General Principles and Processes of Isolation of Elements

A few elements like carbon, sulphur, gold and noble gases, occur in free state while others in combined forms in the earth's crust. The extraction and isolation of an element from its combined form involves various principles of chemistry. A particular element may occur in a variety of compounds. The process of metallurgy and isolation should be such that it is chemically feasible and commercially viable. Still, some general principles are common to all the extraction processes of metals. For obtaining a particular metal, first we look for minerals which are naturally occurring chemical substances in the earth's crust obtainable by mining. Out of many minerals in which a metal may be found, only a few are viable to be used as sources of that metal. Such minerals are known as ores. Rarely, an ore contains only a desired substance. It is usually contaminated with earthly or undesired materials known as gangue. The extraction and isolation of metals from ores involve the following major steps: • Concentration of the ore, • Isolation of the metal from its concentrated ore, and

• Purification of the metal. The entire scientific and technological process used for isolation of the metal from its ores is known as **metallurgy**.

Occurrence of Metals: Elements vary in abundance. Among metals, aluminium is the most abundant. It is the third most abundant element in earth's crust (8.3% approx. by weight). It is a major component of many igneous minerals including mica and clays. Many gemstones are impure forms of Al2O3 and the impurities range from Cr (in 'ruby') to Co (in 'sapphire'). Iron is the second most abundant metal in the earth's crust. It forms a variety of compounds and their

various uses make it a very important element. It is one of the essential elements in biological systems as well. The principal ores of aluminium, iron, copper and zinc have been given below: Composition Metal Ores Aluminium Bauxite $AlO_x(OH)_{3-2x}$ [where 0 < x < 1] [Al₂(OH)₄ Si₂O₅] Kaolinite (a form of clay) Iron Haematite Fe₂O₂

For the purpose of extraction, bauxite is chosen for aluminium. For iron, usually the

Magnetite Fe₃O₄ Siderite FeCO₂ Iron pyrites FeS_2 $CuFeS_2$ Copper pyrites Copper CuCO₂.Cu(OH)₂ Malachite Cuprite Cu_2O Copper glance Cu₂S Zinc blende or Sphalerite ZnS Zinc Calamine ZnCO₂ Zincite ZnO oxide ores which are abundant and do not produce polluting gases (like SO₂ that is produced in case iron pyrites) are taken. For copper and zinc, any of the listed ores may be used depending upon availability and other relevant factors. Before proceeding for concentration, ores are graded and crushed to reasonable size. Concentration of Ores: Removal of the unwanted materials (e.g., sand, clays, etc.) from the ore is known as concentration, dressing or benefaction. It involves several steps and selection of these steps depends upon the differences in physical properties of the compound of the metal present and that of the gangue. The type of the metal, the available facilities and the environmental factors are also taken into consideration. Some of the important procedures are described below.

particles and froth stabilisers (e. g., cresols, aniline) stabilise the froth.

Compreses

Sulphide particles

Hydraulic Washing: This is based on the differences in gravities of the ore and the gangue particles. It is therefore a type of gravity separation. In one such process, an upward stream of running water is used to wash the powdered ore. The lighter gangue particles are washed away and the heavier ores are left **Magnetic Separation:** This is based on differences in magnetic properties of the ore components.

Magnetic roller Magnetic particles

If either the ore or the gangue (one of these two) is capable of being attracted by a magnetic field, then such separations are carried out (e.g., in case of iron ores). The ground ore is carried on a conveyer belt which passes over a magnetic roller.

