Course: **Power System Operation and Control**

Course Code: **23A0271T**

Branch(s): **EEE**

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Designation: **Assistant Professor**

Department: **Electrical and Electronics Engineering**

Title of the Course: POWER SYSTEM OPERATION AND CONTROL
Category: PCC
Couse Code: 23A0271T
Year : IV
Semester: I
Branch/es: EEE

Course Objectives:

- Optimal Operation of Thermal Power Stations.
- Hydrothermal Scheduling.
- Modelling of Turbines and Generators.
- Load frequency control of Single Area and Two Area Systems.
- The Shunt and Series Reactive Power Compensations in Power Systems.

Unit 1 Optimum Operation Thermal Power Station: 10

Optimum Operation of Thermal Power Station:

Heat Rate Curve – Cost Curve – Incremental Fuel Rate – Incremental Fuel Cost and Production Cost, Input – Output Characteristics of Thermal Power Stations and Hydro Power Stations. Optimum Generation Allocation of Thermal Units without Transmission Line Losses and Optimum Generation Allocation with effect of Transmission Line Losses. Transmission Line Loss Formula, Loss coefficients, Numerical Problems.

Unit 2 Economic Operation of Hydro – Thermal Scheduling: 08

Optimum Operation of Hydrothermal Power Stations:

Hydrothermal Coordination Methods – Optimal power flow problem formulation for loss and cost minimization, Solution of optimal power flow problem using Newton‘s method and Linear Programming technique – Numerical problems.

Unit 3 Load Frequency Control: 08

Modelling of Turbine & Governor:

The first order Turbine model, Block Diagram representation of Steam Turbines and approximate Linear models, Mathematical Modelling of Speed Governing Systems – Derivation of small Signal Transfer function – Block Diagram.

Single Area Load Frequency Control:

Necessity of Keeping Frequency constant, Definition of Control Area – Single Area Control – Block Diagram representation of an Isolated Power System – Steady State Analysis – Dynamic Response – Controlled & Uncontrolled case.

Two Area Load Frequency Control:

Load Frequency control of Two Area system – Controlled and Uncontrolled case, Tie – Line Bias Control. Proportional Plus Integral Control of Single Area and Its Block Diagram Representation, Steady State Response – Load Frequency Control and Economic Dispatch Control.

Unit 4 Reactive Power Control: 08

Overview of Reactive Power Control – Reactive Power Compensation in Transmission Systems – Advantages and Disadvantages of Different Types of Compensating Equipment for Transmission Systems; Load Compensation – Specifications of Load Compensator, Uncompensated and Compensated Transmission Lines: Shunt and Series Compensation.

Unit 5 Power System Deregulation: 08

Principle of economics, utility functions, power exchanges, electricity market models, market power indices, ancillary services, transmission and distribution charges, principles of transmission charges, transmission pricing methods, demand-side management, regulatory framework – Numerical problems.

Prescribed Text Books:

1. Modern Power System Analysis, D.P.Kothari and I.J.Nagrath, Tata McGraw Hill Publishing Company Ltd.,
2. Electric Energy Systems Theory: An Introduction, Olle I. Elgerd, TMH Publishing Company Ltd., New Delhi, 2nd edition, 1983.

References Books:

1. Power Generation, Operation and Control, Allen J. Wood and Bruce F. Wollenberg, John Wiley & Sons, Inc., New York, 2nd edition, 1996.
2. Reactive Power Control in Electric Systems, T J E Miller, John Wiley & Sons, New York, 1982.
3. Power System Analysis Operation and Control, Abhijit Chakrabarti and Sunita Halder, PHI Learning Pvt. Ltd.,, 3rd Edition, 2010.

Course Outcomes:

At the end of the course, the student will be able to Blooms Level of Learning

1. To Understand the Thermal Station Characteristics and Economic Dispatch L3

Problem of Thermal Units and Understand the Optimal Scheduling of Hydro Thermal Station with minimization of cost of Thermal station
2. To Develop the First Order Models of Turbine, Governor and Generator Load Model L4
3. To Evaluate the Steady State & Dynamic Analysis of Single Area and Two Area Load Frequency Control L3
4. Analyze Series and Shunt reactive power Compensation in transmission and Load systems L3
5. To understand the aspects of power system deregulation L2

Web Resources:

1. https://archive.nptel.ac.in/courses/108/104/108104052/
2. http://kcl.digimat.in/nptel/courses/video/108104191/L01.htm

CO-PO Mapping:

Course Outcomes	Engineering Knowledge	Problem Analysis	Design/Development of solutions	Conduct investigations of complex problems	Engineering tool usage	The engineer and the world	Ethics	Individual and collaborative team work	Communication	Project management and finance	Life-long learning	PSO1	PSO2
CO1	3	3	3	2	-	1	-	-	-	-	-	3	-
CO2	3	3	3	1	-	1	-	-	-	-	-	3	1
CO3	3	3	3	3	-	1	-	-	-	-	-	3	2
CO4	3	3	3	-	-	1	-	-	-	-	-	3	-
CO5	3	-	2	-	2	1	-	-	-	-	-	3	-

UNIT-1

The optimal system operation involved the consideration of economy of operation, system security, emissions at certain fossil-fuel plants, releases of water at hydrogeneration, etc. Ecoppmic operation is also called "Economic dispatch problem"

The economic load dispatch problem involves the solution of different problems.

* Unit-commitment (or) pre-dispatch problem
* On-line economic dispatch

→ Unit commitment problem, wherein it is required to select optimally out of the available generating sources to operate, to meet the expected load and provide a specified margin of operating reserve over a specified period of time.

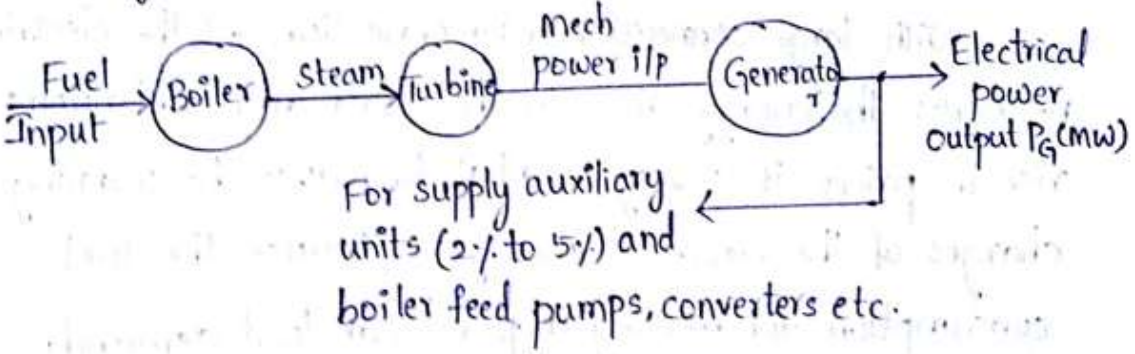
→ The on-line economic dispatch wherein it is required to distribute the load among the generating units actually parallel with system in such manner as to minimize the total cost of supply the m-to-minute requirements of system

With large ~~connection~~ interconnection of the electric networks the energy crisis in the world and continuous rise in prices, it is very essential to reduce the running charges of the electrical energy. i.e., reduce the fuel consumption for meeting a particular load demand.

→ The main aim in the economic dispatch problem is "to minimize the total cost of generating real power (production cost) at various stations while satisfying the loads and losses in transmission links".
In case of economic load dispatch, the generations are not fixed but they are allowed to take values again within certain limits so as to meet a particular load demand with minimum fuel consumption.

The total cost of generation is a function of individual generation of the sources which can take values within certain constraints, the cost of generation will depend upon the system constraint for particular

In analyzing, economic operation of a thermal plant unit input-output modelling characteristics are significant.
For this function.
Consider a single unit consisting of a boiler, a turbine and a generator as shown.



This unit has to supply power not only to the load connected to power system but also the local needs for auxiliaries in the station, which may vary from 2% to 5%.

→ The power requirements for station auxiliaries are necessary to drive boiler feed pumps, fans and condenser circulating water pumps etc.,

The total input to the thermal unit could be British Thermal Unit (BTU)/hr (or) cal/hr in terms of heat supplied (or) Rs/hr in terms of the cost of fuel (coal or gas)

The total output of the unit at the generator bus will be either KW (or) MW

System constraints:

There are two types of constraints

→ Equality constraints .... are basic load flow eqns.

→ Inequality constraints

Inequality constraints are of two types

i) Hard type ... Eg: tapping range of an on-load tap-changing t/f

ii) Soft type ... Eg: Nodal voltages and phase angles

Inequality constraints:

a) Generator constraints

$$P_{G(min)} \leq P_G \leq P_{G_{max}}$$

Max. active power generation of a source is limited by thermal consideration and

Min. active power generation of the system is limited by flame instability of a boiler.

Similarly

Maximum and minimum reactive power generation of a source are limited.

$$Q_{G(min)} \leq Q_G \leq Q_{G(max)}$$

The maximum reactive power is limited because of overheating of the rotor.

The minimum reactive power is limited because of stability limit of the machine.

→ Hence the generator reactive power can't be outside the range stated by inequality.

b) Voltage constraints:

The voltage magnitudes and phase angles at various modes should vary within certain limits.

The voltage magnitudes should vary within certain limits because otherwise most of the equipments connected to the system will not operate satisfactorily (or) additional use of voltage regulating devices will make the system uneconomical.

$$|V_{P(min)}| \leq |V_P| \leq |V_{P(max)}|$$

$$\delta_{P(min)} \leq \delta_P \leq \delta_{P(max)}$$

c) Running spare capacity constraints:

These constraints are required to meet

i) Forced outages of one or more alternators on the system.

ii) Unexpected load on the system.

$$G \geq P_G + P_{so}$$

where P_{Go} is some specified power and G is the total generation. A well planned system is one in which this space capacity P_{so} is minimum.

d) Transformer tap settings:

If an auto-transformer is used, minimum tap setting could be zero and maximum be one.

i.e., $0 \leq t \leq 1.0$

Similarly, for a two winding transformer if tappings are provided on the secondary side

$$0 \leq t \leq n$$

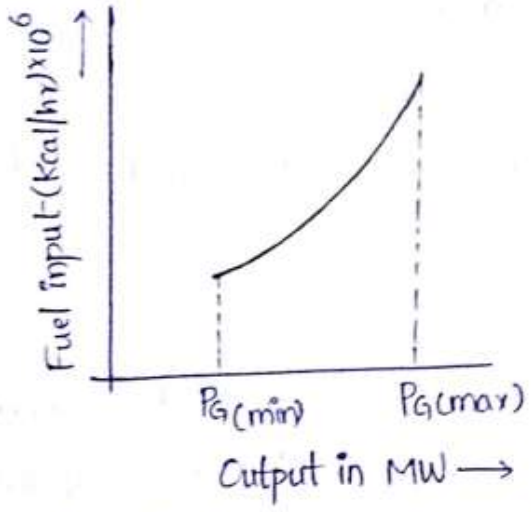
where n is the transformation ratio.

e) Transmission line constraints:

The flow of active and reactive power through the transmission line is limited by the thermal capacity or capability of the circuit and is expressed as $C_p \leq C_{p_{max}}$ where $C_{p_{max}}$ is the maximum capacity loading capacity of pth line.

f) Network security constraints

To formulate economic dispatch problem, it is necessary to know "input-output characteristics of each unit".

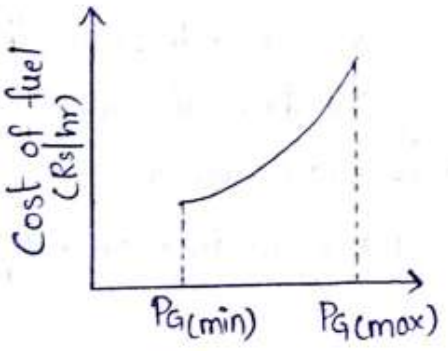


The steam turbine-generating unit curve consists of minimum and maximum limits in operation, which depend upon the steam cycle used, thermal characteristics of material, the operating temperature etc.

Cost-curve:

To convert input-output curves into cost curves, the fuel input per hr is multiplied with cost of the fuel (expressed in Rs/million K·cal)

$$\frac{\text{K·cal}\times 10^6}{\text{hr}} \times \text{Rs/million K·cal} = \text{Rs/hr}$$



Incremental fuel cost curve

From the input - output curves, the incremental fuel cost (IFC) curve can be obtained.

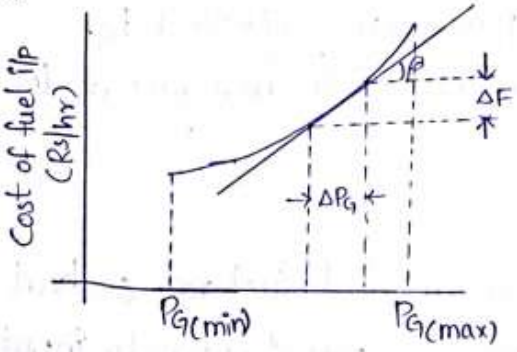
IFC is defined as "ratio of a small change in the input to the corresponding small change in the output.

$$\text{Incremental fuel cost} = \frac{\Delta \text{input}}{\Delta \text{output}} = \frac{\Delta F}{\Delta P_G}$$

where Δ represents small changes.

As Δ quantities become progressively smaller, it is seen that IFC is $\frac{d(\text{input})}{d(\text{output})}$ and is expressed as Rs/MWh.

A typical plot of IFC versus output power is shown



The IFC is now obtained as

$(IC)_i$ = slope of fuel cost curve

$$\text{temp} = \frac{\Delta F}{\Delta P_G} \text{ in Rs/MWhr}$$

$$= \lim_{P_{Gi} \to 0} \frac{\partial F_i}{\partial P_{Gi}} = \frac{dF_i}{dP_{Gi}}$$

Mathematically IFC curve expression can be obtained from the expression of cost curve.

$$C_i = \frac{1}{2} a_i P_{Gi}^2 + b_i P_{Gi} + d_i$$

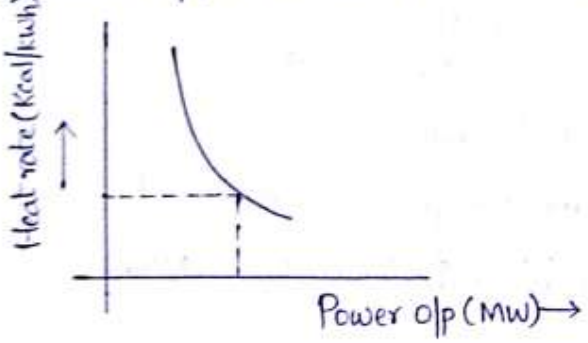
$$IFC = \frac{dc_i}{dP_{Gi}} = (IC)_i$$

$$= a_i P_{Gi} + b_i \quad \text{linear app}$$

$i = 1, 2, \ldots n$

Heat rate curve:

The heat rate characteristic obtained from the plot of net heat rate in BTU/KWh (or) k·cal/kWh vs o/p power (KW) is shown.



→ The thermal unit is most efficient at minimum heat rate which corresponds to a particular generation P_G.

→ The curve indicates an increase in heat rate at low and high power limits.

Incremental efficiency:

The reciprocal of incremental fuel rate (or) heat rate which is defined as "ratio of output energy to input energy" gives a measure of full efficiency for the input.

i.e., Incremental efficiency = $\frac{\text{Output}}{\text{Input}} = \frac{dP_G}{dc}$

Optimization problem - Mathematical formulation

An optimization problem consists of

i) Objective function

ii) Constraint equation.

Objective Function:

The objective function is to minimize the overall cost of production of power generation.

Cost of thermal and nuclear stations is nothing but the cost of fuel.

Let 'n' be no. of units in the system, C_i is cost of power generation of unit 'i'.

Total cost, $C = C_1 + C_2 + \cdots + C_n$

$$= \sum_{i=1}^{n} C_i(P_{Gi})$$

Constraint equations:

(1) Equality constraints:

The sum of real power generation of all the various units must always be equal to the total real-power demand on the system.

i.e., $P_D = \sum_{i=1}^{n} P_{Gi}$ (or)

$$\sum P_{Gi} - P_D = 0$$

This is known as real power balance equation when losses are neglected.

(2) Inequality constraints:

These constraints are considered in an economic power system operation due to physical and operational limitations of the units and components.

* Output power constraints

$$P_{Gi(min)} \leq P_{Gi} \leq P_{Gi(max)}$$

$$Q_{Gi(min)} \leq Q_{Gi} \leq Q_{Gi(max)}$$

* Voltage magnitude and phase angle constraints

$$V_{i(min)} \le V_i \le V_{i(max)} \quad \text{for } i = 1, 2, \ldots n$$

$$\delta_{ij(min)} \le \delta_{ij} \le \delta_{ij(max)} \quad \text{for } i = 1, 2, \ldots n$$

where $j = 1, 2, \ldots m$, $j \neq i$. n is the no. of units

m is no. of loads connected to each unit.

* Dynamic constraints

$$\left|\frac{dP_{Gi}(t)}{dt}\right|_{min} \le \left|\frac{dP_{Gi}(t)}{dt}\right| \le \left|\frac{dP_{Gi}(t)}{dt}\right|_{max}$$

Similarly Q

* Transformer tap position / setting constraints

The problem can be solved by the method Lagrange Multipliers.

$$L = \sum C_i P_{Gi} - \lambda\left[\sum P_{Gi} - P_D\right] \quad ***$$

For minimization, $\frac{dL}{dP_{Gi}} = 0 \Rightarrow \boxed{\frac{dC_i}{dP_{Gi}} = \lambda}$

$$\frac{dL}{dP_{Gi}} = \frac{dC_i}{dP_{Gi}} - \lambda[1 - 0]$$

$$\frac{dC_i}{dP_{Gi}} - \lambda = 0$$

$$\frac{dC_i}{dP_{Gi}} = \lambda \; ; \quad i = 1, 2, \ldots k$$

where $\frac{dC_i}{dP_{Gi}}$ = IFC of the ith generator (Rs/MWh)

$\frac{dC_1}{dP_{G1}} = \frac{dC_2}{dP_{G2}} = \cdots = \frac{dC_k}{dP_{Gk}} = \lambda$ where λ is the lagrangian multiplier

Economic operation of power system:

Optimum generation allocation including the effect of transmission line losses - loss coefficients - transmission line loss formula.

Consider the objective function

$$C = \sum_{i=1}^{n} C_i P_{Gi}$$

Subject to the following

(i) Equality constraints

$$\Sigma P_{Gi} = P_L + P_D \quad (\text{or}) \quad \Sigma P_{Gi} - P_D - P_L = 0 \longrightarrow (1)$$

where P_L is the total transmission losses (MW)

P_D is total real power demand and

P_{Gi} is real power generation at the ith unit

(ii) Inequality constraints.

$$P_{Gi}(min) \leq P_{Gi} \leq P_{Gi}(max)$$

$$Q_{Gi}(min) \leq Q_{Gi} \leq Q_{Gi}(max)$$

$$V_i(min) \leq V_i \leq V_i(max)$$

To solve the problem, we write lagrangian as

$$L = \Sigma C_i(P_{Gi}) - \lambda\left[\sum_{i=1}^{n} P_{Gi} - P_D - P_L\right] \longrightarrow (1)$$

$$P_L = P_L(P_{G1}, P_{G2}, \ldots P_{Gn}) \qquad i = 1, 2, \ldots n$$

For optimum real power dispatch,

$$\frac{dL}{dP_{Gi}} = \frac{dC_i}{dP_{Gi}} - \lambda\left[1 - \frac{dP_L}{dP_{Gi}}\right] = 0 \qquad i = 1, 2, \ldots n \longrightarrow (2)$$

Rearranging above eqn and we get

The output of only one plant can affect the cost at only that plant, we have

$$\frac{\left(\frac{dC_i}{dP_{Gi}}\right)}{\left(1-\frac{dP_L}{dP_{Gi}}\right)} = \lambda \quad (\text{or}) \quad \frac{dC_i}{dP_{Gi}} L_i = \lambda \longrightarrow (3)$$

where $L_i = \frac{1}{1-\frac{\partial P_L}{\partial P_{Gi}}}$ is called penalty factor of ith plant.

→ The Lagrange multiplier λ is in rupees per MW-hr, when fuel cost is in per hour.

From eqn③, minimum fuel cost is obtained when the incremental fuel cost of each plant multiplied by its penalty factor is the same for all the units.

Optimum generation allocation including effect of transmission line losses:

The generating plants are generally located at longer distance from the load centre. In determination of optimum generation allocation (economic load dispatch) transmission losses should be considered. For a system consisting of 'n' no. of plants the cost of fuel input is given by

$$C_i = C_1 + C_2 + \cdots + C_n$$

$$= \sum_{i=1}^{n} C_i(P_{Gi}) \quad [\because \text{objective function}]$$

Subjected to constraints

$$\sum_{i=1}^{n} P_{Gi} = P_D + P_L$$

$$\sum_{i=1}^{n} P_{Gi} - P_D - P_L = 0$$

Lagrangian multiplier method is used to solve optimization problem.

By using Lagrangian we have

$$\mathcal{L} = \sum_{i=1}^{n} C_i(P_{Gi}) - \lambda\left[\sum_{i=1}^{n} P_{Gi} - P_D - P_L\right) = 0$$

$$P_L = f(P_{G_1}, P_{G_2}, \ldots P_{Gn})$$

For optimization

$$\frac{dL}{dP_{Gi}} = 0$$

$$\frac{dL}{dP_{Gi}} = \frac{dC_i}{dP_{Gi}} - \lambda\left[1 - 0 - \frac{dP_L}{dP_{Gi}}\right] = 0$$

$$\frac{dC_i}{dP_{Gi}} = \lambda\left[1 - \frac{dP_L}{dP_{Gi}}\right]$$

$$\lambda = \frac{\frac{dC_i}{dP_{Gi}}}{1 - \frac{dP_L}{dP_{Gi}}} = \frac{IFC}{1 - ITL}$$

IFC = Incremental fuel cost

ITL = Incremental transmission loss

$$\boxed{\lambda = L_i\left(\frac{dC_i}{dP_{Gi}}\right)}$$ → is also known as coordination equation

where L_i = penalty factor

$$L_i = \frac{1}{1 - \frac{dP_L}{dP_{Gi}}}$$

General transmission line loss formula (Loss coefficients)

In addition to the assumptions already made, other assumptions are necessary if B_{ik} coefficients are treated to be constant. Since total load and load distribution between sources vary.

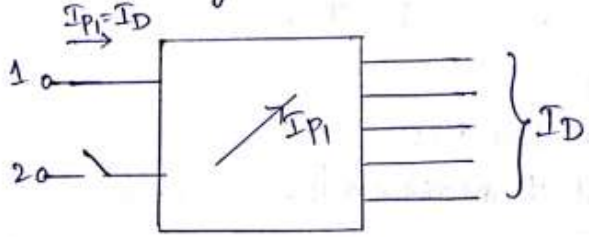
The assumptions are:

1) The power-factor of plants remain constant.
2) All load currents maintain constant ratio to the total current.
3) Voltage magnitudes at all the plants remain constant.
4) Voltage phase angles at plant buses remains fixed.

Fig. shows schematic diagram showing 2 plants connected through a power system network to a no. of loads.



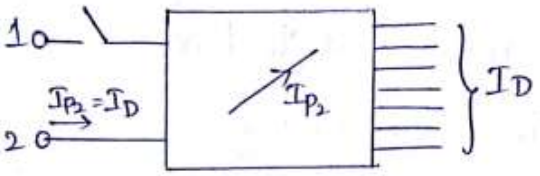
Imagine that the total current I_D is supplied by plant 1 only as shown in fig.



Let us define current distribution factors M_{P_1}, M_{P_2}

Let the current in the line P be I_{P_1}

$$M_{P_1} = \frac{I_{P_1}}{I_D}$$



Let the current in the line P be I_{P_2}

Current distribution factor $M_{P_2} = \frac{I_{P_2}}{I_D}$

The values of current distribution factors depend upon the impedance of lines and their interconnection and are independent of the current I_D.

When both generators 1 & 2 supplying the current into the n/w as shown in fig.

Applying superposition principle,

The current in the line P can be expressed as

$$I_P = I_{P_1} + I_{P_2}$$

$$\boxed{I_P = M_{P_1} I_{G_1} + M_{P_2} I_{G_2}} \longrightarrow (1)$$

where I_{G_1} and I_{G_2} are currents supplied by plants 1 & 2 respectively.

Let us make certain simplifying assumptions at this stage listed below.

i) All the load currents have same phase angle w.r.t. common reference. The load current at i^{th} bus is given by

$$I_{D_i} = |I_{D_i}| \angle \delta_i - \phi_i$$

where

δ_i = the phase angle of bus voltage

ϕ_i = lagging phase angle of the load.

$I_{Di} = |I_{Di}| \angle\theta_i$

θ_i = same for all the load currents at all the times.

ii) Ratio $\frac{X}{R}$ is same for all the network branches.

Let $I_{G_1} = |I_{G_1}| \angle\sigma_1 = |I_{G_1}|(\cos\sigma_1 + j\sin\sigma_1)$

$I_{G_2} = |I_{G_2}| \angle\sigma_2 = |I_{G_2}|(\cos\sigma_2 + j\sin\sigma_2)$

where σ_1 & σ_2 = phase angles of I_{G_1} & I_{G_2} respectively;

Sub I_{G_1} & I_{G_2} in eqn①

$I_P = M_{P_1}|I_{G_1}|(\cos\sigma_1 + j\sin\sigma_1) + M_{P_2}|I_{G_2}|(\cos\sigma_2 + j\sin\sigma_2)$

$|I_P|^2 = \left(M_{P_1}|I_{G_1}|\cos\sigma_1 + M_{P_2}|I_{G_2}|\cos\sigma_2\right)^2 + \left(M_{P_1}|I_{G_1}|\sin\sigma_1 + M_{P_2}|I_{G_2}|\sin\sigma_2\right)^2$

$\because |a+jb|^2 = a^2+b^2$

$= M_{P_1}^2|I_{G_1}|^2\cos^2\sigma_1 + M_{P_2}^2|I_{G_2}|^2\cos^2\sigma_2 + 2M_{P_1}M_{P_2}|I_{G_1}||I_{G_2}|\cos\sigma_1\cos\sigma_2 + M_{P_1}^2|I_{G_1}|^2\sin^2\sigma_1 + M_{P_2}^2|I_{G_2}|^2\sin^2\sigma_2 + 2M_{P_1}M_{P_2}I_{G_1}I_{G_2}\sin\sigma_1\sin\sigma_2$

$|I_P|^2 = M_{P_1}^2|I_{G_1}|^2 + M_{P_2}^2|I_{G_2}|^2 + 2M_{P_1}M_{P_2}|I_{G_1}||I_{G_2}|\cos(\sigma_1 - \sigma_2)$

The plant currents are given by

$|I_{G_1}| = \frac{P_{G_1}}{\sqrt{3}|V_1|\cos\phi_1}$, $|I_{G_2}| = \frac{P_{G_2}}{\sqrt{3}|V_2|\cos\phi_2}$

Power loss $= 3|I_P|^2 R_P$

$= 3R_P\left[M_{P_1}^2 \cdot \frac{P_{G_1}^2}{3(V_1)^2\cos\phi_1)^2} + M_{P_2}^2 \cdot \frac{P_{G_2}^2}{3|V_2|^2(\cos\phi_2)^2} + 2M_{P_1}M_{P_2}\frac{P_{G_1}P_{G_2}}{3|V_1||V_2|\cos\phi_1\cos\phi_2}\cos(\sigma_1 - \sigma_2)\right]$

$$= \frac{M_{P_1}^2 P_{G_1}^2 R_P}{|V_1|^2(\cos\phi_1)^2} + \frac{M_{P_2}^2 P_{G_2}^2 R_P}{|V_2|^2(\cos\phi_2)^2} + \frac{2 M_{P_1} M_{P_2} \cos(\sigma_1 - \sigma_2) P_{G_1} P_{G_2} R_P}{|V_1||V_2| \cos\phi_1 \cos\phi_2}$$

$$\boxed{P_L = B_{11} P_{G_1}^2 + B_{22} P_{G_2}^2 + 2 B_{12} P_{G_1} P_{G_2}}$$

where

$$B_{11} = \frac{1}{|V_1|^2(\cos\phi_1)^2} \sum_{P=1}^{n} M_{P_1}^2 R_P$$

$$B_{22} = \frac{1}{|V_2|^2(\cos\phi_2)^2} \sum_{P=1}^{n} M_{P_2}^2 R_P$$

$$B_{12} = \left[\frac{M_{P_1} M_{P_2} \cos(\sigma_1 - \sigma_2)}{|V_1||V_2| (\cos\phi_1)\cdot(\cos\phi_2)} \right] R_P$$

The terms B_{11}, B_{12} & B_{22} are called loss coefficients (or) B-coefficients.

Note: If voltages are expressed in kV, resistances are in 'Ω' then the units of B-coefficients are MW^{-1}.

