

Fully Standard More were serviced and the serviced serviced serviced and the serviced se	Page
Technology of the Platinum-Group Metals	at rai
1 Review on the Recovery of the Platinum-Group Metals	9. 2.2
1.1 Historical Perspective Period of Discovery 1750 to 1820	1.2.3
First Industrial Period 1820 to 1900 C9. Stell quord-munitary vinus right to replace and industrial Period 1900 to 1900	9 8.S
Second Industrial Period 1900 to 1960	ď
The Modern Period	Plating
1.2 Sources of Platinum Metals	Pallad
Primary Metal	entiu 2
Secondary Metals	port 2
1.3 The Production of PGM Concentrates from Sulphide Ore-bodies	
1.4 Production of PGM Concentrates from Low-Grade Scrap	
and Other Minor Sources Martilling on Drawning with a stage of the Connect Smalling Definition Designation	6
The Copper Smelting – Refining Route	6
The Lead Smelting and Refining Route	7 Pyrc
Auto-Exhaust Catalysts 29bileH lynodisO bits alynodisO to stayle	8
1.5 The Refining of Platinum Metal Concentrates	8
General Principles eavitavised knA bits I/OIA to arayle	8
The Chemistry of Aqueous Chloride Solutions of the Metals The Standard Process	9
The Standard Process	10
Alternative Classical Procedures	13
1.6 Solvent Extraction Techniques	15
General Principles	15
Anion Exchange	15
Ligand Exchange Systems for Pd Extraction	17
Commercial Solvent Extraction Processes	18
.7 Other Modern Techniques	19
Pre-Leach Treatments	
Chlorine/Hydrochloric Acid Leaching	19
on Exchange	20
.8 Analytical Techniques	
melin Handbook	21

Table of Contents	
2 High Purity Platinum-Group Metals	Page 24
2.1 Introduction	N. File
2.2 Production of Pure Platinum-Group Metals	30
2.2.1 Commercial Refining Using Sequential Precipitation	30
2.2.2 Solvent Extraction Methods evilpagara9 isohotelH	30
2.3 Preparation of High Purity Platinum-Group Metal Powders by Chemical Precipitation Methods	38
Platinum 0897 or 0808 Boned languages for	38
Palladium boined meboly	39
Ruthenium	40
Rhodium	41
Osmium	
The Production of PGRI Concentrates from Sulphice One-boules	42
2.4 Melting, Casting and Mechanical Working	42
2.5 Thin Films of High Purity Platinum-Group Metals 2931402 1011M 1011O bits	43
2.5.1 Volatilization Methods	
Pyrolysis of Halides	-44
Pyrolysis of Carbonyls and Carbonyl Halides	47
Pyrolysis of Acetylacetonato and Fluoroacetylacetonato Complexes	47 47
Pyrolysis of Alkyl and Aryl Derivatives	47
2.5.2 Electrodeposition	50
2.5.3 Vacuum Evaporation and Sputtering	51
Of Zone Refinite of Riving County and County	The
2.6 Zone Refining of Platinum-Group Metals	
2.6.1 General Remarks 2.6.2 Platinum 2.6.2 Platinum 2.6.3 General Remarks 2.6.4 General Remarks 2.6.5 General Remarks 2.6.6 General Remarks 2.6.7 General Remarks 2.6.8 General Remarks 2.6.9 General Remarks 2.6.9 General Remarks 2.6.9 General Remarks 2.6.9 General Remarks 2.6.0 General Remarks 2.6.1 General Remarks 2.6.2 General Remarks 2.6.2 General Remarks 2.6.3 General Remarks 2.6.3 General Remarks 2.6.4 General Remarks 2.6.5 General Remarks 2.6.7 General Remarks 2.6.8 General Remarks 2.6.9 General Remarks 2.6.9 General Remarks 2.6.0 General	51
2.6.3 Palladium	54
2.6.4 Rhodium	54
id Exchange Systems for Pd Extraction mulbin 2.6.5	
2.6.6 Ruthenium Extraction Processes	57
2.6.7 OsmiumasupirdzeabaM. rediO	59
2.7 Testing High Purity Platinum-Group Metals	
Review paintised bioA parotrochyHenit	60
Spectrographic Methodsspns.dox.	61
Activation Methods	62