Froth Floatation Method: This method has been in use for removing gangue from sulphide ores. In this process, a suspension of the powdered ore is made with water. To it, collectors and froth stabilisers are added. Collectors (e. g., pine oils, fatty acids, xanthates, etc.) enhance non-wettability of the mineral

ore particles Water Containing pine oil

Forth Bubble Carrying Sulphide

Gangue The mineral particles become wet by oils while the gangue particles by water. A rotating paddle

recovery of the ore particles. Sometimes, it is possible to separate two sulphide ores by adjusting proportion of oil to water or by using 'depressants'. For

example, in case of an ore containing ZnS and PbS, the depressant used is NaCN. It selectively prevents ZnS from coming to the froth but allows PbS to come with the froth. **Leaching:** Leaching is often used if the ore is soluble in some suitable solvent. The following examples illustrate the procedure: (a) Leaching of alumina from bauxite:

The principal ore of aluminium, bauxite, usually contains SiO₂, iron oxides and titanium oxide (TiO₂) as impurities. Concentration is carried out by digesting

agitates the mixture and draws air in it. As a result, froth is formed which carries the mineral particles. The froth is light and is skimmed off. It is then dried for

the powdered ore with a concentrated solution of NaOH at 473 - 523 K and 35 - 36 bar pressure. This way, Al_2O_3 is leached out as sodium aluminate (and SiO₂ too as sodium silicate) leaving the impurities behind: $Al_2O_3(s) + 2NaOH(aq) + 3H_2O(l) \rightarrow 2Na[Al(OH)_4](aq)$ The aluminate in solution is neutralised by passing CO₂ gas and hydrated Al₂O₃ is precipitated. At this stage, the solution is seeded with freshly prepared samples of hydrated Al₂O₃ which induces the precipitation: $2Na[Al(OH)_4](aq) + CO_2(g) \rightarrow Al_2O_3.xH_2O(s) + 2NaHCO_3$ The sodium silicate remains in the solution and hydrated alumina is filtered, dried and heated to give back pure The sodium silicate remains in the solution and hydrated alumina is filtered, dried and heated to give back pure Al2O3:: $Al_2O_3.xH_2O(s) \xrightarrow{1470 \text{ K}} Al_2O_3(s) + xH_2O(g)$

(for O_2) from which the metal is obtained later by replacement:

 $4M(s) + 8CN^{-}(aq) + 2H_2O(aq) + O_2(g) \rightarrow 4[M(CN)_2]^{-}(aq) +$

40H (aq) (M= Ag or Au) $2[M(CN)_2]^{-}(aq) + Zn(s) \rightarrow [Zn(CN)_4]^{2-}(aq) + 2M(s)$ **Extraction of Crude Metal from Concentrated Ore:** The concentrated ore must be converted into a form which is suitable for reduction. Usually the sulphide ore is converted to oxide before reduction. Oxides are

easier to reduce (for the reason see box). Thus isolation of metals from concentrated ore involves two major steps viz.,

(i) Calcination: Calcinaton involves heating when the volatile matter escapes leaving behind the metal oxide:

silicate and copper is produced in the form of copper matte which contains Cu₂S and FeS.

reducing agent for a given metal oxide (MxOy), Gibbs energy interpretations are made.

 $[\Delta G^{\Theta}_{(M,M_xO)}]$

If carbon is taken, there may also be complete oxidation of the element to CO_2 >:

 $[\Delta G_{(C,\ CO)}]$

here. The change in Gibbs energy, ΔG for any process at any specified temperature, is described by the equation:

(b) Other examples: In the metallurgy of silver and that of gold, the respective metal is leached with a dilute solution of NaCN or KCN in the presence of air

 $Fe_2O_3.xH_2O(s) \xrightarrow{\Delta} Fe_2O_3(s) + xH_2O(g)$ $ZnCO_3(s) \xrightarrow{\Delta} ZnO(s) + CO_2(g)$ $CaCO_a.MgCO_a(s) \xrightarrow{\Delta} CaO(s) + MgO(s) + 2CO_2(g)$

(b) reduction of the oxide to metal.