General transmission loss formula:

If the system has k plants supplying the total load through transmission lines then the transmission loss is given by

$$P_L = \sum_{m=1}^{k} \sum_{n=1}^{k} P_{Gm} B_{mn} P_{Gn}$$

If the system has 2 plants i.e., k = 2

$$P_L = \sum_{m=1}^{2} \sum_{n=1}^{2} P_{Gm} B_{mn} P_{Gn}$$

$$= \sum_{n=1}^{2} \left[P_{G_1} B_{1n} P_{Gn} + P_{G_2} B_{2n} P_{Gn} \right]$$

$$= P_{G_1} B_{11} P_{G_1} + P_{G_1} B_{12} P_{G_2} + P_{G_2} B_{21} P_{G_1} + P_{G_2} B_{22} P_{G_2}$$

$$= P_{G_1}^2 B_{11} + P_{G_1} P_{G_2} B_{12} + P_{G_1} P_{G_2} B_{21} + P_{G_2}^2 B_{22}$$

$$\boxed{P_L = B_{11} P_{G_1}^2 + 2 B_{12} P_{G_1} P_{G_2} + B_{22} P_{G_2}^2}$$

For a system consisting of 3 generating plants the transmission loss is given by

$$P_L = \sum_{m=1}^{3} \sum_{n=1}^{3} P_{Gm} B_{mn} P_{Gn}$$

$$= \sum_{n=1}^{3} \left[P_{G_1} B_{1n} P_{Gn} + P_{G_2} B_{2n} P_{Gn} + P_{G_3} B_{3n} P_{Gn} \right]$$

$$= P_{G_1} B_{11} P_{G_1} + P_{G_1} B_{12} P_{G_2} + P_{G_1} B_{13} P_{G_3} + P_{G_2} B_{21} P_{G_1} + P_{G_2} B_{22} P_{G_2} + P_{G_2} B_{23} P_{G_3} + P_{G_3} B_{31} P_{G_1} + P_{G_3} B_{32} P_{G_2} + P_{G_3}^2 B_{33}$$

$$P_L = B_{11} P_{G_1}^2 + B_{22} P_{G_2}^2 + B_{33} P_{G_3}^2 + 2 B_{12} P_{G_1} P_{G_2} + 2 B_{23} P_{G_2} P_{G_3} + 2 B_{31} P_{G_3} P_{G_1}$$

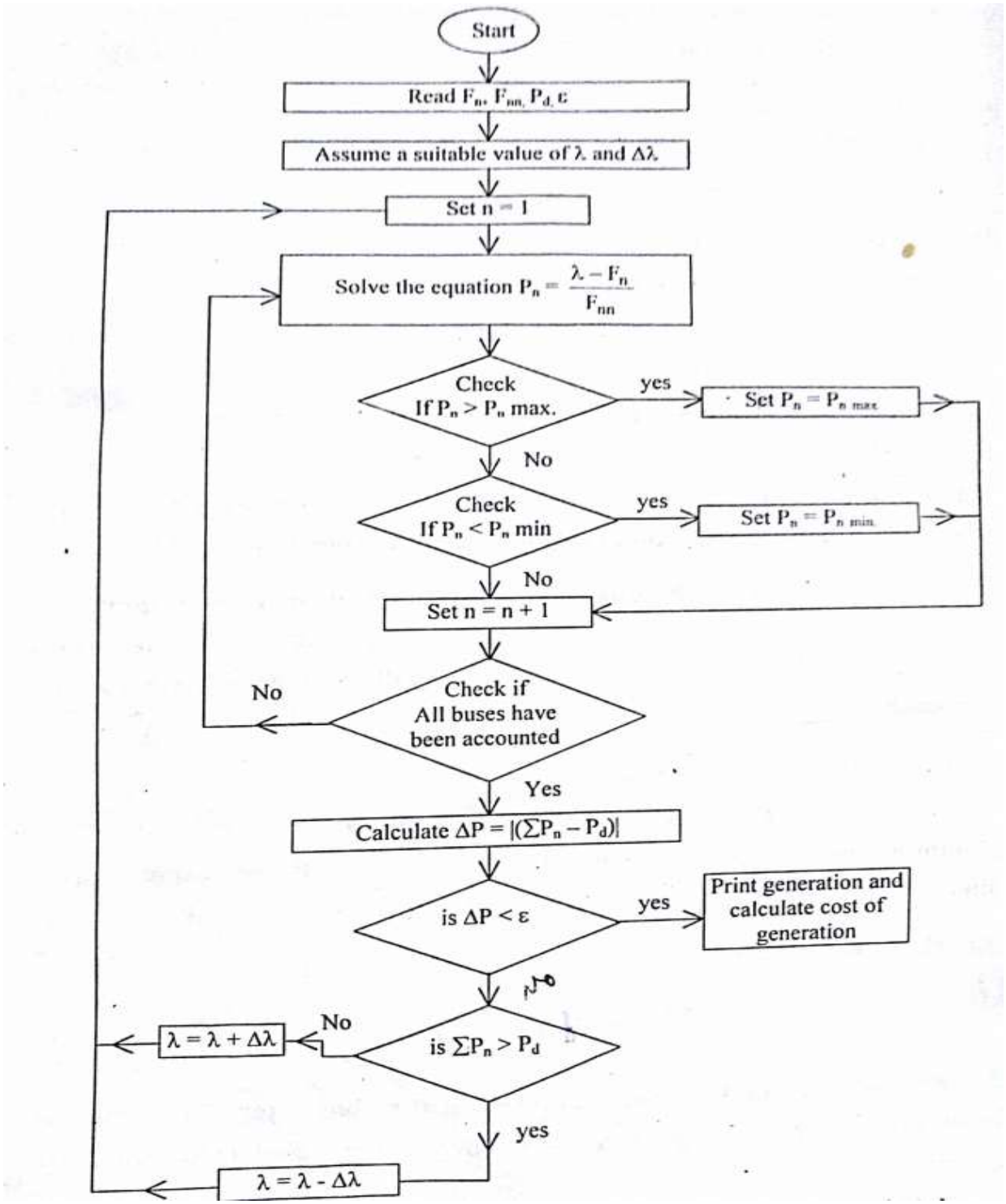
Note:

1. Since the transmission lines are symmetrical, the loss coefficients B_{mn} & B_{nm} are equal

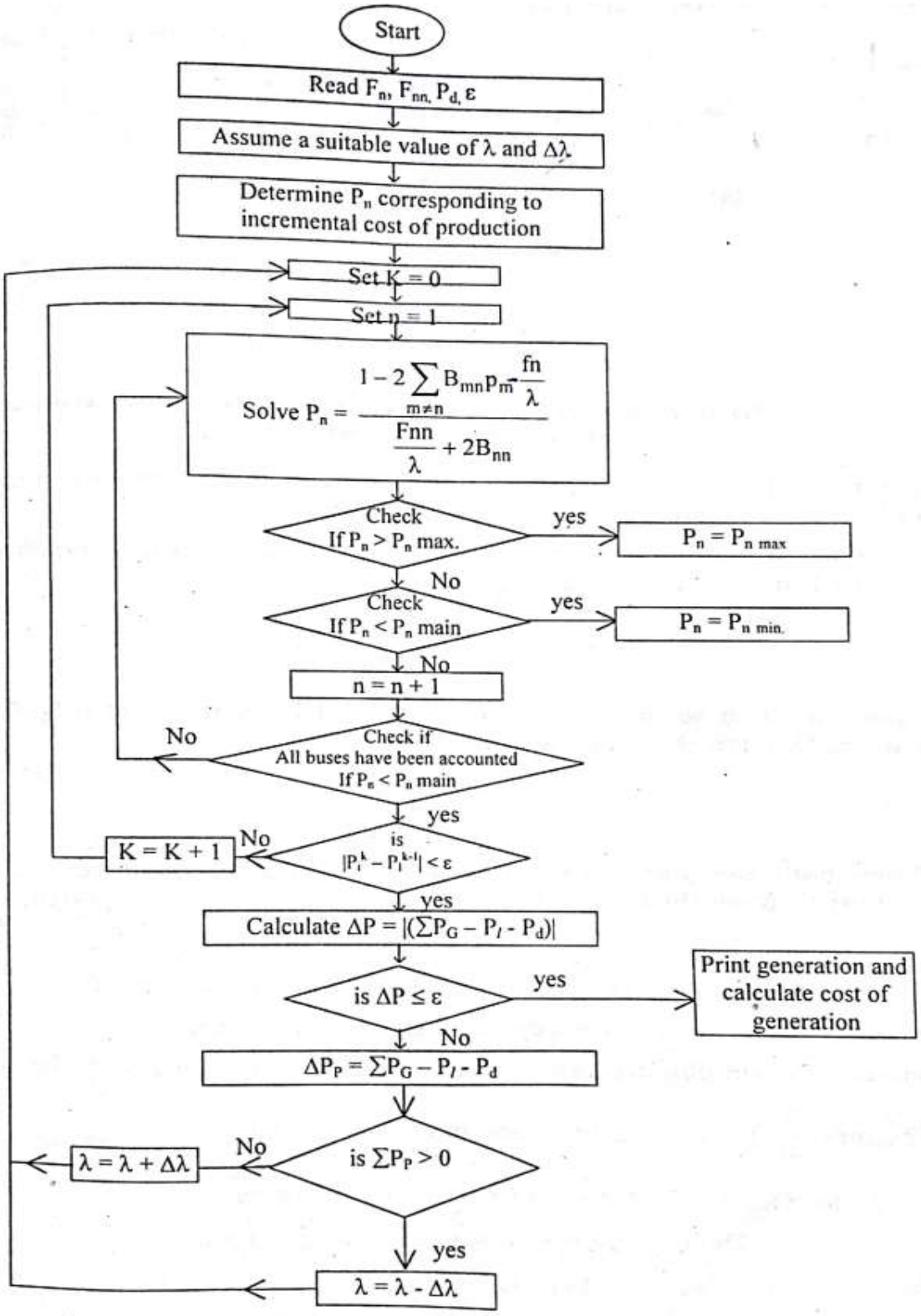
   i.e., $\boxed{B_{mn} = B_{nm}}$

2. B_{mn} (or) B-coefficients are loss coefficients & can be represented in matrix form for power system analysis.

Optimum generation allocation with line losses neglected (or) Economic dispatch neglecting losses:



Optimum generation allocation including the effect of transmission line losses: (or)
Optimum load dispatch including transmission losses:



UNIT-II

Hydrothermal Scheduling

Introduction :

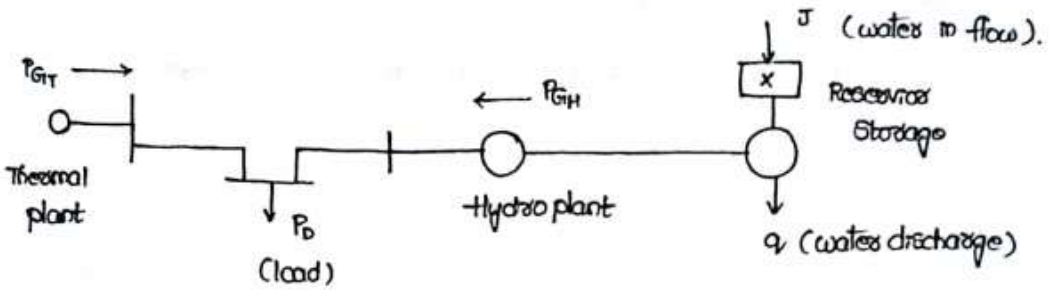
Hydro plants can be started easily and can be assigned load in very short time. However, in case of thermal plants it requires some hours to make boiler, superheaters and turbine systems ready to take the load. For this reason hydroplants can handle fast changing loads effectively. The thermal plants in contrast are slow in response and more suitable to operate as base load plants. Hydroplants are used to operate as peak load plants.

Hydrothermal Scheduling :

The operating cost of thermal plants is very high and at the same time its capital cost is low when compared to hydroelectric plant. The operating cost of hydroelectric plant is low, capatal cost is high. It is economical as well as convinient to run both thermal as well as hydroplants in same grid.

* The problem of minimising operation cost of hydrothermal system can be viewed as minimization of cost of thermal plants under the constraint of the water availability for hydro generation during given period of time. This problem is known as dynamic optimization problem where the time-factor is to be considered.

Consider a simple hydrothermal system as shown in figure



Hydrothermal System Consists of One thermal and One Hydel plant Supplying power to the load Connected at the Centre (or in between the plants & is referred as fundamental System). The Optimization will be Carried Out with real power generation as Control Variable.

Mathematical formulation :

For a Certain period of time 'T' (1 year (or) 1 month (or) 1 day). It is assumed that

(i). Storage of hydro reservior at the beginning and at the Ending of the period are Specified.

(ii). Water in flow to the reservior and the load demand On the System are Known as function of time it Completes Certainity.

The problem is to determine water discharge rate So as to minimize the Cost of the thermal generation 'C_T' i.e

$$C_T = \int_0^T C' [P_{G_T}(t)] \, dt \longrightarrow (1).$$

Under the following Constraints ,

(i). Meeting the load demand

$$P_{G_T}(t) + P_{G_H}(t) - P_D(t) - P_L(t) = 0 \quad ; \quad t \in [0,T]$$

$$\longrightarrow (2).$$

This Equation (2) is Called power balance Equation.

(ii). Water availability

$$x'(T) - x'(0) - \int_0^T J(t) \, dt + \int_0^T q(t) \, dt = 0 \longrightarrow (3).$$

Where, $I(t)$ = Water in flow

$x'(T)$ = water storage

$x'(0)$, $x'(T)$ = water storage at the beginning and at the End of Optimization interval.

$q(t)$ = water discharge.

(ii). The Hydro generation $P_{GH}(t)$ is the function of the water discharge & storage.

$$P_{GH}(t) = f(x'(t), q(t)) \longrightarrow (4).$$

The problem can be handled conveniently by discritization method. The Optimization interval 'T' is subdivided into 'm' sub intervals Each of the time length 'ΔT'. Over Each Subinterval it is assumed that all the Variables remains fixed in Value. Now the problem is represented as

$$\sum C'\left[P_{G_T}(t)\right]\Delta T = \sum_{m=1}^{M} C\left(P_{G_T}^m\right) \longrightarrow (5) \qquad m = 1, 2 \ldots\ldots M$$

q^m $(m = 1, 2, 3 \ldots\ldots M)$.

Under the following Constraints,

(i). power balance Equation

$$P_{G_T}^m + P_{GH}^m - P_D^m - P_L^m = 0 \longrightarrow (6).$$

Where, $P_{G_T}^m$ = Thermal Generation in the mth interval.

P_{GH}^m = Hydel Generation in the mth interval.

P_D^m = Load demand in the mth interval.

P_L^m = Transmission loss in the mth interval.

$$P_L^m = B_{TT}\left(P_{G_T}^m\right)^2 + B_{HH}\left(P_{G_H}^m\right)^2 + 2\,B_{TH}\,P_{G_T}^m\,P_{G_H}^m$$

$$\left[\because P_L = B_{11}P_{G_1}^2 + B_{22}P_{G_2}^2 + 2\,B_{12}P_{G_1}P_{G_2}\right]$$

(ii). Water Continuity Equation

$$x'^m - x'^{(m-1)} - I^m \Delta T + q^m \Delta T = 0$$

x'^m = water storage at the end of mth interval.

I^m = water in flow in the mth interval.

q^m = water discharge in m^{th} interval.

The above Equation can be written as

$$x^m - x^{m-1} - I^m + q^m = 0 \longrightarrow (7)$$

$$x^m = \frac{x'^m}{\Delta T} \; ; \; x^{m-1} = \frac{x'^{(m-1)}}{\Delta T} \qquad m = 1, 2, 3 \ldots\ldots M$$

(iii). Hydro generation in any sub-interval can be expressed as

$$P_{GH}^m = h_0 \{ 1 + 0.5 e (x^m + x^{m-1}) \} (q^m - \rho) \longrightarrow (8).$$

Where, $h_0 = 9.81 \times 10^{-3} h_0'$

h_0' = Basic water head corresponding to dead storage.

$$P_{GH}^m = 9.81 \times 10^{-3} h_{av}^m (q^m - \rho) \; MW$$

Where, $q^m - \rho$ = Effective discharge

h_{av}^m = average head in the mth interval

e = water head correction factor

ρ = Non-Effective discharge (water discharge needed to run hydel plant at no load).

In the above problem, formulation it is convinient to choose water discharges in all the sub intervals. Water discharge is in one of the sub-intervals is a dependent variable as shown below

$$x^m - x^0 - \sum I^m - \sum q^m = 0.$$

Solution Technique :

The Hydro thermal problem is Solved Using non-linear programming technique in Conjunction with the first orders gradient method.

* The Lagrangian 'L' is formulated by developing the Cost function of the Equation (5) with Equality Constraints Equations (6) to (8) through Lagrangian multipliers λ_1^m, λ_2^m, λ_3^m. We Can define lagrangian as

$$L = \sum c(P_{GT}^m) - \lambda_1^m \left[P_{GT}^m + P_{GH}^m - P_D^m - P_L^m \right] + \lambda_2^m \left[x^m - x^{m-1} - J^m + q^m \right] +$$

$$\lambda_3^m \left[P_{GH}^m - h_0 \{ 1 + 0.5\, e\, (x^m + x^{m-1}) \} (q^m - e) \right]. \longrightarrow (9)$$

The Lagrangian multipliers are Obtained by Equating to Zero, the Partial derivatives of the lagrangian with respect to dependent Variables results in the following Equations,

$$\frac{\partial L}{\partial P_{GT}^m} = 0 \Rightarrow \frac{d}{dP_{GT}^m} [C(P_{GT}^m)] - \lambda_1^m \left(1 - \frac{\partial P_L^m}{\partial P_{GT}^m}\right) = 0 \longrightarrow (10).$$

$$\frac{\partial L}{\partial P_{GH}^m} = 0 \Rightarrow 0 - \lambda_1^m \left(1 - \frac{\partial P_L^m}{\partial P_{GT}^m}\right) + \lambda_3^m = 0$$

$$\frac{\partial L}{\partial P_{GT}^m} = \lambda_3^m - \lambda_1^m \left(1 - \frac{\partial P_L^m}{\partial P_{GT}^m}\right) = 0 \longrightarrow (11).$$

$$\frac{\partial L}{\partial x^m} = \lambda_2^{m-1} - \lambda_2^{m+1} - \lambda_3^m \left[0.5\, h_0\, e\, (q^m - e)\right] - \lambda_3^{m+1} \left[0.5\, h_0\, e\, (q^{m+1} - e)\right] = 0$$

$$\longrightarrow (12)$$

Substitute (7) in (11) we get,

$$\frac{\partial L}{\partial q^1} = \lambda_2^1 - \lambda_3^1 h_0 \{ 1 + 0.5\, e\, (2x^0 + J^1 - 2q^1 + e) \} = 0 \longrightarrow (13).$$

The dual Variables for any Sub interval may be Obtained as follows :

(i). Obtain λ_1^m from Equation (10).

(ii). Obtain λ_2^m from Equation (11).

(iii). Obtain λ_3^m from Eq (12) and other Values of λ_2^1 from Eq (13).

$$\frac{\partial L}{\partial q^m} = 0 \Rightarrow \lambda_2^m + \lambda_3^m \left[-h_0 \left(1 + 0.5 e \left[X^m + X^{m-1} \right] \right) \right] = 0 \longrightarrow (14).$$

The above Equation is known as gradient Vector for the Optimality the gradient Vector should be zero.

Hydro thermal Scheduling problem :

It Can be Categorized into two. They are :

(i). Long term Scheduling problem

(ii). Short term Scheduling problem.

Long term Scheduling problem :

In long term Scheduling problem, the Scheduling time period is Considered to be 1 year. During this period, there is a large amount of Variation in the head of water which Cannot be neglected. The dynamics of head Variations is Studied from water Continuity Equation. Due to Variation in the head, determination of Co-ordination Equation becomes Complex.

The long term Scheduling problem is divided into three Categories

(i). Multi-storage Hydro Electric System.

(ii). Cascaded Hydro Electric System.

(iii). Multi-chain Hydro Electric System.

* If the location of Hydro plants are different streams and Each hydro plant has its Own reservior then Such hydroelectric System is Called as " multi storage hydro Electric System ".
* If the location of hydro plants are same streams and then Such hydro electric plant is Called " Cascaded Hydro electric system ".
* If the location of hydro plants are at different streams and Each plant follow the same stream then Such System is " Multichain hydro electric s/m ".

Short term Scheduling problem :

In short term Scheduling problem the Scheduling time is Considered to be 1 day or 1 week . As Scheduling period is Very Small , the head of the water in the reservior Can be assumed Constant. Short term Scheduling problem is divided into 2 Categories

(i). Constant head hydrothermal Scheduling :

The head of water in large capacity reservior is assumed to be Constant Over a period of the Operation.

(ii). Variable head hydrothermal Scheduling :

If the Capacity of the reservior is Small than the head of water in that reservior is Variable . The load demand should meet the total generation with losses. The Co-ordination Equation of short term Scheduling problem is Simple and Easy.

Short term Hydro thermal Scheduling by Kirchmayer's Method :

In this method, Co-ordination Equations are derived in terms of penalty factor of both methods for obtaining the optimal scheduling of the hydro thermal system and hence it is also known as "penalty factor method of Solution" of short term Hydro thermal Scheduling problem.

Let P_{GTi} = Power generation of ith thermal plant (in mw)

P_{GHj} = power generation of jth Hydel plant (in mw)

$\frac{dC_i}{dP_{GTi}}$ = Incremental fuel Cost of ith thermal plant (in Rs/mwhr)

W_j = Quantity of water Used for power generation at j^{th} hydro plant (m^3/s).

$\frac{dW_j}{dP_{GHj}}$ = Incremental water rate of jth hydro plant ($m^3/s/mw$).

$\frac{dP_L}{dP_{GTi}}$ = Incremental Transmission loss of ith thermal plant

$\frac{dP_L}{dP_{GHj}}$ = Incremental Transmission loss of jth Hydel plant.

λ = Lagrangian multiplier

γ_j = Constant that Converts incremental water rate of Hydel plant 'j' into incremental Cost.

n = No:of plants in total.

$(n-\alpha)$ = no:of Hydel plants

α = No:of Thermal plants.

T = Time interval during which the plant Operation is Considered.

Problem formulation :

The Objective function to minimize the Cost of generation is given by

$$\min \sum_{i=1}^{\alpha} \int_{0}^{T} C_i \, dt \longrightarrow (1)$$

Subject to the Equality Constraints we get

$$\sum_{i=1}^{\alpha} P_{GT_i} + \sum_{j=\alpha+1}^{n} P_{GH_j} = P_D + P_L \longrightarrow (2).$$

and $$\int_{0}^{T} W_j \, dt = K_j \quad \text{for } j = \alpha+1, \alpha+2 \ldots\ldots n \longrightarrow (3).$$

Where, K_j = Amount of water in m^3 utilized during the period 'T' in j^{th} hydro-plant.

W_j = Turbine discharge in the jth plant in m^3/s.

The Co-efficient 'γ' must be Selected So as to Use the Specified amount of discharge of water during the Operating period. Now objective function become

$$\min C = \sum_{i=1}^{\alpha} \int_{0}^{T} C_i \, dt + \sum_{j=\alpha+1}^{n} \gamma_j K_j$$

Substitute equation (3) in the above Equation we get,

$$\min C = \sum_{i=1}^{\alpha} \int_{0}^{T} C_i \, dt + \sum_{j=\alpha+1}^{n} \gamma_j \int_{0}^{T} W_j \, dt.$$

Modelling of Turbines :

Modelling :

The process of achieving mathematical Equivalent to the Existing physical System is known as modelling. The following Components Can be modelled to check the performance with help of Computers by means of Suitable Software packages. For Example Pspice, E-tap, E-CADD ... etc to monitor the Generator, Transformers, transmission lines, Turbines, field System, motors ... etc).

* In power System, both active and reactive powers are never steady and they Continously change with the arising and falling load.
* Steam input to the turbo generators (or) water input to hydro generators must to regulated Continously to match active power demand, failing which the machine speed will Vary with Consequent change in frequency and it is Undesirable.
* The maximum permissible change in frequency is ± 2 %.
* The Excitation of the generators must be regulated Continously to match the reactive power demand with the reactive power generation, otherwise Voltages at Various buses may go beyond the permissible limits.
* In modern large interConnected Systems, manual regulation is not feasible and therefore automatic generation and Voltage regulation Equipment is installed On Each generator.

Speed governing System of Turbine :

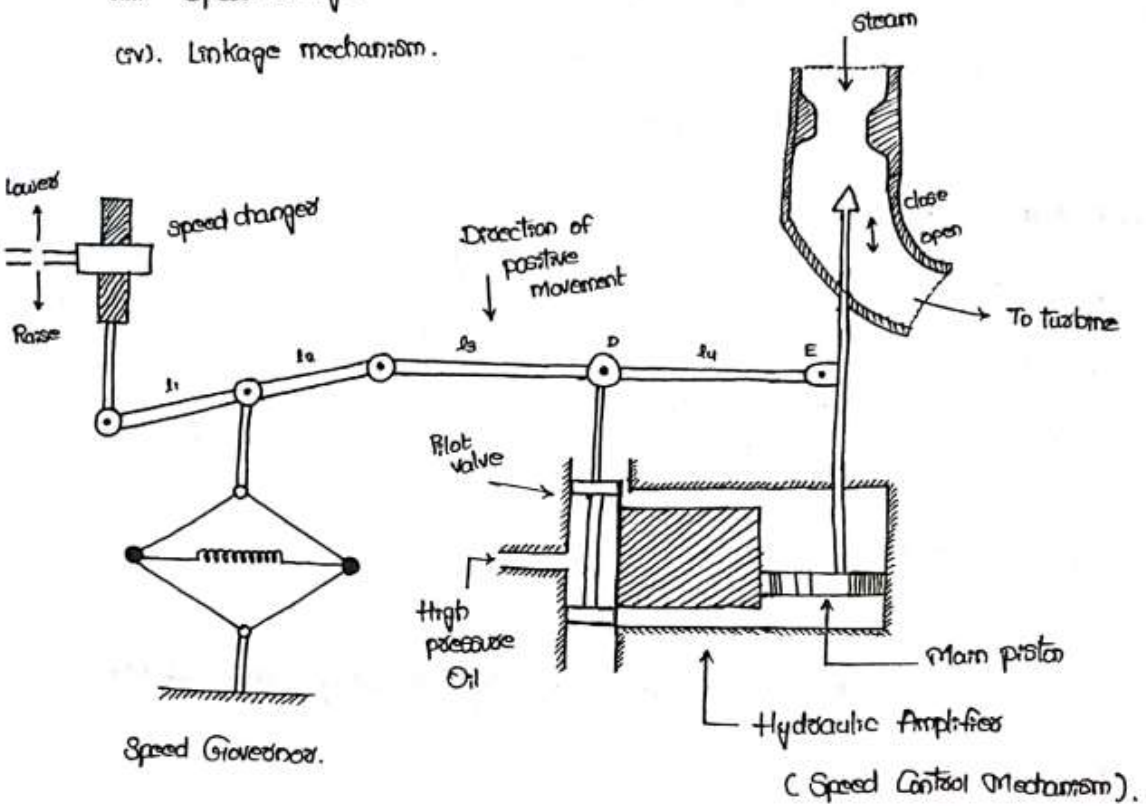
Fig shows the Schematic diagram of a Speed governing System which Controls the real power flow in the power System. The Speed governing System Consists the following parts. They are :

(i). Fly ball Speed governor

(ii). Hydraulic amplifier

(iii). Speed changer

(iv). Linkage mechanism.



Fly ball Speed governor :

* This is the heart of the System which Senses the change in the Speed (frequency). With the increase in Speed the fly ball move Outwards and the point 'B' On linkage mechanism move downwards and Vice Versa.

Hydraulic Amplifier :

* It Comprises pilot Valve and main piston arrangement.
* Low power level pilot Valve movement is Converted into high power level pilot Valve movement. This is necessary in Order to Open or close Steam Valve against high pressure Steam.

Linkage Mechanism :

ABC is a origid link pivoted at 'B' and CDE is another origid link pivoted at 'D'. This link mechanism provides a movement to Control the Valve in proportion to change in Speed. It also provides a feedback from the Steam Valve movement.

Speed changer :

* It provides a steady state power Output setting for turbine.
* Its downward movement Opens the upper pilot Valve. So that more Steam is admitted to the turbine Under steady Conditions. (Hence more steady power Output). The oreverse happens for upward movement of the Speed changer.

Modelling of Speed governor System :

Assume that the System is initially Operating Under steady State Conditions

(i). The linkage mechanism is stationary.
(ii). pilot Valve is closed.
(iii). Steam Valve is Opened by a definite magnitude.
(iv). Turbine is orunning at Constant Speed.

Let f^0 = System frequency

P_G^0 = Generation Output (or) Turbine Output (neglecting Generator losses).

X_E^0 = Steam Valve Setting Corresponding to initial Conditions.

Let the point 'A' on the linkage mechanism moved downwards by an amount ΔX_A. It is a command which causes the turbine output to change (increase in power ΔP_C (commanded power) and can be written mathematically as

$$\boxed{\Delta X_A = K \Delta P_C}$$

* The moment of linkage point 'A' causes a small position changes ΔX_C and ΔX_D of the linkage point 'C' and 'D'.
* If the moment of 'D' upwards by ΔX_D high pressure oil flows into the hydraulic amplifier from the top of main piston thereby steam valve will move downwards by the small distance, ΔX_E which results increase in the power.

Let us model these events mathematically. Two factors contribute to the moment of 'C'. They are :

(i). Increase in frequency 'Δf' causes the flyball to move outwards. So that 'B' move downwards by ΔX_B. This contribution is positive and given by $K_1 \Delta f$.

(ii). The lowering of speed changes by an amount ΔX_A lifts the point 'C' upwards by an amount proportional to ΔX_A.

* Net moment of 'C' is given by

$$\Delta X_C = K_1 \Delta f - K_2 \Delta P_C \quad [\Delta X_A \propto \Delta P_C] \longrightarrow (1).$$

The constants K_1 and K_2 depending upon length of the linkage arms AB and BC. It is proportional to the constants of speed changes and also the speed governor.

* The moment of 'D' is Contributed by moment of 'C' and 'E'. When 'D' moves upwards C & E go downwards.

∴ Net moment of 'D' is given by $\Delta X_D = K_3 \Delta X_C + K_4 \Delta X_E \longrightarrow (2)$.

Constants K_3, K_4 depends On linkage arms CD & DE.

Assume that Volume of Oil admitted to the Cylinder is proportional to the time integral of ΔX_D. The Value of ΔX_E is given by

$$\Delta X_E = K_5 \int_0^t -\Delta X_D \, dt \longrightarrow (3).$$

Constant K_5 depends On the fluid pressure and geometry of Orifice & Cylinder

Taking laplace transform for above (1), (2), (3) Equations.

$$\Delta X_C(s) = K_1 \Delta F(s) - K_2 \Delta P_C(s) \longrightarrow (4)$$

$$\Delta X_D(s) = K_3 \Delta X_C(s) + K_4 \Delta X_E(s) \longrightarrow (5)$$

$$\Delta X_E(s) = \frac{-K_5}{s} \Delta X_D(s) \longrightarrow (6).$$

put (5) in (6), we get

$$\Delta X_E(s) = -\frac{K_5}{s} [K_3 \Delta X_C(s) + K_4 \Delta X_E(s)]$$

$$\Delta X_E(s) + \frac{K_5 K_4}{s} \Delta X_E(s) = -\frac{K_5 K_3}{s} \Delta X_C(s)$$

$$\Delta X_E(s) \left[1 + \frac{K_5 K_4}{s}\right] = -\frac{K_5 K_3}{s} \Delta X_C(s)$$

$$\Delta X_E(s) \left[1 + \frac{K_5 K_4}{s}\right] = -\frac{K_5 K_3}{s} [K_1 \Delta F(s) - K_2 \Delta P_C(s)]$$

$$\Delta X_E(s) \left[1 + \frac{K_5 K_4}{s}\right] = -\frac{K_1 K_3 K_5}{s} \Delta F(s) + \frac{K_2 K_3 K_5}{s} \Delta P_C(s).$$

$$\Delta X_E(s) \left[\frac{s + K_5 K_4}{s}\right] = \left[-K_1 K_3 K_5 \Delta F(s) + K_2 K_3 K_5 \Delta P_C(s)\right] \frac{1}{s}$$

$$\Delta X_E(s) \left[\frac{s}{K_5} + K_4\right] K_5 = [K_3 K_2 \Delta P_C(s) - K_3 K_1 \Delta F(s)] K_5$$

$$\Delta X_E(s)\left[\frac{s}{K_5}+K_4\right] = K_3K_2\,\Delta P_c(s) - K_3K_1\,\Delta F(s)$$

$$\Delta X_E(s) = \frac{K_2K_3\,\Delta P_c(s) - K_3K_1\,\Delta F(s)}{\left[\frac{s}{K_5}+K_4\right]}$$

The above Equation Can also be written as

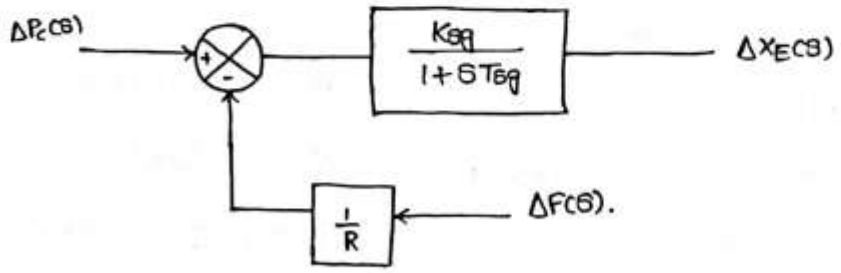
$$\Delta X_E(s) = \frac{K_{eg}}{1+sT_{eg}}\left[\Delta P_c(s) - \Delta F(s)\cdot\frac{1}{R}\right]$$

Where, $R = \frac{K_2}{K_1}$ = Speed oregulation of Goverenor

$K_{eg} = \frac{K_2K_3}{K_4}$ = Gain of Speed Goverenor

$T_{eg} = \frac{1}{K_4K_5}$ = Time Constant of Speed Goverenor.

The block diagram oreprosentation of Speed governing System for the Steam turbine is given by



Note :

* Time Constant of Speed Goverenor (T_{eg}) is less than 100 ms.

Modelling of Turbine :

The model requires relation between changes in power Output of the Steam turbine to the changes in steam Valve Opening ΔX_E.

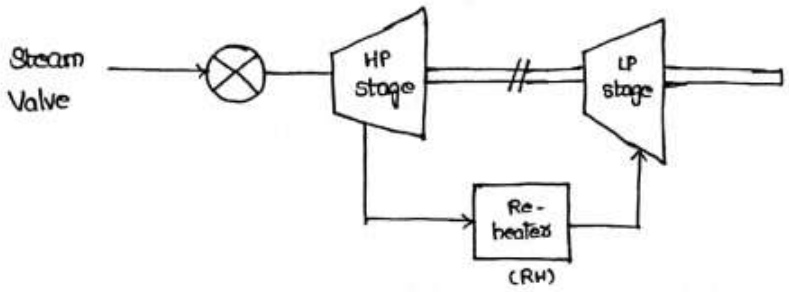
* A steam turbine Converts stored Energy of high temperature and high pressure steam into rotating Energy which inturn Converted into Electrical Energy by the generator. Generally, Steam turbines are Usually of two types

(i). Reheat type

(ii). Non-Reheat type.

(i). Reheat Type Steam turbine :

fig shows two stage turbine with Reheat Unit



* The high pressure and low pressure stages are there in reheated type.
* In Reheated type, the steam when leaving H.P Sections returns to the boiler i.e passed to the reheater (RH) before returning to the I.P (Intermediate pressure) section. Reheating improves Efficiency of the turbine.
* In Speed governing System, the System Valve Opens and large amount of steam Enters the turbine.
* The high pressure steam Enters the HP Section of the turbine. Here in the high pressure Section the pressure will be dropped and temperature decreases.

To get Constant temperature, the Steam from high pressure Section

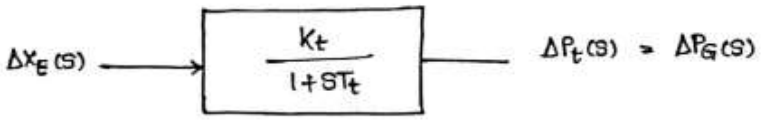
is given to the boiler. Here the pressure is kept Constant and temperature increases to the rated Value. This steam return given to LP Section from boiler.

* The steam from the L.P Section is dumped to Condenser, the steam is Converted to water and is fedback to the boiler. Thus mechanical Output of three Sections is Coupled together to give rated Output Speed. The boiler is named as Reheater.
* Reheater turbines have more than One time Constant.
* The dynamic response of two stage turbine System is influenced by
  (i). Steam which lies between the System inlet Valve and 1st stage of turbine.
  (ii). Reheater storage action due to which Output of low pressure stage to lag behind that of high pressure stage.

  Thus, turbine is transfer function is characterised by '2' time Constants

For Ease of analysis, it will be assumed that turbine Can be modelled to have a single Equivalent time Constant.

* Typically, time Constant 'T_t' lies in the range of 0.2 to 2.5 secs.
* Fig shows transfer function model of the steam turbine



$$\frac{\Delta P_t(s)}{\Delta X_E(s)} = \frac{K_t}{1+sT_t}$$

Note :

* On Opening steam Valve, the steam flow will now reach the turbine Cylinder instantaneously. The time delay Experienced On this is of Order of 2s in steam pipe.