Page	Page
Mass Spectrograph Methods 150 Islam quoto-munits 9 of bouotplased Isahotel and	63
보냈다. 그리는 그리는 사람들이 없는 아이를 살아가면서 얼마를 하는데	64
Gas Analysis	64
Determination of Carbon and Sulphur	64
Electrical Resistivity and Electromotive Force Tests for Purity	65
ent Literature Reviews 98	HgQ.
3 Electrodeposition of the Platinum-Group Metals	
3.1 Electrodeposition of Ruthenium	66
3.2 Electrodeposition of Rhodium	68
Introduction	68
Applications of Rhodium Electrodeposits	
Historical Survey on Rhodium Plating	70
Aqueous Electrolytes for Electrodenosition of Bhodium	70
Electrolytes for Deposition of Rhodium Alloys	75
Electrolytes for Electroless Deposition of Rhodium	76
Electrodeposition of Rhodium from Molten Salts	
laney and Urushibera Skeletal Catalysts 127	
3.3 Electrodeposition of Palladium	9 77
General Properties of Palladium Electrodeposits 19M 2018 10 2018 19 19 19 19 19 19 19 19 19 19 19 19 19	78
Hydrogen in Electrodeposited Palladium	
Palladium Electrodeposition Processes	80
Acid PdCl ₂ /HCl and Pd(NO ₃) ₂ /H ₂ SO ₄ Electrolytes	80
The Alkaline Pd(NH ₃) ₄ Cl ₂ Electrolyte	80
The Alkaline Pd(NH ₃) ₂ (NO ₂) ₂ Electrolyte (P-Salt Type)	81
Other Alkaline Pd Electrolytes	81
Electrodeposition of Pd Alloys	82
3.4 Electrodeposition of Osmium	85
3.5 Electrodeposition of Iridium	86
3.6 Electrodeposition of Platinum	87
Electrolytes Based on Platinic Chloride	444.0
Electrolytes Based on Diammino-dinitritoplatinum ("P" Salt)	00
Electrolytes Based on Alkali-Hydroxyplatinates	
Electrodeposition of Pt from Salt Melts	
Ensured operation of the following sentence of the contraction of the	90
4 Platinum-Group Metals, Alloys and Compounds in Catalysis 1997 1997 1997	92
tinum-Group Metal Hydride Preparation	92

Page	Page
4.1.1 The Historical Background to Platinum-Group Metal Catalysis	, ug
1110 11131 1 61100 10 1030	
The Second Period, 1836 Onwards	
The Third Industrial Period, 1838 Onwards	. 214.050A 295
4.1.2 Catalyst Literature	nonenimatera
Congress, Symposia etc., Reports 13 101 etae T entre avitomorbal and vrivita	97
Patent Literature Reviews	ser isonto 98
Periodical Liferature	
Reviews of Platinum Metal Catalysis	100
Reviews of Platinum Metal Catalysis	in Electrodi
4.1.3 Platinum-Group Metal Catalysis	104
industrial Platinum Metal Catalysis	
4.2 The Unsupported Metals and Motal Alloys	
4.2 The Unsupported Metals and Metal Alloys	120
4.2.1 Preparation and Physical Properties of Industrial Unsupported Metal and	
Metal Alloy Catalysts	100
Toparation of Wetat Diocks	104
Hydrogen Heduction of Oxides and Hydroxides 10 1001200000 applications	1 2011/04/1405
Reduction of Aqueous Solutions of Metal Salts	126
Raney and Urushibara Skeletal Catalysts	127
Promoted Metal Blacks Pseudo-organometallic or Solvated Metal Atom Dispersed (SMAD) Catalysts	50.12.13 129
Nascent Platinum Metal Blacks	131
Physical Properties of Platinum-Group Metal Blacks	132
Gauze Manufacture and Properties	140
Electrode Fabrication	143
4.2.2 Reactions	
Hydrogenation	
Hydrogenation Dehydrogenation Ovidation	
Oxidation astylubeld by s	164
Energy Generation	107
· · · · · · · · · · · · · · · · · · ·	
4.3 Supported Metal and Metal Alloy Catalysts	174
4.3.1 Preparation and Reactions mulbil to dollarous	oostoelä 437
Catalysis in the Petroleum Industry	174
Automobile Emission Catalysts	175
Control of Toxic Emissions, Odour and Solvent Streams	101
Catalytic Combustion	
Fischer-Tropsch Catalysis for Liquid Fuels	193
1.4 Platinum-Group Metal Compounds in Catalysis STEM 1182 months to assure	
distribution of the catalysis	
I.4.1 Platinum-Group Metal Hydrides	203
tatiliani-Group wetat nyuride Preparation	206
Reactions	200
	Gmelin Handbook Pt Suppl. Vol. A 1