(a) conversion to oxide, and

involving sulphide ores are:

FeO + SiO₂ → FeSiO₃

 $\Delta G = \Delta H - T\Delta S$

the usual way:

 $xM(s \text{ or } l) + \frac{1}{2}O_2(g) \rightarrow M_xO(s)$

 $CO(g) + \frac{1}{2}O_2(g) \rightarrow CO_2(g)$ [$\Delta G_{(CO, CO_2)}$]

 $M_xO(s) + \frac{1}{2}C(s) \rightarrow xM(s \text{ or } l) + \frac{1}{2}CO_2(g)$

Ellingham Diagram:

 $2xM(s) + O_2(g) \rightarrow 2M_xO(s)$

 $M_xO + A_{red} = xM + AO_{ox}$

 $FeO(s) + C(s) \rightarrow Fe(s/l) + CO(g)$

 $\text{FeO(s)} \rightarrow \text{Fe(s)} + \frac{1}{2} O_2(g) \quad [\Delta G_{\text{(FeO, Fe)}}]$

 $C(s) + \frac{1}{2}O_2(g) \rightarrow CO(g) \quad [\Delta G_{(C,C)}]$

 $\Delta G_{(C, CO)} + \Delta G_{(FeO, Fe)} = \Delta_r G$

curve in the following Figure.

 $4Cu + O_2 \rightarrow 2Cu_2O$

800°C

 $1073 \ K$

At 500 – 800 K (lower temperature range in the blast furnace)–

At 900 – 1500 K (higher temperature range in the blast furnace):

carbon content (about 3%) and is extremely hard and brittle.

with haematite. This haematite oxidises carbon to carbon monoxide:

(b) Extraction of copper from cuprous oxide [copper(I) oxide]:

673 K

 $3 \text{ Fe}_2\text{O}_3 + \text{CO} \rightarrow 2 \text{ Fe}_3\text{O}_4 + \text{CO}_2$ $Fe_3O_4 + 4 CO \rightarrow 3Fe + 4 CO_2$ $Fe_2O_3 + CO \rightarrow 2FeO + CO_2$

FeO + CO → Fe + CO₂

 $C + CO_2 \rightarrow 2 CO$

Further Reductions:

 $Fe_2O_3 + 3 C \rightarrow 2 Fe + 3 CO$

passing through rollers.

1200°C

1473 K

1600°C

1873 K

corresponding intersections in the ΔrG^{0} vs T plots. These reactions can be summarised as follows:

2000°C

2273 K

-100

-900

273 K

-1000 -1100-1200

lower line. If the difference is large, the reduction is easier.

 $C(s) + \frac{1}{2}O_2(g) \rightarrow CO(g)$

 $2ZnS + 3O_2 \rightarrow 2ZnO + 2SO_2$ $2PbS + 3O_2 \rightarrow 2PbO + 2SO_2$ $2Cu_2S + 3O_2 \rightarrow 2Cu_2O + 2SO_2$

The sulphide ores of copper are heated in reverberatory furnace. If the ore contains iron, it is mixed with silica before heating. Iron oxide 'slags of '* as iron

(b) Reduction of oxide to the metal: Reduction of the metal oxide usually involves heating it with some other substance acting as a reducing agent (C or CO

(ii) Roasting: In roasting, the ore is heated in a regular supply of air in a furnace at a temperature below the melting point of the metal. Some of the reactions

or even another metal). The reducing agent (e.g., carbon) combines with the oxygen of the metal oxide. $M_xO_v + yC \rightarrow xM + y CO$ Some metal oxides get reduced easily while others are very difficult to be reduced (reduction means electron gain or electronation). In any case, heating is required. To understand the variation in the temperature requirement for thermal reductions (pyrometallurgy) and to predict which element will suit as the

The SO₂ produced is utilised for manufacturing H₂SO₄.