(ii). Non-Reheat Turbine Model :

* In non-reheat type turbine, there is neither an Intermediate pressure section nor Reheater Unit. In this Case, the Steam is directly passed to the LP Section.

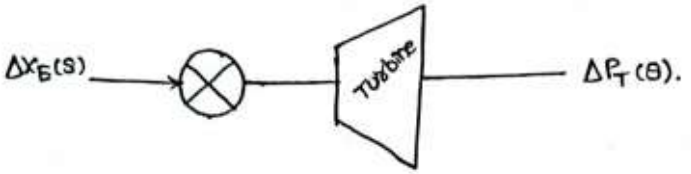
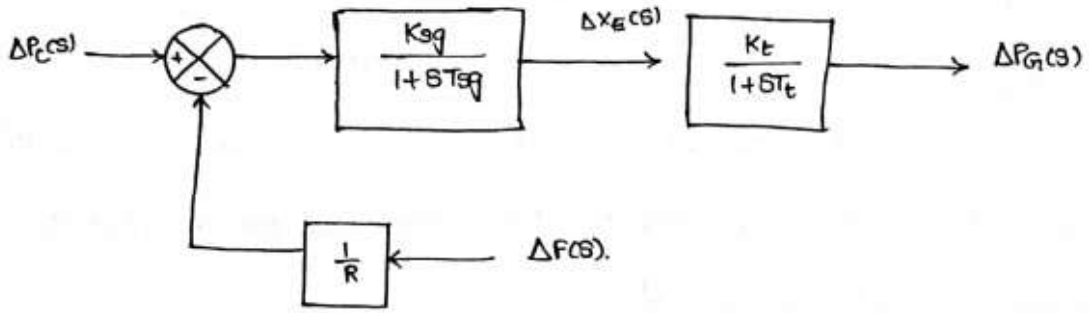


Fig. shows linearised model of non-Reheat turbine along with Speed governing mechanism.



From the figure, the Combined transfer-function of turbine and Speed governor mechanism will be $\frac{K_{sg} \cdot K_t}{(1+sT_{sg})(1+sT_t)}$.

$$\therefore \quad \frac{\Delta P_{G1}(s)}{\left[\Delta P_C(s) - \frac{1}{R}\Delta F(s)\right]} = \frac{K_{sg}\,K_t}{(1+sT_{sg})(1+sT_t)}$$

Generator Load Model :

The generator - load model gives the relation between the change in the frequency (Δf) as a result of change in generation when the load changes by a small amount.

Let ΔP_D be change in load as a result the generation also swings by an amount ΔP_G. The net power surplus at the busbar is $(\Delta P_G - \Delta P_D)$. This surplus power can be absorbed by the system into two different ways. They are :

(i). By increasing the stored kinetic energy of generator rotors at a rate $\frac{dw}{dt}$. Where 'w' is the new value of kinetic energy.

Let 'w°' be the kinetic energy before change in load occurs. at the normal speed and frequency f°.

Let 'w' be the kinetic energy when frequency is $f^\circ + \Delta f$.

* Since kinetic energy (KE) is proportional to square of the speed of the governor, therefore two kinetic energies can be correlated as

$$\frac{w}{w_0} = \left(\frac{f_0 + \Delta f}{f^\circ}\right)^2$$

$$w = w_0 \left(1 + \frac{2\Delta f}{f^\circ} + \text{.....Higher Order terms}\right)$$

Neglecting higher order terms. Since $\frac{\Delta f}{f^\circ}$ is small.

$$w = w^\circ \left(1 + \frac{2\Delta f}{f^\circ}\right)$$

Differentiating above expression w.r.t 't' we get

$$\frac{dw}{dt} = \frac{2w^\circ}{f^\circ} \cdot \frac{d}{dt}(\Delta f).$$

Let H = Inertia Constant of a generator (MW-s/MVA)

P = Rating of the turbo generator (MVA).

3. LOAD FREQUENCY CONTROL

Single Area Load frequency control

Necessity of keeping frequency constant. Definitions of control area - single area control - Block diagram representation of isolated power system - steady state analysis - Dynamic response - Uncontrolled case.

In a power system, both active and reactive power demands are never steady and they continuously change with rising or falling trend.

Steam input to turbo-generators (or) water-input to hydro generators must be regulated continuously to match the active power demand, failing which consequent change in frequency and it may be highly undesirable.

The excitation of the generators must be regulated continuously to match the reactive power demand with reactive power generation, otherwise, the voltages at various system buses may go beyond the prescribed limits.

→ The maximum possible change in frequency is ±5%

Necessity of keeping frequency constant:

1. Loads are usually designed to operate at a particular frequency. If it is not maintained at the normal value, then frequency of operation causes electrical loads to deviate from the desired output.
2. For synchronous operation of various units in the power system network, it is necessary to maintain frequency constant.
3. Steam turbine blades are designed to operate in a narrow band of frequencies.

Deviation of frequency beyond this band may cause gradual (or) immediate turbine damage.
So, control equipment take action in case of under/over frequency.

4. Frequency affects the amount of power transmitted through interconnecting lines.

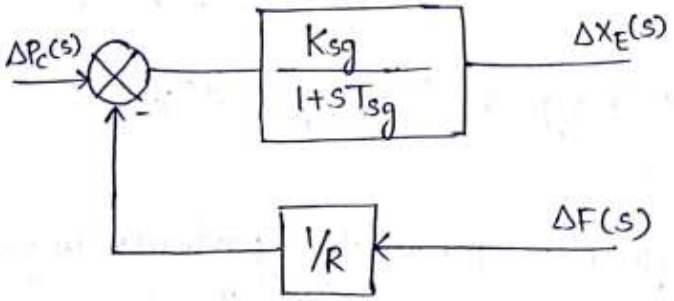
5. Electrical clocks which are driven by synchronous motors will lose or gain time in case of change in frequency.

Block diagram representation of an isolated power system

A complete block diagram representation of an isolated power system comprising speed governor, turbine and generator - load model.

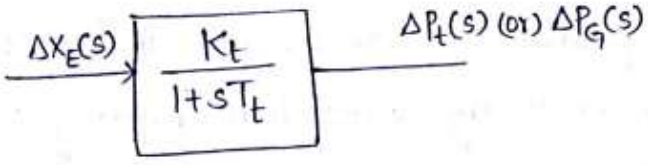
Speed governor model:

The block diagram of speed governor is shown.



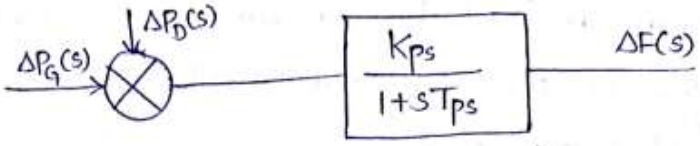
Turbine model:

The transfer function of non-reheat type is shown:



Generator - load model:

The transfer function of generator-load model is shown



By combining the three block diagrams i.e.,
Speed-governor model,
turbine model and
generator-load model
we get complete block diagram representation of LFC.

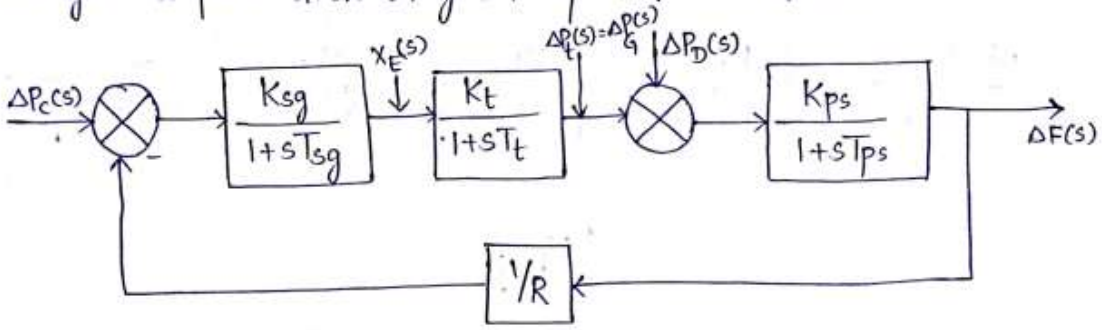


Fig. Block diagram representation of load frequency control (Isolated power system)

Steady state Analysis:

There are two incremental inputs to the load frequency control system.

ΔP_C – change in speed changer setting

ΔP_D – change in load demand.

Note: Relation between T_{sg}, T_t and T_{ps}

$$T_{sg} < T_t < T_{ps}$$

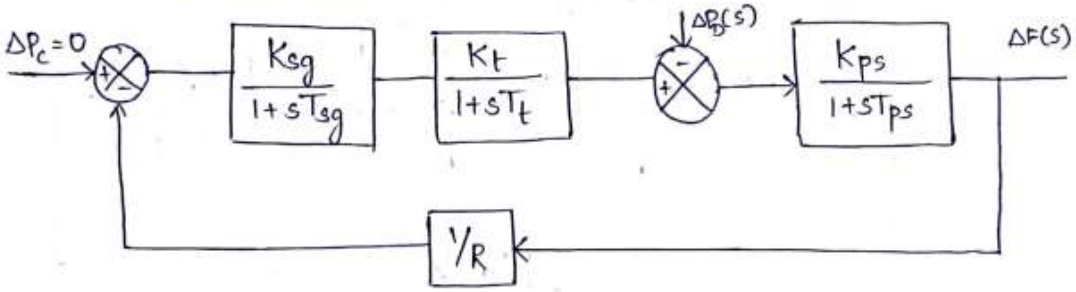
Constant speed changer position with variable load demand i.e., $\Delta P_c = 0$.

Consider a case wherein speed changer has a fixed setting, which means $\Delta P_c = 0$ and the load demand changes. Such operation is known as free governor operation (or) uncontrolled case since the speed changer is not manipulated.

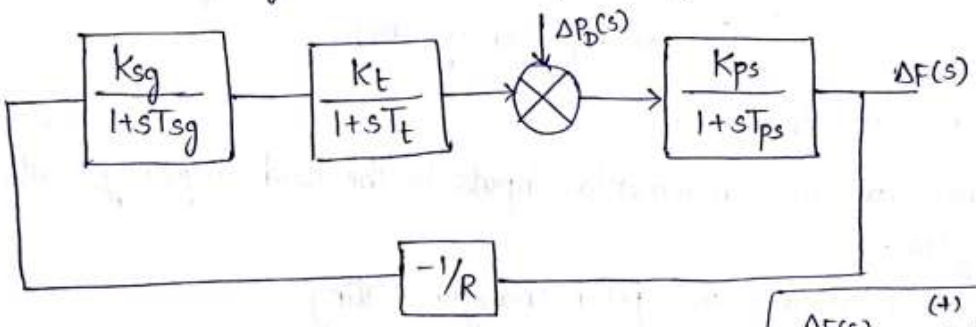
For a sudden step change of load demand

$$\Delta P_D(s) = \frac{\Delta P_D}{s}$$

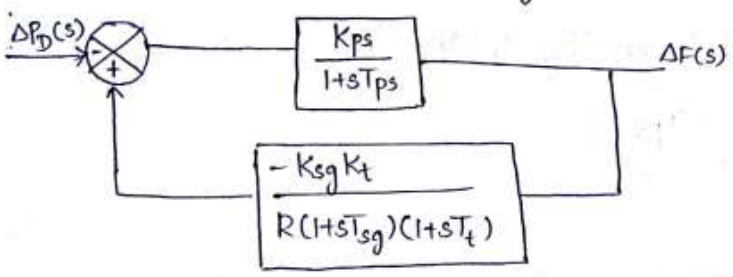
For such an operation, steady state change of frequency $\Delta F(s)$ is to be estimated from the block diagram.



The block diagram of isolated power system now becomes..

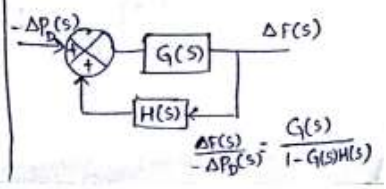


Combining blocks in series, we get



$$\frac{\Delta F(s)}{\Delta P_D(s)} = \frac{G(s)}{1 - G(s)H(s)} \quad (+)$$

$$= \frac{G(s)}{1 + G(s)H(s)} \quad (-)$$



$$\frac{\Delta F(s)}{-\Delta P_D(s)} = \frac{G(s)}{1 - G(s)H(s)}$$

$$\left.\frac{\Delta F(s)}{\Delta P_D(s)}\right|_{\Delta P_C(s)=0} = -\left[\frac{\left(\frac{K_{ps}}{1+sT_{ps}}\right)}{1+\frac{K_{sg}\cdot K_t\cdot K_{ps}}{(1+sT_{sg})(1+sT_t)(1+sT_{ps})}\cdot\frac{1}{R}}\right]$$

For a sudden step change of load demand

$$\Delta P_D(s) = \frac{\Delta P_D}{s}$$

Multiply with $(1+sT_{ps})$ both numerator and denominator

$$\Delta F(s) = -\left[\frac{K_{ps}}{(1+sT_{ps})+\frac{K_{sg}\cdot K_t\cdot K_{ps}}{(1+sT_{sg})(1+sT_t)}\cdot\frac{1}{R}}\right]\times\frac{\Delta P_D}{s}$$

Apply final value theorem

The steady state frequency error is given by

$$\Delta f\,|_{\text{steady state}} = \lim_{s\to 0} s\cdot\Delta F(s)$$

$$\Delta f = \lim_{s\to 0} s\left\{-\frac{K_{ps}}{(1+sT_{ps})+\frac{K_{sg}K_tK_{ps}}{(1+sT_{sg})(1+sT_t)}\cdot\frac{1}{R}}\right\}\cdot\frac{\Delta P_D}{s}$$

$$= \left(\frac{-K_{ps}}{1+\frac{K_{sg}K_tK_{ps}}{R}}\right)\Delta P_D \longrightarrow (2)$$

where K_{sg}, K_t, K_{ps} are gains of speed governor, turbine and power system respectively.

The gain K_{sg} of the speed governor is easily adjustable by changing the lengths of various links of the linkage mechanism.

K_{sg} is adjusted such that $K_{sg}K_t = 1$

$$\Delta f = -\left(\frac{K_{ps}}{1+\frac{K_{ps}}{R}}\right)\Delta P_D$$

From the dynamics of generator-load model.

$K_{ps} = \frac{1}{D}$, where $D = \frac{\partial P_D}{\partial f}$ MW/Hz

$= \left(\frac{\partial P_D}{\partial f}\right)\Big/_{P.U.}$ in P.U.

$$\Delta f = -\left(\frac{1/D}{1+\frac{1}{D}\cdot\frac{1}{R}}\right)\Delta P_D$$

Multiply with D both in numerator and denominator

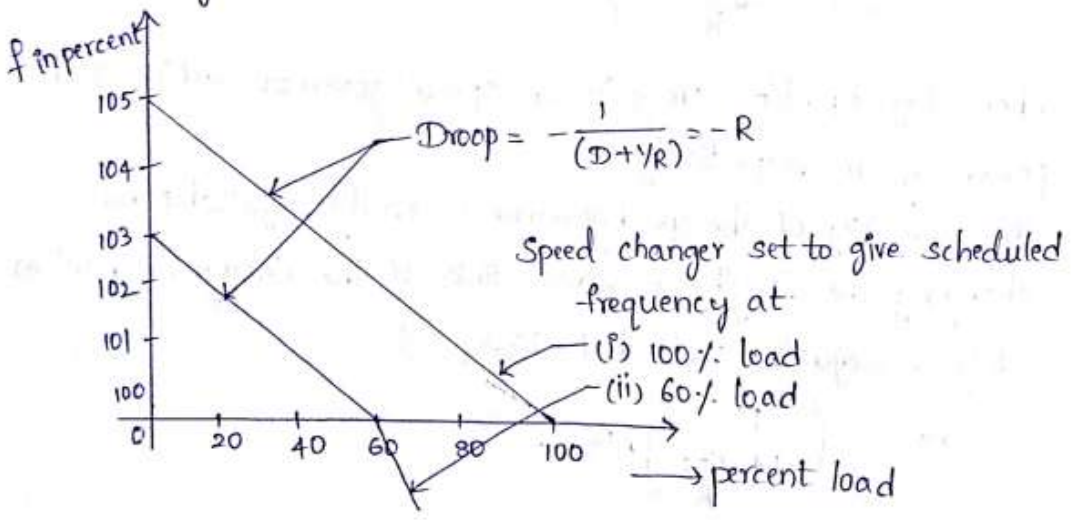
$$= -\left(\frac{1}{D+1/R}\right)\Delta P_D$$

$$\boxed{\Delta f = -\frac{1}{\beta}\Delta P_D}$$

where $\beta = (D+1/R)$ and is known as area frequency response characteristics (~~ARFC~~) (AFRC) (or) area frequency regulation characteristics.

$$\Delta f = -\left(\frac{1}{D+1/R}\right)\Delta P_D$$

The above equation gives steady state response of frequency to change in demand.



Power system parameter D is generally much smaller than $1/R$ ($D = 0.01$ pu ; $1/R = 1/3$) so that D can be neglected in comparison.

Above equation is reduced to

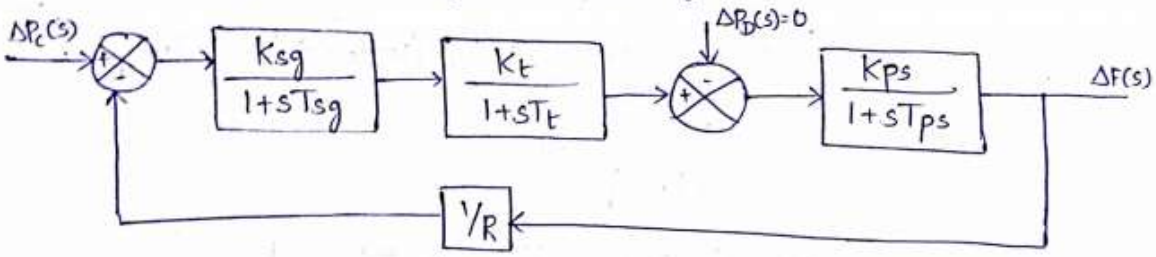
$$\Delta f = -R.(\Delta P_D)$$

The droop of the load-frequency curve is mainly determined by R, the speed governor regulation.

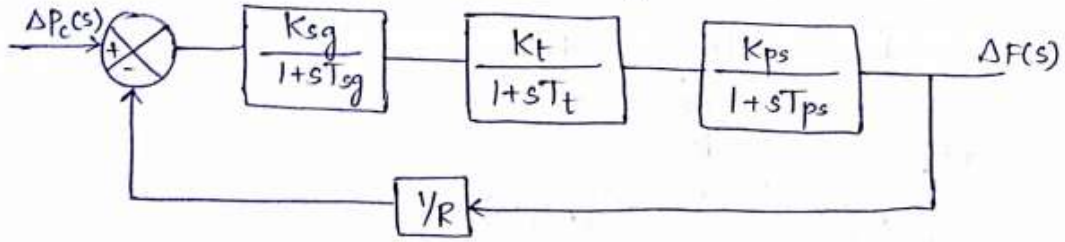
Case 2: Constant load-demand position with variable speed changer (i.e., $\Delta P_D = 0$)

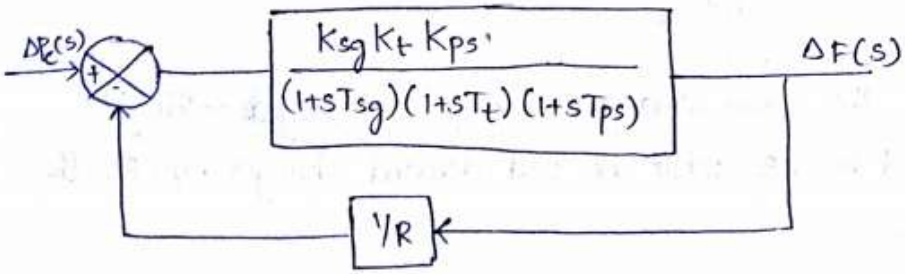
Consider steady effect of changing speed changer setting ($\Delta P_C(s) = \frac{\Delta P_C}{s}$) with load demand remaining fixed (i.e., $\Delta P_D = 0$).

→ The steady state change in frequency is obtained as follows.



The block diagram of isolated power system now becomes,





The transfer function is given by

$$\frac{\Delta F(s)}{\Delta P_c(s)} = \frac{\left(\frac{K_{sg} \cdot K_t \cdot K_{ps}}{(1+sT_{sg})(1+sT_t)(1+sT_{ps})}\right)}{1+\left(\frac{K_{sg} \cdot K_t \cdot K_{ps}}{(1+sT_{sg})(1+sT_t)(1+sT_{ps})}\right)\left[\frac{1}{R}\right]}$$

For a sudden step change in speed changer, $\Delta P_c(s) = \frac{\Delta P_c}{s}$

Multiply numerator and denominator with $(1+sT_{sg})(1+sT_t)(1+sT_{ps})$

$$\Delta F(s) = \left[\frac{K_{sg} K_t K_{ps}}{(1+sT_{sg})(1+sT_t)(1+sT_{ps}) + \left(\frac{K_{sg} K_t K_{ps}}{R}\right)}\right] \times \frac{\Delta P_c}{s}$$

The steady state frequency error is given by

$$\Delta f\Big|_{\Delta P_D = 0} = \lim_{s \to 0} s \cdot F(s)$$

$$= \lim_{s \to 0} s \left(\frac{K_{sg} K_t K_{ps}}{(1+sT_{sg})(1+sT_t)(1+sT_{ps}) + (K_{sg} K_t K_{ps})/R}\right) \cdot \frac{\Delta P_c}{s}$$

$$\Delta f = \frac{K_{sg} K_t K_{ps}}{1 + (K_{sg} \cdot K_t \cdot K_{ps})/R} \cdot \Delta P_c$$

Let us consider $K_{sg} \cdot K_t \simeq 1$, $K_{ps} = 1/B$ where $B = \frac{\partial P_D}{\partial f}$

The above equation becomes

$$\Delta f = \frac{1/B}{1 + \frac{(1/B)}{R}} \cdot \Delta P_c$$

$$\boxed{\Delta f = \left(\frac{1}{B + 1/R}\right) \Delta P_c}$$

Note: If the speed changer setting is changed setting is changed by ΔP_c while the load demand changes by ΔP_D, the

· steady frequency change is obtained by superposition.

i.e., $\Delta f = \left(\frac{1}{B + 1/R}\right)(\Delta P_C - \Delta P_D)$

Dynamic Response - (Uncontrolled case i.e., $\Delta P_C = 0$)

The dynamic response is how fast the frequency changes as a function of time immediately after disturbance before it reaches the new steady-state condition.

Comparing the relative values of time constants, we reduce the third order LFC model to first order

For a practical LFC system

$$T_{sg} < T_t << T_{ps}$$

Typical values are: $T_{sg} = 0.4s$; $T_t = 0.5s$, $T_{ps} = 20s$

If T_{sg} and T_t are considered negligible compared to T_{ps} and by adjusting $K_{sg} K_t \simeq 1$, the block diagram of LFC of the power system of an isolated system is reduced to first order system as shown

For uncontrolled case, $\Delta P_C(s) = 0$

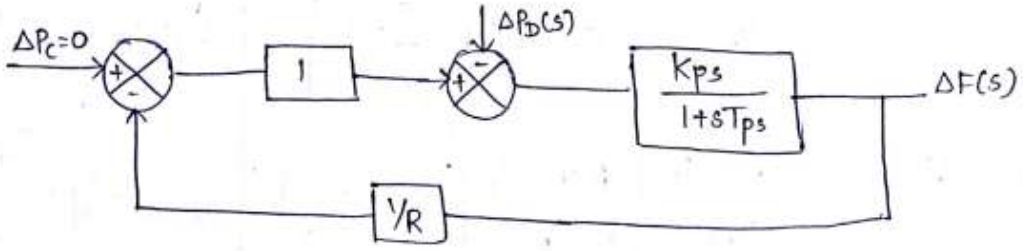
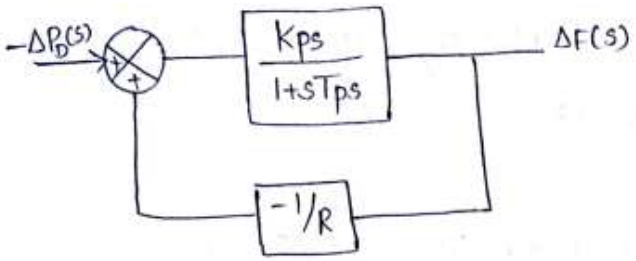


Fig. First order approximate block diagram of LFC of an isolated area



$$\frac{\Delta F(s)}{-\Delta P_D(s)} = \frac{\frac{K_{ps}}{1+sT_{ps}}}{1-\left(\frac{K_{ps}}{1+sT_{ps}}\right)\left(\frac{-1}{R}\right)} = \frac{\frac{K_{ps}}{1+sT_{ps}}}{1+\left(\frac{K_{ps}}{1+sT_{ps}}\right)\left(\frac{1}{R}\right)}$$

Multiplying both numerator and denominator by $(1+sT_{ps})$

$$\Delta F(s) = \left[\frac{K_{ps}}{(1+sT_{ps})+\frac{K_{ps}}{R}}\right]\frac{-\Delta P_D}{s}$$

$$= \left[\frac{-K_{ps}}{sT_{ps}+\left(\frac{R+K_{ps}}{R}\right)}\right]\frac{\Delta P_D}{s}$$

Dividing by T_{ps}

$$= \left[\frac{\left(\frac{-K_{ps}}{T_{ps}}\right)}{s+\left(\frac{R+K_{ps}}{RT_{ps}}\right)}\right]\frac{\Delta P_D}{s}$$

$$= \frac{-K_{ps}\cdot\Delta P_D}{T_{ps}}\left[\frac{1}{s\left\{s+\underbrace{\left(\frac{R+K_{ps}}{RT_{ps}}\right)}_{a}\right\}}\right]$$

$$\Delta F(s) = \frac{-K_{ps}\cdot\Delta P_D}{T_{ps}}\left(\frac{RT_{ps}}{R+K_{ps}}\right)\left[\frac{1}{s}-\frac{1}{s+a}\right]$$

$$\frac{1}{s(s+a)} = \frac{A}{s}+\frac{B}{s+a}$$

$$= \frac{1/a}{s}+\frac{-1/a}{s+a}$$

$$\frac{1}{s(s+a)} = \frac{1}{a}\left[\frac{1}{s}-\frac{1}{s+a}\right]$$

By taking inverse laplace

$$\Delta f(t) = L^{-1}\left[\Delta F(s)\right]$$

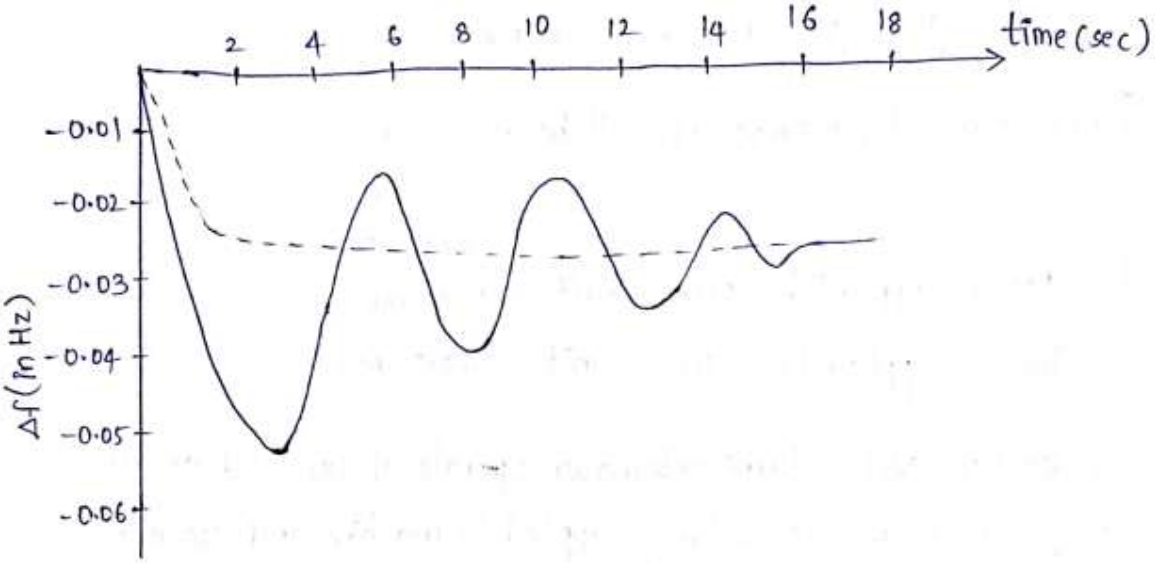
$$= L^{-1}\left[\frac{-RK_{ps}\cdot\Delta P_D}{R+K_{ps}}\cdot\left(\frac{1}{s}-\frac{1}{s+a}\right)\right]$$

$$\Delta f(t) = \frac{-R\,K_{ps}}{R+K_{ps}}\left[1-e^{-at}\right]\cdot\Delta P_D \qquad \text{where } a = \frac{R+K_{ps}}{R\,T_{ps}}$$

***

$$\Delta f(t) = \frac{-R K_{ps}}{R+K_{ps}}\left[1-e^{\frac{-t}{T_{ps}}\left(\frac{R+K_{ps}}{R}\right)}\right]\Delta P_D$$

The dynamic response of change in frequency for a step-change in load is as follows



① Two turbo alternators rated for 110 MW and 210 MW have governor droop characteristics of 5% from no-load to full-load. They are connected in parallel to share a load of 250 MW. Determine the load shared by each machine. Assuming free governor action.

Sol:- Since the two machines are working in parallel, the percent drop in frequency from the machines due to different loadings must be same.

Let load on generator 1 of 110 MW unit = x

Load on generator 2 of 210 MW unit = $(250-x)$

Percent drop in speed $= \frac{5}{110}x$ ⟶ ①

Percent drop in speed $= \frac{5}{210}(250-x)$ ⟶ ②

As percent drop in frequency must be same

From eqns ① & ②

$$\frac{5x}{110} = \frac{5}{210}(250-x)$$

$$\frac{x}{11} = \frac{250-x}{21} \Rightarrow x = 85.93 \text{ MW}$$

Power shared by 210 MW unit will be $= 250 - x$

$= 250 - 85.93$

$= 164.07$ MW

i.e., power supplied by 110 MW unit $= 85.93$ MW

Power supplied by 210 MW unit $= 164.07$ MW

② A 100 MVA, 50 Hz turbo alternator operate at no-load at 3000 rpm. A load of 25 MW is suddenly applied to machine and steam valve to the turbine commence to open after 0.6 sec due to time-lag in the governor system. Assuming inertia constant H of 4.5 MW-sec/KVA of generator capacity, calculate the frequency to which the generated voltage droops before the steam flow commences to increase to meet the new load.

Sol:- By definition, $H = \frac{\text{Stored energy (K.E.)}}{\text{Capacity of machine}}$

The energy stored at no-load $= 4.5 \times 100 = 450$ MJ $= W^0$

W^0 = Inertia constant x rating of turbo generator

Before the steam valve open,

The energy lost by the rotor $= 25 \times 0.6 = 15\ MJ = W$

As a result of this, there is a reduction in speed of the rotor and therefore reduction in frequency.

$$\left(\frac{f}{f^0}\right)^2 = \left(\frac{W}{W^0}\right)$$

W = new value

W^0 = old value

$$f = f^0\sqrt{\left(\frac{W}{W^0}\right)}$$

$(f^0)^2 \longrightarrow W^0$

$(f)^2 \longrightarrow W$

$$= 50 \times \sqrt{\frac{(450-15)}{450}}$$

$$= 49.16\ Hz$$

$$\text{Frequency deviation} = \frac{50 - 49.16}{50} \times 100$$

$$= 1.68$$

(3) Two generators rated 200 mw and 400 mw are operating in parallel. The droop characteristics of their governors are 4% and 5% respectively from no-load to full-load. Assuming that the generators are operating at 50 Hz at no-load. how would a load of 600 mw be shared between them? What will be the system frequency at this load? Assume free governor operation.

Sol:- Since the generators are in parallel, they will operate at the same frequency at steady load.

Let load on generator 1 of 200 mw $= x$ mw

and load on generator 2 of 400 mw $= (600 - x)$ mw

Let

Reduction in frequency $= \Delta f$

Reduction in frequency of 200 MW will be $\left(\frac{4}{200}\right)x$

Reduction in frequency of 400 MW will be $\left(\frac{5}{400}\right)(600-x)$

Equating above eqns, we get

$x = 231$ MW (load on gen 1)

$600 - x = 369$ MW (load on gen 2)

System frequency $= 50 - \frac{0.04 \times 50}{200} \times 231$

$= 47.69$ Hz

④ A 100 MVA synchronous generator operates on full-load at frequency of 50 Hz. The load is suddenly reduced to 50 MW. Due to time lag in governor system, the steam value begins to close after 0.4 sec. Determine the change in frequency that occurs in this time. Assume H = 5 KW-sec/KVA

Sol:- Given,

H = 5 KW-sec/KVA of generator capacity

K.E. stored in rotating parts of generator and turbine (rotor)

K.E. = H.P.

