

# Trust Management and Security Access Controls in High Payload System Architecture

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# **ABSTRACT**

Enterprise services are commonly deployed on Internet facing applications and Mobile Apps. There is a need to have an Infrastructure and Application Framework to protect the information flow web layer and mobile apps. Trust management is being recognized in the industry along with Federated Single Sign on to cater the need of data protection at rest and in transits. Web layer need to be very light to have better performance. However, there are multiple products and services are available to measure and control security problems. In Web based architecture still companies are maintain credentials and access level in the database. As a matter of fact connectivity to database needs to have a secure channel and database manages the information that is being access via web layer (browsers, Webservice, Restful APIs etc). In multiple scenarios Service registry acts as metadata repository for external partners interface with Webservice using WS -**Security or Security Assertion Markup Language** (SAML). Web Services Trust Language (WS -Trust) plays a vital role in providing secure messaging and adds on to security token exchange within different trust domains for better credentials management. Web and Application Layer need to have WS Trust association to handle request and response of security tokens, managing trusts within the extended **Enterprise** Architecture establishing trust relationships within Intra Organization and external partners, vendors and suppliers.

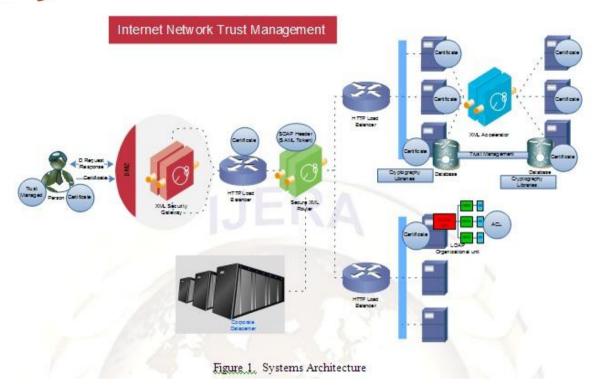
This paper addresses the fit/gap and design approach to integrate Trust management model and Security Access Control with Database. Proposed Trust model illustrate the optimal security model and Asymmetric/ Cryptography. This Symmetric **Trust** management architecture model enhances the existing models in the current Database systems. Strategic model approach guarantee the enterprise wide security controls for PHI / PI datasets in database as well as complete data model and data at rest. This paper also illustrates the modeling technique and implement of open

source Data repositories, metamodel and design and performance constraints in Data Architecture.

**Keywords:** Trust management, relational database, security controls, Data Protection.

# 1.INTRODUCTION

Large organizations in public and private sector needs to offer data intensive services to external party over the internet, and this organization might be at multiple geographical regions. Some organization have MPLS lines to build the eVPN cloud for Intranet and users are allowed to access Information systems while being on Intranet and other users access it via internet. With the growing social applications and dynamics of user base plays a vital role in Trust management and Authentication & Authorization. Many organization still practice the user management at each server and application, technical team spend many hours for user management tasks like user id creation, deletion, locks, access rights and password reset. This can be achieved by self service portal and for better security it is advisable to have two-factor authentication. Infact even better is to have Federated Single Signon and Portal for users to get enrolled and based on that Identity can provisioned to other systems and application access levels based on Role base security matrix. Single Sign On approach do have constraints with proprietary software, COTS products and Legacy application primarily because authentication token sharing is limited within the application instead of enterprise wide authentication and authorization. Token based access control and trust management ... (Bibl). Database acts as the hub for keeping ID based information and data exchange with web interfaces. Database access control based on multi level security (Ying Guang 2011), Fine Granularity (et al Zhang 2012), Security Label( et al Xinqiang Ma 2010). Fine grained access control uses built in database access control and optimize the prarmeterized views interaction with web tier applications and decreases web database attack surface. The approach presented in this paper supports the authentication and authorization and support ID management vision standards.



#### 1.1 Scenario

This is an extended Enterprise Architecture for the deployment of Trust management components and semantic services. As the application is light weight, user's browser makes a request to access application; underlying process request/ response goes through the Perimeter firewall (NATs the local IPs to Internet routable IPs) and DMZ firewall, XML gateway looks for Soap over HTTPs header and sends to the respective web server and application server and ultimately data gets transfer from database over the lightweight context aware XML semantic transaction services. Some of the Client machines can have user certificates which get authenticated at the application server to accept client user certificates, Trust Management and Certificates are deployed on the web servers.

Figure 1 illustrates a enterprise wide systems architecture along with interoperability of services and integration of Trust management and Database. In this architecture, systems interaction follows listed steps:-

1. User's machine invoke the application browser session based on user's certificate that gets authenticated at the application server or gets the Assymmetric keys from the server , browser connects to the XML gateway and maintains source IP persistence , session is being managed by the session management server, certificates get validated at the XML gateway and Load balancer decrypt the HTTPs header and reads the HTTP

header to determine the destination and applies application port translation. Again

2. Browser forwards certificates to the XML Gateway, and XML Gateway forwards to Load balancer and Load Balancer forwar



d it to Web servers which in turn forwards the certificate to Database. Database verifies the certificates, issuing authority, certificate owner, certificate path and domain names, verify the validity of the certificate and looks for server certificate, Intermediate certificate and Root Certificate for certificate chaining. The challenge request and response are based on the cryptographic module within the RDBMS.

3. Challenge request and response is based on Elliptic Curve Cryptography.

Definition 1 : An Elliptic Curve E over a field K is defined by an equation



$$E: y^2 + a_1 xy + a_3 y = x^3 + a_2 x^2 + a_4 x + a_6$$
 (1)

Where  $a_1, a_2, a_3, a_4, a_6 \in K$  and  $\Delta \neq 0$ ,

where  $\Delta$  is the discri min ant of E and is defined as follows:

$$\Delta = -d_2^2 d_8 - 8d_4^3 - 27d_6^2 + 9d_2 d_4 d_6$$
$$d_2 = a_1^2 + 4a_2$$

$$d_4 = 2a_4 + a_1a_3$$

$$d_6 = a_2^2 + 4a_6$$

$$d_8 = a_1^2 a_6 + 4a_2 a_6 - a_1 a_3 a_4 + a_2 a_3^2 - a_4^2$$

If L is any extension field of K, then the set of L-rational points of E is

$$E(L) = \{(x, y) \in L \times L : y^2 + a_1 xy + a_3 y - x^3 - a_2 x^2 - a_4 x - a_6 = 0\} \cup \{\infty\}$$

# Where $\infty$ is the point at infinity.

In a challenge –reponse mechanism , Client Browser (the verifier) has input to a composition of a protocol message and the composition involves a cryptographic operation performed by web server so that client browser verify the lively correspondence of Web server via its input. The usual form of Browser input can be a random number generated by Client machine and passed to web server beforehand. Let  $N_B$  denote a nonce generated by client machine.