Thermodynamic principles of Metallurgy

 $(\Delta H < T\Delta S)$ and then ΔG will become –ve.

where, ΔH is the enthalpy change and ΔS is the entropy change for the process. For any reaction, this change could also be explained through the equation: $\Delta G^{0} = -RTlnK$ where, K is the equilibrium constant of the 'reactant – product' system at the temperature, T. A negative ΔG implies a +ve K in equation 6.15. And this can happen only when reaction proceeds towards products. From these facts we can make the following conclusions:

Some basic concepts of thermodynamics help us in understanding the theory of metallurgical transformations. Gibbs energy is the most significant term

the process of interpretation involves coupling of the two reactions, getting the sum of their ΔG and looking for its magnitude and sign. Such coupling is easily understood through Gibbs energy (ΔG^{0}) vs T plots for formation of the oxides. The reducing agent forms its oxide when the metal oxide is reduced. The role of reducing agent is to provide ΔG^0 negative and large enough to make the sum of ΔG^{0} of the two reactions (oxidation of the reducing agent and reduction of the metal oxide) negative. As we know, during reduction, the oxide of a metal decomposes: $M_xO(s) \rightarrow xM$ (solid or liq) + $\frac{1}{2}O_2$ (g)

The reducing agent takes away the oxygen. Equation 6.16 can be visualised as reverse of the oxidation of the metal. And then, the Δf GV value is written in

1. When the value of ΔG is negative, only then the reaction will proceed. If ΔS is positive, on increasing the temperature (T), the value of $T\Delta S$ would increase

2. If reactants and products of two reactions are put together in a system and the net ΔG of the two possible reactions is –ve, the overall reaction will occur. So

 $\frac{1}{2}\,C(s)\,+\,\frac{1}{2}\,O_2(g)\,\rightarrow\,\frac{1}{2}\,CO_2(g)\qquad\qquad [\frac{1}{2}\,\,\Delta G_{(c,\,co_2)}]$ On subtracting the reduction equation from one of the three oxidation equations, we get: $M_xO(s) + C(s) \rightarrow xM(s \text{ or } l) + CO(g)$ $M_xO(s) + CO(g) \rightarrow xM(s \text{ or } l) + CO_2(g)$

obtained by similar subtraction of the corresponding $\Delta_r G^o$ values. As we have seen, heating (i.e., increasing T) favours a negative value of $\Delta_r G^o$. Therefore, the temperature is chosen such that the sum of $\Delta_r G^o$ in the two combined redox process is negative. In $\Delta_r G^0$ vs T plots, this is indicated by the point of intersection of the two curves (curve for M_xO and that for the oxidation of the reducing substance). After that point, the $\Delta_r G^o$ value becomes more negative for the combined process including the reduction of M_xO . The

difference in the two $\Delta_r G^o$ values after that point determines whether reductions of the oxide of the upper line is feasible by the element represented by the

The graphical representation of Gibbs energy was first used by H.J.T.Ellingham. This provides a sound basis for considering the choice of reducing agent in

In this reaction, the gaseous amount (hence molecular randomness) is decreasing from left to right due to the consumption of gases leading to a –ve value of ΔS which changes the sign of the second term in equation. Subsequently ΔG shifts towards higher side despite rising T (normally, ΔG decreases i.e., goes

(b) Each plot is a straight line except when some change in phase (s→liq or liq→g) takes place. The temperature at which such change occurs, is indicated

to lower side with increasing temperature). The result is +ve slope in the curve for most of the reactions shown above for formation of $M_xO(s)$.

These reactions describe the actual reduction of the metal oxide, M_xO that we want to accomplish. The Δ_rG^o values for these reactions in general, can be

the reduction of oxides. This is known as Ellingham Diagram. Such diagrams help us in predicting the feasibility of thermal reduction of an ore. The criterion of feasibility is that at a given temperature, Gibbs energy of the reaction must be negative. (a) Ellingham diagram normally consists of plots of $\Delta_f G^o$ vs T for formation of oxides of elements i.e., for the reaction,

by an increase in the slope on +ve side (e.g., in the Zn, ZnO plot, the melting is indicated by an abrupt change in the curve).