$= 5$ KW-sec/KVA $\cdot 100 \times 10^3$ KVA

$= 500 \times 10^3$ K.W-sec $= 500$ MJ

Energy gained by the rotor = 50 MW. 0.4 sec

$= 20$ MW-sec $= 20$ MJ

Stored K.E. $\propto$ (frequency)^2

$f = f^\circ \sqrt{\frac{W}{W_0}} = 50 \sqrt{\frac{500+20}{500}}$

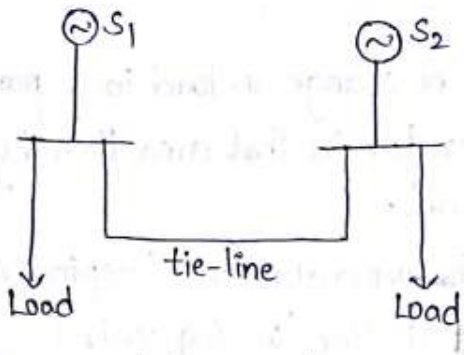
$f = 51$ Hz

Two-area load frequency control:

Load frequency control of 2-area system - Uncontrolled case and controlled case, tie-line bias control

Tie-line bias control:

If the system consists of a single machine connected to a group of loads the speed and frequency change in accordance with the governor characteristic as the load change. If system consists of two machines running in parallel as shown.



The possibility of sharing the load by two machines is as follows:

Tie-line: The line which connects two control areas (or two generators, here) is called tie-line.

If the change in load is either S_1 or S_2, but the generation of S_1 alone is regulated to adjust this change so as to have constant frequency. The method of regulation is known as "flat frequency regulation". Under such conditions S_2 is said to be operating at base load.

The major drawback of flat frequency regulation is that S_1

must absorb all load changes for the entire system. Therefore the tie-line between two stations would have to absorb all load changes.

Since the generator at S_2 would maintain its output constant. (The operation of generation S_2 on base load has the advantages when S_2 is much more efficient than the other station and it is desirable to obtain maximum output)

→ The other possibility of sharing the change in load is both S_1 and S_2 would regulate their generators to maintain the frequency constant. This is known as "parallel-frequency regulation".

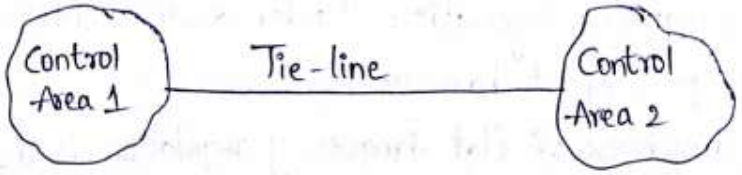
→ The third possibility is that the change in load in a particular area is taken care by the generator in that area there-by tie-line loading remains constant.

This method of regulating the generation for keeping constant frequency is known as "Flat tie-line loading control".

Load frequency control of two-area system:

An extended power system can be divided into a number of load frequency control (LFC) areas which are interconnected by tie-lines. Such an operation is called "pool operation".

Consider two control areas connected by a single tie-line as shown



The control objective is to regulate the frequency of each area and simultaneously regulate the tie-line power as per inter-area power agreement.

In case of an isolated control area, proportional plus Integral Controller (PI) will be installed so as to give zero steady state error (i.e., $\Delta f_{(ss)} = 0$)

In two-area control case, PI controller will be installed to give zero steady state error in tie-line power (i.e., $\Delta P_{tie} = 0$) in addition to PI controller to give zero steady state error in frequency.

For the sake of convenience, each control area can be represented by an equivalent governor, turbine and generator system.

Mathematical Treatment:

In an isolated control area case,

The incremental power ($\Delta P_G - \Delta P_D$) is accounted for the rate of increase of stored kinetic energy and increase in area load caused by increase in frequency.

Since a tie line transports power in or out of an area, this fact must be ~~accounted for~~ accounted for The incremental power balance equation of each area.

→ Power transported out of area 1 is given by

$$P_{tie,1} = \frac{|V_1||V_2|}{X_{12}} \sin(\delta_1^\circ - \delta_2^\circ) \longrightarrow (1)$$

where δ_1°, δ_2° = power angles of equivalent machines of two areas.

X_{12} = tie-line reactance b/w control areas 1 & 2.

→ If there is change in load demands of two areas, there will be incremental changes in power angles ($\Delta\delta_1, \Delta\delta_2$) then change in incremental tie-line power.

$$\Delta P_{tie,1} = \frac{|V_1||V_2|}{X_{12}} \cos(\delta_1^\circ - \delta_2^\circ)(\Delta\delta_1 - \Delta\delta_2)$$

$$\Delta P_{tie,1} = \frac{|V_1||V_2|}{P_{r_1} X_{12}} \cos(\delta_1^\circ - \delta_2^\circ)(\Delta\delta_1 - \Delta\delta_2)$$

where P_r = rating of the turbo generator (num)

$$\Delta P_{tie,1} = T_{12}(\Delta\delta_1 - \Delta\delta_2)$$

where $T_{12} = \frac{|V_1||V_2|}{P_{r_1} X_{12}} \cos(\delta_1^\circ - \delta_2^\circ)$

= synchronizing coefficient.

Since incremental power angles are integrals of incremental frequencies we can write above eqn as

$$\Delta P_{tie,1} = T_{12}\left(\int \omega_1 dt - \int \omega_2 dt\right)$$

$$\boxed{\Delta P_{tie,1} = 2\pi T_{12}\left(\int \Delta f_1 dt - \int \Delta f_2 dt\right)} \longrightarrow (2)$$

where Δf_1 and Δf_2 are incremental changes of areas 1 and 2 respectively.

Similarly,

→ The incremental tie-line power out of area 2 is given by

$$\Delta P_{tie,2} = 2\pi T_{21}\left(\int \Delta f_2 dt - \int \Delta f_1 dt\right)$$

where

$$T_{21} = \frac{|V_2||V_1|}{P_{r_2} X_{21}} \cos(\delta_2^\circ - \delta_1^\circ) = \frac{P_{r_1} T_{12}}{P_{r_2}} = a_{12} T_{12}$$

$$\Delta P_{tie,i} = -2\pi a_{12} T_{12} \left(\int \Delta f_1 dt - \int \Delta f_2 dt \right) \longrightarrow (3)$$

→ The incremental power balance equation for area 1 can be written as

$$\Delta P_{G_1} - \Delta P_{D1} = \frac{2H}{f_1^\circ} \frac{d}{dt} (\Delta f_1) + B \Delta f_1 + \Delta P_{tie,1} \longrightarrow (4)$$

→ D (or) $B\left[\frac{\partial P_D}{\partial f}\right] \Delta F$

D is positive for a predominately motor load

All quantities other than frequency are in p.u.

Taking laplace transform of reorganizing, we get

$$\Delta P_{G_1}(s) - \Delta P_{D_1}(s) = \frac{2H}{f_1^\circ} s \cdot \Delta F_1(s) + B_1 \Delta F_1(s) + \Delta P_{tie,1}(s)$$

$$\Delta P_{G_1}(s) - \Delta P_{D_1}(s) - \Delta P_{tie,1}(s) = \left(\frac{2H}{f_1^\circ} s + B_1\right) \Delta F_1(s)$$

$$\Delta F_1(s) = \frac{\Delta P_{G_1}(s) - \Delta P_{D_1}(s) - \Delta P_{tie,1}(s)}{\left(\frac{2H}{f_1^\circ} s + B\right)}$$

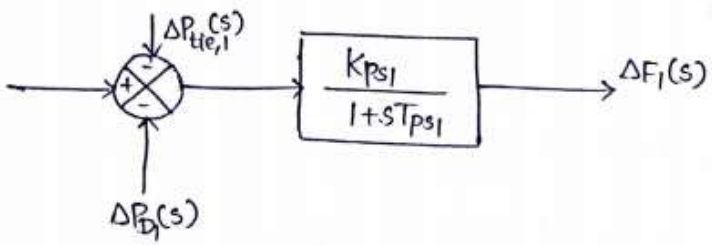
$\because B\left(s \cdot \frac{2H}{Bf_1^\circ} + 1\right)$

$K_{PS1} = 1/B_1$

$T_{PS1} = 2H/Bf_1'$

$$\Delta F_1(s) = \left[\Delta P_{G_1}(s) - \Delta P_{D_1}(s) - \Delta P_{tie,1}(s)\right] * \left(\frac{K_{PS}}{1 + sT_{PS}}\right) \longrightarrow (5)$$

Comparing above equation of isolated control area case, the only one change is the appearance of signal $\Delta P_{tie,1}(s)$ and is represented as



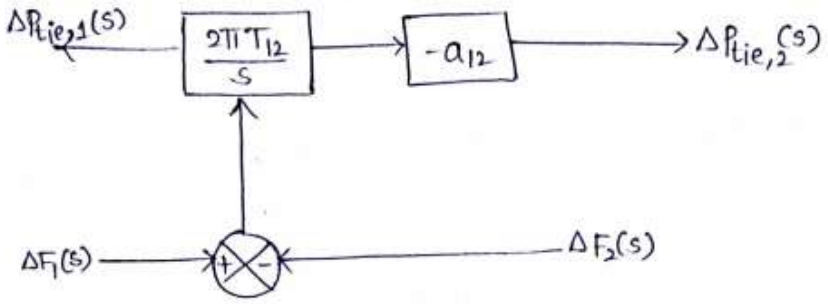
Taking laplace transform of eqn ②, we get

$$\Delta P_{tie,1}(s) = \left(\frac{2\pi T_{12}}{s}\right)[\Delta f_1(s) - \Delta f_2(s)]$$

Taking laplace transform of eqn ③, we get

$$\Delta P_{tie,2}(s) = -\frac{2\pi}{s} a_{12} T_{12} [\Delta f_1(s) - \Delta f_2(s)]$$

The corresponding block diagram is shown in figure.



Area control Error (ACE)

In case of an isolated area,

ACE is the change in area frequency which when used in integral control loop forced the steady state frequency error to zero.

In case of two-area control,

The steady state tie line power error can be made zero by introducing another integral control loop (one for each area). This is accomplished by a single integrating block by redefining ACE as a linear combination of incremental frequency and tie-line power.

For control area 1

$$ACE_1 = \Delta P_{tie,1} + b_1 \Delta f_1$$

where the constant b_1 is called "area frequency bias".

The above eqn can be expressed in laplace transform as

$$ACE_1(s) = \Delta P_{tie,1}(s) + b_1 \Delta F_1(s)$$

Similarly

For control area 2,

$$ACE_2(s) = \Delta P_{tie,2}(s) + b_2 \Delta F_2(s)$$

Fig. shows composite block diagram of two-area LFC (by combining basic block diagrams of two control areas)

$\Delta P_{tie,1} + b_1 \Delta f_1 = 0$ (input of integrating block $-\frac{K i_1}{s}$)

$\Delta P_{tie,2} + b_2 \Delta f_2 = 0$ (input of integrating block $-\frac{K i_2}{s}$

$\Delta f_1 - \Delta f_2 = 0$ (input of integrating block $-\frac{2\pi T_{12}}{s}$)

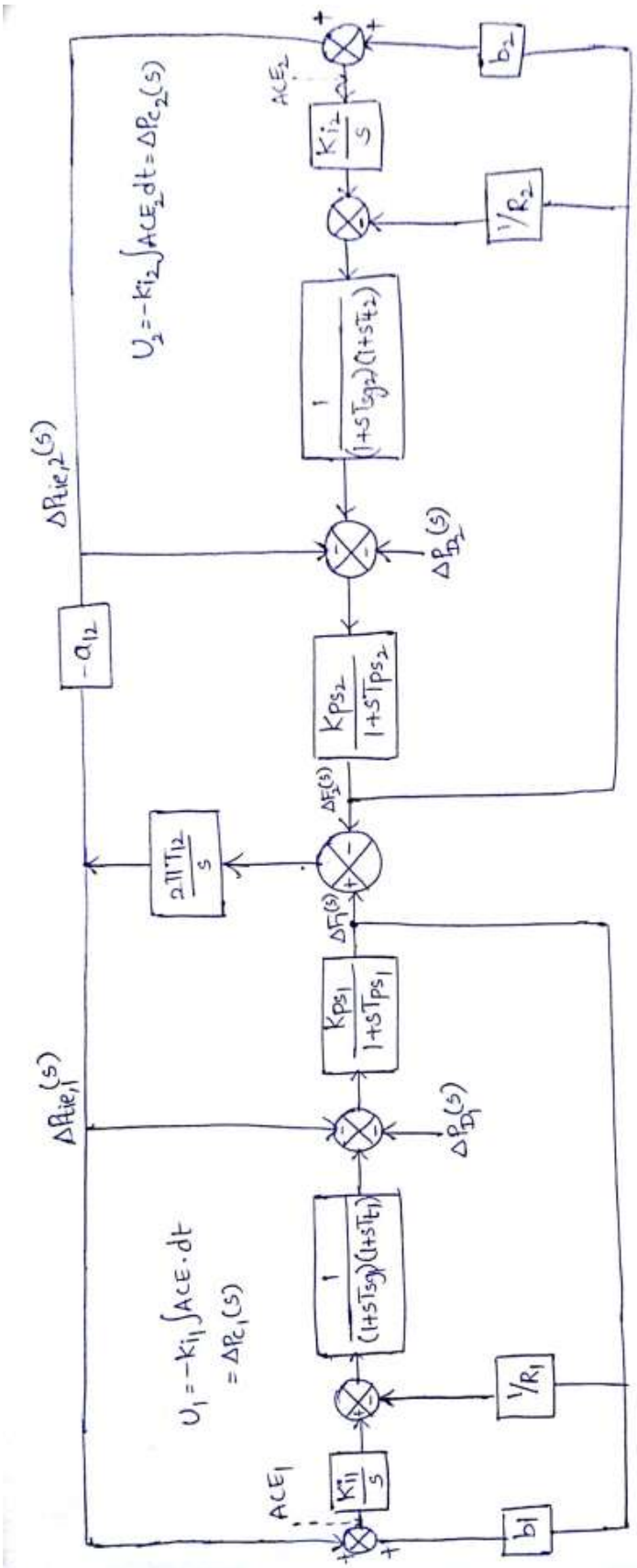


Fig. Composite block diagram of two-area load frequency control
(Controlled case)
(Feedback loops provided with integral of respective area control errors)

Dynamic response of two-area system:

To describe the dynamic response of the two-area system, a system of seventh order differential equations is required. The solution of these equations would be tedious.

The analysis of two-area system is based on the following assumptions.

1. Consider the case of two (equal) identical areas.
2. The time constants of speed governor and turbine are $T_{sg} = T_t = 0$ for both the areas.
3. The damping constants of two areas are neglected.
   i.e., $B_1 = B_2 = 0$

We know that,

Incremental power balance equation for area 1 & 2 is given by

$$\Delta P_{G_1} - \Delta P_{D_1} = \frac{2H_1}{f^\circ}\frac{d}{dt}(\Delta f_1) + B_1 \Delta f_1 + \Delta P_{tie,1} \longrightarrow (1)$$

$$\Delta P_{G_2} - \Delta P_{D_2} = \frac{2H_2}{f^\circ}\frac{d}{dt}(\Delta f_2) + B_2 \Delta f_2 + \Delta P_{tie,2} \longrightarrow (2)$$

With above assumptions, eqns (1) & (2) can be written as

$$\Delta P_{G_1} - \Delta P_{D_1} = \frac{2H_1}{f^\circ}\cdot\frac{d}{dt}(\Delta f_1) + \Delta P_{tie,1} \longrightarrow (3)$$

$$\Delta P_{G_2} - \Delta P_{D_2} = \frac{2H_2}{f^\circ}\frac{d}{dt}(\Delta f_2) + \Delta P_{tie,2} \longrightarrow (4)$$

Taking laplace transform of eqns (3) & (4) and re-arranging, we get

$$\Delta F_1(s) = \frac{f^\circ}{2H_1 s}\left[\Delta P_{G_1}(s) - \Delta P_{D_1}(s) - \Delta P_{tie,1}(s)\right] \longrightarrow (5)$$

$$\Delta F_2(s) = \frac{f^\circ}{2H_2 s}\left[\Delta P_{G_2}(s) - \Delta P_{D_2}(s) - \Delta P_{tie,2}(s)\right] \longrightarrow (6)$$

From the block-diagram, we have

$$\Delta P_{G_1}(s) = \frac{-1}{R}\Delta F_1(s) \longrightarrow (7)$$

$$\Delta P_{G_2}(s) = \frac{-1}{R}\Delta F_2(s) \longrightarrow (8)$$

($\because R_1 = R_2 = R$ for identical areas)

We know that

$$\Delta P_{tie,1}(s) = \frac{2\pi T_{12}}{s}\left[\Delta F_1(s) - \Delta F_2(s)\right] \longrightarrow (9)$$

$$\Delta P_{tie,2}(s) = -\frac{2\pi a_{12} T_{12}}{s}\left[\Delta F_1(s) - \Delta F_2(s)\right] \longrightarrow (10)$$

The above eqn can be written as

$$\Delta P_{tie,2}(s) = -a_{12}\cdot \Delta P_{tie,1}$$

$$\boxed{\Delta P_{tie,2} = -\Delta P_{tie,1}} \longrightarrow (11)$$

($\because a_{12} = \frac{P_{r1}}{P_{r2}} = 1$, since two areas are identical)

By solving eqns (5) to (11), we get

$$\Delta P_{tie,1}(s) = \frac{\Delta P_{D_2}(s) - \Delta P_{D_1}(s)}{\left[s^2 + \left(\frac{f^\circ}{2RH}\right)s + \frac{2\pi f^\circ T_{12}}{H}\right]} \cdot \frac{\pi f^\circ T_{12}}{H}$$

From the above equation, the following observations can be made.

i) The denominator is of the form

$$s^2 + 2\alpha s + \omega^2$$

where $\alpha = \frac{f^\circ}{4RH}$; $\omega = \sqrt{\frac{2\pi f^\circ T_{12}}{H}}$

The values of α and ω^2 are real and positive.

Hence, it can be concluded that from the roots of the characteristic equation, the response is stable and damped

ii) If $\alpha = \omega_n$, system is critically damped.

ii) If $\alpha > \omega_n$, system becomes overdamped.

iii) If $\alpha < \omega_n$, system becomes underdamped.

The roots are $s_{1,2} = -\alpha \pm j\sqrt{\omega_n^2 - \alpha^2}$

$$= -\alpha + j\omega_n\sqrt{1 - \left(\frac{\alpha}{\omega_n}\right)^2}$$

$$s_{1,2} = -\alpha \pm j\omega_0$$

where α = damping factor (or) decrement of attenuation

ω_0 = damped angular frequency

$$= \sqrt{\frac{2\pi f^2 T_{12}}{H} - \left(B + \frac{1}{R}\right)^2 \frac{f0^2}{16H^2}}$$

State-Variable model- Two area system:

Dynamic response is difficult to obtain by transfer function approach because of

→ Complexity of blocks

→ Multiple inputs (ΔP_{D_1}, ΔP_{D_2})

→ Multiple outputs ($\Delta P_{tie,1}$, $\Delta P_{tie,2}$, Δf_1, Δf_2)

A more organized and more conveniently carried out analysis is through the state space approach.

Formulation of state-space model for two-area system is as follows.

The state variables are:

$x_1 = \Delta f_1$; $x_4 = \Delta f_2$; $x_8 = \int ACE_1\, dt$

$x_2 = \Delta P_{G1}$; $x_5 = \Delta P_{G2}$; $x_9 = \int ACE_2\, dt$

The control variables are denoted by u_1 and u_2

$u_1 = \Delta P_{C_1}$; $u_2 = \Delta P_{C_2}$

The disturbance variables are denoted by w_1 and w_2

$w_1 = \Delta P_{D_1}$; $w_2 = \Delta P_{D_2}$

→ For formulating the state variable model for this purpose the conventional feedback loops are opened and each time constant is represented by a separate blocks as shown in fig.

→ Before presenting the optimal design, we must formulate the state model. This is achieved below by writing the differential equations describing each individual blocks as shown in fig. in terms of state variables.

(Note: The differential equations are written by replacing s by $\frac{d}{dt}$)

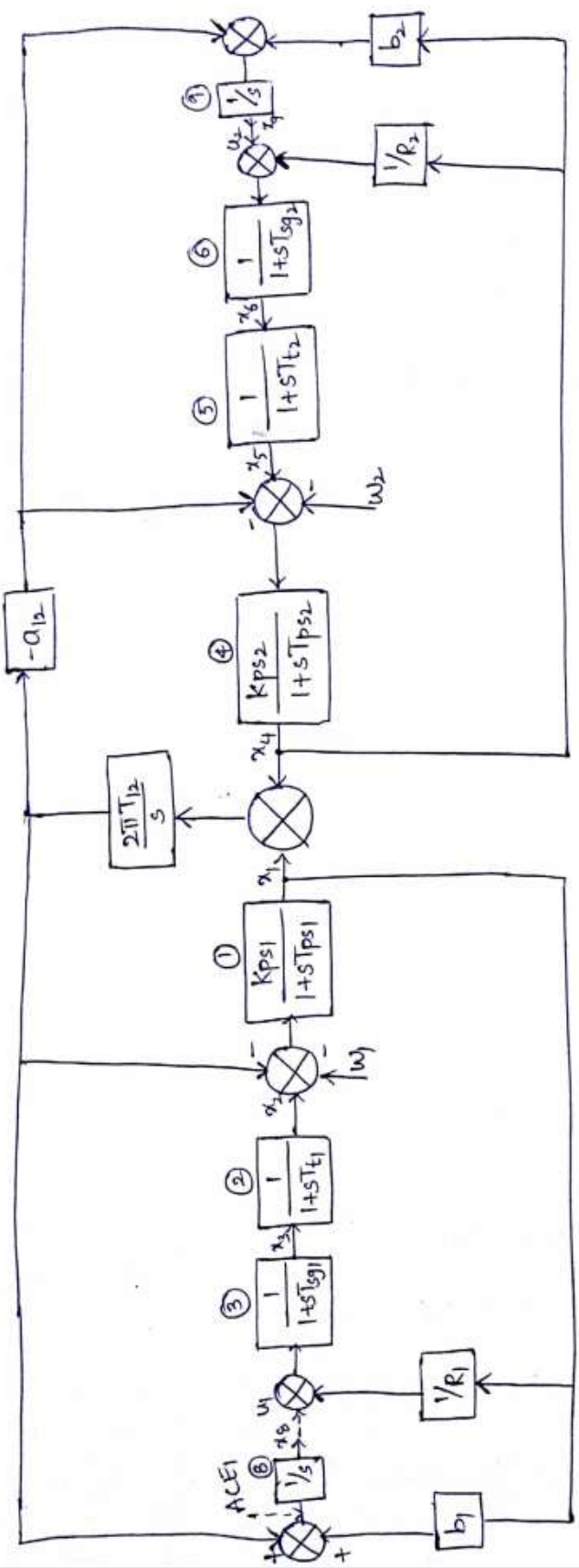


Fig. State space model of two-area system.

For block-1:

$$x_1 = \frac{K_{ps1}}{1+sT_{ps1}}(x_2 - x_7 - w_1)$$

$$x_1 + sx_1T_{ps1} = K_{ps1}(x_2 - x_7 - w_1)$$

$$\dot{x}_1 = \frac{1}{T_{ps1}}\left[K_{ps1}(x_2 - x_7 - w_1) - x_1\right]$$

$$\dot{x}_1 = \frac{-1}{T_{ps1}}x_1 + \frac{K_{ps1}}{T_{ps1}}x_2 - \frac{K_{ps1}}{T_{ps1}}x_7 - w_1\frac{K_{ps1}}{T_{ps1}} \longrightarrow (1)$$

For block 2:

$$x_2 = \frac{1}{1+sT_{t1}} \cdot x_3$$

$$x_2 + sx_2T_{t1} = x_3$$

$$\dot{x}_2T_{t1} = x_3 - x_2$$

$$\dot{x}_2 = \frac{-1}{T_{t1}}x_2 + \frac{1}{T_{t1}}x_3 \longrightarrow (2)$$

For block 3:

$$x_3 = \frac{1}{1+sT_{sg1}}\left(u_1 - \frac{1}{R_1}x_1\right)$$

$$x_3 + sx_3T_{sg1} = u_1 - \frac{1}{R_1}x_1$$

$$\dot{x}_3T_{sg1} = \frac{-1}{R_1}x_1 - x_3 + u_1$$

$$\dot{x}_3 = \frac{-1}{R_1}\cdot\frac{1}{T_{sg1}}\cdot x_1 - \frac{1}{T_{sg1}}x_3 + \frac{1}{T_{sg1}}u_1 \longrightarrow (3)$$

For block 4:

$$x_4 = \frac{K_{ps2}}{1+sT_{ps2}}(x_5 + a_{12}x_7 - w_2)$$

$$x_4 + sx_4T_{ps2} = \left[K_{ps2}(x_5 + a_{12}x_7 - w_2)\right]$$

$$\dot{x}_4T_{ps2} = \left[K_{ps2}(x_5 + a_{12}x_7 - w_2)\right] - x_4$$

$$\dot{x}_4 = \frac{K_{ps2}}{T_{ps2}}\cdot x_5 + \frac{K_{ps2}}{T_{ps2}}a_{12}x_7 - \frac{K_{ps2}}{T_{ps2}}w_2 - \frac{1}{T_{ps2}}\cdot x_4$$

For block 5:

$$x_5 = \left(\frac{1}{1+sT_{t2}}\right)\cdot x_6$$

$$x_5 + sx_5 \cdot T_{t2} = x_6$$

$$\dot{x}_5 T_{t2} = -x_5 + x_6$$

$$\dot{x}_5 = \frac{-1}{T_{t2}} x_5 + \frac{1}{T_{t2}} x_6$$

For block 6:

$$\dot{x}_6 = \left(\frac{1}{1+sT_{sg2}}\right)\left(U_2 - \frac{1}{R_2} x_4\right)$$

$$x_6 + sx_6 T_{sg2} = \frac{-1}{R_2} x_4 + U_2$$

$$\dot{x}_6 T_{sg2} = \frac{-1}{R_2} x_4 - x_6 + U_2$$

$$\dot{x}_6 = \frac{-1}{T_{sg2} R_2} x_4 - \frac{1}{T_{sg2}} x_6 + \frac{1}{T_{sg2}} U_2$$

For block 7:

$$x_7 = \frac{2\pi T_{12}}{s}(x_1 - x_4)$$

$$sx_7 = 2\pi T_{12}(x_1 - x_4)$$

$$\dot{x}_7 = 2\pi T_{12} x_1 - 2\pi T_{12} x_4$$

For block 8:

$$x_8 = \frac{1}{s}(b_1 x_1 + x_7)$$

$$sx_8 = b_1 x_1 + x_7$$

$$\dot{x}_8 = b_1 x_1 + x_7$$

For block - 9

$$x_9 = \frac{1}{s}(b_2 x_4 - a_{12} x_7)$$

$$sx_9 = b_2 x_4 - a_{12} x_7$$

$$\dot{x}_9 = b_2 x_4 - a_{12} x_7$$

The nine equations can be organized in following vector matrix form.

i.e., $\dot{x} = Ax + Bu + Fw$

where $x = [x_1\ x_2 \ldots x_9]^T$ = state vector

$u = [u_1\ u_2]^T$ = control vector

$w = [w_1\ w_2]^T$ = disturbance vector

while the matrices A, B and F are defined below.

$$A = \begin{array}{c} \\ 1 \\ 2 \\ 3 \\ 4 \\ 5 \\ 6 \\ 7 \\ 8 \\ 9 \end{array} \begin{array}{c} \begin{array}{ccccccccc} 1 & 2 & 3 & 4 & 5 & 6 & 7 & 8 & 9 \end{array} \\ \begin{bmatrix} \frac{-1}{T_{ps1}} & \frac{K_{ps1}}{T_{ps1}} & 0 & 0 & 0 & 0 & \frac{-K_{ps1}}{T_{ps1}} & 0 & 0 \\ 0 & \frac{-1}{T_{t1}} & \frac{1}{T_{t1}} & 0 & 0 & 0 & 0 & 0 & 0 \\ \frac{-1}{R_1 T_{sg1}} & 0 & \frac{-1}{T_{sg1}} & 0 & 0 & 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & \frac{-1}{T_{ps2}} & \frac{K_{ps2}}{T_{ps2}} & 0 & \frac{a_{10} K_{ps2}}{T_{ps2}} & 0 & 0 \\ 0 & 0 & 0 & 0 & \frac{-1}{T_{t2}} & \frac{1}{T_{t2}} & 0 & 0 & 0 \\ 0 & 0 & 0 & \frac{-1}{R_2 T_{sg2}} & 0 & \frac{-1}{T_{sg2}} & 0 & 0 & 0 \\ 2\pi T_{12} & 0 & 0 & -2\pi T_{12} & 0 & 0 & 0 & 0 & 0 \\ b_1 & 0 & 0 & 0 & 0 & 0 & 1 & 0 & 0 \\ 0 & 0 & 0 & b_2 & 0 & 0 & -a_{12} & 0 & 0 \end{bmatrix} \end{array}$$

$$B^T = \begin{bmatrix} 0 & 0 & \frac{1}{T_{sg1}} & 0 & 0 & 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 & 0 & \frac{1}{T_{sg2}} & 0 & 0 & 0 \end{bmatrix}$$

$$F^T = \begin{bmatrix} \frac{-K_{ps1}}{T_{ps1}} & 0 & 0 & 0 & 0 & 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & \frac{-K_{ps2}}{T_{ps2}} & 0 & 0 & 0 & 0 & 0 \end{bmatrix}$$

Load frequency controllers:

Proportional plus integral control of single area and its block diagram representation, steady state response - Load frequency control and Economic dispatch control.

Basic concept of control area

In the early days electric power systems were usually operated as individual units because they started as isolated systems. The demand for large blocks of power and increased reliability suggested the interconnection of neighbouring plants. Interconnection is advantageous because fewer machine are required as a reserve for operation at peak loads and fewer machines running without load are required to take care of sudden, unexpected jumps in load (spinning reserve). Therefore, all generating plants are interconnected to form a regional grid, state grid and the national grid.

Load dispatch centres are required for the control of power flow in these grids.

It is feasible to divide a very large power system, say a national grid into sub-areas in which all the generators are assumed to be tightly coupled i.e., they swing in unision with change in load (or) due to speed changer settings.

Definition of control area:

The area in which all the generators are tightly coupled together so as to form a coherent group.

i.e., all the generators respond in unison to changes in load (or) speed changer settings. Such area is called "control area".

→ In control area, the frequency is assumed to be same throughout in static as well as dynamic conditions.

→ For purpose of developing a suitable control ~~systems~~ strategy, control area can be reduced to a single speed-governor, turbo governor and load system.

Area control Error (ACE)

The change in frequency in a given control area is known as "Area Control Error".

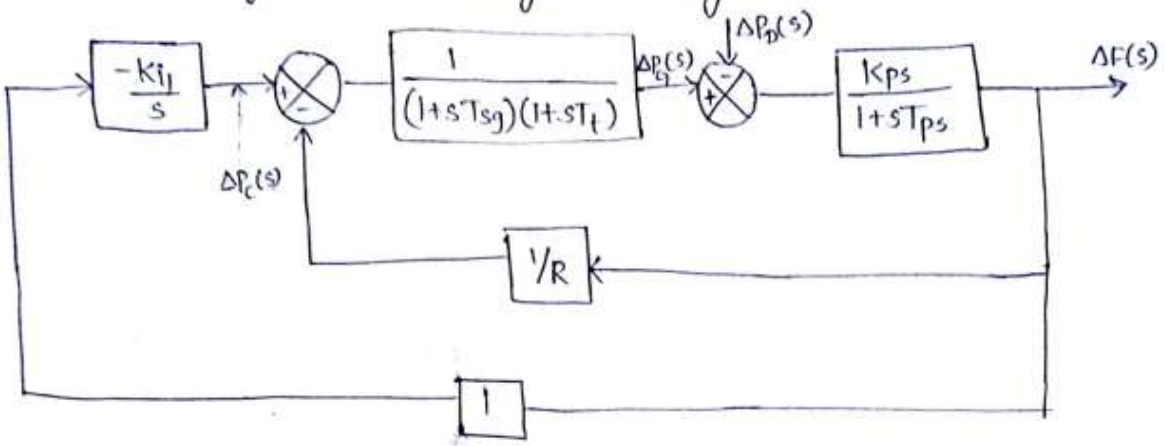
Proportional plus integral load frequency control:

As so much change in frequency cannot be tolerated. In fact, it is expected that steady change in frequency will be zero. by introducing integral controller. The steady state frequency can be brought back to prescribed value (or scheduled value) by adjusting speed changer settings as the load on system is changed.

→ The speed changer setting be adjusted automatically by monitoring the frequency changes.

For this purpose, a signal from Δf is fed through an integrator (i.e., $\frac{K_i}{s}$) to the speed changer.