Client machine decrypts the cipher chunk and  $\begin{cases} accepts & if webserver sees \ N_B \\ rejects & otherwise \end{cases}$ 

Here, the first message transmission is often called client browser challenge to the webserver, and the second message transmission is thereby called webserver response to the client browse. Client browser is in a position of an initiator while Webserver is in a position to a responder.

# A One Time Password Scheme

Here "one — time" means that the passwords transmitted from a given (Client Browser) U to (Web Server) H do not repeat, however they are computationally related one another. Now, a password eavesdropped from a protocol run is no good for further use, and hence the password eavesdropping problem is successfully prevented.

In the user initialization time, a password entry of (Client Browser) U is set to

$$ID_{U}, f^{n}(P_{U})$$
) where
$$f^{n}(P_{U}) \stackrel{def}{=} \underbrace{f(\dots \dots (f(P_{U}))\dots)}_{n}$$

For a large integer n. The user (Client Browser) U still memorizes  $P_U$  as in the case of the password Authentication protocol.

When (Client Browser) U and (Web Server)H engages in the first run of password authentication, upon prompted by "Password" (message in Password Authentication Protocol), a computing device of (client Browser) U, such as a client platform or a Webservice, will ask U to key in  $P_U$ , and will then compute  $f^{n-1}(P_U)$  by repeatedly applying f (n-1) times. This can be efficiently done even for a large n (e.g., n = 2000), The result will be sent to (Web Server) H for Authentication

# **Premise:**

User (Client Browser) U and Host (Web Server) H have setup U's initial password entry  $(ID_U, f^n(P_U), n)$  where f is a cryptographic hash function; (Client Browser) U memorizes password  $P_U$ ;

The current password entry of (Client Browser) U in (Web server) H is  $(ID_U, f^n(P_U), c)$  for  $1 \le_c \le_n$ 

# Goal:

(Client Browser) U authenticates to (Web Server) H without transmitting  $P_U$  in cleartext

$$U \rightarrow H:ID_U;$$
  
 $H \rightarrow U:c$ , "Input password:";  
 $U \rightarrow H:Q=f^{c-1}(P_U);$ 

H finds entry  $(ID_U, f^c(P_U), c)$  from its archive;

Access is granted if  $f(Q) = f^{c-1}(P_U)$ , and U's password entry is updated to  $(ID_U, Q, c-1)$ 

The integrated Trust management services and the database allow the end to end secured data architecture for the management of access controls.

This design provides the security measure controls that protect application and web server from vulnerability. This system under Trust management is protected from Attacks, as the credentials and keys are secured for the user and web application interface.



We consider the following anycast field equations defined over an open bounded piece of network and /or feature space  $\Omega \subset R^d$ . They describe the dynamics of the mean anycast of each of p node populations.

$$\begin{cases} (\frac{d}{dt} + l_i)V_i(t,r) = \sum_{j=1}^p \int_{\Omega} J_{ij}(r,r)S[(V_j(t-\tau_{ij}(r,r),r) - h_{ij})]dr \\ + I_i^{ext}(r,t), & t \ge 0, 1 \le i \le p, \\ V_i(t,r) = \phi_i(t,r) & t \in [-T,0] \end{cases}$$
 (1)

We give an interpretation of the various parameters and functions that appear in (1),  $\Omega$  is finite piece of nodes and/or feature space and is represented as an open bounded set of  $R^d$ . The vector r and r represent points in  $\Omega$ . The function  $S: R \to (0,1)$  is the normalized sigmoid function:

$$S(z) = \frac{1}{1 + e^{-z}} \tag{2}$$

It describes the relation between the input rate  $v_i$  of population i as a function of the packets potential, for example,  $V_i = v_i = S[\sigma_i(V_i - h_i)].$ We note V the p- dimensional vector  $(V_1,...,V_p)$ . The p function  $\phi_i, i=1,...,p$ , represent the initial conditions, see below. We note  $\phi$  the p – dimensional vector  $(\phi_1,...,\phi_n)$ . The p function  $I_i^{ext}$ , i = 1,..., p, represent external factors from other network areas. We note  $I^{ext}$  the dimensional vector  $(I_1^{ext},...,I_n^{ext})$ . The  $p \times p$  matrix of functions  $J = \{J_{ij}\}_{i,j=1,\dots,p}$ represents the connectivity between populations iand j, see below. The p real values  $h_i$ , i = 1,..., p, determine the threshold of activity for each population, that is, the value of the nodes potential corresponding to 50% of the maximal activity. The p real positive values  $\sigma_i$ , i = 1, ..., p, determine the slopes of the sigmoids at the origin. Finally the p real positive values  $l_i$ , i = 1,..., p, determine the speed at which each anycast node potential decreases exponentially toward its real We also introduce the  $S: \mathbb{R}^p \to \mathbb{R}^p$ . defined by  $S(x) = [S(\sigma_1(x_1 - h_1)), ..., S(\sigma_p - h_p))],$ the diagonal  $p \times p$  matrix  $L_0 = diag(l_1,...,l_p)$ .