2. The interpretation of ΔG^{0} is based on K ($\Delta G^{0} = -RT \ln K$). Thus it is presumed that the reactants and products are in equilibrium:

reduces the oxide and why this furnace is chosen. One of the main reduction steps in this process is:

When both the reactions take place to yield the overall equation, the net Gibbs energy change becomes:

If reduction is being carried out through the above equation, the oxidation of the reducing agent (e.g., C or CO) will be there:

(c) There is a point in a curve below which ΔG is negative (So M_xO is stable). Above this point, M_xO will decompose on its own. (d) In an Ellingham diagram, the plots of ΔG^0 for oxidation (and therefore reduction of the corresponding species) of common metals and some reducing agents are given. The values of $\Delta_f G^0$, etc.(for formation of oxides) at different temperatures are depicted which make the interpretation easy. (e) Similar diagrams are also constructed for sulfides and halides and it becomes clear why reductions of M_xO is difficult. There, the Δ_fG^O of M_xO is not compensated. **Limitations of Ellingham Diagram**

1. The graph simply indicates whether a reaction is possible or not i.e., the tendency of reduction with a reducing agent is indicated. This is so because it is based only on the thermodynamic concepts. It does not say about the kinetics of the reduction process (Cannot answer questions like how fast it could be?).

This is not always true because the reactant/product may be solid. [However it explains how the reactions are sluggish when every species is in solid state

and smooth when the ore melts down. It is interesting to note here that ΔH (enthalpy change) and the ΔS (entropy change) values for any chemical reaction remain nearly constant even on varying temperature. So the only dominant variable in equation becomes T. However, ΔS depends much on the physical state of the compound. Since entropy depends on disorder or randomness in the system, it will increase if a compound melts $(s\rightarrow l)$ or vapourises $(l\rightarrow g)$ since molecular randomness increases on changing the phase from solid to liquid or from liquid to gas]. **Applications** (a) Extraction of iron from its oxides: Oxide ores of iron, after concentration through calcination/roasting (to remove water, to decompose carbonates and to oxidise sulphides) are mixed with

limestone and coke and fed into a Blast furnace from its top. Here, the oxide is reduced to the metal. Thermodynamics helps us to understand how coke

Naturally, the resultant reaction will take place when the right hand side in the above equation is negative. In ΔG^{o} vs T plot representing reaction, the plot goes upward and that representing the change $C \rightarrow CO$ (C,CO) goes downward. At temperatures above 1073K (approx.), the C,CO line comes below the Fe, FeO line $[\Delta G_{(C,CO)} < \Delta G_{(Fe,FeO)}]$. So in this range, coke will be reducing the FeO and will itself be oxidised to CO. In a similar way the reduction of

Fe₃O₄ and Fe₂O₃ at relatively lower temperatures by CO can be explained on the basis of lower lying points of intersection of their curves with the CO, CO₂

It can be seen as a couple of two simpler reactions. In one, the reduction of FeO is taking place and in the other, C is being oxidised to CO:

-200 -300 $C + O_2 \rightarrow CO_2$ AG"/kJ mol' of O, -400 → 2CO2 2C + O2 → 2CO → 2ZnO -700 -800

In the Blast furnace, reduction of iron oxides takes place in different temperature ranges. Hot air is blown from the bottom of the furnace and coke is burnt to give temperature upto about 2200K in the lower portion itself. The burning of coke therefore supplies most of the heat required in the process. The CO and heat moves to upper part of the furnace. In upper part, the temperature is lower and the iron oxides (Fe₂O₃ and Fe₃O₄) coming from the top are reduced in

steps to FeO. Thus, the reduction reactions taking place in the lower temperature range and in the higher temperature range, depend on the points of

Limestone is also decomposed to CaO which removes silicate impurity of the ore as slag. The slag is in molten state and separates out from iron.