The system now modifies to a proportional plus integral controller. The system block diagram configuration is shown



Frequency sensor

From the block diagram,

$$\Delta P_c(s) = \frac{-K_i}{s}\Delta F(s) \longrightarrow (1)$$

$$\left[\frac{-1}{R}\Delta F(s) - \frac{K_I}{s}\Delta F(s)\right]\cdot\frac{1}{(1+sT_{sg})(1+sT_t)} = \Delta P_G(s)$$

$$-\left[\frac{1}{R}+\frac{K_I}{s}\right]\cdot\frac{1}{(1+sT_{sg})(1+sT_t)}\cdot\Delta F(s) = \Delta P_G(s) \longrightarrow (2)$$

$$\Delta F(s) = \left(\frac{K_{ps}}{1+sT_{ps}}\right)\left(\Delta P_G(s) - \Delta P_D(s)\right) \longrightarrow (3)$$

$$\Delta F(s) = -\left(\frac{K_{ps}}{1+sT_{ps}}\right)\cdot\Delta P_D(s) + \left(\frac{K_{ps}}{1+sT_{ps}}\right)\Delta P_G(s)$$

$$= -\left(\frac{K_{ps}}{1+sT_{ps}}\right)\cdot\Delta P_D(s) + \left(\frac{K_{ps}}{1+sT_{ps}}\right)\cdot\left\{-\left[\frac{1}{R}+\frac{K_i}{s}\right]\frac{1}{(1+sT_{sg})(1+sT_t)}\Delta F(s)\right\}$$

$$\Delta F(s)\left\{1+\left[\left(\frac{K_{ps}}{1+sT_{ps}}\right)\left(\frac{1}{R}+\frac{K_i}{s}\right)\frac{1}{(1+sT_{sg})(1+sT_t)}\right]\right\} = \frac{-K_{ps}}{1+sT_{ps}}\cdot\Delta P_D(s)$$

$$\Delta F(s) = \frac{\left(\frac{-K_{ps}}{1+sT_{ps}}\right)}{1+\left[\left(\frac{K_{ps}}{1+sT_{ps}}\right)\left(\frac{1}{R}+\frac{K_i}{s}\right)\cdot\frac{1}{(1+sT_{sg})(1+sT_t)}\right]}\cdot\Delta P_D(s) \longrightarrow (4)$$

For a step (or sudden) change of load demand,

$$\Delta P_D(s) = \frac{\Delta P_D}{s}$$

The eqn (3) can be written as

$$\Delta F(s) = \frac{\left(\frac{-K_{ps}}{1+sT_{ps}}\right)}{1+\left[\left(\frac{K_{ps}}{1+sT_{ps}}\right)\left(\frac{1}{R}+\frac{K_i}{s}\right)\cdot\frac{1}{(1+sT_{sg})(1+sT_t)}\right]}\cdot\frac{\Delta P_D}{s}$$

The steady state change (error) in frequency is given by

$$\Delta f\Big|_{\text{steady state}} = \lim_{s\to 0} s\cdot\Delta F(s)$$

$$= 0$$

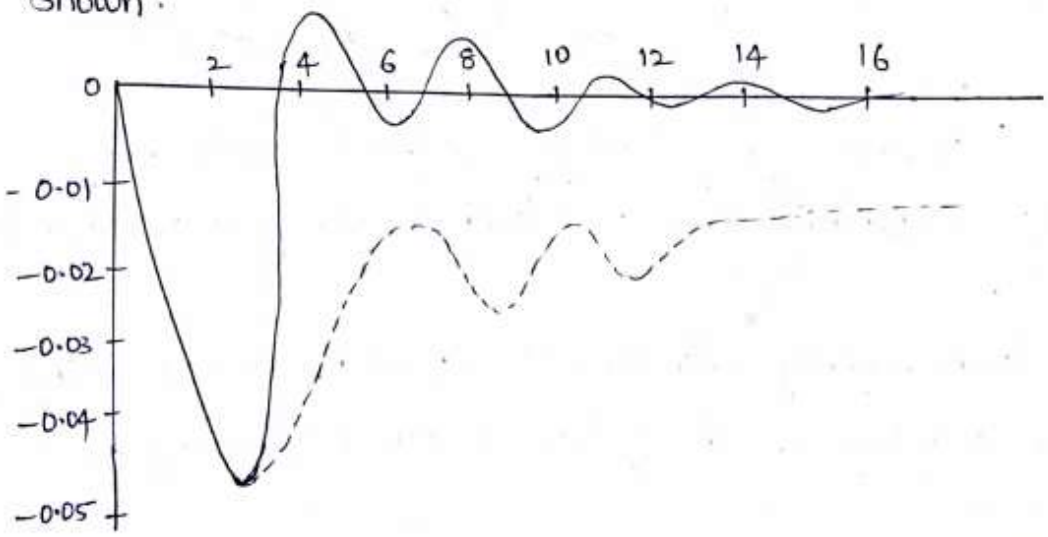
LOAD FREQUENCY CONTROL AND ECONOMIC DISPATCH CONTROL

The following are the basic requirements needed for the control strategy.

1. The control on the system frequency is obtained through a closed-loop.
   The stability is always a problem with a closed loop control. Hence, the first requirement is stability of CLC
2. The frequency deviation due to a step-load change should return to zero. The control that offers zero frequency deviation is called 'Isochronous control".
3. The control ~~magnitude~~ strategy should keep the magnitude of transient frequency error (or deviation) to a minimum

because no control can eliminate transient frequency error.
The integral of frequency error should not exceed a certain maximum value.
4. The individual generators of the control area should divide (share) the load such that optimum economy is obtained.
The first three requirements can be met with aid of the integral control.

→ The order of load frequency controller of an isolated system is being of fourth order due to addition of integral loop.
The dynamic response of proportional plus integral controller can be obtained through digital computer as shown.



From the graph, it is concluded that the response of system is fast and zero steady state error can be achieved.

Load frequency control and Economic dispatch control

Economic load dispatch and LFC play a vital role in modern power system.

In LFC,

Zero steady-state frequency error, fast and dynamic response were achieved by integral controller action. But this control is independent of economic dispatch i.e., there is no control over the economic loadings of the individual units of the control area. Some control over loading of individual units can be obtained by adjusting the gain factor (k_i) of the integral signal of ACE as fed to individual units. But this is not satisfactory solution.

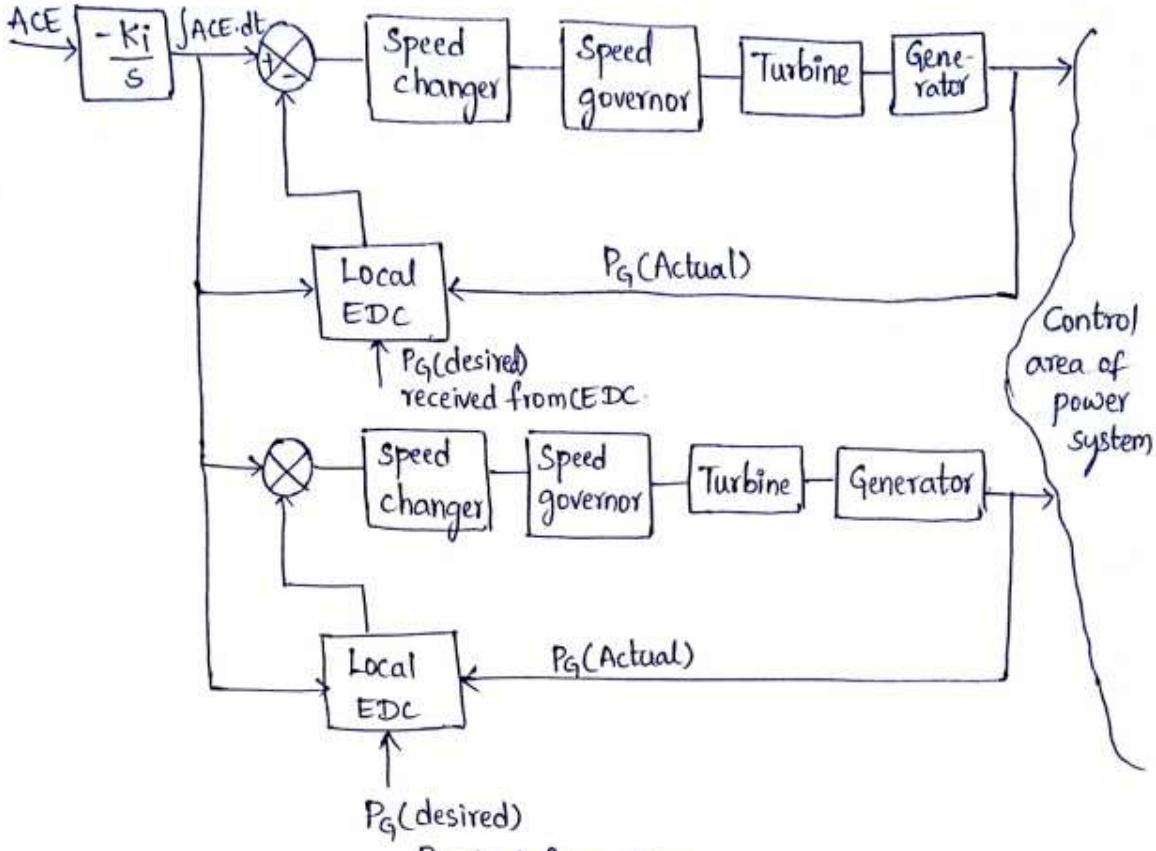
(In fact, some of the units in the process may even get overloaded).

A suitable and satisfactory solution is obtained by using independent controls of load-frequency and economic dispatch.

→ The load frequency controller (LFC) provides as fast-acting control and regulates the system around an operating point, where as

→ The economic load dispatch (EDC) provides a slow-acting control, which adjust speed-changer settings every minute in accordance with a command signal generated by central economic dispatch computer.

The schematic diagram shows both controls (LFC & EDC) for typical two units of a control area



CEDC – Central Economic Dispatch Controller

→ The signal to change the speed changer setting is constructed in accordance with economic dispatch error (i.e., P_G(desired) − P_G(actual)) suitably modified by signal representing integral ACE at that instant of time.

→ The signal P_G(desired) is computed by CEDC and is transmitted to the local EDC installed at each station.

→ The system thus operates with economic dispatch error only for very short periods of time before it is readjusted.

UNIT-IV

REACTIVE POWER CONTROL

5.1 Overview of Reactive power control:

The economics of ac power transmission have always forced the planning engineer to transmit as much power as possible through a given transmission line. First, the dependence of load centers on the continuity of electrical supplies has become critical events. This means that the security (or) Reliability, of transmission circuits has needed to be continuously improved. Modern compensation methods have helped to make these improvements possible. Second, there has been extensive development of remote hydroelectric resources. The development of compensation schemes has helped to make ac transmission technically and economically competitive even in an age when the transmission alternative has made great strides also. The third planning constraint has been the difficulty of acquiring right – of – way for new transmission circuits. Increased pressure to maximize the utilization of both new and existing lines has helped to motivate the development and application of compensation systems.

The transmission of electrical power by ac is possible only if the following two fundamental requirements are satisfied.

1) Major synchronous machines must remain stably in synchronism.

2) Voltages must be kept near to their rated values.

Reactive power has been recognized as a significant factor in the design and operation of alternating current electric power systems for a long time. It has been observed that the impedances of the network components are predominantly reactive, the transmission of active power requires a difference in angular phase between the voltages at the sending and receiving points. The voltages at all the key points of the network. This control may be accomplished in large-part by the supply

(or) consumption of reactive power at these points. Reactive power have long been recognized they have recently acquired increased importance for at least two reasons, first the increasing pressures to utilize transmission capacity as much as possible and second the development of newer static types of control lebel reactive power compensator.

Thus a direct way of increasing power transfer capacity in transmission systems and of reducing voltage drop in distribution systems is to compensate part of the series inductive reactance by series capcitor. Operating problems have been encountered and applications to distribution systems have become rare, but the series capacitor remains the best way to increase transmission capacity in many cases. For purely sinusoidal single frequency voltages and currents, the concept is simple reactive current is the component out of phase with the voltage and there is a simple right triangle relation between active reactive and apparent power.

The need for adjustable reactive power compensation can be divided into three basic classes.

1) The need to maintain the stability of synchronous machines.
2) The need to control voltage within acceptable bounds about the desired steady state value to provide quality service to consumer loads.
3) The need to regulate voltage profiles in the network to prevent unnecessary flows of reactive power on transmission lines.

The static controlled reactive power sources almost always produce harmonics. The simple concept reactive power being the product of voltage and the out of phase component of current, where the reactive power by any direct measurement is practically zero, but the power factor is less than unity.

Reactive power compensation in transmission systems:

Reactive power flow (Q) is closely related with the voltage control. The apparent power S(KVA) is given by $S = P \pm jQ$

Where S = Apparent power, KVA
P = Real power, KW
Q = Reactive power, KVAr.

The various equipments in the network 'absorb' (or) 'generate' reactive power.

They are 1) Synchronous machine.
2) Shunt capacitors.

1) Synchronous machine:

An over excited synchronous motor working on no load, generates leading reactive power. It is a flexible reactive power source, which can generate variable amount of reactor power by changing the excitation.

a) Transmission of reactive power:

Over a long distance having additional losses for large size conductor and cost of the temperature.

Synchronous condenser:

Locate the synchronous condenser nearer to load constant, to reduce the losses in the system.

Industrial load lagging p.f provides static capacitors at the consumer's premises to supply the leading reactive power.

2) Shunt capacitors: A fixed capacitor is connected across the load at the consumers premises.

Volt amperes reactive are absorbed by inductive loads and Q for inductive loads is considered positive. Voltaamperes are supplied by capacitive loads and Q for capactive load is considered negative.

In complex notations:

Complex power S is the product of voltage V and complex conjugate of I

i.e. $S = VI^*$ where $S = P + jQ$.

Power P controls the active power which is converted into mechanical/heat (or) some other form.....and influence frequency f.

Reactive power Q is exchanged between inductive and capacitive loads in the network and influences the voltage in the network. Reactive power flow increases losses. Hence compensation is provided at each bus. The control of various bus voltage is achieved by supplying absorbing the reactive power requirements (KVAr) of respective busbars by means of series (or) shunt compensation. Compensation of reactive power means supplying/absorbing reactive vot-amperes.

Effect of reactive power flow on voltage at sending end and receiving end of transmission line:

P = power transfer watts per phase

Q = reactive power transfer VARs per phase

V_s = Sending end voltage volts,

V_R = receiving end voltage

$\Delta V = V_s - V_R$

= Drop in the line voltage

$R + jX$ = Series impedance of line/Ph

. The relation between V_S, V_R and P, Q is given by the equation.

$$\Delta V = V_S - V_R = \frac{RP + XQ}{V_R}$$

If the resistance R is neglected i.e. X >> R, then

$$\Delta V = V_S - V_R = \frac{XQ}{V_R}$$

Hence voltage drop in line depends mainly on the flow of reactive power Q. The power angle δ between V_R and V_S is proportional to

$$\delta \propto \frac{XP - RQ}{V_R} = \frac{XP}{V_R}$$

IF X >> R, angle δ depends mainly on P.

Thus,

Voltage is mainly controlled by reactive power flow power angle δ is mainly controlled by real power flow. For voltage control, the flow of reactive power through the transmission line should be controlled. The flow of reactive power is controlled by injecting required VAR's into the system by means of

- Static shunt capacitors / reactors (SVS)
- Series capacitors
- Synchronous condensers (Compensators)

5.3 Advantages and disadvantages of different types of compensating equipment for transmission systems:

	Compensating equipment	Advantages	Disadvantages
1)	Switched shunt reactor	Simple in principle and construction	Fixed in value
2)	Switched shunt capacitor	Simple in principle and construction	Fixed in value switching transients.
3)	Series capacitor	Simple in principle performance relatively insensitive to location	Requires overvoltage protection and subharmonic filters, limited overload capability.
4)	Synchronous condenser	Has useful overload capability, fully controllable, low harmonics.	High maintenance requirement, slow control response, performance sensitive to location, requires strong foundations.
5)	Polyphase saturated reactor	Very rugged construction, large overload capability, No effect on fault level, low harmonics	Essentially fixed in value performance sensitive to location, Noisy.

6)	Thyristor – controlled reactor (TCR)	Fast response, Fully controllable, No effect on fault level can be rapidly repaired	Generates harmonics, performance sensitive to location
7)	Thyristor-switched capacitor (TSC)	Can be rapidly repaired after failures, No harmonics.	No inherent absorbing capability to limit overvoltages, complex buswork and controls, Low frequency resonances with system, Performance sensitive to location.

5.4 Load compensation:

The compensation in power system is basically designed for absorption (or) generation of reactive power. There are two methods of compensation.

1) Load compensation

2) Line compensation

Compensator is connected at the terminals of the load is called load compensation. Compensator may be located at midpoint of the line (or) some suitable location uniformly distributed is called line compensation.

In an ideal ac power system, the voltage and frequency at every supply point would be constant and free harmonics, and the power factor would be unity. These parameters would be independent of the size and characteristics of consumer's loads. In these – phase systems, the degree to which the phase currents and voltage are balanced must also be included in the notion of quality of supply.

Load compensation is the management of reactive power to improve the quality of supply in an ac power systems. The term load compensation is used where the reactive power management is effected for single load, the compensating equipment usually being installed on the consumer's own permises near to the load. In load compensation there are three main objectives.

1) Power factor correction.

2) Improvement of voltage regulation

3) Load balancing

1) Power factor connection:

The supply utilities also have good reasons for not transmitting unnecessary reactive power from generators to loads, their generators and distribution networks cannot be used at full efficiency, and the control of voltage in the supply can become more difficult.

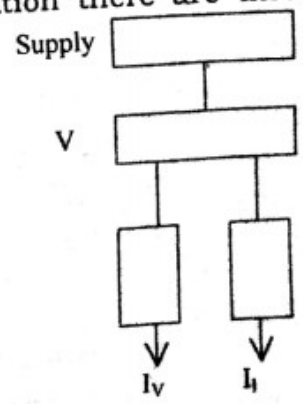


Fig.5.1

$$\text{Load current } I_l = V_{yl}$$

$$= V(G_l + jB_l)$$

$$= VG_l + jVB_l$$

$$I_l = I_R + jI_X.$$

It is an inductive load, where B_l is negative (-ve) and IX is also negative. (-ve).

$$\text{Complex power} = VI_l^*$$

$$= V[I_R - jI_X]$$

$$= VI_R - jVI_X$$

$$\text{Complex} = P_R + jQ_l$$

Where, Q_l is positive (+ve) when I_X is negative for inductive load.

To provide compensation and improvement in power factor connect compensator across the load having a suitable reactive admittances such that over all power factor becomes unity.

2) Improvement of voltage regulation:

Voltage regulation becomes an important. All loads vary their demand for reactive power, although their differ widely in their range and rate of variation. The limits may vary from typically ± 5% averaged over a period of a few minutes (or) hours, to the much more constraints imposed where large, rapidly varying loads could produce voltage dips to the operation of protective equipment.

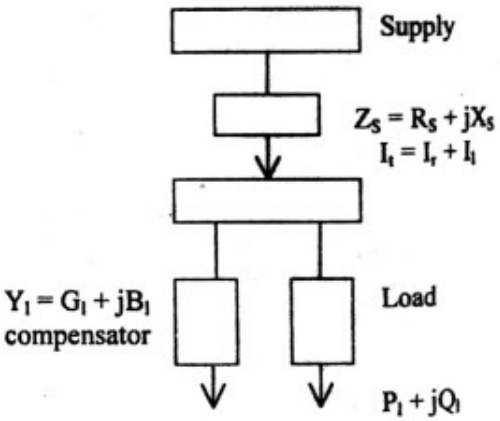


Fig.5.2.

$$dv = E - V = Z_s I_L = (R_S + jX_S)I_L$$

$$VI_L = P_L + jQ_L$$

$$\Rightarrow I_L = \frac{P_L + jQ_L}{V}$$

$$\therefore dv \text{ (or) } \Delta V = (R_S + jX_S)I_L$$

$$= (R_S + jX_S)\left(\frac{P_L + jQ_L}{V}\right)$$

$$= \frac{P_L R_S + jP_L X_S + jQ_L R_L + j^2 Q_L X_S}{V}$$

$$= \frac{P_L R_L - Q_L X_S}{V} + j\left(\frac{P_L X_S + Q_L R_L}{V}\right)$$

$$\Delta V = \Delta V_R + j\Delta V_X.$$

$$\text{where } \Delta V_R = \frac{P_L R_S - Q_L X_S}{V}$$

$$\text{and } \Delta V_X = \frac{P_L X_S + Q_L R_S}{V}$$

By adding a compensating voltage for improving voltage regulation, a purely reactive compensator is designed. The votlage change ΔV was a compensation voltage ΔV_R inphase with voltage and ΔV_X in perpendicular with voltage. The change in voltage 'V' depends on the real and reactive of the load i.e. $P_L + jQ_L$.

3. Load balancing:

ac power systems are three-phase, and are designed for balanced operation. Unbalanced operation gives rise to components of current in the wrong phase-sequence.

5.5 Specifications of load compensator:

The parameters and factors need to be considered when specifying a load compensator are summarized in the following.

1) Maximum continuous reactive power requirement, both absorbing and generating.
2) Over load rating and duration.
3) Rated voltage and limits of voltage between which the reactive power ratings must not be exceeded.
4) Frequency and its variation.
5) Votlage regulation requried.
6) Special control requirements.
7) Response time of the compensator for a specified disturbance
8) Maintenance, spare parts, provision for future expansion.
9) Reliability of components
10) Performance with unbalanced supply voltages and / or with unbalanced load.

5.6 Uncompensated transmission lines:

A transmission line is characterized by four distributed circuit parameters:

1) Series resistance (r)
2) inductance (l)
3) Shunt conductance (g) and
4) capacitance (c)

All four parameters are functions of the line design, of the conductor size, type, spacing, height above ground, frequency and temperature. The characteristic behavior of the line is dominated by the series inductance and the shunt capacitance.

The fundamental equation governing the propagation of energy of a transmission line is the wave equation.

$$\frac{d^2V}{dx^2} = \Gamma^2 V \text{.....(5.1)}$$

with $\Gamma^2 = (r + jwl)(g + jwc)$

Frequency is assumed fixed, and V is the phasor voltage $\hat{V}\ e^{jwt}/\sqrt{2}$ at any point on the line. If the line is assumed lossless, the general solution of equation (5.1)

$$V(x) = V_r \cos\beta(a - x) + jZ_0 I_r \sin\beta(a - x) \text{ .....(5.2)}$$

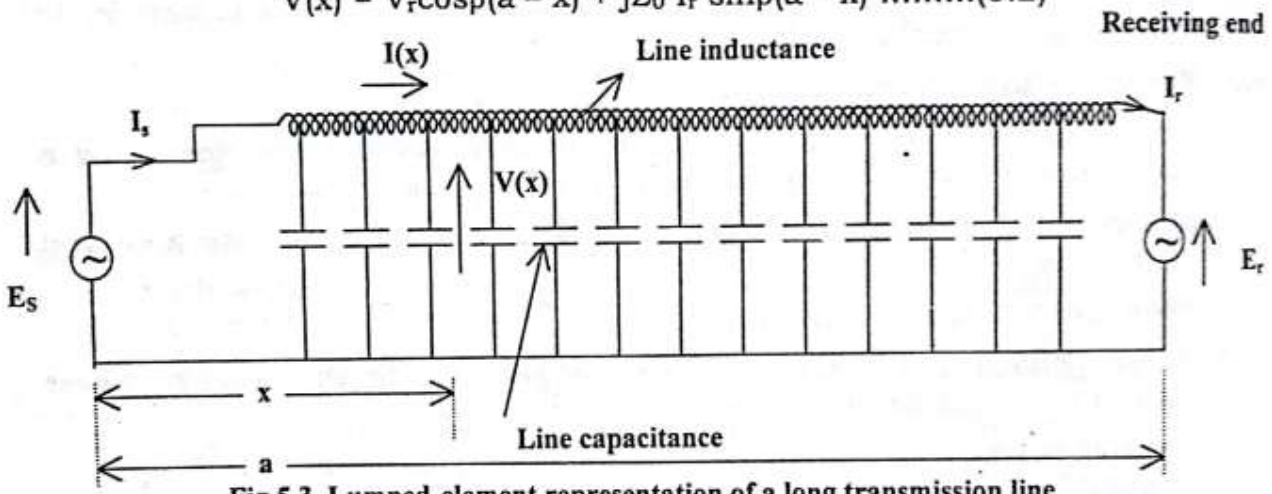


Fig.5.3. Lumped-element representation of a long transmission line

$$I(x) = j\left(\frac{v_r}{Z_0}\right)\sin\beta(a - x) + I_r \cos\beta(a - x) \text{.....(5.3)}$$

Where β is derived from the propagation constant Γ by putting r = g = 0

$$\Gamma = j\beta$$

$$\text{and } p = w\sqrt{lc} \text{ .....(5.4)}$$

V and I are form standing waves because of the sinusoidal variation of both their real and imaginary parts of the line.

Velocity of light = $u = 3 * 10^8$ m/sec

The constant Z_0 in equation (5.2) is the surge impedance

$$Z_0 = \sqrt{\frac{l}{c}} \text{ .....(5.4)}$$

This impedance is also called the characteristic impedence.

For overhead lines, the positive – sequence value is 400 Ω.

The surge impedance is the apparent impedance of an infinitely long line, the ratio of voltage to current at any point along it.

$$\text{If } \frac{V_r}{I_r} = Z_0, \text{ then}$$

$$Z(x) = \frac{V(x)}{I(x)}$$

$$= \frac{V_r \cos\beta(a-x) + jZ_0 I_r sn\beta(a-x)}{j\left(\frac{V_r}{Z_0}\right)\sin\beta(a-x) + I_r \cos\beta(a-x)}$$

$$= \frac{Z_0 I_r \cos\beta(a-x) + jZ_0 I_r \sin\beta(a-x)}{jI_r \sin\beta(a-x) + I_r \cos\beta(a-x)} = \frac{Z_0 I_r[\cos\beta(a-x) + j\sin\beta(a-x)]}{I_r[\cos\beta(a-x) + j\sin\beta(a-x)]}$$

$$= Z_0 \text{.....(5.5)}$$

The line is said to have a flat voltage profile. While V and I are in phase with each other all along the line, both are rotated in phase.

The natural load (or) surge Impedance load SIL is

$$P_0 = \frac{V_o^2}{Z_0} \text{ .....(5.6)}$$

Where V_0 is the nominal (or) rated voltage of the line. If V_0 is the line-to neutral voltage. Therefore, natural load of the uncompensated line increases with the square of the voltage. An advantage of operating the line at the natural load is that because of the flat voltage profile, the insulation is uniformly stressed at all points.

Voltage and current profiles: (Uncompensated line on open-circuit)

In equation (5.2) with $I_r = 0$

$$\therefore V(x) = V_r \cos\beta(a - x) \text{ .....(5.7)}$$

and in equation (5.3) with $I_r = 0$

$$\therefore I_x = j\left(\frac{V_r}{Z_0}\right)\sin\beta\,(a - x) \text{ .....(5.8)}$$

The voltage and current at the sending end are given by these equations with $x = 0$

$$E_s = V_r \cos\theta \text{ .....(5.9)}$$

$$I_S = j\left(\frac{V_r}{Z_0}\right)\sin\theta. \qquad \text{Where } \theta = \beta(a-x)$$

$$= j\left(\frac{E_S}{\cos\theta Z_0}\right)\sin\theta$$

$$= j\left(\frac{E_S}{Z_0}\right)\tan\theta \ldots\ldots\ldots(5.10)$$

E_S and V_r are in phase, which is consistent with the fact that there is no power transfer.

The voltage profile is $V(x) = V_r \cos\beta(a-x)$

$$= \frac{E_S}{\cos\theta}\cos\beta(a-x)$$

$$V(x) = E_s \frac{\cos\beta(a-x)}{\cos\theta} \ldots\ldots(5.11)$$

Similarly the current profile is

$$I(x) = j\left(\frac{E_S}{Z_0}\right)\frac{\sin\beta(a-x)}{\cos\theta} \ldots\ldots\ldots(5.12)$$

The terminal voltage are controlled to have the same magnitude, $E_S = E_r$; with no power transfer the electrical conditions are the same at both ends. Therefore by symmetry $I_S = -I_r$.

The reactive power absorption capability of synchronous generators is limited for two reasons. First, underexcited operation increases the heating of the ends of the stator core. Second, the reduced field current reduces the internal emf of the generators, and this impairs stability.

In general, in the absence of compensating equipment, the synchronous machines must absorb (or) generate the difference between the reactiave power of the line and the line and that of the local load.

5.7 Compensated transmission lines:

The theory of compensation basically related with the following

1) Supply systems
2) load
3) Compensater.

1) Supply systems:

Is modelled as an equivalent circuit with a voltage source in series with a series impedance.

2) Load:

Is considered as a constant, impedance (or) admittance.

3) Compensator: Is modelled as a variable impednace (or) admittance (or) a variable source of reactive power (or) reactive current

A compensation system ideally performs the following functions.

1) It improves stability by increasing the maximum transmissible power.

2) It helps produce a substantially flat voltage profile at all levels of power transmission.

3) It provides an economical means for meeting the reactive power requirements of the transmission system.

Compensated lines enable the transmission of the natural load over greater distances, and shorter compensated lines can carry loads greater than the natural load. A flat voltage profile can be achieved if the effective surge impedance of the line is modified so as to have a virtual value, Z_0', for which the corresponding virtual natural load $\frac{V_0^2}{Z_0'}$ is equal to the actual load.

Campensations has divides into four types.

1) Passive compensation

2) Active compensation

3) Thyristor controlled capacitor

4) Thyristor controlled reactor

Passive compensators include shunt reactors, capacitors and series capacitors. These devices may be either permenently connected (or) switched. These compensators are used only for surge impedance compensation and line length compensation.

Active compensators are synchronous condensers only. These compensators are usually shunt-connected devices, which have the property of tending to maintain a substantially constant voltage at their terminals.

They are usually capable of continuous variation and rapid response.

In equipment development, activity is concentrated on the static reactive power controller (or) static compensator, to improve its efficiency, reliability, and response characteristics.

The surge impedance Z_0 of an uncompensated is

$$Z_0 = \sqrt{\frac{l}{C}} = \sqrt{\frac{jwl}{jwc}}$$

A fundamental frequency this can be written as $\sqrt{x_l x_c}$

$$\therefore Z_0 = \sqrt{\frac{l}{c}} = \sqrt{x_l x_c} \quad \text{.....(5.13)}$$

If a uniformly distributed shunt compensating inductance l_{vsh} is introduced, the effective value of the shunt capacitive admittance per mile becomes

$$(jwc)^1 = jwc + \frac{1}{jwl_{vsh}}$$

$$= jwc(1 - K_{sh}) \quad .....(5.14)$$

where $k_{sh} \rightarrow$ is the degree of shunt compensation.

$$K_{sh} = \frac{1}{w^2 l_{vsh} c}$$

Substituting for (jwe)′ in equation (5.13), the surge impedance has the effective (or) virtual value

$$Z_0' = \frac{Z_0}{\sqrt{1-K_{sh}}} \quad .....(5.15)$$

Shunt inductive compensation, therefore, increases the virtual surge impedance, whereas shunt capacitive compensation reduces it. In a similar way the effect of uniformly distributed series capacitance

$$Z_0' = Z_0 \sqrt{1-K_{se}} \quad .....(5.16)$$

Where K_{se} is the degree of series compensation

The parameters K_{sh} and K_{se} are a useful measure of the reactive power ratings required of the compensating equipment.

Combining the effects of shunt and series compensation,

$$Z_0' = Z_0 \sqrt{\frac{1-K_{se}}{1-K_{sh}}} \quad .....(5.17)$$

Corresponding to the virtual surge impedance Z_0' is a virtual natural load P_0'.

Given by $\frac{V_0^2}{Z_0'}$, so that

$$P_0' = P_0 \sqrt{\frac{1-K_{sh}}{1-K_{se}}} \quad .....(5.18)$$

The wave number β is also modified and has the virtual value

$$\beta' = \beta \sqrt{(1-K_{sh})(1-K_{se})} \quad .....(5.19)$$

The electrical length θ is modified according to

$$\theta' = \theta \sqrt{(1-K_{sh})(1-K_{se})} \quad .....(5.20)$$

Where $\theta = a\beta$ and $\theta' = a\beta'$

5.8 Shunt compensation:

Shunt compensation with reactors increases the virtual surge impedance and reduces the virtual natural load. In practice, shunt compensating reactors cannot be uniformly distributed. They are connected at the ends of the line and at intermediate points – usually at intermediate switching substations. In the case of very long lines, at least some of the shunt reactors are permanently connected to the line in order to give maximum security against overvoltage in the event of a sudden rejection of load (or) open circuiting of the line.

If there is a sudden load-rejection (or) open-circuiting of the line, it may be necessary to disconnect them very quickly, to prevent them from increasing the voltage.