Is the intrinsic dynamics of the population given by the linear response of data transfer.  $(\frac{d}{dt} + l_i)$  is replaced by  $(\frac{d}{dt} + l_i)^2$  to use the alpha function

response. We use  $(\frac{d}{dt} + l_i)$  for simplicity although our analysis applies to more general intrinsic dynamics. For the sake, of generality, the propagation delays are not assumed to be identical for all populations, hence they are described by a matrix  $\tau(r,r)$  whose element  $\tau_{ii}(r,r)$  is the propagation delay between population j at r and population i at r. The reason for this assumption is that it is still unclear from anycast if propagation delays are independent of the populations. We assume for technical reasons that  $\tau$  is continuous, that is  $\tau \in C^0(\overline{\Omega}^2, R_{\perp}^{p \times p})$ . Moreover packet data indicate that au is not a symmetric function i.e.,  $\tau_{ii}(r,r) \neq \tau_{ii}(r,r)$ , thus no assumption is made about this symmetry unless otherwise stated. In order to compute the righthand side of (1), we need to know the node potential factor V on interval [-T,0]. The value of T is obtained by considering the maximal delay:

$$\tau_m = \max_{i,j(r,r\in\Omega\times\overline{\Omega})} \tau_{i,j}(r,r)$$
 (3)

Hence we choose  $T = \tau_m$ 

### A. Mathematical Framework

A convenient functional setting for the non-delayed packet field equations is to use the space  $F = L^2(\Omega, \mathbb{R}^p)$  which is a Hilbert space endowed with the usual inner product:

$$\langle V, U \rangle_F = \sum_{i=1}^p \int_{\Omega} V_i(r) U_i(r) dr$$
 (1)

To give a meaning to (1), we defined the history space  $C = C^0([-\tau_m, 0], F)$  with  $\|\phi\| = \sup_{t \in [-\tau_m, 0]} \|\phi(t)\| F$ , which is the Banach phase space associated with equation (3). Using the notation  $V_t(\theta) = V(t+\theta), \theta \in [-\tau_m, 0]$ , we write (1) as

$$\begin{cases} V(t) = -L_0 V(t) + L_1 S(V_t) + I^{ext}(t), \\ V_0 = \phi \in C, \end{cases}$$
 (2)

Where



$$\begin{cases} L_1: C \to F, \\ \phi \to \int_{\Omega} J(., r) \phi(r, -\tau(., r)) dr \end{cases}$$

Is the linear continuous operator satisfying  $\|L_1\| \leq \|J\|_{L^2(\Omega^2,R^{p\times p})}$ . Notice that most of the papers on this subject assume  $\Omega$  infinite, hence requiring  $\tau_m = \infty$ .

Proposition 1.0 If the following assumptions are satisfied.

1. 
$$J \in L^2(\Omega^2, R^{p \times p}),$$

2. The external current 
$$I^{ext} \in C^0(R, F)$$
,

3. 
$$\tau \in C^0(\overline{\Omega^2}, R_+^{p \times p}), \sup_{\overline{\Omega^2}} \tau \leq \tau_m$$
.

Then for any  $\phi \in C$ , there exists a unique solution  $V \in C^1([0,\infty), F) \cap C^0([-\tau_m,\infty,F))$  to (3)

Notice that this result gives existence on  $R_+$ , finite-time explosion is impossible for this delayed differential equation. Nevertheless, a particular solution could grow indefinitely, we now prove that this cannot happen.

#### **B.** Boundedness of Solutions

A valid model of neural networks should only feature bounded packet node potentials.

**Theorem 1.0** All the trajectories are ultimately bounded by the same constant R if  $I \equiv \max_{t \in R^+} \left\| I^{ext}(t) \right\|_{F} < \infty$ .

Proof :Let us defined  $f: R \times C \to R^+$  as  $f(t,V_t) \stackrel{\text{def}}{=} \left\langle -L_0 V_t(0) + L_1 S(V_t) + I^{\text{ext}}(t), V(t) \right\rangle_F = \frac{1}{2} \frac{d \left\| V \right\|_F^2}{dt}$ 

We note  $l = \min_{i=1,\dots p} l_i$ 

$$f(t,V_t) \le -l \|V(t)\|_E^2 + (\sqrt{p|\Omega|} \|J\|_E + I) \|V(t)\|_E$$

Thus, if

$$\left\|V(t)\right\|_{F} \geq 2\frac{\sqrt{p|\Omega|}.\left\|J\right\|_{F}+I}{l} \stackrel{def}{=} R, f(t,V_{t}) \leq -\frac{lR^{2}}{2} \stackrel{def}{=} -\delta < 0$$

Let us show that the open route of F of center 0 and radius  $R, B_R$ , is stable under the dynamics of equation. We know that V(t) is defined for all  $t \ge 0s$  and that f < 0 on  $\partial B_R$ , the

boundary of  $B_R$ . We consider three cases for the initial condition  $V_0$ . If  $\|V_0\|_C < R$  and set  $T = \sup\{t \mid \forall s \in [0,t], V(s) \in \overline{B_R}\}$ . Suppose that  $T \in R$ , then V(T) is defined and belongs to  $\overline{B_R}$ , the closure of  $B_R$ , because  $\overline{B_R}$  is closed, in effect to  $\partial B_R$ , we also have  $\frac{d}{dt}\|V\|_F^2|_{t=T} = f(T,V_T) \le -\delta < 0$  because  $V(T) \in \partial B_R$ . Thus we deduce that for  $\varepsilon > 0$  and small enough,  $V(T+\varepsilon) \in \overline{B_R}$  which contradicts the definition of T. Thus  $T \notin R$  and  $\overline{B_R}$  is stable.

Because f<0 on  $\partial B_R, V(0) \in \partial B_R$  implies that  $\forall t>0, V(t) \in B_R$ . Finally we consider the case  $V(0) \in C\overline{B_R}$ . Suppose that  $\forall t>0, V(t) \notin \overline{B_R}$ , then  $\forall t>0, V(t) \notin \overline{B_R}$ , then  $\forall t>0, \frac{d}{dt} \|V\|_F^2 \leq -2\delta$ , thus  $\|V(t)\|_F$  is monotonically decreasing and reaches the value of R in finite time when V(t) reaches  $\partial B_R$ . This contradicts our assumption. Thus

**Proposition 1.1:** Let s and t be measured simple functions on X. for  $E \in M$ , define

$$\phi(E) = \int_{E} s \, d\mu \tag{1}$$

 $\exists T > 0 \mid V(T) \in B_R$ .