		Page
4.4.2	Platinum-Group Metal Borides	213
	eparation of Boride Catalysts	215
Ch	emical and Physical Properties	217
	actions "nodagened bhooses" edric	219
4.4.3	Platinum-Group Metal Compounds with Group IV Elements	224
	rbides	
	aphite Intercalate Compounds	227
A 10 Sept 100 S	icides	-
Sili	icates	235
4.4.4	Platinum-Group Metal Compounds with Group V Elements	240
4.4.5	Platinum-Group Metal Compounds with Group VIA Elements	243
	ides	247
	Oxides as Catalytic Metal Precursors	248
	Oxides as Catalysts	255
	as Oxidation Catalysts	261
. (Oxides as Electrocatalysts	276
Ch	alcogenides	280
	Preparation	283
F	Reactions	287
4.4.6	Platinum-Group Metal Compounds with Halogens	299
	eparation	300
	actions	300
	dustrial Processes	308
	Homogeneous Catalysis	310
-	droformylation	310
	rbonylationmogeneous Hydrogenation	312
	drosilation	314
	efin Oligomerisation and Telomerisation	315
De	velopments in Homogeneous Catalysis	316
E N	Indical Has of Catastatic Blatinum Compounds	318
5 N	ledical Use of Cytostatic Platinum Compounds	
5.1	Detection of the Biological Activity of Platinum Compounds	318
5.2	Antitumor Effect of cis-Diamminedichloroplatinum(II) in Animals	318
5.3	Clinical Trials with cis-Diamminedichloroplatinum(II)	320
5.3.1	Genito-Urinary Tumors	320
5.3.2	Carcinomas of the Head and Neck	322
5.3.3	Bronchogenic Tumors	322
5.3.4	Miscellaneous Tumors	322
5.4	Organ Distribution and Pharmacokinetics of cis-Diamminedichloroplatinum(II)	327

	Page
5.5	Side Effects and Toxicity of cis-Diamminedichloroplatinum(II) . 94949-9449-9328
5.6	Structure-Activity Relation and Cytostatic Platinum Complexes 10 no 15 per 15 p
Stru	cture-Activity Relation
Plat	inum Complexes of the "Second Generation"
Oth	er New-Developed Cytostatic Platinum Compounds
5.7	Cytostatic Complexes of Platinum Metals Other than Platinum
5.8	Outlook Stopmass Viguoto https://doi.org/10.100/10.0000000000000000000000000000
Tab	le of Conversion Factors AlV Guoup Metal Compounds with Group AlV Guoup 339
	Oxides Oxides as Catalytic Metal Precursors
	Poxides as Catalysts.
	Ruthenates, Perruthanates, Ruthenium Tetroxide and Osmium Tetroxide
261	as Oxidation Catelysts
276 280	Oxides as Electronalalysis Chalcopenides
	Prepara on
	Reactions
	4.4.6 Pratinum-Group Metal Compounds with Halogens
	Preparation
	Reactions Industrial Processes
3.10	4.4.7 Homogeneous Catalysis
310	Hydroformylation
312	Carbonylation Homogeneous Hydrogenation
314	Hydrosilation
315	Olefin Oligomérisation and Telemensation
316	Developments in Homogeneous Catalysis
318	5 Medical Use of Cytostatic Platinum Compounds
818	5.1 Detection of the Biological Activity of Platinum Compounds
818	5.2 Anthumor Effect of cls-Diamminedichloroptatinum(II) in Animats
	5.3 Clinical Trials with cis-Diamminedichloropiatinum(II)
320	
322	5.3.2 Carcinomas of the Read and Neck
322	5.3.3 Bronchogenic Tumors
322	5.3.4 Miscelleneous Tumors
	5.4 Organ Distribution and Programmenting of the Distributed definition of the