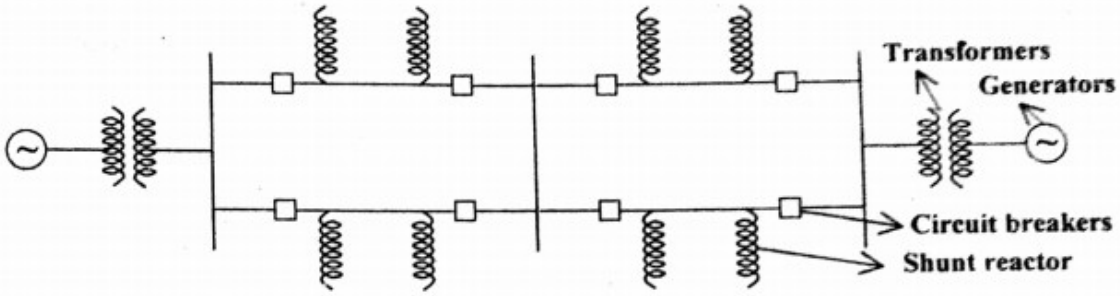


Figure 5.4. Arrangement of shunt reactors on a long distance high voltage ac transmission line

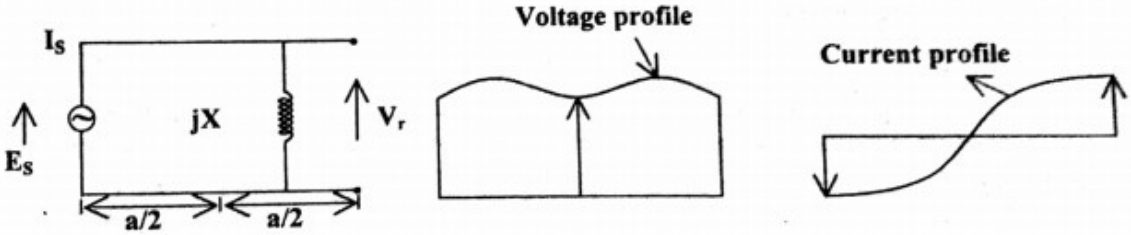


Figure 5.4. Voltage and current profiles of a shunt – compensated line at no load

Consider the simple circuit in figure 5.8 which has a single shunt reactor of reactance X at the receiving end and a pure voltage source E_S at the sending end. The receiving end voltage is given by

$$V_r = j\,X\,I_r \quad \text{.....(5.21)}$$

From equation (5.2) $E_S = V_r \cos\beta a + jZ_0 I_r \sin\beta a$

$$= V_r\left[\cos\theta + \frac{Z_0}{X}\sin\theta\right] \quad \text{.....(5.22)}$$

E_s and V_r are, therefore, in phase, which is consistent with the fact that no real power is being transmitted. The receiving end voltage to be equal to the sending end voltage, X must be given by

$$X = Z_0 \frac{\sin\theta}{1-\cos\theta} \quad \text{.....(5.23)}$$

The sending end current is given by equation (5.3)

$$I_S = j\frac{E_S}{Z_0} \sin\theta + I_r\cos\theta \quad \text{.....(5.24)}$$

Making use of equations (5.21) (5.22) and (5.23), this can be rearranged

$$I_S = j\frac{E_S}{Z_0} \frac{1-\cos\theta}{\sin\theta}$$

$$= j\frac{E_s}{X}$$

$$= - I_r \quad \text{.....(5.25)}$$

Since $E_S = V_r$. This means that the generator at the sending end behaves exactly like the shunt reactor at the receiving end in that both absorb the same amount of reactive power

$$Q_S - Q_r$$

$$= \frac{E_s^2}{X}$$

$$= \frac{E_s^2}{Z_0}\left[\frac{1-\cos\theta}{\sin\theta}\right] \quad \text{.....(5.26)}$$

The charging current divides equally between the two halves of the line.

5.9 Series compensation:

Series compensation is to cancel part of the reactance of the line by means of series capacitors. This increases the maximum power, reduces the transmission angle at a given level of power transfer, and increases the virtual natural load. Reducing the transfer reactance between the ends of a line, the series capacitor finds two main classes of application.

1) It can be used to increase the power transfer on a line of any length. Sometimes a series capacitor is used to increase the load share on one of two (or) more parallel lines
2) It can be used to enable power to be transmitted stably over a greater distance than is possible without compensation.

The circuit would also be series resonant at the fundamental frequency, and it would be difficult to control transient voltages and currents during disturbances. The high voltage and reactive power absorption at the ends of the line can be corrected by means of shunt reactors.

The voltage on either side of the series capacitor is nearly 1.0 pu over practically the whole range of power transfer, implying that the shunt reactors could be permanently connected without disadvantage. The reactive absorption at the terminals is considerable reduced, and at high

values of p the terminal power factors become lagging, which may be transient stability because the internal generator voltages are increased.

5.10 Introduction to voltage stability problem in power systems:

Voltage stability is sometimes also called load stability. Power system stability may be defined as that property of the system which enables the synchronous machines of the system to respond to a disturbance from a normal operating condition so as to return to a condition where their operation is again normal. Stability studies are usually classified into three types depending upon the nature and order of magnitude of the disturbance.

These are 1) Steady state stability
2) Dynamic stability
3) Transient state stability

The terms voltage instability and voltage collapse are often used interchangable. The voltage instability is a dynamic process where in contrast to rotor angle stability, voltage dynamics mainly involves loads and the means for voltage control. Voltage instability limit is not directly correlated to the network maximum power transfer limit.

The concept of voltage stability is related to transient stability of a power system. The analysis of voltage stability normally requires simulation of the system modelled by non-linear differential logebriac equations. Comparsions of these types of stability is listed below.

		Steady state stability	Dynamic stability	Transient stability
1)	Oscillations	No	Small	Large
2)	Type of disturbance	Small, slow and gradual	Random load fluctuations	Large, sudden
3)	Mathematical model	Non-linear algebric equation	Linear differential equation	Non-linear differential equation
		$P_c \frac{\lvert E_1 \rvert \lvert E_2 \rvert}{X} \sin\delta$	$\frac{Md^2\delta}{dt^2} = Pa$	$\frac{Md^2\delta}{dt^2} = P_I - P_m \sin\delta$

The concept of small disturbance voltage stability is related to steady state stability and can be analysed using small signal model of the system. Voltage security is the ability of a system, not only to operate stably, but also do remain stable following credible contingencies (or) load increases.

Certain situations in power system cause problems in reactive power flow which lead to system voltage collapse. Some of the situations that can occur are listed and explained below.

1) Long transmission lines
2) Radial transmission lines
3) Shortage of local reactive power

1) Long transmission lines:

In power systems, long lines with voltage uncontrolled buses at the receiving ends create major voltage problems during light load (or) heavy load conditions.

2) Radial transmission lines:

In a power system, most of the parallel EHV networks are composed of radial transmission lines. Any loss of an EHV line in the network causes an enhancement in system reactance. Under certain conditions the increases in reactive power delivered by the lines to the load for a given drop in voltage, is less than the increase in reactive power required by the load for the same voltage drop.

3) Shortage of local reactive power:

There may occur a disorganised combination of outage and maintenance schedule that may cause localised reactive power shortage leading to voltage control problems.

The voltage stability analysis for a given system state involves the following two aspects

i) Proximity to voltage instability
ii) Mechanism of voltage instability

i) Proximity of voltage instability: Instability may be measured in terms of physical quantities, such as load level, real power flow through a critical interface, and reactive power reserve. Possible contingencies such as a line outage, loss of a generating unit (or) a reactive power source must be given due consideration.

ii) Mechanism of voltage instability: The slower forms of voltage instability are normally analysed as steady state problems using power flow simulation as the primary study method. The two methods (i.e. pv curves and VQ curves) give steady state load ability limits which are related to voltage stability. Pv curves are useful for conceptual analysis of voltage stability and for study of radial systems.

5.11 Static compensators:

Static compensators are those which have no rotating parts and are used for surge impedance compensation and for compensation by sectioning a long transmission line. These are also used for load compensation where in these maintain constant voltage

i) Under slowly varying load changes
ii) Load rejection, outages of generator and line
iii) Under rapidly varying loads.

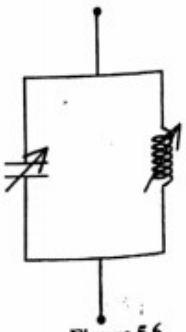


Figure 5.6

Fig. 5.6 shows an ideal static reactive power compensator. In actual practice, the reactive current is limited in both lagging and leading regions because of the current carrying capacity of the compensator.

Some of the important compensators used in transmission and distribution networks are

i) Thyristor controlled reactor (TCR)
ii) Thyristor switched capacitors (TSC)
iii) Saturated reactors.

i) Thyristor controlled reactor (TCR):

Figure 5.7 shows a basic thyrister controlled reactor. The controlled element is the reactor and the controlling element is the thyristor controller consisting of two oppositely poled thyristors which conduct every alternate half cycles of the supply frequency. The reactive power absorbed by the reactor will depend upon the instant of switching on the voltage wave.

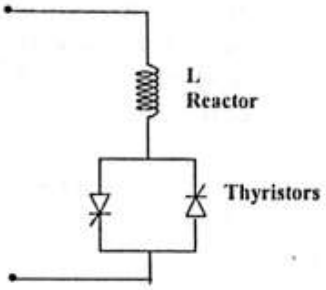


Fig. 5.7. Basic thyristor controlled reactor (TCR)

If the voltage is passing through its peak value at the instant of switching of thyristor, and as the delay angle increases between 90^0 to 180^0. The fundamental component of current will be maximum when delay angle is 90^0 decreases with increase in delay angle between 90^0 and 180^0.

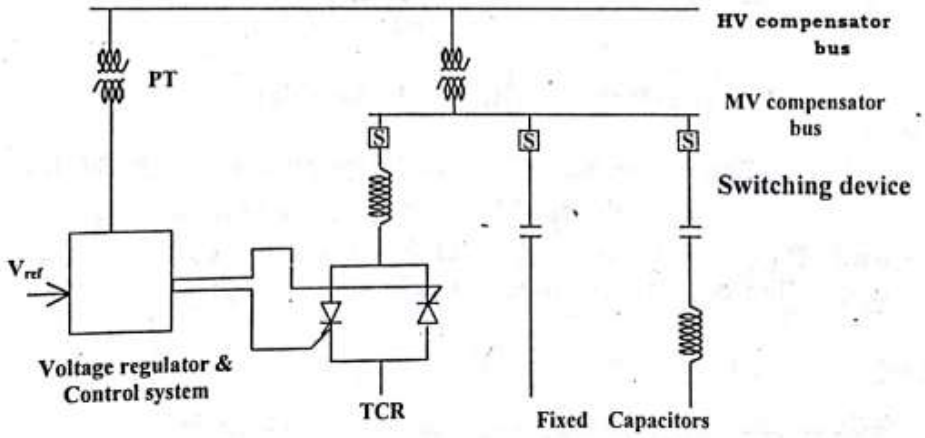


Figure 5.8 Thyristor controlled reactor (TCR)

To find the following features in the above Ckt.

1) A fixed shunt capacitor is parallel with the controlled susceptance. The fixed capacitors are usually tuned with small reactors to harmonic frequencies to absorb harmonics generated by the controlled susceptance (or) to avoid harmful resonances.
2) A step down transformer with significantly affects the performance of the transformer especially with respect to losses, harmonics and over voltages.

Increasing the delay angle (or) reducing the conduction angle decreases the power loss in both the reactors and the thyristors and generators harmic currents. If both the thyristors are fired at the same angle, all odd order harmonics are produced.

ii) Thyristor switched capacitors: (TSC)

Figure 5.10. shown the Thyristor switched capacitors (TSC) and figure 5.11 shows the current and voltage wave forms under transient free switching in and switching out of capacitor.

Assuming the time origin to coincide with a positive going zero crossing of the voltage, the delay angle 'α' and the conduction angle 'ρ' so that the conduction period lies between α and α + P.

i.e. $v = V_m \sin\omega t$

Therefore, instantaneous current i is given as

$$i = \frac{1}{X_l}\int_{\alpha}^{\omega t} V_m \sin\omega t \; d(\omega t)$$

where X_l is the inductive reactance of the reactor at fundamental frequency

$$i = \frac{V_m}{X_l}[\cos\alpha - \cos\omega t] \quad \text{and} \quad I = 0 \quad \text{for } \alpha + \rho < \omega t < \alpha + \pi$$

Using Fourier series, the fundamental component of current is given be

$$I_1 = \frac{\rho - \sin\rho}{\pi X_l} \cdot \frac{V_m}{\sqrt{2}}$$

ρ is related to α through the relation $\alpha + \frac{\rho}{2} = \pi$

The above equation can be written as $I_1 = B_l(\rho).V$

Where $B_l(\rho) \rightarrow$ is the rms voltage.

And $B_l(\rho) \rightarrow$ is an adjustable fundamental frequency susceptance controlled by the conduction angle according to the low

$$B_l(\rho) = \frac{\rho - \sin\rho}{\pi X_l}$$

The susceptance is maximum when ρ = π and its value is given as

$$B_l(\pi) = \frac{\pi - 0}{\pi X_l} = \frac{1}{X_l}$$

And its value is zero when ρ = 0

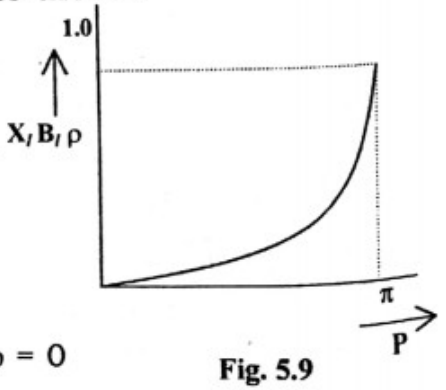


Fig. 5.9

The variation of B_l as a function of ρ is shown in figure 5.9.

Increasing the delay angle (or) reducing the conduction angel decreases the power loss in both the reactors and the thyristors and and generators harmonic currents. If both the thyristors are fired at the same angle, all odd order harmonics are produced.

ii) Thyristor switched capacitors (TSC)

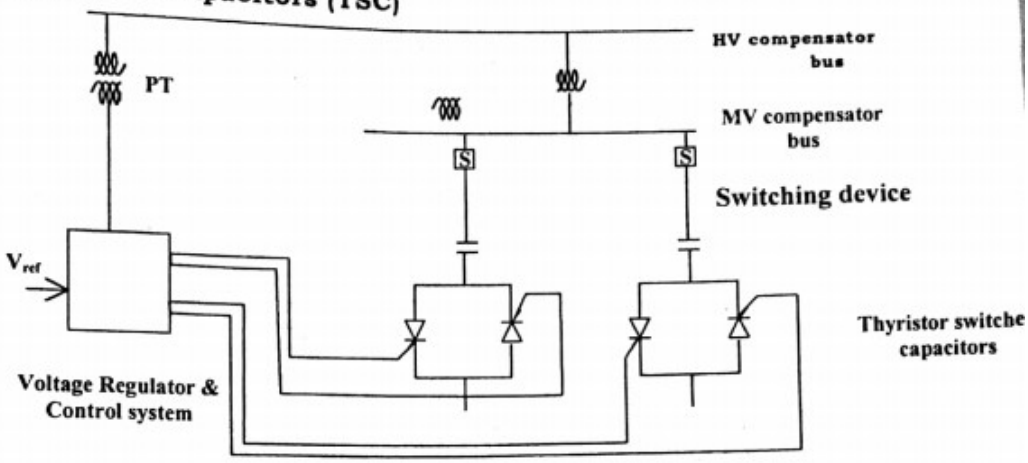
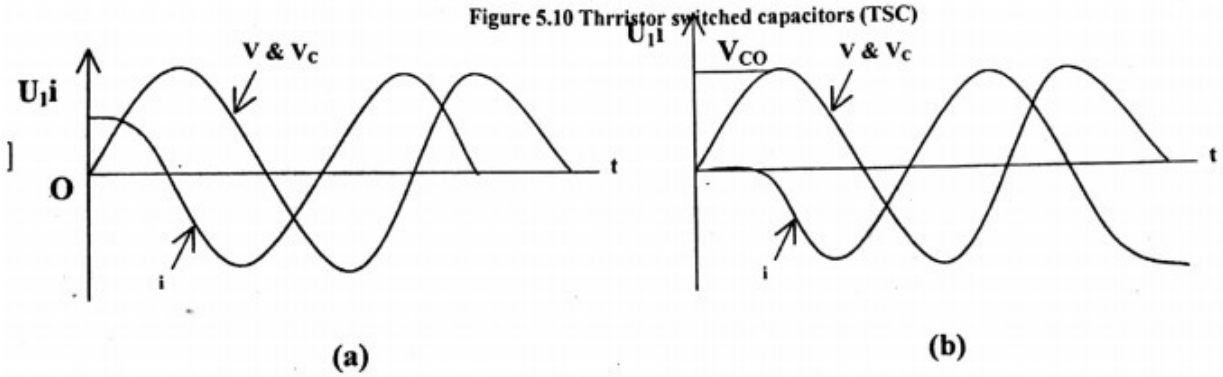


Figure 5.10 Thrristor switched capacitors (TSC)



(a) (b)

Fig. 5.11. Transient free switching operation

(a) → Switching in (b) → Switching out

Fig 5.10 shown the Thyristor switched capacitors (TSC) and figure 5.11 shows the current and voltage wave forms under transient free switching in and switching out of capacitor.

Each capacitor always conducts for an integral number of cycles.

Normally the capacitors used are of the same capacity except one of the capacitors which has its susceptance half the susceptance of other capacitors. If there are (n – 1) number of equal capacitances and one of half value, 2n number of combinations of capacitors is possible. In order to have transient free switching in (or) switching out of capacitor is charged to either the positive (or) the negative system peak voltage.

iii) Saturated reactions:

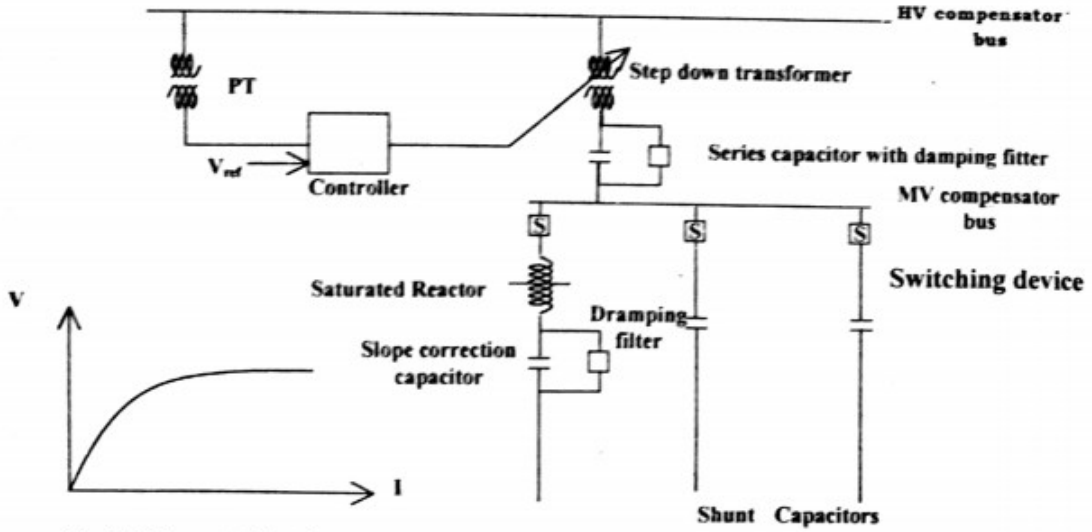


Figure 5.12 Saturated reactors

Fig. 5.13 Characteristic of a saturated reactor compensator

The plain saturated reactor is unsuitable for use in transmission systems as the voltage (or) the current contain lot of harmonics. Figure 5.13 shows a typical volt-amp characteristic of a saturated reactor, the slope of which depends upon the after-saturation inductance of the winding. The slope of the characteristic varies between 5 and 15%. The characteristic is linear above about 10% of the rated current. The presence of slope-correction capacitors in series with the saturated reactor has been recognised as a possible source of ferro-resonant sub-harmonic oscillations. Therefore, in the design of static reactive compensators this phenomenon is always suppressed by a sub-harmonic damping by-pass circuit incorporating an adeqately dimensioned resistor and a power frequency tuned –rejector circuit as shown in figure.

5.12 Series capacitors:

Both series and shunt capacitor banks are usuful tools in improving system efficiency and power transfer capacity. Shunt capacitors generate reactive power and help to reduce the amount of reactive power that flows through the network. For maximum effectiveness in reducing line losses and voltage control, they are typically installed near the load.

Series capacitors are applied to compensate for the inductive reactance of the transmission line. They may be installed remote from the load. Their benefits are

1) improved system steady state stability
2) improved system transient stability
3) reduced transmission losses.
4) adjustment of line loadings

The voltage drop on a transmission line is approximately.

Proportional to the inductive reactance X, the reduction of this reactance is an obviously powerful tool in improving voltage conditions. This is done by providing capacitors in series with the line and is known as series compensation.

In series compensation of the line K = $\frac{Xc}{X} * 100$

Where K = is know as percentage compenation

X_C = capacitive reactance of the series capacitors in series with the line

X = reactance of the transmission line.

Figure 5.14 shows the short radial feeder with series capacitor.

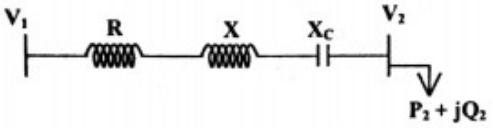


Fig. 5.14. Radial feeder with series capacitro

Voltage drop (without series capacitor) $\Delta V = \frac{P_2R + Q_2X}{|V_2|}$

Voltage drop (with series capacitor) $\Delta V' = \frac{P_2R + Q_2(X - X_C)}{|V_2|}$

The capacitor raises the voltage = $\Delta V_C = \Delta V - \Delta V' = \frac{Q_2}{|V_2|} X_C$

The capacitive reactance X_C used is generally less than the inductive reactance of the transmission line.

.13 The reactive power balance and its effect on system voltage:

There is a strong inter-relation in the power system between reactive power flow and the system voltage. Let us consider the two bus system shown in figure 5.15. Load $P_2 + jQ_2$ is connected to bus 2 and power is transferred from bus 1 through a short transmission line having resistance R and reactance X. Let us consider that the sending end voltage V_1 phase to neutral is kept constant by the field control of the generator G_1 and | or top changing transformer T.

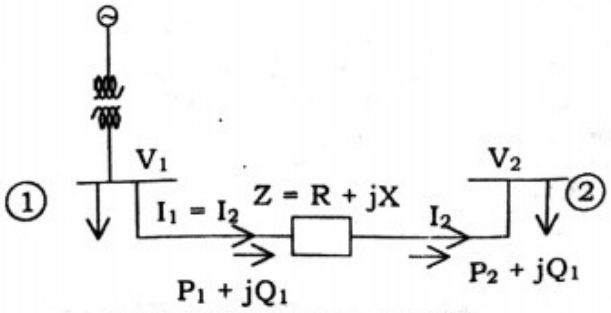


Fig. 5.15. Power flow from bus 1 to bus 2

The transmission line $R \leq X$, neglect the active power loss ΔP_L simplifying the problem neglect the reactive power loss. ΔQ_L of the line also.

Above assumption, $S_2 = P_2 + jQ_2$

$= P_1 + jQ_1$

and the $V_2 = V_1 - I_1 (jX)$

$V_1 I_1^* = P_2 + jQ_2$

(or) $V_1^* I_1 = P_2 - jQ_2$

$$\Rightarrow I_1 = \frac{P_2 - jQ_2}{V_1^*} = \frac{P_2 - jQ_2}{V_1}$$

($V_1^* = V_1$ being the reference vector)

$$\therefore V_2 = V - j \frac{P_2 - jQ_2}{V_1} x$$

$$= \left(V_1 - \frac{Q_2 X}{V_1}\right) - j \frac{P_2}{V_1} X$$

$= (V_1 - \Delta V) - j\delta V$

where $\Delta V = \frac{Q_2}{V_1} X$

and $\delta V = \frac{P_2}{V_1} X$

When the reactive component of the load is decreased by connecting a static capacitor across the load and under this condition the new reactive load is

$$Q_2^1 = (Q_2 - Q_c)$$

Where Q_C is the reactive power of the static capacitor. A small change in the magnitude of the bus voltage, the real power at the bus does not change appreciably and similarly a small change in phase angle of the bus voltage, the reactive power does not change appreciably.

5.14 Combined use of tap changing transformers and reactive power injection:

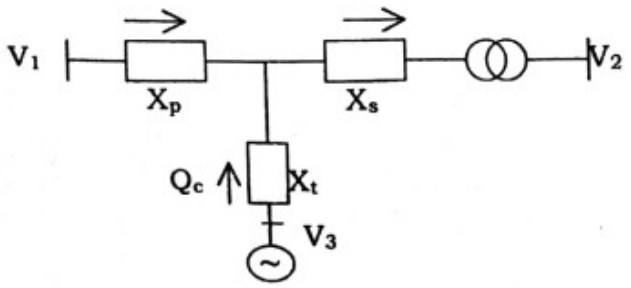


Fig 5.16. Equivalent circuit of combined use of tap changing transformers and reactive power injection

Equivalent of the transformer in terms of equivalent primary, secondary and tertiary reactances is shown in figure 5.16 along with off-nominal setting t at the receiving end. V_1 and V_2 are the phase voltages. Normally, tap setting are provided in steps for the range of ± 20%

$$\text{Voltage drop } \Delta V = \frac{P_2 R + (Q_2 - Qc) X_p}{|V_n|}$$

Where $V_n \rightarrow$ is the phase voltage at the star point of the equivalent circuit of the three winding transformer.

$$= \frac{(Q_2 - Q_c) X_p}{|V_n|} \quad \text{.....(5.27)}$$

Resistance R of the transmission line and that of three-winding transformer is neglected.

Similarly, quadrature voltage drop $\delta V = \frac{P_2 X_p}{|V_n|}$ .....(5.28)

Using equations (5.27) and (5.28)

$$|V_1|^2 = (|V_n| + \Delta V)^2 + |\delta V|^2$$

$$= \left[|V_n| + \frac{(Q_2 - Q_c) X_p}{|V_n|}\right]^2 + \left[\frac{P_2 X_p}{|V_n|}\right]^2$$

$$\therefore |V_1|^2 = \left[\frac{|V_n|^2 + (Q_2 - Q_C) X_p}{|V_n|}\right]^2 + \left[\frac{P_2 X_p}{|V_n|}\right]^2$$

$$|V_1|^2 = \frac{1}{|V_n|^2} [|V_n|^2 + (Q_2 - Q_c)X_p]^2 + (P_2 X_p)^2$$

$$= \frac{1}{|V_n|^2} [|V_n|^4 + ((Q_2 - Q_c)X_p]^2 + 2|V_n|^2 + (Q_2 - Q_c)X_p] + (P_2 X_p)^2$$

Solving $|V_n|$

$$|V_n|^2 = \left[\frac{|V_1|^2}{2} - (Q_2 - Q_c) X_p\right] \pm \sqrt{\frac{|V_1|^4}{4} - X_p |V_1|^2 (Q_2 - Q_c) - X_p^2 P_2^2} .....$$

5.15 Shunt reactors connected in high voltage transmission system:

Shunt capacitance in lines, and particularly in cables, actually have a surplus of reactive power during light load(or) no load conditions. During light load and under no load conditions, to connect Q-consuming elements.

Generally the shunt reactors are used in long EHV transmission line and the practical and economic reasons lead to concentrate these compensating elements at a few points along the line. If the linear shunt

reactors are switched out under heavy load conditions, the maximum power transfer capability is considerably increased, but voltage changes due to sudden load rejection are likely to be unacceptably high.

5.16 Line compensations:

Increases the steady state stability limit of the power system network. For increasing steady state stability limits, modify the reactance of the transmission network. Series compensation (or) a series capacitor to reduce the effective reactance of the transmission system.

1) **Line length compensation:** For any transmission line, three critical loadings
   a) Natural impedance loading
   b) Thermal loading
   c) Steady state stability limit.

   For short lines, it is observed that the steady state stability limit will be greater than thermal loading limit.
2) **Medium lines compensation:** For un-compensated lines the voltage level is the main consideration for limiting the amount of power transfer.
3) **Long lines compensation:** Steady state stability limits the amount of power transfer. In the case of long lines it is essential that should try to employ suitable compensation to increase the steady state stability link.
4) **Ferranti effect:** In the case of long lines, under no load & light load conditions, the receiving end voltage is higher than sending end voltage. Variation in voltage levels for all operating conditions should be within permissible limit.
5) **Flat voltage profile:** Under excited operation of synchronous machines is to be avoided. Reactive power management is to be using suitable compensates.

5.17. Problems:

1) A load of (66 + j60) MVA at the receiving end is being transmitted via a single circuit 220 KV line, having resistance of 21 ohms and reactance of 34 ohms. The sending end voltage is maintained at 220 KV. The operating conditions of power consumers require that at this load voltage drop across the line should not exceed 5 percent. In order to reduce voltage drop, standard 1 ϕ, 0.66 KV, 40 KVAR capacitors are to be switched in series in each phase of the line. Determine the required number of capacitors, rated voltage and installed capacitors of the capacitor link. The losses in the line are neglected.

Solution:

Resistance (R) = 21 Ω

Reactance (X) = 34 Ω

P_2 = 22 MW

Q_2 = 20 MVAR

$$|V_1| = \frac{220}{\sqrt{3}} = 127 \text{ KV}$$

Voltage drop (without series capacitor) = $\Delta V = \frac{P_2 R + Q_2 X}{|V_2|}$

$$|V_1| = |V_2|$$

$$\therefore \Delta V = \frac{(22*21)+(20*34)}{127} = 8.99 \text{ KV}$$

Permissible voltage drop ΔV_{perm} = 5% * 127

$$= \frac{5}{100}*127 = 6.35 \text{ KV}$$

Voltage drop(with series capacitor) = $\therefore \Delta V' = \frac{P_2 R + (X - X_c) Q_2}{|V_2|}$

But voltage drop with series capacitor should be equal to the permissible voltage drop.

$$\therefore \Delta V_{perm} = \Delta V'$$

$$\therefore \Delta V_{perm} = \frac{P_2 R + (X - X_c) Q_2}{|V_2|}$$

$$6.35 = \frac{(22*21)+20(34 - X_c)}{127}$$

$$\Rightarrow X_C = 16.65\ \Omega$$

given rated load, the current in the line is

$$|I| = \frac{\sqrt{(66)^2 + (60)^2}}{\sqrt{3}*220} = 235.5 \text{ A}$$

The rate current of the given capacitors $I_C = \frac{40}{0.66} = 60.6$ A

erefore, the number of capacitors connected in parallel in one-phase must be greater than $m = \frac{235.5}{6.6} = 3.88 \approx 4$

$\therefore$ The number of parallel connected capacitors = 4

The reactance of the given capacitor $X_{C1} = \frac{660}{60.6} = 10.9\ \Omega$

The reactance of each capacitor and the number of their parallel circuits m, determine the number of series connected capacitors n using the equation

$$\frac{X_{cl}}{m} * n = X_c$$

$$\Rightarrow \frac{10.9}{4} * n = 16.65$$

$$\Rightarrow n = 6.1 \approx 7$$

$\therefore$ The total number of capacitors in all the three phases are

$$= 3 * 4 * 7 = 84$$

The installed capacity of the capacitor bank is $Q_{cb} = 84 * 40 = 3360$ KVAR

The rated voltage of the capacitors bank is $V_{cb} = 0.66 * n$

$= 0.66 * 7$

$= 4.62$ KV

The rated current of the capacitors bank is $I_{cb} = I_c * m$

$= 6.06 * 4$

$= 242.4$ AV

Selected number of capacitors, the actual reactance of the bank is]

$$X_{cb} = \frac{X_{cl}}{m} * n$$

$$= \frac{10.9}{4} * 7 = 19.075\ \Omega$$

The voltage drop in the line is $\Delta V = \frac{(22*21)+20(34-19.075)}{127} = 6$ KV

2. With a 100 MVA generator operating at 85% power factor logging

(i) How many MVAr will be produced?

(ii) What mega watt load should the machine be limited so that its MVA rating will not be exceeded?

Solution:

MVA rating of generator = 100 * 0.85 = 85 MW

$$\phi = \text{Cos}^{-1}(0.85) = 31.788$$

(i) MVAr = 85 * Tan(31.788) = 52.87 MVAr

(ii) MW rating of generator = 85 MW

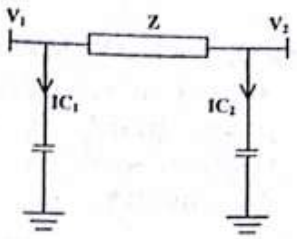
3. A single circuit three phase 220 KV, line operates on no load, voltage at the receiving end of the line is 205 KV. Find the sending end voltage, if the line has resistance of 21.7 ohms, reactance of 85.2 ohms and the total susceptance as $5.32 * 10^{-4}$ mho. The transmission line is to be represented by π method.

Solution:

$$R = 21.7\ \Omega$$

$$X = 85.2\ \Omega$$

$$\text{Susceptance} = 5.32 * 10^{-4}\ \text{mho}$$



$$|V_1| = |V_2| - \frac{Q_c}{|V_2|} X$$

$$|V_2| = \frac{205}{\sqrt{3}} = 118.35\ \text{KV}$$

$$Q_C = |V_2|^2 \frac{Wc}{2} = (118.35)^2 * \frac{5.32}{2}\ \text{mho} = 3.716\ \text{MVAr}$$

$$\therefore |V_1| = 118.35 - \frac{3.716}{118.35} * 85.2$$

$$= 118.35 - 2.69 = 115.67\ \text{KV}$$

Sending end voltage, line – to – line = $\sqrt{3} * 115.67 = 200.34$ KV

4. The load at the receiving end of a three – phase, over head line is 25 MW, power factor 0.8 lagging, at a line voltage of 33 KV. A synchronous compensator is situated at the receiving end and the voltage at both ends of the line is maintained at 33 KV. Calculate the MVAr of the compensator. The line has resistance 5 ohm per phase and inductive reactance 20 ohm per phase.