Then  $\phi$  is a measure on M.  $\int_{V} (s+t)d\mu = \int_{V} s \, d\mu + \int_{V} t \, d\mu \qquad (2)$ 

**Proof**: If s and if  $E_1, E_2,...$  are disjoint members of M whose union is E, the countable additivity of  $\mu$  shows that

$$\phi(E) = \sum_{i=1}^{n} \alpha_i \mu(A_i \cap E) = \sum_{i=1}^{n} \alpha_i \sum_{r=1}^{\infty} \mu(A_i \cap E_r)$$
$$= \sum_{r=1}^{\infty} \sum_{i=1}^{n} \alpha_i \mu(A_i \cap E_r) = \sum_{r=1}^{\infty} \phi(E_r)$$

Also,  $\varphi(\phi)=0$ , so that  $\varphi$  is not identically  $\infty$ . Next, let s be as before, let  $\beta_1,...,\beta_m$  be the distinct values of t,and let  $B_j=\{x:t(x)=\beta_j\}$  If



$$E_{ij} = A_i \cap B_j,$$
 the 
$$\int_{E_{ij}} (s+t) d\mu = (\alpha_i + \beta_j) \mu(E_{ij})$$
 and 
$$\int_{E_{ij}} s d\mu + \int_{E_{ij}} t d\mu = \alpha_i \mu(E_{ij}) + \beta_j \mu(E_{ij})$$

Thus (2) holds with  $E_{ij}$  in place of X. Since X is the disjoint union of the sets  $E_{ij}$   $(1 \le i \le n, 1 \le j \le m)$ , the first half of our proposition implies that (2) holds.

**Theorem 1.1:** If K is a compact set in the plane whose complement is connected, if f is a continuous complex function on K which is holomorphic in the interior of, and if  $\varepsilon > 0$ , then there exists a polynomial P such that  $|f(z) = P(z)| < \varepsilon$  for all  $z \varepsilon K$ . If the interior of K is empty, then part of the hypothesis is vacuously satisfied, and the conclusion holds for every  $f \varepsilon C(K)$ . Note that K need to be connected.

*Proof:* By Tietze's theorem, f can be extended to a continuous function in the plane, with compact support. We fix one such extension and denote it again by f. For any  $\delta > 0$ , let  $\omega(\delta)$  be the supremum of the numbers  $\left| f(z_2) - f(z_1) \right|$  Where  $z_1$  and  $z_2$  are subject to the condition  $\left| z_2 - z_1 \right| \leq \delta$ . Since f is uniformly continous, we have  $\lim_{\delta \to 0} \omega(\delta) = 0$  (1) From now on,

 $\delta$  will be fixed. We shall prove that there is a polynomial P such that

$$|f(z)-P(z)| < 10,000 \omega(\delta) \quad (z\varepsilon K)$$
 (2)

By (1), this proves the theorem. Our first objective is the construction of a function  $\Phi \mathcal{E} C_c(R^2)$ , such that for all z

$$|f(z) - \Phi(z)| \le \omega(\delta),$$
 (3)

$$|(\partial\Phi)(z)| < \frac{2\omega(\delta)}{\delta},$$
 (4)

And

$$\Phi(z) = -\frac{1}{\pi} \iint_{\mathcal{X}} \frac{(\partial \Phi)(\zeta)}{\zeta - z} d\zeta d\eta \qquad (\zeta = \xi + i\eta), \qquad (5)$$

Where X is the set of all points in the support of  $\Phi$  whose distance from the complement of K does not  $\delta$ . (Thus X contains no point which is "far within" K.) We construct  $\Phi$  as the

convolution of f with a smoothing function A. Put a(r) = 0 if  $r > \delta$ , put

$$a(r) = \frac{3}{\pi \delta^2} (1 - \frac{r^2}{\delta^2})^2$$
  $(0 \le r \le \delta),$  (6)

And define

$$A(z) = a(|z|) \tag{7}$$

For all complex z . It is clear that  $A \varepsilon C_c(R^2)$  . We claim that

$$\iint_{R^s} A = 1,\tag{8}$$

$$\iint_{\mathbb{R}^2} \partial A = 0, \tag{9}$$

$$\iint_{\mathbb{R}^3} \left| \partial A \right| = \frac{24}{15\delta} < \frac{2}{\delta},\tag{10}$$

The constants are so adjusted in (6) that (8) holds. (Compute the integral in polar coordinates), (9) holds simply because A has compact support. To compute (10), express  $\partial A$  in polar coordinates, and

note that 
$$\partial A/\partial \theta = 0$$
,

$$\partial A/\partial r = -a'$$

Now define

$$\Phi(z) = \iint_{\mathbb{R}^2} f(z - \zeta) A d\zeta d\eta = \iint_{\mathbb{R}^2} A(z - \zeta) f(\zeta) d\zeta d\eta$$
 (11)

Since f and A have compact support, so does  $\Phi$  . Since

$$\Phi(z) - f(z)$$

$$= \iint_{\mathbb{R}^2} [f(z - \zeta) - f(z)] A(\xi) d\xi d\eta \quad (12)$$

And 
$$A(\zeta) = 0$$
 if  $|\zeta| > \delta$ , (3) follows from (8).