The iron obtained from Blast furnace contains about 4% carbon and many impurities in smaller amount (e.g., S, P, Si, Mn). This is known as pig iron and cast into variety of shapes. Cast iron is different from pig iron and is made by melting pig iron with scrap iron and coke using hot air blast. It has slightly lower

Wrought iron or malleable iron is the purest form of commercial iron and is prepared from cast iron by oxidising impurities in a reverberatory furnace lined

In the graph of $\Delta_r G^o$ vs T for formation of oxides shown above, the Cu₂O line is almost at the top. So it is quite easy to reduce oxide ores of copper directly to the metal by heating with coke (both the lines of C, CO and C, CO₂ are at much lower positions in the graph particularly after 500 - 600K). However most of

Limestone is added as a flux and sulphur, silicon and phosphorus are oxidised and passed into the slag. The metal is removed and freed from the slag by

the ores are sulphide and some may also contain iron. The sulphide ores are roasted/smelted to give oxides: $2Cu_2S + 3O_2 \rightarrow 2Cu_2O + 2SO_2$ The oxide can then be easily reduced to metallic copper using coke: $Cu_2O + C \rightarrow 2 Cu + CO$ In actual process, the ore is heated in a reverberatory furnace after mixing with silica. In the furnace, iron oxide 'slags of' as iron silicate and copper is produced in the form of *copper matte*. This contains Cu₂S and FeS. FeO + SiO₂ → FeSiO₃ Copper matte is then charged into silica lined convertor. Some silica is also added and hot air blast is blown to convert the remaining FeS, FeO and Cu₂S/Cu₂O to the metallic copper. Following reactions take place: $2\text{FeS} + 3\text{O}_2 \rightarrow 2\text{FeO} + 2\text{SO}_2$ FeO + SiO₂ → FeSiO₃ $2Cu_2S + 3O_2 \rightarrow 2Cu_2O + 2SO_2$ $2Cu_2O + Cu_2S \rightarrow 6Cu + SO_2$ The solidified copper obtained has blistered appearance due to the evolution of SO₂ and so it is called **blister copper**. (c) Extraction of zinc from zinc oxide The reduction of zinc oxide is done using coke. The temperature in this case is higher than that in case of copper. For the purpose of heating, the oxide is made into brickettes with coke and clay. $ZnO + C \xrightarrow{coke, 1673K} Zn + CO$

The metal is distilled off and collected by rapid chilling.

This process of electrolysis is widely known as Hall-Heroult process.

 $C(s) + O^{2-} (melt) \rightarrow CO(g) + 2e^{-}$

Copper from Low Grade Ores and Scraps

 $2C\Gamma(aq) + 2H_2O(l) \rightarrow 2OH(aq) + H_2(g) + Cl_2(g)$

 $Cu^{2+}(aq) + H_2(g) \rightarrow Cu(s) + 2H^{+}(aq)$

Oxidation Reduction

 $C(s) + 2O^{2-} (melt) \rightarrow CO_2 (g) + 4e^{-}$

Cathode: Al^{3+} (melt) + $3e^{-} \rightarrow Al(l)$

extraction of chlorine from brine (chlorine is abundant in sea water as common salt).

depending upon the differences in properties of the metal and the impurity. Some of them are listed below.

Electrochemical Principles of Metallurgy

 $\Delta G^{o} = - nFE^{o}$

Aluminium

Anode:

 H_2

 $2Al_2O_3 + 3C \rightarrow 4Al + 3CO_2$

above equation, then the less reactive metal will come out of the solution and the more reactive metal will go to the solution, e.g., Cu^{2+} (aq) + Fe(s) \rightarrow Cu(s) + Fe²⁺ (aq) In simple electrolysis, the Mⁿ⁺ ions are discharged at negative electrodes (cathodes) and deposited there. Precautions are taken considering the reactivity of the metal produced and suitable materials are used as electrodes. Sometimes a flux is added for making the molten mass more conducting.

matrix is electrolysed. Steel vessel with lining of carbon acts as cathode and graphite anode is used. The overall reaction may be written as:

methods are based on electrochemical principles which could be understood through the equation,

We have seen how principles of thermodyamics are applied to pyrometallurgy. Similar principles are effective in the reductions of metal ions in solution or molten state. Here they are reduced by electrolysis or by adding some reducing element. In the reduction of a molten metal salt, electrolysis is done. Such

here n is the number of electrons and E^o is the electrode potential of the redox couple formed in the system. More reactive metals have large negative values of