Technology of the Platinum-Group Metals

1 Review on the Recovery of the Platinum-Group Metals

1.1 Historical Perspective

No definite date can be ascribed to the discovery of platinum; rather it became gradually known to the Western World as a distinct metal of unique properties over a period of two centuries, between 1500 and 1700. The first source to be exploited was the placer deposits in New Granada (now Colombia) where it was initially regarded as a nuisance in the gold extraction operations. By about 1750 however it was beginning to be exploited for its own sake and had found uses for decorative purposes.

The development of the technology of these metals can be divided into four stages:

Period of Discovery, 1750 to 1820

During this period major progress was made in separating the various platinum metals and in establishing many of the basic characteristics of their chemistry. Thus, for example Scheffer established in 1751 the fact that NH⁴ ions precipitated Pt as an insoluble salt from Cl⁻ solutions — a technique used in its recovery and refining up to the present day. By 1804 various techniques had been used to isolate five of the six metals (Pt, Pd, Rh, Os, Ir) from Colombian ores.

At the same time much attention was also paid to the physical metallurgy of the new metals, especially of Pt and to methods of fabricating them.

The techniques of extraction, refining, and fabrication developed during this period thus laid the foundation for the industrial exploitation of the metals.

First Industrial Period 1820 to 1900 13 SM COURT INW ESUPERIOR I SEEM IN IN INC.

This period is distinguished by four major developments in the industrial applications of the platinum metals. These were:

- (a) The development of industrial-scale integrated refining processes for the metals. Starting with Breant in 1822, these processes included steps such as aqua regia leaching, distillation of Os, precipitation of Pt as (NH₄)₂PtCl₆ and precipitation of Pd as PdCl₂(NH₃)₂.
- (b) The establishment of industrial companies whose major business was the extraction, refining, and fabrication of the platinum metals. These include Johnson Matthey, Hereaus, and Desmontis Quennessen early in the period, and Bishop and Baker-Engelhard later on.
- (c) The establishment of international trade in the metals and their ores, firstly from Colombia alone, but later and almost exclusively from new deposits in Russia.
- (d) The development of satisfactory methods of fabricating the metals and their usage as corrosion-resisting linings, especially in sulphuric acid boilers – the major use for Pt till after the turn of the century.

Gmelin Handbook Pt Suppl. Vol. A 1

Second Industrial Period 1900 to 1960

Platinum metals were obtained from a wide variety of sources during this period. Placer deposits in Colombia and Russia were still exploited, but the recovery of byproduct platinum metals from Cu-Ni ores from Sudbury, Canada became an important source from about 1920 and by the late 1930s this source was dominant in the market. This situation continued until the late 1940s, when deposits in South Africa and, somewhat later, in Russia, also of the Cu-Ni sulphide type, became of increasing importance.

During this period the refining processes developed during the nineteenth century became standardised to a large extent. This is best illustrated by the Inco process which with minor variations was almost universally used until the 1970s.

Because of the change in the nature of the ores from which PGM were derived, new methods also had to be developed for the early stages of concentration. These techniques only reached their full maturity with the development of the South African and Russian deposits, however.