The difference quotients of A converge boundedly to the corresponding partial derivatives, since  $A\varepsilon C_c(R^2)$ . Hence the last expression in (11) may be differentiated under the integral sign, and we obtain

$$(\partial \Phi)(z) = \iint_{\mathbb{R}^2} (\overline{\partial A})(z - \zeta) f(\zeta) d\xi d\eta$$

$$= \iint_{\mathbb{R}^2} f(z - \zeta)(\partial A)(\zeta) d\xi d\eta$$

$$= \iint_{\mathbb{R}^2} [f(z - \zeta) - f(z)](\partial A)(\zeta) d\xi d\eta \qquad (13)$$



The last equality depends on (9). Now (10) and (13) give (4). If we write (13) with  $\Phi_x$  and  $\Phi_y$  in place of  $\partial \Phi$ , we see that  $\Phi$  has continuous partial derivatives, if we can show that  $\partial \Phi = 0$  in G, where G is the set of all  $z \in K$  whose distance from the complement of K exceeds  $\delta$ . We shall do this by showing that

$$\Phi(z) = f(z) \qquad (z \in G); \qquad (14)$$

Note that  $\partial f = 0$  in G, since f is holomorphic there. Now if  $z \in G$ , then  $z - \zeta$  is in the interior of K for all  $\zeta$  with  $|\zeta| < \delta$ . The mean value property for harmonic functions therefore gives, by the first equation in (11),

$$\Phi(z) = \int_0^\delta a(r)rdr \int_0^{2\pi} f(z - re^{i\theta})d\theta$$
$$= 2\pi f(z) \int_0^\delta a(r)rdr = f(z) \iint_{\mathbb{R}^2} A = f(z)$$
(15)

For all  $z \in G$ , we have now proved (3), (4), and (5) The definition of X shows that X is compact and that X can be covered by finitely many open discs  $D_1,...,D_n$ , of radius  $2\delta$ , whose centers are not in K. Since  $S^2-K$  is connected, the center of each  $D_j$  can be joined to  $\infty$  by a polygonal path in  $S^2-K$ . It follows that each  $D_j$  contains a compact connected set  $E_j$ , of diameter at least  $2\delta$ , so that  $S^2-E_j$  is connected and so that  $K\cap E_j=\phi$ . with  $r=2\delta$ . There are functions  $g_j\mathcal{E}H(S^2-E_j)$  and constants  $b_j$  so that the inequalities.

$$\begin{aligned} \left|Q_{j}(\zeta,z)\right| &< \frac{50}{\delta}, \qquad (16) \\ \left|Q_{j}(\zeta,z) - \frac{1}{z - \zeta}\right| &< \frac{4,000\delta^{2}}{\left|z - \zeta\right|^{2}} \qquad (17) \\ \text{Hold for } z \notin E_{j} \text{ and } \zeta \in D_{j}, \text{ if} \\ Q_{j}(\zeta,z) &= g_{j}(z) + (\zeta - b_{j})g_{j}^{2}(z) \qquad (18) \\ \text{Let } \Omega \text{ be the complement of } E_{1} \cup ... \cup E_{n}. \text{ Then } \\ \Omega \text{ is an open set which contains } K. \text{ Put } \\ X_{1} &= X \cap D_{1} \qquad \text{and} \\ X_{j} &= (X \cap D_{j}) - (X_{1} \cup ... \cup X_{j-1}), \qquad \text{for} \end{aligned}$$

 $2 \le j \le n$ ,

Define

$$R(\zeta, z) = Q_i(\zeta, z) \qquad (\zeta \varepsilon X_i, z \varepsilon \Omega) \tag{19}$$

And

$$F(z) = \frac{1}{\pi} \iint_{X} (\partial \Phi)(\zeta) R(\zeta, z) d\zeta d\eta \qquad (20)$$

$$(z \in \Omega)$$

Since,

$$F(z) = \sum_{j=1}^{\infty} \frac{1}{\pi} \iint_{X_{-}} (\partial \Phi)(\zeta) Q_{j}(\zeta, z) d\xi d\eta, \qquad (21)$$

(18) shows that F is a finite linear combination of the functions  $g_j$  and  $g_j^2$ . Hence  $F \varepsilon H(\Omega)$ . By (20), (4), and (5) we have

$$|F(z) - \Phi(z)| < \frac{2\omega(\delta)}{\pi \delta} \iint_{Y} |R(\zeta, z)|$$

$$-\frac{1}{z-\zeta}|d\xi d\eta \quad (z \in \Omega) \quad (22)$$

Observe that the inequalities (16) and (17) are valid with R in place of  $Q_j$  if  $\zeta \in X$  and  $z \in \Omega$ . Now fix  $z \in \Omega$ ., put  $\zeta = z + \rho e^{i\theta}$ , and estimate the integrand in (22) by (16) if  $\rho < 4\delta$ , by (17) if  $4\delta \le \rho$ . The integral in (22) is then seen to be less than the sum of

$$2\pi \int_0^{4\delta} \left(\frac{50}{\delta} + \frac{1}{\rho}\right) \rho d\rho = 808\pi\delta \tag{23}$$

And

$$2\pi \int_{4\delta}^{\infty} \frac{4,000\delta^2}{\rho^2} \rho d\rho = 2,000\pi\delta.$$
 (24)

Hence (22) yields

$$|F(z) - \Phi(z)| < 6{,}000\omega(\delta)$$
  $(z \in \Omega)$  (25)

Since  $F \in H(\Omega), K \subset \Omega$ , and

 $S^2 - K$  is connected, Runge's theorem shows that F can be uniformly approximated on K by polynomials. Hence (3) and (25) show that (2) can be satisfied. This completes the proof.

**Lemma 1.0 :** Suppose  $f \in C_c(\mathbb{R}^2)$ , the space of all continuously differentiable functions in the plane, with compact support. Put

$$\partial = \frac{1}{2} \left( \frac{\partial}{\partial x} + i \frac{\partial}{\partial y} \right) \tag{1}$$

Then the following "Cauchy formula" holds:



$$f(z) = -\frac{1}{\pi} \iint_{R^2} \frac{(\partial f)(\zeta)}{\zeta - z} d\xi d\eta$$
$$(\zeta = \xi + i\eta) \tag{2}$$

**Proof:** This may be deduced from Green's theorem. However, here is a simple direct proof:

Put 
$$\varphi(r,\theta) = f(z + re^{i\theta}), r > 0, \theta$$
 real

If  $\zeta = z + re^{i\theta}$ , the chain rule gives

$$(\partial f)(\zeta) = \frac{1}{2}e^{i\theta} \left[ \frac{\partial}{\partial r} + \frac{i}{r} \frac{\partial}{\partial \theta} \right] \varphi(r,\theta)$$
 (3)