Solution:

The synchronous compensator of Q_C MVAr is in parallel with the load, then

$$|V_1| = \sqrt{\left[|V_{ll} + \frac{P_2R + (Q_2 - Q_c)X}{|V_2|}\right]^2 + \left[\frac{P_2X - (Q_2 - Q_c)R}{|V_2|}\right]^2}$$

Total load = 25 MW

$$P_2 = \frac{25}{3} = 8.33\ \textbf{MW / Phase}$$

$$Q_2 = 8.33 * \text{Tan}^{-1}(\phi)$$

Where $\phi = \text{Cos}^{-1}(0.8)$

$$Q_2 = 8.33 * \frac{3}{4} = 6.25\ \text{MVAr / Phase}$$

$$|V_1| = |V_2| = \frac{33}{\sqrt{3}} = 19\ \text{KV}$$

$$R = 5\ \Omega\ \text{/ Phase}$$

$$X = 20\ \Omega\ \text{/ Phase}$$

Substituting the above given values, then the above expression is

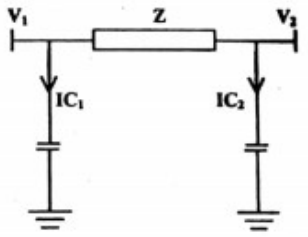
$$19 = \sqrt{\left[19+\frac{8.33*5)+(6.25-Q_c)20}{19}\right]^2+\left[\frac{8.33*20-(6.25-Q_c)5}{19}\right]^2}$$

$\Rightarrow Q_c$ = 11.033 MVAr; Total reactive power = 3 * 11.0333 . 33.099 MVAr

5. A load of (15 +j10) MVA is supplied with power from the busbars of power plant via a three phase 110 KV line, 100 KM lagging. The transmission line is represented by π - method and has the following parameters

$$R = 26.4\ \Omega$$
$$X = 33.9\ \Omega$$
$$B = 219 * 10^{-6}\ \Omega$$

Voltage across the power plant has bars V_1 = 116 KV. Find the power consumed from the power plant bus bars.



Solution:

$$S_2 = P_2 + jQ_2$$
$$= \frac{1}{3}(15 + j10)$$
$$= 5 + j3.33 \text{ MVA}$$

Rated voltage per phase $|V_r| = \frac{110}{\sqrt{3}} = 63.5$ KV

$$Q_c = |V_r|^2 \frac{WC}{2}$$

$$= \frac{(63.5)^2 * 219 * 10^{-6}}{2} = 0.441 \text{ MVAr}$$

Power at the receiving end of the line $S = S_2 - Q_c$
$= 5 + j3.33 - j0.441 = (5 + j2.89)$ MVA

Power loss in the line $\Delta P_L = \frac{|S|^2}{|V_r|^2} * R$

$$= \frac{|(5+j2.89)^2|}{(63.5)^2} * 26.4$$
$$= 0.218 \text{ MW}$$

$$\Delta Q_L = \frac{|S|^2}{|Vr^2} * (X)$$

$$= \frac{|(5+j2.89)^2|}{(63.5)^2} * 33.9 = 0.279 \text{ MVAr}$$

Power at the sending end of the line $S^1 = S + \Delta P_L + \Delta Q_L$
$= (5 + j2.89) + 0.218 + j0.279$

$= (5.22 + j3.169)$ MVA

$\therefore$ Total power consumed from the power plant bus bars

$$S_1^1 = 3S_1$$

$$= 3(5.22 + j2.73) = (15.66 + j\,8.19) \text{ MVA}$$

6. A single circuit three phase 220 KV line runs at **no load**. Voltage at the receiving end of the line is 210 KV. **Find** the **sending** end voltage, if the line has resistance of 20.5 ohms, reactance of **81.3** ohms and the total susceptance as $5.45 * 10^{-4}$ mho. The tran**smission** line is to be represented by π - method.

Solution:

$$|V_2| = \frac{210}{\sqrt{3}} = 121.2 \text{ KV}$$

$$Q_c = |V_2|^2 * \frac{WC}{2} = (121.2)^2 * \frac{5.45*10^{-4}}{2} = 4 \textbf{ MVAr}$$

$$X = 81.3\ \Omega$$

$$\therefore |V_1| = |V_2| - \frac{Q_c}{|V_2|}X$$

$$= 121.2 - \frac{4}{121.2} * 81.3 = 118.51$$

Sending end voltage to line $= \sqrt{3} * (V_1)$

$$= \sqrt{3} * 118.51$$

$$= 205.2 \text{ KV}$$

7. Find the capacity of static VAR compensator to be installed at a bus with $\pm$ 5 voltage fluctuation. The short circuit capacity is 5000 MVA.

Solution:

The switching of static shunt compensator,

ΔV = Voltage fluctuation

ΔQ = Reactive power variation

S = System short circuit capacity

$$\text{Then } \Delta V = \frac{\Delta Q}{S}$$

$$\Rightarrow \Delta Q = \Delta V * S$$

$$= 0.05 * 5000 = 250 \text{ MVAr}$$

The capacity of a static VAR compensator is (+ 250 MVAr)

8. A short line having an impedance of (1.5 + j2.5) Ω interconnects two plants A and B each plant operates at 11 KV and the two voltage are equal in magnitude and phase. It is proposed to transfer 20 MW power at 0.8 p.f. from plant A to B. Find the voltage boost needed at plant A to achieve this power transfer.

Solution:

$$I = \frac{20*10^3}{\sqrt{3}*11*0.8} = 1312.2 \angle -36.87^0$$

$$= (1049.76 - j787.32)A$$

Voltage drop in the inter connector = (1049.76 – j787.32) (1.5 + j2.5)

$$= (3542.94 + j1443.42)V$$

The voltage boost needed at station A(plant A) is

$$= (3542.94 + j1443.42)\ V$$

9. A load bus is composed of induction motor where the nominal reactive power is 1 Pu. The shunt compensation is K_{sh}. Find the reactive power sensitivity at the bus with respect to change in voltage.

Solution:

$$\frac{dQnet}{dv} = 2V - 2V\ K_{sh}$$

$$V = 1.0\ Pu$$

$$K_{sh} = 0.8$$

$$\therefore \frac{dQ_{net}}{dv} = (2*1) - (2*1*0.8)$$

$$= 2 - 1.6 = 0.4\ Pu$$

10. The line is open circuited with a receiving end voltage of 220 KV, find the rms value and phase angle of the following

a) The incident and reflected voltages to neutral at the receiving end.

b) The incident and reflected voltages to neutral at 200 KM from the receiving end.

$$\alpha = 0.163 * 10^{-3}$$

$$\beta = 1.068 * 10^{-3}$$

Solution:

a) At the receiving end

For open circuit $I_R = 0$

$$\text{Incident voltage} = \frac{V_R + Z_C I_R}{2} = \frac{V_R + Z_c(0)}{2} = \frac{V_R}{2}$$

$$= \frac{220/\sqrt{3}}{2} = 63.51\angle 0^0 \text{ KV}$$

$$\text{Reflected voltage} = \frac{V_R - Z_C I_R}{2} = \frac{V_R}{2} = 63.51\angle 0^0 \text{ KV}$$

b) At 200 KM from the receiving end

$$\text{incident voltage} = \frac{V_R}{2} e^{\alpha x} e^{j\beta x}\Big|_{x=200 \text{ km}}$$

$$= 63.51 \exp(0.163 * 10^3 * 200) * \exp(j1.068 * 10^{-3} * 200)$$

$$= 65.62 \angle 12.2^0 \text{ kv}$$

$$\text{Reflected voltage} = \frac{V_R}{2} e^{-\alpha x} e^{-j\beta x}\Big|_{x=200 \text{ km}}$$

$$= 63.51\ e^{-0.0326}\ e^{-j0.2135}$$

$$= 61.47 \angle -12.2^0 \text{ KV}$$

11. A load of 30 MW, 45 MVAr is connected to a line where X to R ratio is 5 and the short circuit capacity of the load bus is 250 MVA. The supply voltage is 11 KV and the load is star connected. Determine the load bus voltage

Solution:

$$\text{The load per phase} = 10 + j15 \text{ MVA}$$

$$\text{Short circuit current} = \frac{250}{3} \text{ MVA}$$

$$\text{Voltage line} - \text{neutral} = \frac{11}{\sqrt{3}} \text{ KV}$$

$$\text{The equivalent short circuit impedance} = \frac{(11/\sqrt{3})}{250/3} = 0.484\Omega$$

$$\phi_{sc} = \text{Tan}^{-1}(5) = 78.69^0$$

$$\therefore R = 0.0949\ \Omega$$

$$X = 0.4746\ \Omega$$

$$\therefore \left(\frac{11}{\sqrt{3}}\right)^2 = \left(V\cos\phi + \frac{P}{V}R\right)^2 + \left(V\sin\phi + \frac{Q}{V}X\right)^2$$

$$= \left(V * 0.5547 + \frac{10 * 10^6}{V} * 0.0949\right)^2 + \left(V * 0.832 + \frac{15 * 10^6}{V} * 0.4746\right)^2$$

$$\frac{121}{3} = \left(0.5547V + \frac{10}{V} * 0.0949\right)^2 + \left(0.832V + \frac{15 * 0.4746}{V}\right)^2$$

$$\Rightarrow 40.33\ V^2 = 0.3077\ V^4 + 0.9 + 1.053V^2 + 0.69V^4 + 50.68 + 11.846\ V^2$$

$$\Rightarrow 0.9977\ V^4 - 27.43\ V^2 + 51.59 = 0$$

$$\Rightarrow V^4 - 27.5\ V^2 + 51.6 = 0$$

$$\Rightarrow V^2 = \frac{27.5 \pm \sqrt{756.25 - 206.32}}{2}$$

$$= \frac{27.5 + 23.45}{2} = 25.47$$

$$V = 5.047\ KV$$

The drop in volts is = 6.350 – 5.047 = 1.303 KV

$$\therefore \frac{\Delta V}{V} = \frac{1}{Ssc}(P\cos\phi + Q\sin\phi) + j\frac{1}{S_{sc}}(P\sin\phi sc - Q\cos\phi sc)$$

$$= \frac{1}{250}(30 * 0.196 + 45 * 0.98) + j\frac{1}{250}(30 * 0.98 - 45 * 0.196)$$

$$= \frac{1}{250}[(5.88 + 44) + j(29.4 - 8.82)]$$

$$= \frac{53.6}{250} = 0.2158$$

UNIT- V

1.1 INTRODUCTION

For many decades, vertically integrated electric utilities monopolized the way they controlled, sold and distributed electricity to customers in their service territories. In this monopoly, each utility managed three main components of the system: generation, transmission and distribution. Analogous to perceived competitions in airline, telephone, and natural gas industries which demonstrated that vertically integrated monopolies could not provide services as efficiently as competitive firms, the electric power industry plans to improve its efficiency by providing a more reliable energy at least cost to customers. A competition is guaranteed by establishing a restructured environment in which customers could choose to buy from different suppliers and change suppliers as they wish in order to pay market-based rates.

To implement competition, vertically integrated utilities are required to unbundle their retail services into generation, transmission and distribution; generation utilities will no longer have a monopoly, small businesses will be free to sign contract for buying power from cheaper sources, and utilities will be obligated to deliver or wheel power over existing lines for a fee that is the same as the cost (non-discriminatory) delivering the utility's own power without power production costs. The vertically integrated system is steadily restructuring to a more market based system in which competition will replace the role of regulation in setting the price of electric power.

1.2 RESTRUCTURING MODELS

Three major models are being discussed as alternatives to the current vertically integrated monopoly. The three models are:

a) PoolCo Model
b) Bilateral Contracts Model
c) Hybrid Model

Elements of a certain electric power industry define the nature of competition and models or institutions that support the competition process. In adopting a model, the following issues are being debated regularly:

- Who will maintain the control of transmission grid?
- What types of transactions are allowed?
- What level of competition does a system warrant?

A PoolCo is defined as a centralized marketplace that clears the market for buyers and sellers where electric power sellers/buyers submit bids and prices into the pool for the amounts of energy that they are willing to sell/buy. The ISO or similar entities (e.g. PX) will forecast the demand for the following day and receive bids that will satisfy the demand at the lowest cost and prices for electricity on the basis of the most expensive generator in operation (marginal generator). On the other hand, in the second model, bilateral trades are negotiable and terms and conditions of contracts are set by traders without interference with system operators.

1.2.1 PoolCo Model:

The PoolCo model is comprised of competitive power providers as obligatory members of an independently owned regional power pool, vertically integrated distribution companies, vertically integrated transmission companies and a single and separate entity responsible for: establishing bidding procedures, scheduling and dispatching generation resources, acquiring necessary ancillary services to assure system reliability, administering the settlements process and ensuring non-discriminatory access to the transmission grid. PoolCo does not own any generation or transmission components and centrally dispatches all generating units within the service jurisdiction of the pool. PoolCo controls the maintenance of transmission grid and encourages an efficient operation by assessing non-discriminatory fees to generators and distributors to cover its operating expenses.

In a PoolCo, sellers and buyers submit their bids to inject power into and out of the pool. Sellers compete for the right to inject power into the grid, not for specific customers. If a power provider bids too high, it may not be able to sell power. On the other hand, buyers compete for buying power and if their bids are too low, they may not be getting any power. In this market, low cost generators would essentially be rewarded. Power pools would implement the economic dispatch and produce a single (spot) price for electricity, giving participants a clear signal for consumption and investment decisions. Winning bidders are paid the spot price that is equal to the highest bid of the winners. Since the spot price may exceed the actual running of selected bidders, bidders are encouraged to expand their market share which will force high cost generators to exit the market. Market dynamics will drive the spot price to a competitive level that is equal to the marginal cost of most efficient firms.

To give the reader an idea on how price signals could play an important role in a restructured environment, we consider the following example.

Example 1.1: Impact of Price Signals on Demand Consumption

Figures 1.1 and 1.2 show the idea of price signals and their impact on consumption behavior. Let's assume that an Industrial Customer (IndustCo) consumption pattern in a vertically integrated electric industry, in a certain day, would look like that shown in Figure 1.1. In this figure, the price of electricity in Rs/KWh is fixed, and the IndustCo has no incentives to change its consumption pattern. While in a competition-based market, when the IndustCo receives a real time price signal, it could change its consumption pattern in response to the price (See Figure 1.2). In Figure 1.2, the IndustCo reduces its usage at times high prices. As the price jumps from 6Rs/KWh at 5 a.m. to 8Rs/KWh at 9 a.m. the IndustCo starts decreasing its consumption. As the price continues to increase from 8Rs/KWh at 9 a.m. to 16Rs/KWh at 6 p.m. the IndustCo continues decreasing its consumption. When the price decreases after 6 p.m. the IndustCo increases its usage of electricity.

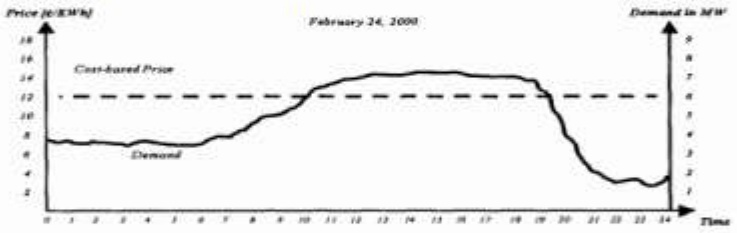


Figure 1.1 Behavior of a Demand in a Vertically Integrated Power Market

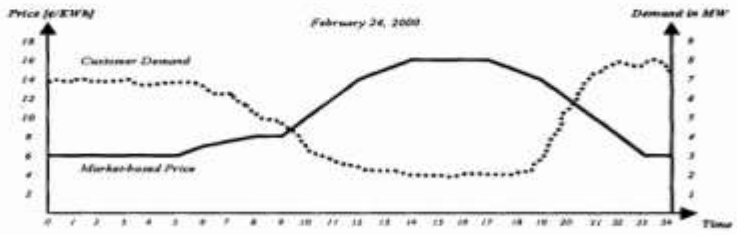


Figure 1.2 Response of a Demand to Price Signals

Although buyers and sellers in a PoolCo are prevented from making individual contracts for power, participants may hold optional financial instruments called Contracts for Differences (CFDs). These contracts are long-term price hedging bilateral contracts between generators and distribution utilities or retail customers. These contracts allow a physical dispatch of individual generating units by their owners and allow consumers to establish long-term prices. When used, a power seller is paid a fixed amount over time that is a combination of short-term market price and an adjustment for the difference. CFDs are established as a mechanism to stabilize power costs to customers and revenues for generators.

These contracts are suggested due to the fact that the spot price set by PoolCo fluctuates over a wide range and is difficult to forecast over long periods. Using CFDs, any differences between the spot price and the contract price would be offset by cash payments by generators to customers; in other words, by holding these contracts, customers gain protection against unexpected spot price increases and generators could obtain a greater revenue stability.

1.2.2 Bilateral Contracts (Direct Access) Model.

The Bilateral Contracts model has two main characteristics that would distinguish it from the PoolCo model. These two characteristics are: the ISO's role is more limited; and buyers and sellers could negotiate directly it/the marketplace. In this model, small customers' aggregation is essential to ensure that they would benefit from competition.

This model permits direct contracts between customers and generators without entering into pooling arrangements. By establishing non-discriminatory access and pricing rules for transmission and distribution systems, direct sales of power over a utility's transmission and distribution systems are guaranteed. Wholesale suppliers would pay transmission charges to a transmission company to acquire access to the transmission grid and pays similar charges to a distribution company to acquire access to the local distribution grid. In this model, a distribution company may function as an aggregator for a large number of retail customers in supplying a long-term capacity. Also, the generation portion of a former integrated utility may function as a supplier or other independent generating companies, and transmission system would serve as a common carrier to contracted parties that would permit mutual benefits and customer's choice. Any two contracted parties would agree on contract terms such as price, quantity and locations, and generation providers would inform the ISO on how its hourly generators would be dispatched.

1.2.3 Hybrid Model:

The hybrid model combines various features of the previous two models. The hybrid model differs from the PoolCo model as utilizing the PX is not obligatory and customers are allowed to sign bilateral contracts and choose suppliers from the pool. The pool would serve all participants (buyers and sellers) who choose not to sign bilateral contracts. The Califomia model is an example of the hybrid model. This structure has advantages over a mandatory pool as it provides end-users with the maximum flexibility to purchase from either the pool or directly from suppliers. A customer who would choose a PX option with CFDs could acquire the economic equivalent of bilateral contracts.

The existence of the pool can efficiently identify individual customer's energy requirements and simplify the balancing process of energy supply. The hybrid model would enable market participants to choose between the two options based on provided prices and services. The hybrid model is very costly to set up because of separate entities required for operating the PX and the transmission system. In the following, we learn more about the functions of an ISO in a restructured power system.

1.3 INDEPENDENT SYSTEM OPERATOR (ISO)

1.3.1 Background:

In a vertically integrated monopoly, utilities created regional centrally dispatched power pools to coordinate the operation and planning of generation and transmission among their members in order to improve operating efficiencies by selecting the least-cost mix of generating and transmission capacity, coordinating maintenance of units, and sharing operating reserve requirements and thus lowering the cost to end-use electricity customers. The centrally dispatched power pools are classified as:

- **Tight power pools:** they have customarily functioned as control areas for their members; perform functions such as unit commitment, dispatch and transaction scheduling services. Examples of this kind of pools are New York Power Pool (NYPP), New England Power Pool (NEPOOL), Pennsylvania, New Jersey, Maryland (PJM) Interconnection, Colorado Power Pool and Texas Municipal Power Pool.
- **Loose power pools:** they have generally had more limited roles, and in contrast to tight pools, have a low level of coordination in operation and planning. The most significant role of these

pools has provided support in emergency conditions. Loose power pools did not provide control area services.

- **Affiliate power pools:** in this kind of pools, generated power which was owned by the various companies was dispatched as a single utility. Pools have had extensive agreements on governing the cost of generation services and use of transmission systems.

Power pools controlling access to regional transmission systems made it difficult for non-members to use pool members' transmission facilities by establishing complex operating rules and financial arrangements. Also, restrictive membership and governance of pools were practiced occasionally in a way that large utilities prevented changes in policies and rules of the pool which led to closing pool membership to outsiders. Unfair industry practices generally impacted the growth of a competitive generation market and were motive forces for the FERC to order transmission owners to provide other parties an open access to transmission grids.

A competitive generation market and retail direct access necessitated an independent operational control of the grid. However, the independent operation of the grid was not guaranteed without an independent entity, the so-called Independent System Operator (ISO). An ISO is independent of individual market participants such as transmission owners, generators, distribution companies and end-users. The basic purpose of an ISO is to ensure a fair and non-discriminatory access to transmission services and ancillary services, maintain the real-time operation of the system and facilitate its reliability requirements.

1.3.2 The Role of ISO:

The primary objective of the ISO is not dispatching or re-dispatching generation, but matching electricity supply with demand as necessary to ensure reliability. ISO should control generation to the extent necessary to maintain reliability and optimize transmission efficiency. The ISO would continually evaluate the condition of transmission system and either approve or deny requests for transmission service.

In its Order No. 888, FERC developed eleven principles as guidelines to the electric industry restructuring to form a properly constituted ISO, through which public utilities could comply with FERC's non-discriminatory transmission tariff requirements. The eleven principles for ISOs are:

1) The ISO's governance should be structured in a fair and non- discriminatory manner.
2) An ISO and its employees should have no financial interest in the economic performance of any power market participant. An ISO should adopt and enforce strict conflict of interest standards.
3) An ISO should provide open access to the transmission system and all services under its control at non-pancaked rates pursuant to a single, unbundled, grid-wide tariff that applies to all eligible users in a non-discriminatory manner.
4) An ISO should have the primary responsibility in ensuring short-term reliability of grid operations. Its role in this responsibility should be well defined and comply with applicable standards set by NERC and the regional reliability council.
5) An ISO should have control over the operation of interconnected transmission facilities within its region.
6) An ISO should identify constraints on the system and be able take operational actions to relieve those constraints within the trading rules established by the governing body. These rules should promote efficient trading.
7) An ISO should have appropriate incentives for efficient management and administration and should procure services needed for such management and administration in an open market.
8) An ISO's transmission and ancillary services pricing policies should promote the efficient use of and investment in generation, transmission, and consumption. An ISO or a Regional Transmission Group (RTG) of which the ISO is a member should conduct such studies as may be necessary to identify operational problems or appropriate expansions.

9) An ISO should make transmission system information publicly available on a timely basis via an electronic information network consistent with the Commission's requirements.
10) An ISO should develop mechanisms to coordinate its activities with neighboring control areas.
11) An ISO should establish an Alternative Dispute Resolution (ADR) process to resolve disputes in the first instance.

According to the FERC Order 888, the ISO is authorized to maintain transmission system reliability in real-time. To comply with the FERC Order 888, each ISO may take one of the following structures:

The first structure is mainly concerned with maintaining the transmission reliability by operating the power market to the extent that the ISO would schedule transfers in a constrained transmission system. An example of this proposal is the Midwest ISO.

The second proposal for an ISO includes a PX that is integral to the ISO's operation. In some proposals such as those of the UK and the PJM interconnection, the PX would function within the same organization and under the control of the ISO; the ISO is responsible for dispatching all generators and would set the price of energy at each hour based on the highest price bid in the market.

In its Order 888, FERC also defined six ancillary services that must be provided by or made available through transmission providers. These ancillary services include:

1) Scheduling, Control and Dispatch Services
2) Reactive Supply and Voltage Control
3) Regulation and Frequency Response Services
4) Energy Imbalance Service
5) Operating Reserve, Spinning and Supplemental Reserve Services
6) Transmission Constraint Mitigation

As Order 888 implies, transmission customers may self-provide these services or buy them through one of the following methods.

(i) Providers of these services advertise their availability via the OASIS or commercial exchanges
(ii) The ISO provides these services in real-time and charges transmission users.

To make these services available, the ISO contracts with service providers so that the services are available under the ISO's request. When a service provider is called by the ISO, the provider is paid extra to cover its operating costs. Capacity resources are contracted seasonally by the ISO and providers send their bids to an auction operated by the ISO. The ISO chooses successful providers based on a least-cost bid basis. When determining the winners, the ISO takes into account factors such as time and locational constraints and the expected use of resources. If the ISO finds that spot market services are less expensive than contracted ones, the ISO experiences its authority by acquiring these services from the energy spot market.

The following example illustrates the role of ISO in providing operating reserves.

Example 1.2: (Operating Reserves)

Assume that a hydroelectric generator, which has a 30 MW capacity and can be brought up and running in 4 minutes, offers its 30 MW capacity in the operating reserve market. It submits a reserve price offer of \$2.5/MW for each hour of the entire next day, along with an energy offer price of \$42.5/MWh. Also, assume that there are two customers, C1 and C2 that can cut back quickly their usage of electricity. C can cut up to 15 MW in 5 minutes, so it decides to submit a bid to the operating reserve market of 15 MW at \$1.5/MW for each hour of the entire next day and a maximum energy bid price of \$55/MWh. C2 can cut up to 10 MW in 5 minutes, so it decides to submit a bid to the operating reserve market of 10 MW at \$2/MW for each hour of the entire next day and a maximum energy bid price of \$70/MWh.

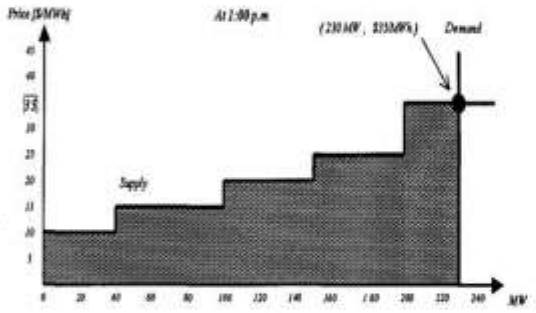
The ISO anticipates that it will need a large operating reserve, say 55 MW, so it accepts the generator's offer and the customers' bids. Assume that the MCP of the operating reserves is \$2.5/MW/h. Each winning participant in the operating reserve market will be paid the MCP to stand by in case of

contingency. The generator will be paid $1800 (or $2.5/MW/h×30MW×24h), C will be paid $900 (or $2.5/MW/h×15MW×24h), and C2 will be paid $600 (or $2.5/MW/h×10 MW×24 h).

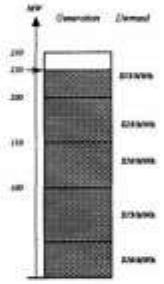
In the next day, say at 1:00 p.m., let's assume that supply and demand equilibrium is initially established at 230 MW (in the spot market). The spot market price of electricity at this point is $35/MWh.

The net supply curve of all generating units is as shown in Figure 1.3. The supply segment of $20/MWh is offered by a single generating unit that has a capacity of 50 MW. As segment) that is dispatched last (in of its total capacity of 50 MW, and shown in Figure 1.3, the unit (supply the spot market) is producing 30 MW has available capacity of 20 MW.

Now suppose the generating unit that is producing 50 MW at $20/MWh tripped out due to severe weather conditions at 1:10 p.m. which disconnected the line joining this generating unit to the system. The supply curve after the unit outage is shown in Figure 1.4. At 1:10 p.m., demand is still 230 MW, and the supply has a shortage of 50 MW. At this point, the ISO has to replace the sudden loss of capacity in the remainder of the hour (from 1:10-2:00 p.m.), before more capacity can be dispatched in the spot market. So the ISO has to reduce the demand if possible or/and provide the required power from operating reserves.

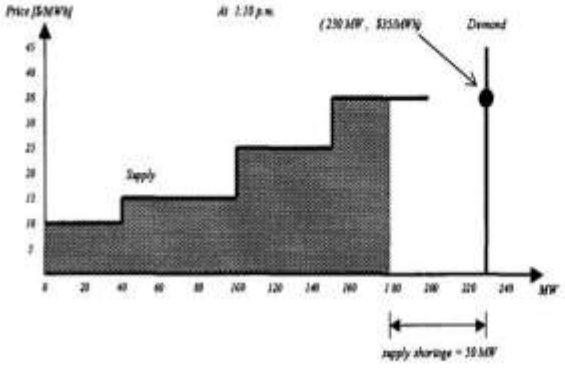


1.3.a Supply and Demand Curves of Initial Situation

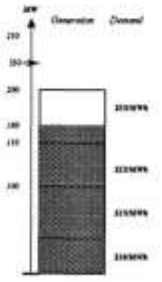


1.3.b Total Production and Production of each Generating Unit of the Initial Situation

Figure 1.3 Initial Situation



1.4.a Supply and Demand Curves after the Unit Outage



1.4.b Total Production and Production of each Generating Unit after the Outage

Figure 1.4 Situation after the Unit Outage

The ISO dispatches the 30 MW generator (see Figure 1.5), and the generator is paid $42.5/MWh. To restore the balance, the ISO also dispatches off 15 MW from C1 and 5 MW from C2 (See Figure 1.6). The ISO pays $20/MWh (or $55/MWh-$35/MWh) to C~ and $35/MWh (or $70/MWh-$35/MWh) to C2

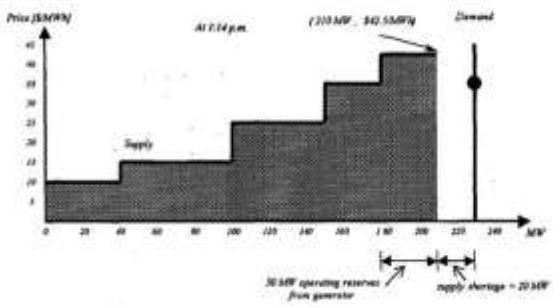
for the electricity that it has dispatched off. The payments to the three participants (generator, C1 and C2) continue until replacement energy can be provided from the spot market.

1.4 POWER EXCHANGE (PX)

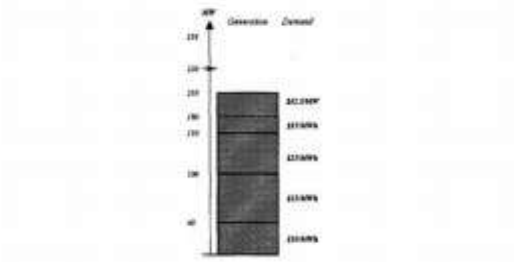
Even though short-term and long-term financial energy transactions could be in bilateral forms in the electricity industry where contracted parties agree individually for certain terms such as price, availability and quality of products, industry restructuring proposals have concluded the necessity of creating a new marketplace to trade energy and other services in a competitive manner. This marketplace is termed Power Exchange (PX) or, as sometimes called, spot price pool. This marketplace permits different participants to sell and buy energy and other services in a competitive way based on quantity bids and prices. Participants include utilities, power marketers, brokers, load aggregators, retailers, large industrial customers and co-generators.

PX is a new independent, non-government and non-profit entity which accept schedules for loads and generation resources. It provides a competitive marketplace by running an electronic auction where market participants buy and sell electricity and can do business quickly and easily. Through an electronic auction, PX establishes an MCP for each hour of the following day for trades between buyers (demands) and sellers (supplies). In this marketplace, PX does not deal with small consumers. Add to that, PX manages settlement and credit arrangements for scheduling and balancing of loads and generation resources. As a main objective in its work, PX guarantees an equal and non-discriminatory access and competitive opportunity to all participants. It is claimed that participants entering the PX get more cost-effectiveness by removing the complexities of arranging generation, transmission and energy purchases.

In general, the PX includes a day-ahead market and an hour-ahead market. Here we discuss these markets in general, and later we elaborate on them when we discuss some market models in the United States.

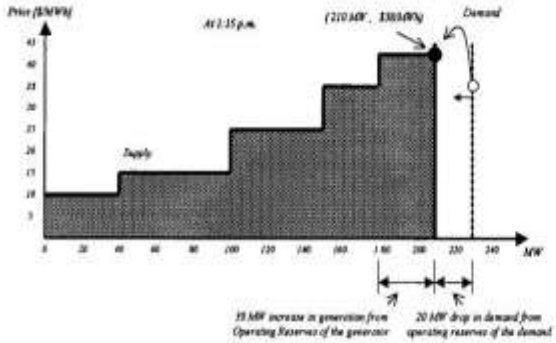


1.5.a Supply and Demand Curves after the ISO dispatches the 30 MW generator

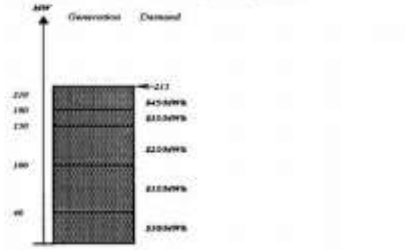


1.5.b Total Production and Production of each Generating Unit after the ISO dispatches the 30 MW generator

***Figure 1.5** Situation after Using Available Operating Reserves*



1.6.a Supply and Demand Curves after the ISO dispatches off the 20 MW



1.6.b Total Production and Production of each Generating Unit after the ISO dispatches off the 20 MW

***Figure 1.6** Situation after Drop of C_1 and C_2 Demand*

PX accepts supply and demand bids to determine a MCP for each of the 24 periods in the trading day. Computers aggregate all valid (approved) supply bids and demand bids into an energy supply curve and an energy demand curve. MCP is determined at the intersection of the two curves and all trades are executed at the MCP, in other words, the MCP is the balance price at the market equilibrium for the aggregated supply and demand graphs. Figure 1.7 shows the determination of MCPs for certain hours when demand (Di) varies. Generators are encouraged to bid according to their operating costs because bidding lower would lead to financial losses if MCP is lower than the operating cost and bidding higher could cause units to run less frequently or not run at all.