Uses of the PGM continued to grow during this period and their catalytic properties began to be exploited, firstly in the production of H_2SO_4 , then the oxidation of NH_3 to HNO_3 , and thereafter in a whole range of new industrial processes.

The Modern Period behivib ad nec statem seaft to votonical arti-

This period bas been marked by the vast increase in the range and amount of usage of the PGM. Accompanying this has been the virtual elimination of primary sources of PGM other than Cu-Ni sulphide deposits, and the dominance of the Russian and South African deposits.

At the same time the recovery of the metals from secondary materials, i.e. their recycling, has become a major factor in the industry. The secondary refining industry is probably almost as large as the primary, and will probably continue to expand in the future.

Technologies have been developed to a high level of sophistication to extract and concentrate the metals from ores containing only a few parts per million into concentrates.

Beginning in about 1970, the principles and techniques of modern inorganic chemistry have been used very effectively to analyse the conventional process for the refining of the concentrates and to improve these methods. A feature of newer refining processes has been the introduction of solvent extraction and ion-exchange methods to the industry, and it appears probable that these techniques will become standard within the industry in the near future.

1.2 Sources of Platinum Metals

Primary Metal

As outlined previously, various sources have at various times dominated the supply of PGM. From about 1960 onwards, PGM associated with Cu-Ni deposits became by far the dominant source, and today about 98% of the total world production of primary PGM is derived from only three sources, i.e. sulphide deposits in Canada, USSR, and South Africa.

These three sources have in common a low PGM grade (2 to 10 ppm), and in general a very close association between the PGM minerals and the base metal sulphide minerals. This means that, in general, recovery of the PGM separately form the base metal sulphides is not

possible by physical methods such as gravity separation. Thus, the major amount of PGM is recovered as a byproduct of the processes used for recovery of the much larger amounts of base metals that accompany them.

In other respects these three deposits are somewhat different. In the case of the Russian and Canadian deposits, the ratio of (Cu + Ni)/PGM is so high that these deposits are in essence mined for their base metal content, the PGM being merely byproducts. In the Bushveld Igneous Complex (BIC) deposits in South Africa, the reverse is true, $\sim 60\%$ of the revenue being obtained from the PGM, and < 40% from the base metals.

In addition, Pt is the major PGM found in South African ores, whereas Pd is the major in Russia and Canada. South Africa therefore dominates the world Pt market whereas the USSR dominates the Pd market.

A further source of supply is beginning to be exploited in South Africa. This is the "UG-2" chromitite seam, underlying the main Merensky Reef. This contains a low sulphide mineralisation in a chromitite seam, and, associated with these sulphides are PGM. The ratio of individual PGM in this ore is quite distinct to that of the Merensky Reef, and it is in particular much richer in Ru and Rh than the Merensky.

The reserves of PGM available in South Africa are reputed to be in excess of 42000 t. This includes those in the Merensky Reef and the UG-2 reef, where exploitation has only just begun.

Secondary Metals is own once some for the POM are only of minor economic sites and a second rentral references.

The consumption of PGM by industry increased by a factor of ~ 50 between 1930 and 1980. This vast increase in consumption has been matched by the growth of the secondary metal industry since most of the metals are used in products from which they may eventually be recovered for recycling. The major uses of the metals today are as follows:

electrical and electronic usage 40%; chemical industry (catalysts) 30%; auto-exhaust catalysts 10%; petrochemical catalysts 10%; fabricated ware 10%.

Recycled material from these uses falls into one of two categories:

- (I) high and medium grade scrap includes supported and gauze catalysts, fabricated ware, etc., and usually has a PGM content of > 10%, or is associated with an easily separated substrate (e.g. carbon supported catalysts). These materials can be recycled directly to a secondary refinery for treatment and recovery of the individual PGM.
- (II) low-grade scrap. This includes much material from the electronics industry in the way of obsolete equipment and production scrap, and the PGM are usually present in low amounts on mixed metallic/non-metallic materials. Certain low-grade catalysts, e.g. auto-exhaust catalists, alumina-supported petroleum catalists, etc. also fall into this category, as well as low-grade byproducts from the secondary refiners themselves. These materials are usually treated by Pd or Cu smelters, and are removed from the base metal circuits in the form of byproducts where they may be sufficiently concentrated to allow final refining to take place.