The right side of (2) is therefore equal to the limit, as  $\varepsilon \to 0$ , of

$$-\frac{1}{2}\int_{\varepsilon}^{\infty}\int_{0}^{2\pi} \left(\frac{\partial\varphi}{\partial r} + \frac{i}{r}\frac{\partial\varphi}{\partial\theta}\right) d\theta dr \tag{4}$$

For each  $r>0, \varphi$  is periodic in  $\theta$ , with period  $2\pi$ . The integral of  $\partial \varphi/\partial \theta$  is therefore 0, and (4) becomes

$$-\frac{1}{2\pi} \int_{0}^{2\pi} d\theta \int_{\varepsilon}^{\infty} \frac{\partial \varphi}{\partial r} dr = \frac{1}{2\pi} \int_{0}^{2\pi} \varphi(\varepsilon, \theta) d\theta \tag{5}$$

As  $\varepsilon \to 0$ ,  $\varphi(\varepsilon, \theta) \to f(z)$  uniformly. This gives (2)

If  $X^{\alpha} \in a$  and  $X^{\beta} \in k[X_1,...X_n]$ , then  $X^{\alpha}X^{\beta} = X^{\alpha+\beta} \in a$ , and so A satisfies the condition (\*). Conversely,

$$(\sum_{\alpha \in A} c_{\alpha} X^{\alpha})(\sum_{\beta \in \mathbb{L}^{n}} d_{\beta} X^{\beta}) = \sum_{\alpha, \beta} c_{\alpha} d_{\beta} X^{\alpha + \beta} \qquad (finite sums),$$

and so if A satisfies (\*), then the subspace generated by the monomials  $X^{\alpha}, \alpha \in a$ , is an ideal. The proposition gives a classification of the monomial ideals in  $k[X_1,...X_n]$ : they are in one to one correspondence with the subsets A of  $\square^n$  satisfying (\*). For example, the monomial ideals in k[X] are exactly the ideals  $(X^n), n \geq 1$ , and the zero ideal (corresponding to the empty set A). We write  $\langle X^{\alpha} | \alpha \in A \rangle$  for the ideal corresponding to A (subspace generated by the  $X^{\alpha}, \alpha \in a$ ).

LEMMA 1.1. Let S be a subset of  $\square^n$ . The the ideal a generated by  $X^\alpha, \alpha \in S$  is the monomial ideal corresponding to

$$A = \left\{ \beta \in \square^n \mid \beta - \alpha \in \square^n, \quad some \ \alpha \in S \right\}$$
Thus, a monomial is in  $a$  if and only if it is

Thus, a monomial is in a if and only if it is divisible by one of the  $X^{\alpha}, \alpha \in S$ 

PROOF. Clearly A satisfies (\*), and  $a \subset \langle X^{\beta} \mid \beta \in A \rangle$ . Conversely, if  $\beta \in A$ , then  $\beta - \alpha \in \square^n$  for some  $\alpha \in S$ , and  $X^{\beta} = X^{\alpha} X^{\beta - \alpha} \in a$ . The last statement follows from the fact that  $X^{\alpha} \mid X^{\beta} \Leftrightarrow \beta - \alpha \in \square^n$ . Let  $A \subset \square^n$  satisfy (\*). From the geometry of A, it is clear that there is a finite set of elements  $S = \{\alpha_1, ... \alpha_s\}$  of A such that  $A = \{\beta \in \square^n \mid \beta - \alpha_i \in \square^2, some \alpha_i \in S\}$ 

(The  $\alpha_i$ 's are the corners of A) Moreover,  $a = \langle X^{\alpha} \mid \alpha \in A \rangle$  is generated by the monomials  $X^{\alpha_i}, \alpha_i \in S$ .

DEFINITION 1.0. For a nonzero ideal a in  $k[X_1,...,X_n]$ , we let (LT(a)) be the ideal generated by  $\{LT(f) | f \in a\}$ 

LEMMA 1.2 Let a be a nonzero ideal in  $k[X_1,...,X_n]$ ; then (LT(a)) is a monomial ideal, and it equals  $(LT(g_1),...,LT(g_n))$  for some  $g_1,...,g_n \in a$ .

PROOF. Since (LT(a)) can also be described as the ideal generated by the leading monomials (rather than the leading terms) of elements of a.

**THEOREM 1.2.** Every *ideal* a in  $k[X_1,...,X_n]$  is finitely generated; more precisely,  $a=(g_1,...,g_s)$  where  $g_1,...,g_s$  are any elements of a whose leading terms generate LT(a)

**PROOF.** Let  $f \in a$ . On applying the division algorithm, we find  $f = a_1g_1 + ... + a_sg_s + r, \qquad a_i, r \in k\big[X_1, ..., X_n\big]$ , where either r = 0 or no monomial occurring in it is divisible by any  $LT(g_i)$ . But



$$\begin{split} r &= f - \sum a_i g_i \in a \,, \qquad \text{and} \qquad \text{therefore} \\ LT(r) &\in LT(a) = (LT(g_1), ..., LT(g_s)) \,, \\ \text{implies that every monomial occurring in } r \quad \text{is} \\ \text{divisible by one in } LT(g_i) \,. \quad \text{Thus } r = 0 \,, \text{ and} \\ g &\in (g_1, ..., g_s) \,. \end{split}$$

**DEFINITION 1.1.** A finite subset  $S = \{g_1, | ..., g_s\}$  of an ideal a is a standard ( (Grobner) bases for a if  $(LT(g_1), ..., LT(g_s)) = LT(a)$ . In other words, S is a standard basis if the leading term of every element of a is divisible by at least one of the leading terms of the  $g_i$ .

THEOREM 1.3 The ring  $k[X_1,...,X_n]$  is Noetherian i.e., every ideal is finitely generated.