1.5 MARKET OPERATIONS:

1.5.1 Day-Ahead and Hour-Ahead Markets:

In the day-ahead market and for each hour of the 24-hour scheduling day, sellers bid a schedule of supply at various prices, buyers bid a schedule of demand at various prices, and MCP is determined for each hour. Then, sellers specify the resources for the sold power, and buyers specify the delivery points for the purchased power. PX schedules supply and demand with the ISO for delivery. Supply and demand are adjusted to account for congestion and ancillary services and then PX finalizes the schedules.

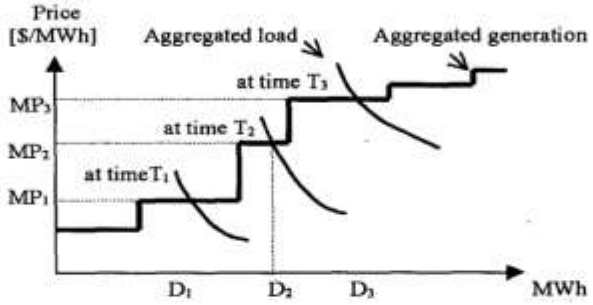


Figure 1.7 Process of Determining MCP in PoolCo

The hour-ahead market is similar to day-ahead, except trades are for 1 hour, and the available transfer capability (ATC) is reduced to include day-ahead trades, and bids are not iterative in this market. Once the MCP is determined in the PX, market participants submit additional data to the PX. The data would include individual schedules by generating unit; take out point for demand, adjustment bids for congestion management and ancillary service bids. After this stage, the ISO and the PX know the injection points of individual generating units to the transmission system. A schedule may include imports and/or exports. To account for transmission losses, generator's schedules are adjusted where real losses are only known after all metered data are processed.

1.5.2 Elastic and Inelastic Markets.

An inelastic market does not provide sufficient signals or incentives to a consumer to adjust its demand in response to the price, i.e., the consumer does not have any motivation to adjust its demand for electrical energy to adapt to market conditions. In a market that has a demand; MCP for energy is determined by the price structure of supply offers. The concept of inelastic demand is directly related to the concept of firm load, which formed the basis of the electricity industry for many decades before the introduction of open access and energy markets. Customers use the concept of elastic demand when they are exposed to and aware of the price of energy and arrange their affairs in such a fashion as to reduce their demand as the price of the next available offer exceeds a certain level.

The following example illustrates how the elasticity of demand in a market has some serious impacts on the energy market and the power system itself, and demonstrates how the pool price cap (price ceiling) energy markets with inelastic and elastic demand may play important role. The example also illustrates the need for capacity reserves.

1.6 MARKET POWER:

One of the main anti-competitive practices or difficulties that may prevent competition in the electric power industry, especially in generation, is market power. When an owner of a generation facility in a restructured industry is able to exert significant influences (monopoly) on pricing or availability of electricity, we say that market power exists, and if so it prevents both competition and customer choice. Market power may be defined as owning the ability by a seller, or a group of sellers, to drive price over a competitive level, control the total output or exclude competitors from a relevant market for a significant period of time. Other than price, any entity that exercises market power would reduce competition in power production, service quality and technological innovation. The net result of practicing market power is a transfer of wealth from buyers to sellers or a misallocation of resources.

There are two types of market power:

1-Vertical Market Power: It arises from a single-firm's or affiliate's ownership of two or more steps in a production and market delivery process where one of the steps provides the firm with a control of a bottleneck in the process. The bottleneck facility is a point on the system, such as a transmission line, through which electricity must pass to get to its intended buyers. A control of a bottleneck process enables the firm to give preference to itself or its affiliate over competitive firms.

2- Horizontal Market Power: It is the ability of a dominant firm or group of firms to control production to restrict output and thereby raise prices. It arises from a firm's local control or ownership concentration of a single process step in productive assets within a defined market area. If such concentration is sufficient with respect to certain other market conditions, the firm can influence the supply-demand equilibrium, and hence prices, simply by withholding production. This type of market power cannot be resolved by the ISO.

Concentration in a market measures the market dominance (degree of monopoly experienced by a firm in a competitive market) using market share data, i.e., how many firms exist in the market, and what are their relative sizes determine the market dominance.

Example 1.6: (How is Market Power Exercised?)

(A) Exercising Market Power when a Power Supply has a Large Market Share,

Figure 1.14 shows a 3-bus system with three generating companies, one at each bus. The three generating companies are $GenCo_1$ with a maximum capacity of 120 MW, $GenCo_2$ with a maximum capacity of 250 MW, and $GenCo_3$ with a maximum capacity of 630 MW.

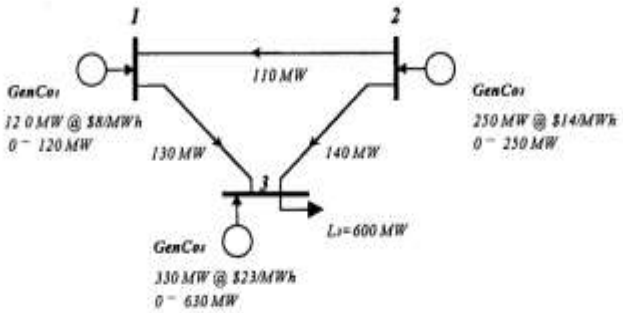


Figure 1.14 Exercising Market Power when a Power Supply has Large Market Share

The HHI for this situation is

$$HHI = \sum_{1}^{3} S_i^2 = (\frac{120}{1000})^2 + (\frac{250}{1000})^2 + (\frac{630}{1000})^2$$

$$= 0.0144 + 0.0625 + 0.3969 = 0.4738 \text{ (in per unit basis)}$$

or,

$$HHI = \sum_{1}^{3} S_i^2 = (\frac{120}{1000} \times 100)^2 + (\frac{250}{1000} \times 100)^2 + (\frac{630}{1000} \times 100)^2$$

$$= 144 + 625 + 3969 = 4738 \text{ (in percent basis)}$$

Note that $GenCo_3$ owns the maximum share of the total generation capacity (=630MW/1000MW=63%). By ignoring the limitation transmission lines, $GenCo_3$ monopolizes the market, because L_3 needs much more than the total capacity of the other cheaper resources ($GenCo_1$ can generate up to 120 MW at \$8/MWh and $GenCo_2$ can generate up to 250 MW at \$14/MWh). It means when $GenCo_3$ wants to exercise its market power, it can ask for any price for its electric power production to fulfill L_3's need.

(B) Exercising Market Power when Transmission System is Congested

Figure 1.15 shows a 3-bus system with three generating companies, one at each bus. The three generating companies are $GenCo_1$ with a maximum capacity of 250 MW, $GenCo_2$ with a maximum capacity of 350 MW, and $GenCo_3$ with a maximum capacity of 400 MW. The HHI for this situation is

$$HHI = \sum_{1}^{3} S_i^2$$

$$= (\frac{250}{1000})^2 + (\frac{350}{1000})^2 + (\frac{400}{1000})^2$$

$$= 0.0625 + 0.1225 + 0.1600 = 0.345 \text{ (in per unit basis)}$$

or,

$$HHI = \sum_{1}^{3} S_i^2 = (\frac{250}{1000} \times 100)^2 + (\frac{350}{1000} \times 100)^2 + (\frac{400}{1000} \times 100)^2$$

$$= 625 + 1225 + 1600 = 3450 \text{ (in percent basis)}$$

In this case $GenCo_1$, $GenCo_2$ and $GenCo_3$ own, respectively, 25%, 35%, and 40% of the total generation capacity. Transmission line limits are imposed on the system in this case, as shown in Figure 1.15. Even though the cheapest resources ($GenCo_1$and $GenCo_2$) have a total capacity of 600 MW, which is adequate to cover the 600 MW required by L_3, the limitations of the transmission lines do not permit L_3 to use $GenC0_1$ and $GenC0_2$. This situation may lead to exercising market power by $GenCo_3$ by imposing a higher than competitive price.

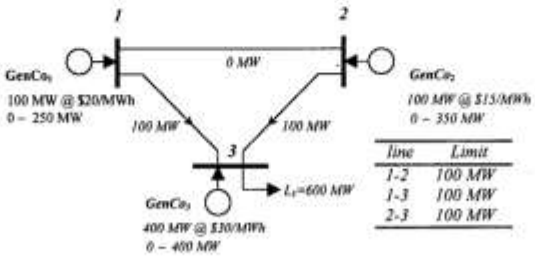


Figure 1.15 Exercising Market Power when Transmission is Congested

1.7 STRANDED COSTS:

A major and a debatable issue associated with the electric utility restructuring is the issue of stranded costs; how to be determined, how to be recovered and who pays for recovery. Stranded cost is a terminology created under restructuring process. Multiplicity of definitions and interpretations of stranded costs confused people working on restructuring, but in general this term refers to the difference

between costs that are expected to be recovered under the rate regulation and those recoverable in a competitive market.

In a vertically integrated monopoly, utilities are used to cover their costs of doing business in rates charged to customers. Costs include operating costs and invested capital costs, where utilities cover these costs and considerable returns on their capitals through rates imposed on customers. But when restructuring is proposed to open market-based competition, and due to the fact that market-based prices are uncertain and sometimes less than vertically integrated rates, financial obligations of vertically integrated utilities may become unrecoverable in a competitive market and the level of revenue earned by a utility may no longer be adequate to cover its costs. If market prices are lower than vertically integrated rates, as many expect, utilities could be faced with investments that are unrecoverable in the competitive market.

Stranded costs still need a more clear definition (what costs should be strandable? what costs are unrecoverable? and to what extent (totally or partially) should be recovered?) and quantification. On the other hand, the duration of recovery or who will pay for recovery is not clear yet and varies from model to model in the United States. In this regard, the big question is whether different participants should pay for uneconomical previous investments.

1.8 TRANSMISSION PRICING

FERC recognized that transmission grid is the key issue to competition, and issued guidelines to price the transmission. The guidelines are summarized such that the transmission pricing would:

(i) Meet traditional revenue requirements of transmission owners
(ii) Reflect comparability: i.e. a transmission owner would charge itself on the same basis that it would charge others for the same service
(iii) Promote economic efficiency
(iv) Promote fairness
(v) Be practical

Even though transmission costs are small as compared to power production expenses and represent a small percent of major investor owned utilities operating expenses, a transmission system is the most important key to competition because it would provide price signals that can create efficiencies in the power generation market. The true price signals are used as criteria for adding transmission capacity, generation capacity, or future loads. Adding transmission capacity to relieve transmission constraints could allow high-cost generation to be replaced by less expensive generation, which would result in additional savings to consumers.

1.8.1 Contract Path Method:

It has been used between transacted parties to price transmission where power flows are assumed to flow over a predefined path(s). Despite its ease, this technique was claimed be a bad implementation of true transmission pricing as power flows would very seldom correspond to predefined paths. Physically, electrons could flow in a network over parallel paths25 (loop flows) owned several utilities that may not be on the contract path. As a result, transmission owners may not be compensated for the actual use of their facilities. Added to parallel flows, the pancakin~6 of transmission rates is another shortcoming of this method.

As a solution to the pancaking effect, zonal pricing schemes have been proposed by most ISOs. Using a zonal scheme, the ISO-controlled transmission system is divided into zones and a transmission user would pay rates based on energy prices in zones where power is injected or withdrawn. When the zonal approach is used, rates are calculated regardless of paths between the two zones or how many other zones are crossed.

1.8.2 The MW-Mile Method:

We illustrate this method by an example.

Several ISOs are using a MW-Mile approach to price transmission. The MW-Mile rate is basically based on the distance between transacted parties (from the generating source to the load) and

flow in each line resulted from the transaction. This approach takes into account parallel power flows and eliminates the previous problem that transmission owners were not compensated for using their facilities. This approach does not give credit for counter-flows on transmission lines. The method is complicated because every change in transmission lines or transmission equipment requires a recalculation of flows and charges in all lines.

Example 1.8: (MW-Mile Charges)

Figure 1.19 illustrates a 3-bus system and a situation of five transactions between generating units and loads. Table 1.8 shows the system data. Let variable $L_{n,j}$ refer to a load that exists at bus n and supplied by generator j. Assume that a load of 150 MW exists at bus 2 which is supplied by generator 1 ($L_{2,1}$=60 MW) and generator 3 ($L_{2,3}$=90 MW). Also another load of 600 MW exists at bus 3, which is supplied by the three generators, 240 from generator 1, 200 MW from Generator 2, and 160 MW from generator 3 (or $L_{3,1}$=240 MW, $L_{3,2}$=200 MW, $L_{3,3}$=160 MW).

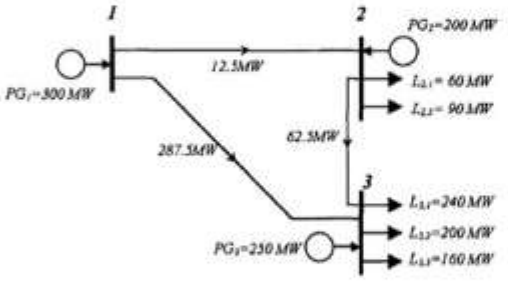


Figure 1.19 3-bus System with Five Transactions

Table 1.8 System Data of the Example

Line	Resistance [Ω]	Reactance [Ω]	Length [Mile]	R [$/Mile]
1–2	0.0	0.30	20.0	5.0
1–3	0.0	0.10	10.0	23.0
2–3	0.0	0.40	40.0	10.0

We assume the system is lossless and we use dc-load flow equations. Furthermore, bus 3 is the reference bus ($\delta_3 = 0.0$) and P_i is the net injection (net generation - net load) at bus i. For this system, we find voltage angles as follows:

$$\begin{bmatrix} 1/0.3+1/0.1 & -1/0.3 \\ -1/0.3 & 1/0.3+1/0.4 \end{bmatrix}\begin{bmatrix}\delta_1 \\ \delta_2\end{bmatrix}=\begin{bmatrix}P_1 \\ P_2\end{bmatrix}$$

$$\begin{bmatrix}\delta_1 \\ \delta_2\end{bmatrix}=\begin{bmatrix} 13.3333 & -3.3333 \\ -3.3333 & 5.8333\end{bmatrix}^{-1}\begin{bmatrix}P_1 \\ P_2\end{bmatrix}=\frac{1}{66.6666}\begin{bmatrix}5.8333 & 3.3333 \\ 3.3333 & 13.3333\end{bmatrix}\begin{bmatrix}P_1 \\ P_2\end{bmatrix} \quad \text{(a)}$$

$$=\begin{bmatrix}0.0875 & 0.0500 \\ 0.0500 & 0.2000\end{bmatrix}\begin{bmatrix}P_1 \\ P_2\end{bmatrix}$$

or

Let x_{mn} refer to the reactance of a line connecting buses m and n, and f_{mn} refer to the flow from m to n. Then,

$$f_{mn} = \frac{\delta_m - \delta_n}{x_{mn}} \qquad \text{(b)}$$

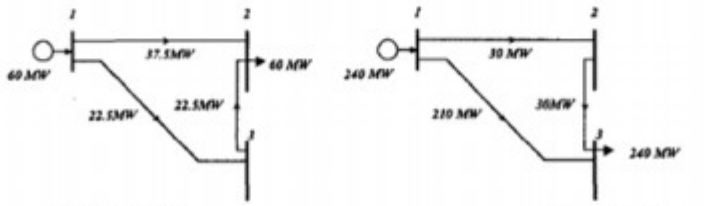
After finding voltage angles, we calculate line flows using Equation b. The net line flows are shown in Figure 1.19.

If $f_{m-n,j}$ is the loading of line m-n due to transaction j, D_{m-n} is the length of line m-n in miles, and R_{m-n} is the required revenue per unit length of line m-n ($/mile), the MW-Mile method uses the following equation to find charges for line m-n corresponding to transaction j (i.e. $C_{m-n,j}$):

$$C_{m-n,j} = \frac{f_{m-n,j}\ D_{m-n}\ R_{m-n}}{f_{m-n}} = \frac{f_{m-n,j}\ Z_{m-n}}{f_{m-n}}$$

where z_{m-n} is the required revenue of line m-n in $, i.e., $z_{m-n} = D_{m-n}\ R_{m-n}$.

Line flows due to each transaction are shown in Figures 1.20–1.22 and detailed in Table 1.9. Note that the total flow in each line given in Table 1.9 is equal to the total flow in each line given in Figure 1.19. Table 1.9 shows MW-Mile charges for each transaction's contributions to line flows as well as the total charges paid by each transaction for its contributions. The table also shows charges paid by all transactions in each line.



a. PG_1 and $L_{2,1}$ b. PG_1 and $L_{3,1}$

Figure 1.20 *Transactions of PG_1*

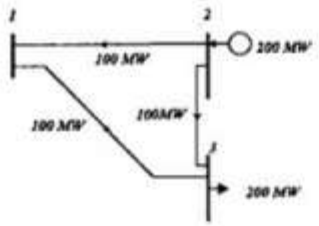
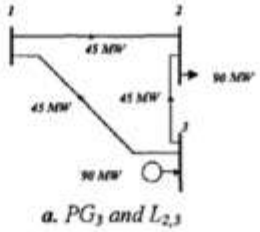
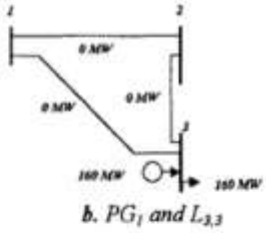


Figure 1.21 *Transaction of PG_2 and $L_{3,2}$*



a. PG_3 and $L_{2,3}$



b. PG_1 and $L_{3,3}$

Figure 1.22 *Transactions of PG_3*

Table 1.9 Impact of each Transaction on Line Flows

	Transaction j	$f_{1\text{-}2,j}$ (1→2)	Cost [$]	$f_{1\text{-}3,j}$ (1→3)	Cost [$]	$f_{2\text{-}3,j}$ (2→3)	Cost [$]	Total [$]
1.	60 MW : $PG_1 \rightarrow L_{2,1}$	37.5	300.	22.5	18.	-22.5	144.	199.5
2.	240 MW : $PG_1 \rightarrow L_{3,1}$	30.0	240.	210.	168	30.	192.	600.0
3.	200 MW : $PG_2 \rightarrow L_{3,2}$	-100.	800.	100.	80	100.	640.	1,520.
4.	90 MW : $PG_3 \rightarrow L_{2,3}$	45.0	360.	-45.0	36	-45.	288.	684.0
5.	160 MW : $PG_3 \rightarrow L_{3,3}$	0.0	0.0	0.0	0.0	0.0	0.0	0.0
	Total	12.5 MW	$1,700	287.5 MW	$302	62.5 MW	1,264.	$3,003.

1.9 CONGESTION PRICING:

The condition where overloads in transmission lines or transformers occur is called congestion. Congestion could prevent system operators from dispatching additional power from a specific generator. Congestion could be caused for various reasons, such as transmission line outages, generator outages, changes in energy demand and uncoordinated transactions. Congestion may result in preventing new contracts, infeasibility in existing and new contracts, additional outages, monopoly of prices in some regions of power systems and damages to system components. Congestion may be prevented to some extent (preventive actions) by means of reservations, rights and congestion pricing. Also, congestion can be corrected by applying controls (corrective actions) such as phase shifters, tap transformers, reactive power control, re-dispatch of generation and curtailment of loads.

FERC has set guidelines for a workable market approach to mange congestion, which are:

(1) The approach should establish clear and trade able rights for transmission usage,

(2) The approach should promote efficient regional dispatch,

(3) The approach should support the emergence of secondary markets for transmission rights,

(4) The approach should provide market participants with the opportunity to hedge locational differences in energy prices,

(5) Congestion pricing method should seek to ensure that the generators that are dispatched in the presence of transmission constraints are those that can serve system loads at least cost, and

(6) The method should ensure that the transmission capacity is used by market participants who value that use most highly.

As such, FERC declares that some approaches appear to have more advantages than others. Even though LMP scan be costly and difficult to implement, especially by entities that have not previously operated as tight power pools, FERC suggests that markets that are based on LMP and financial rights for

firm transmission service will provide an efficient congestion management framework, and this is due to the following facts:

I- LMP assigns congestion charges directly to transmission customers in a fashion that agrees with each customer's actual use of the system and the actual dispatch that its transactions cause.

II- LMP facilitates the creation of financial transmission rights, which enable customers to pay known transmission rates and to hedge against congestion charges.

III- Financial rights entitles their holders to receive a share of congestion revenues, and consequently the availability of such rights congestion pricing resolve the problem of the over recovery of transmission costs.

To solve the congestion problem, several alternatives could be considered such as re-dispatching existing generators or dispatching generators outside the congested area to supply power. The latter alternative is referred to as out-of-merit dispatch. In both alternatives, congestion has costs based on differences in energy prices between locations. In a vertically integrated monopoly, congestion costs were either ignored or hidden as bundled into the transmission charges that in turn were considered as a shortcoming in previous transmission pricing schemes. It was considered a shortcoming because it did not provide a true price signal for efficient allocation of transmission resources or allocated congestion costs to transmission customers who were not causing the congestion.

1.9.1 Congestion Pricing Methods:

All new restructuring proposals are taking congestion costs into account by developing appropriate approaches to measure congestion costs and allocate these costs to transmission system users in a fair way that reflects actual use of the transmission system. These approach evolved in three basic methods based on:

1- **Costs of out-of-merit dispatch:** This is appropriate to systems with less significant transmission congestion problems. In this approach, congestion costs are allocated to each load on the transmission system based on its load ratio share (i.e., individual load expressed as a percent of total load).

2- **Locational Marginal Prices (LMPs):** This technique is based the cost of supplying energy to the next increment of load at a specific location on the transmission grid. It determines the price that buyers would pay for energy in a competitive market at specific locations and measures congestion costs by considering the difference in LMPsb etween two locations. In this approach LMPsa re calculated at all nodes of the transmission system based on bids provided to the PX.

3- **Usage charges of inter-zonal lines:** In this approach, the ISO region is divided into congestion zones based on the historical behavior of constrained transmission paths. Violations of transmission lines between zones (inter-zonal lines) are severe while in the congestion zone transmission constraints are small.

All transmission users who use the inter-zonal pay usage charges. These charges will be determined from bids submitted voluntarily by market participants to decrease or increase (adjust) power generation. Adjustment bids reflect a participant's willingness to increase or decrease power generation at a specified cost. An example of this approach is the case of California.

1.9.2 Transmission Rights:

These rights are used to guarantee an efficient use of transmission system capacity and to allocate transmission capacity to users who value it the most. These rights are tradable rights referred to as the right to use transmission capacity and represent a claim on the physical usage of the transmission system. Moreover, these rights enable utilities to purchase existing transmission rights more cheaply than expanding the system, thereby avoiding unneeded investments. Efficient usage of the transmission can be improved by willingness to offer capacity reservations to those who value them more.

Another form of these rights is the concept of financial rights (some times called Fixed Transmission Right), which is equivalent to the physical rights. This form is proposed because it is easier for trading and less costly because the usage of a transmission system need not be tied to ownership

Demand Side Management

Concept of DSM

Demand side management is used to describe the actions of a Utility, beyond the customer's meter, with the objective of altering the end-use of electricity.

In other words, DSM is the implementation of those measures that help the customers to use electricity more efficiently and so reduce the utility costs of customer.

DSM can be achieved through;

(i) Improving the efficiency of various end users through better housekeeping, correcting energy leakages, system conversion losses etc.

(ii) Developing and promoting energy efficient technologies, and

(iii) Demand management through adopting soft options like higher prices during peak hours, concessional rates during off-peak hours, seasonal tariffs, interruptible tariffs etc.

Need and objectives of DSM

DSM is the modification of consumer demand for energy through various methods such as financial incentives and education.

Goal of DSM is to encourage the consumer to use less energy during peak hours, or to move the time of energy use to off-peak times such as night time and weekends.

Peak demand management does not necessarily decrease total energy consumption, but could be expected to reduce the need for investments in networks and/or power plants.

The term DSM was coined during the time of the 1973 energy crisis and 1979 energy crisis.

~~One of~~ Demand for any ~~a~~ commodity can be modified by actions of market players and goverment. Energy demand management implies actions that influence demand for energy.

DSM is originally adopted in energy; but today DSM is applied widely to utility including water and gas as well.

In centrally planned economies subsidizing energy was one of the main economic development tools. subsidies to the energy supply industry are still common in some countries.

Contrary to the historical situation, energy prices and availability ~~are~~ are expected to deteriorate.

One of the main goals of DSM is to be able to charge the consumer based on the true price of the utilities at that time.

If consumers are charged more during peak demand and less during off-peak hours, then supply and demand would theoretically encourage the customer to use less electricity during peak hours. Thus main goal of DSM is achieved.

Hence, we can say that DSM is "planning, implementation and evaluation of utility activities designed to encourage customers to modify their electricity consumption pattern, both w.r.t. the timing and level of demand (kW) and energy (kWh).

DSM Benefits

Following table outlines the benefits of the DSM initiatives.

Customer Benefits	Societal Benefits	Utility Benefits
1. Satisfy electricity demands	Reduce environmental degradation	Lower Cost of Service
2. Reduce/Stabilize Costs	Conserve Resources	Improve operating efficiency, flexibility
3. Improve value of Service	Protect global environment	Reduce capital needs
4. Maintain/improve life style and productivity	Maximize customer welfare	Improve customer service.

DSM Planning and Implementation

DSM programs are utility and consumer specific. The various steps involved in implementing DSM programme are as follows:

1. Load Research
2. Define load shape objectives
3. Assess program implementation strategy
4. Implementation
5. Monitoring and evaluation

1. Load Research

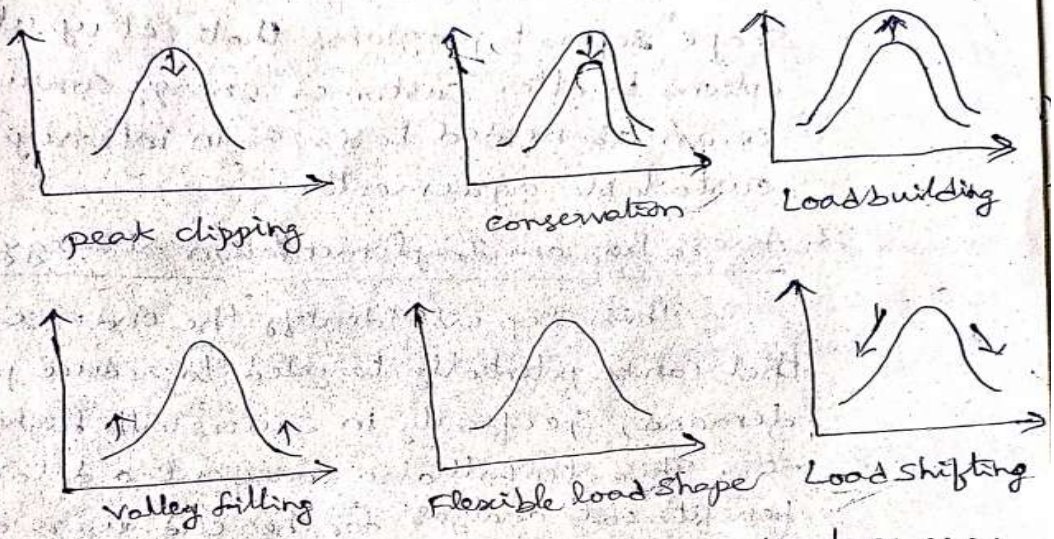
This stage in the DSM implementation will typically assess the customer base, tariff, load profile on an hourly basis and will identify the sectors contributing to the load shape.

This step also identifies the tariff classes in the utility, current recovery from different sectors and current subsidy offered to different sectors.

2. Define Load shape objectives

Based on the results of the load research in the utility, DSM engineers will define the load shape objectives for the current situation.

Various load shape objectives - Peak clipping, Valley filling, Load shifting and load building are possible. These are represented in following figures.



Specific descriptions of load shape objectives are;

(i) Peak clipping :-

The reduction of utility load primarily during periods of peak demand.

(ii) Valley filling :-

The improvement of system load factor by building load in off-peak periods.

(iii) Load shifting :-

The reduction of utility loads during periods of peak demand while at the same time building load in off-peak periods. Typically, it does not substantially alter total electricity sales.

(iv) Conservation :-

The reduction of utility loads, more or less equally, during all or most hours of the day.

(v) Load building :-

The increase of utility loads, more or less equally, during all or most hours of the day.

Provision of a more "flexible utility load shape" refers to programs that set up utility options to alter customer energy consumption on an as-needed basis, as in interruptible / curtailable agreements.

3. Assess Program Implementation Strategy

This step will identify the end-use application that can be potentially targeted to reduce peak demand, specifically in sectors with higher subsidies. This step will also carry out a detailed benefit-cost analysis for the end-users and the utilities, including analysis on societal as well as environmental benefits.

4. Implementation

Implementation stage will design the programme for specific end-use applications, will promote the programme to the target audience through marketing approaches such as advertising, bills, and inserts, and focused group meetings as in case of the industrial sector.

5. Monitoring and Evaluation

This step will track the program design and implementation and will compare the same with proposed DSM goal set by the utility. A detailed benefit-cost analysis in this case will include

identifying the avoided supply cost for the utility and total programme cost for the utilities and benefits to the participants including the reduced bills or incentives to the end-users.

DSM Technology options

Identification of suitable and practical DSM options requires study of users and end-uses of electricity.

An understanding of end-uses of electricity helps to identify end-use options that offer maximum DSM potential.

The need for more specific option, ~~which for end-use and/or which~~ for the purpose of DSM implementation, would require identification of alternatives.

The base technology generally refers to the standard or most commonly used technology within the geographical boundaries of a utility.

In contrast, an alternative technology is the ~~id~~ candidate efficient technology intended to replace base technology in order to achieve DSM objectives.

Alternate DSM technologies

Domestic	Commercial	Industrial
CFL's	CFL's	CFL's
High efficiency fluorescents (HEF)	H.E.F.	H.E.F.
Day lighting	Day lighting	Day Lighting
Low loss ballasts	Low loss ballasts	Low loss ballasts
High efficiency Ac's	H.E. Ac's	H.E. Ac's
~~H~~ Refrigerators	H.E. Refrigerators	H.E. Refrigerators
Gas cooking	A.C. Maintenance	A.C. Maintenance

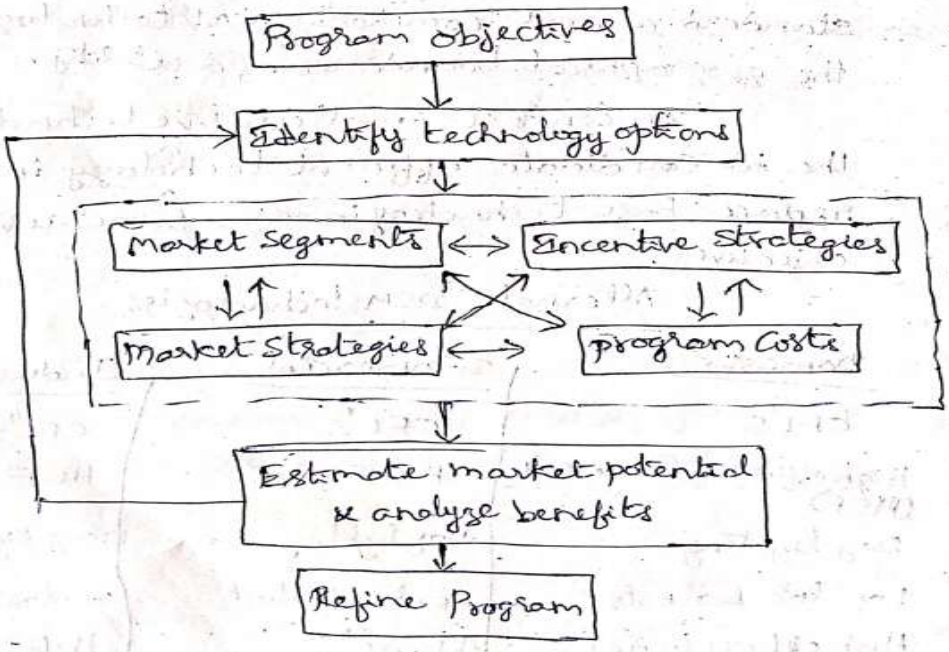
Domestic	Commercial	Industrial
Efficient Rice Cookers	A.C. Timers	A.C. Timers
orientation - New homes	Efficient Security Lighting	Efficient Security lighting
Solar hot water Systems	H.E. Motors	H.E. Motors
	TOD tariffs	TOD tariffs
	Interruptible tariffs	Interruptible tariffs
	EE building code	EE building code
	Solar hot water systems	Cogeneration
		power factor correction
		Variable Speed drives

DSM Programme Design Guidelines

Typically a DSM programme design includes the following:

(i) Determination of the "base" and "DSM" technology

(ii) " " " targeted market sector or Segment



DSM programme design process

(iii) Identifying all potential barriers and possible solutions.
(iv) Conduct a technology cost effective analysis.
(v) Estimation of market penetration over the programme duration.
(vi) Evaluating programme marketing strategies.
(vii) Estimation of staffing resources and programme costs.
(viii) Developing an implementation plan.
(ix) Developing procedures for programme monitoring and evaluation.