In the metallurgy of aluminium, purified Al₂O₃ is mixed with Na₃AlF₆ or CaF₂ which lowers the melting point of the mix and brings conductivity. The fused

Thus electrolysis of the molten mass is carried out in an electrolytic cell using carbon electrodes. The oxygen liberated at anode reacts with the carbon of anode producing CO and CO₂. This way for each kg of aluminium produced, about 0.5 kg of carbon anode is burnt away. The electrolytic reactions are:

Copper is extracted by hydrometallurgy from low grade ores. It is leached out using acid or bacteria. The solution containing Cu²⁺ is treated with scrap iron or

Besides reductions, some extractions are based on oxidation particularly for non-metals. A very common example of extraction based on oxidation is the

The ΔG^{o} for this reaction is + 422 kJ. When it is converted to E^{o} (using $\Delta G^{o} = -nFE^{o}$), we get $E^{o} = -2.2$ V. Naturally, it will require an external e.m.f. that is greater than 2.2 V. But the electrolysis requires an excess potential to overcome some other hindering reactions. Thus, Cl₂ is obtained by electrolysis giving out

the electrode potential. So their reduction is difficult. If the difference of two E^{0} values corresponds to a positive E0 and consequently negative ΔG^{0} in the

H₂ and aqueous NaOH as byproducts. Electrolysis of molten NaCl is also carried out. But in that case, Na metal is produced and not NaOH. As studied earlier, extraction of gold and silver involves leaching the metal with CN⁻. This is also an oxidation reaction (Ag \rightarrow Ag⁺ or Au \rightarrow Au⁺). The metal is later recovered by displacement method. $4Au(s) + 8CN(aq) + 2H_2O(aq) + O_2(g) \rightarrow$ $4[Au(CN)_2](aq) + 4OH(aq)$

 $2[\mathrm{Au}(\mathrm{CN})_2]^-(\mathrm{aq}) + \mathrm{Zn}(\mathrm{s}) \to 2\mathrm{Au}(\mathrm{s}) + [\mathrm{Zn}(\mathrm{CN})_4]^{2-}(\mathrm{aq})$

(e) Vapour phase refining (f) Chromatographic methods

In this reaction zinc acts as a reducing agent.

(a) Distillation (b) Liquation

(c) Electrolysis (d) Zone refining

These are described in detail here.

 $Cu \rightarrow Cu^{2+} + 2 e^{-}$

metal. So, the two requirements are:

 $ZrI_4 \rightarrow Zr + 2I_2$

Cathode: $Cu^{2+} + 2e^{-} \rightarrow Cu$

This process is also explained using the concept of electrode potential, over potential, and Gibbs energy which you have seen in previous sections. The reactions are: Anode: $M \rightarrow M^{n+} + ne^{-}$ Cathode: $M^{n+} + ne^{-} \rightarrow M$ Copper is refined using an electrolytic method. Anodes are of impure copper and pure copper strips are taken as cathode. The electrolyte is acidified solution of copper sulphate and the net result of electrolysis is the transfer of copper in pure form from the anode to the cathode:

Impurities from the blister copper deposit as anode mud which contains antimony, selenium, tellurium, silver, gold and platinum; recovery of these elements

(d) Zone refining: This method is based on the principle that the impurities are more soluble in the melt than in the solid state of the metal. A circular mobile heater is fixed at one end of a rod of the impure metal The molten zone moves along with the heater which is moved forward. As the heater moves forward, the pure metal crystallises out of the melt and the impurities pass on into the adjacent molten zone. The process is repeated several times and the heater is moved in the same direction. At one end, impurities get concentrated. This end is cut off. This method is very useful for producing semiconductor and other metals of

(e) Vapour phase refining: In this method, the metal is converted into its volatile compound and collected elsewhere. It is then decomposed to give pure

(a) **Distillation:** This is very useful for low boiling metals like zinc and mercury. The impure metal is evaporated to obtain the pure metal as distillate.