By 1981 it is estimated that \sim 100 t p.a. of PGM scrap in one form or another was being recycled via secondary metal refiners. This may be compared with total world production of new metal of \sim 200 t p.a. in that year.

1.3 The Production of PGM Concentrates from Sulphide Ore-bodies is said and seed and seed and seed are seed as the seed as the

Almost all ores that are treated today for their PGM content contain less than 10 ppm PGM in association with a relatively large amount of Cu, Ni, and Fe sulphides. The initial steps in the recovery of the PGM are thus very closely related to the recovery of the associated base metals, and the processing only becomes distinct once a concentrate containing the PGM can be separated from the base metals.

The following description of the early stages in the processing of the ore applies to the South African producers. The technology used by Russian and Canadian producers is similar in principle but does vary somewhat in detail, as a result of the slightly differing mineralogy of these deposits.

The ore is first crushed and then milled to the desired "liberation" size. On South African ores of a relatively fine-grained character, this is usually $75\% < 70 \,\mu\text{m}$. The finely ground ore may then be treated by a gravity separation technique to remove some of the PGM ($\pm 20\%$) as a rich As/Te heavy mineral concentrate, although this is by no means essential. This concentrate is rich enough to be treated directly by the PGM refinery.

The sulphide minerals are then recovered from the ore by froth flotation. Usually ~ 5 to 10% of the ore appears in the concentrate, and this contains $\pm 80\%$ of the PGM and Cu and Ni sulphides. The grade of PGM in the concentrate is thus ~ 100 ppm and the nickel content 5 to 10%.

The flotation process may be varied to include differential flotation of Ni and Cu minerals. Thus, at Inco, it has been found possible to selectively float chalcopyrite, pendlandite, and pyrrhotite to produce separate concentrates for processing. This may be a distinct advantage for further processing, especially if the PGM are only of minor economic significance, but this does not apply to South African ores.

Further upgrading of the concentrate usually takes place by smelting. This was formerly done in blast furnaces, but all new plants have been built on the basis of electric smelting.

The concentrate, dried and sometimes agglomerated, is fed via chutes between the electrodes to form a cold top.

The electrodes usually in an in-line pattern, are spaced ~ 1.5 to 2 m apart, and heating takes place by means of the ohmic resistance of the molten slag layer between the electrodes. The sulphide minerals melt and collect at the bottom of the furnace as a matte layer, which is tapped off intermittently while slag runs over an end weir continuously.

Typical operation is at a temperature of 1250 to 1350°C. The composition of the matte produced – "green matte" is dependent on the Ni and Cu content of the concentrate. In South Africa a typical composition would be Fe 40%, Ni 20%, Cu 10%, S 30%, PGM 500 to 1000 ppm. Recovery of valuable metals in the matte is very high – usually >98%, so that the slag is normally discarded, although at Rustenburg Platinum Mines (RPM) a slag grinding/flotation circuit was formerly used.

Although the operation is in essence a melting, some incidental oxidation of sulphur takes place, and the furnace off-gases contain a low concentration, 1 to 2%, of SO₂. This is of considerable concern from the atmospheric pollution that results. Furnace gas is also very dusty and is cleaned before discharge to atmosphere via a tall stock, the dust being returned to the furnace.

The matte phase formed in this operation is an excellent collector of the precious metals, and losses in the slag are almost entirely due to mechanical causes (entrainment of matte prills). It is also an excellent collector of various other minor constituents of the ore, such as Se, Te, As, Sb, Pb, Co, Bi, etc., and these are thus concentrated along with the PGM.

Gmelin Handbook Pt Suppl. Vol. A 1