(b) Liquation: In this method a low melting metal like tin can be made to flow on a sloping surface. In this way it is separated from higher melting impurities. (c) Electrolytic refining: In this method, the impure metal is made to act as anode. A strip of the same metal in pure form is used as cathode. They are put in a suitable electrolytic bath containing soluble salt of the same metal. The more basic metal remains in the solution and the less basic ones go to the anode mud.

Refining: A metal extracted by any method is usually contaminated with some impurity. For obtaining metals of high purity, several techniques are used

(ii) the volatile compound should be easily decomposable, so that the recovery is easy. Following examples will illustrate this technique. Mond Process for Refining Nickel: In this process, nickel is heated in a stream of carbon monoxide forming a volatile complex, nickel tetracarbonyl: $Ni + 4CO \xrightarrow{330-350 \text{ K}} Ni(CO)_4$ The carbonyl is subjected to higher temperature so that it is decomposed giving the pure metal:

may meet the cost of refining. Zinc may also be refined this way.

very high purity, e.g., germanium, silicon, boron, gallium and indium.

(i) the metal should form a volatile compound with an available reagent,

Ni(CO)₄ 450-470 K Ni + 4CO van Arkel Method for Refining Zirconium or Titanium: This method is very useful for removing all the oxygen and nitrogen present in the form of impurity in certain metals like Zr and Ti. The crude metal is heated in an evacuated vessel with iodine. The metal iodide being more covalent, volatilises: $Zr + 2I_2 \rightarrow ZrI_4$ The metal iodide is decomposed on a tungsten filament, electrically heated to about 1800K. The pure metal is thus deposited on the filament.

adsorbent material and also on the process of passage of the moving medium, the chromatographic method* is given the name. In one such method the column of Al2O3 is prepared in a glass tube and the moving medium containing a solution of the components is in liquid form. This is an example of column chromatography. This is very useful for purification of the elements which are available in minute quantities and the impurities are not very different in chemical properties from the element to be purified. There are several chromatographic techniques such as paper chromatography, column chromatography, gas chromatography, etc.

(f) Chromatographic methods: This method is based on the principle that different components of a mixture are differently adsorbed on an adsorbent. The mixture is put in a liquid or gaseous medium which is moved through the adsorbent. Different components are adsorbed at different levels on the column. Later the adsorbed components are removed (eluted) by using suitable solvents (eluant). Depending upon the physical state of the moving medium and the

Uses of Aluminium, Copper, Zinc and Iron: Aluminium foils are used as wrappers for chocolates. The fine dust of the metal is used in paints and lacquers. Aluminium, being highly reactive, is also used in the extraction of chromium and manganese from their oxides. Wires of aluminium are used as electricity conductors. Alloys containing aluminium, being light, are very useful. Copper is used for making wires used in electrical industry and for water and steam pipes. It is also used in several alloys that are rather tougher than the metal itself, e.g., brass (with zinc), bronze (with tin) and coinage alloy (with nickel). Zinc is used for galvanising iron. It is also used in large quantities in batteries, as a constituent of many alloys, e.g., brass, (Cu 60%, Zn 40%) and german silver (Cu 25-30%, Zn 25-30%, Ni 40–50%). Zinc dust is used as a reducing agent in the manufacture of dye-stuffs, paints, etc. Cast iron, which is the most important form of iron, is used for casting stoves, railway sleepers, gutter pipes, toys, etc. It is used in the manufacture of wrought iron and steel. Wrought iron is used in making anchors, wires, bolts, chains and agricultural implements. Steel finds a number of uses. Alloy steel is obtained

when other metals are added to it. Nickel steel is used for making cables, automobiles and aeroplane parts, pendulum, measuring tapes, chrome steel for

cutting tools and crushing machines, and stainless steel for cycles, automobiles, utensils, pens, etc.