**PROOF.** For n=1, k[X] is a principal ideal domain, which means that every ideal is generated by single element. We shall prove the theorem by induction on n. Note that the obvious map  $k[X_1,...X_{n-1}][X_n] \rightarrow k[X_1,...X_n]$  is an isomorphism – this simply says that every polynomial f in n variables  $X_1,...X_n$  can be expressed uniquely as a polynomial in  $X_n$  with coefficients in  $k[X_1,...,X_n]$ :

$$f(X_1,...X_n) = a_0(X_1,...X_{n-1})X_n^r + ... + a_r(X_1,...X_{n-1})$$

Thus the next lemma will complete the proof

**LEMMA 1.3.** If A is Noetherian, then so also is A[X]

PROOF. For a polynomial

$$f(X) = a_0 X^r + a_1 X^{r-1} + ... + a_r, \quad a_i \in A, \quad a_0 \neq 0,$$

r is called the degree of f, and  $a_0$  is its leading coefficient. We call 0 the leading coefficient of the polynomial 0. Let a be an ideal in A[X]. The leading coefficients of the polynomials in a form an ideal a in A, and since A is Noetherian, a will be finitely generated. Let  $g_1, \ldots, g_m$  be elements of a whose leading coefficients generate a, and let a be the maximum degree of a. Now

let  $f \in a$ , and suppose f has degree s > r, say,  $f = aX^s + ...$  Then  $a \in a$ , and so we can write  $a = \sum b_i a_i$ ,  $b_i \in A$ ,  $a_i = leading \ coefficient \ of \ g_i$ 

 $f - \sum b_i g_i X^{s-r_i}$ ,  $r_i = \deg(g_i)$ , has < deg(f) . By continuing in this way, we find that  $mod(g_1,...g_m)$  With  $f \equiv f$ polynomial of degree t < r For each d < r, let  $a_d$  be the subset of A consisting of 0 and the leading coefficients of all polynomials in a of degree d; it is again an ideal in  $g_{d,1},...,g_{d,m_d}$  be polynomials of degree d whose leading coefficients generate  $a_d$ . Then the same argument as above shows that any polynomial  $f_d$  in of degree dcan be  $mod(g_{d,1},...g_{d,m_d})$  With  $f_{d-1}$  $f_d \equiv f_{d-1}$ of degree  $\leq d-1$ . On applying this remark repeatedly we  $f_t \in (g_{r-1,1}, ...g_{r-1,m_{r-1}}, ...g_{0,1}, ...g_{0,m_0})$  Hence

$$f_t \in (g_1, ..., g_m g_{r-1,1}, ..., g_{r-1,m_{r-1}}, ..., g_{0,1}, ..., g_{0,m_0})$$

and so the polynomials  $g_1, ..., g_{0,m_0}$  generate a

One of the great successes of category in computer science has been the development of a "unified theory" of the constructions underlying denotational semantics. In the untyped  $\lambda$ -calculus, any term may appear in the function position of an application. This means that a model D of the  $\lambda$ -calculus must have the property that given a term t whose interpretation is  $d \in D$ , Also, the interpretation of a functional abstraction like  $\lambda x \cdot x$  is most conveniently defined as a function from DtoD, which must then be regarded as an element of D. Let  $\psi: [D \to D] \to D$  be the function that picks out elements of D to represent elements of  $[D \rightarrow D]$ and  $\phi: D \to [D \to D]$  be the function that maps elements of D to functions of D. Since  $\psi(f)$  is intended to represent the function f as an element of D, it makes sense to require that  $\phi(\psi(f)) = f$ ,



that is,  $\psi \circ \psi = id_{[D \to D]}$  Furthermore, we often want to view every element of D as representing some function from D to D and require that elements representing the same function be equal – that is

$$\psi(\varphi(d)) = d$$

or

$$\psi \circ \phi = id_{D}$$

condition called latter extensionality. These conditions together imply that  $\phi$  and  $\psi$  are inverses--- that is, D is isomorphic to the space of functions from D to D that can be the interpretations of functional abstractions:  $D \cong [D \to D]$  .Let us suppose we are working with the untyped  $\lambda$  – calculus, we need a solution ot the equation  $D \cong A + [D \to D]$ , where A is some predetermined domain containing interpretations for elements of C. Each element of D corresponds to either an element of A or an element of  $|D \rightarrow D|$ , with a tag. This equation can be solved by finding least fixed points of the function  $F(X) = A + |X \rightarrow X|$  from domains to domains --- that is, finding domains X such that  $X \cong A + [X \rightarrow X]$ , and such that for any domain Y also satisfying this equation, there is an embedding of X to Y --- a pair of maps

$$X \bigcap_{f^R}^f Y$$

Such that

$$f^R o f = id_X$$

$$f \circ f^R \subseteq id_Y$$

Where 
$$f \subseteq g$$
 means that

f approximates g in some ordering representing their information content. The key shift of perspective from the domain-theoretic to the more general category-theoretic approach lies in considering F not as a function on domains, but as a functor on a category of domains. Instead of a least fixed point of the function, F.

**Definition 1.3**: Let K be a category and  $F: K \to K$  as a functor. A fixed point of F is a pair (A,a), where A is a K-object and  $a: F(A) \to A$  is an isomorphism. A prefixed point of F is a pair (A,a), where A is a K-object and a is any arrow from F(A) to A

**Definition 1.4:** An  $\omega$ -chain in a category K is a diagram of the following form:

$$\Delta = D_o \xrightarrow{f_o} D_1 \xrightarrow{f_1} D_2 \xrightarrow{f_2} \dots$$

Recall that a cocone  $\mu$  of an  $\omega-chain$   $\Delta$  is a K-object X and a collection of K—arrows  $\left\{\mu_i:D_i\to X\,|\,i\ge 0\right\}$  such that  $\mu_i=\mu_{i+1}o\,f_i$  for all  $i\ge 0$ . We sometimes write  $\mu:\Delta\to X$  as a reminder of the arrangement of  $\mu$ 's components Similarly, a colimit  $\mu:\Delta\to X$  is a cocone with the property that if  $\nu:\Delta\to X'$  is also a cocone then there exists a unique mediating arrow  $k:X\to X'$  such that for all  $i\ge 0$ ,,  $v_i=k\,o\,\mu_i$ . Colimits of  $\omega-chains$  are sometimes referred to as  $\omega-co\,\limits$ . Dually, an  $\omega^{op}-chain$  in K is a diagram of the following form:

 $\Delta = D_o \overset{f_o}{\longleftarrow} D_1 \overset{f_1}{\longleftarrow} D_2 \overset{f_2}{\longleftarrow} \dots$  $\mu: X \to \Delta$  of an  $\omega^{op}$  - chain  $\Delta$  is a **K**-object X and a collection of **K**-arrows  $\{\mu_i : D_i \mid i \geq 0\}$ such that for all  $i \ge 0$ ,  $\mu_i = f_i \circ \mu_{i+1}$ . An  $\omega^{op}$ limit of an  $\omega^{op}$  – chain  $\Delta$  is a cone  $\mu: X \to \Delta$ with the property that if  $\nu: X' \to \Delta$  is also a cone, then there exists a unique mediating arrow  $k: X' \to X$  such that for all  $i \ge 0$ ,  $\mu_i \circ k = v_i$ . We write  $\perp_k$  (or just  $\perp$ ) for the distinguish initial object of K, when it has one, and  $\bot \rightarrow A$  for the unique arrow from  $\perp$  to each K-object A. It is also convenient to write  $\Delta^- = D_1 \xrightarrow{f_1} D_2 \xrightarrow{f_2} \dots$  to denote all of  $\Delta$  except  $D_o$  and  $f_0$ . By analogy,  $\mu^-$  is  $\{\mu_i \mid i \geq 1\}$ . For the images of  $\Delta$  and  $\mu$  $F(f_o)$  $F(f_1)$  $F(\Delta) = F(D_0) \longrightarrow F(D_1) \longrightarrow F(D_2) \longrightarrow \dots$ and  $F(\mu) = \{F(\mu_i) | i \ge 0\}$ 

We write  $F^i$  for the *i*-fold iterated composition of F — that is,  $F^o(f) = f$ ,  $F^1(f) = F(f)$ ,  $F^2(f) = F(F(f))$ , etc. With these definitions we can state that every monitonic function on a complete lattice has a least fixed point:

**Lemma 1.4.** Let K be a category with initial object  $\bot$  and let  $F: K \to K$  be a functor. Define the  $\omega - chain \Delta$  by



$$\Delta = \perp \xrightarrow{f(\bot)} F(\bot) \xrightarrow{F(\bot) \to F(\bot)} F^2(\bot) \xrightarrow{F^2(\bot \to F(\bot))} \dots \dots \dots$$
 If both  $\mu : \Delta \to D$  and  $F(\mu) : F(\Delta) \to F(D)$  are colimits, then (D,d) is an intial F-algebra, where  $d : F(D) \to D$  is the mediating arrow from  $F(\mu)$  to the cocone  $\mu^-$ 

Theorem 1.4 Let a DAG G given in which each node is a random variable, and let a discrete conditional probability distribution of each node given values of its parents in G be specified. Then the product of these conditional distributions yields a joint probability distribution P of the variables, and (G.P) satisfies the Markov condition.

**Proof.** Order the nodes according to an ancestral ordering. Let  $X_1, X_2, \dots, X_n$  be the resultant ordering. Next define.

$$P(x_1, x_2,...x_n) = P(x_n | pa_n) P(x_{n-1} | Pa_{n-1})...$$
  
.. $P(x_2 | pa_2) P(x_1 | pa_1),$ 

Where  $PA_i$  is the set of parents of  $X_i$  of in G and  $P(x_i \mid pa_i)$  is the specified conditional probability distribution. First we show this does indeed yield a joint probability distribution. Clearly,  $0 \le P(x_1, x_2, ... x_n) \le 1$  for all values of the variables. Therefore, to show we have a joint distribution, as the variables range through all their possible values, is equal to one. To that end, Specified conditional distributions are the conditional distributions they notationally represent in the joint distribution. Finally, we show the Markov condition is satisfied. To do this, we need show for  $1 \le k \le n$  that

whenever
$$P(pa_k) \neq 0, \text{if } P(nd_k \mid pa_k) \neq 0$$

$$and P(x_k \mid pa_k) \neq 0$$

then 
$$P(x_k \mid nd_k, pa_k) = P(x_k \mid pa_k),$$

Where  $ND_k$  is the set of nondescendents of  $X_k$  of in G. Since  $PA_k \subseteq ND_k$ , we need only show  $P(x_k \mid nd_k) = P(x_k \mid pa_k)$ . First for a given k, order the nodes so that all and only nondescendents of  $X_k$  precede  $X_k$  in the ordering. Note that this ordering depends on k, whereas the ordering in the first part of the proof does not. Clearly then

$$\begin{split} ND_k &= \left\{X_1, X_2, .... X_{k-1}\right\} \\ Let \\ D_k &= \left\{X_{k+1}, X_{k+2}, .... X_n\right\} \\ \text{follows } \sum\nolimits_{d_k} \end{split}$$

We define the  $m^{th}$  cyclotomic field to be the field  $Q[x]/(\Phi_m(x))$  Where  $\Phi_m(x)$  is the  $m^{th}$  cyclotomic polynomial.  $Q[x]/(\Phi_m(x))$   $\Phi_m(x)$  has degree  $\varphi(m)$  over Q since  $\Phi_m(x)$  has degree  $\varphi(m)$ . The roots of  $\Phi_m(x)$  are just the primitive  $m^{th}$  roots of unity, so the complex embeddings of  $Q[x]/(\Phi_m(x))$  are simply the  $\varphi(m)$  maps  $\sigma_k:Q[x]/(\Phi_m(x))\mapsto C$ ,  $1\leq k\prec m, (k,m)=1$ , where  $\sigma_k(x)=\xi_m^k$ ,

 $\xi_m$  being our fixed choice of primitive  $m^{th}$  root of unity. Note that  $\xi_m^k \in Q(\xi_m)$  for every k; it follows that  $Q(\xi_m) = Q(\xi_m^k)$  for all k relatively prime to m. In particular, the images of the  $\sigma_i$  coincide, so  $Q[x]/(\Phi_m(x))$  is Galois over Q. This means that we can write  $Q(\xi_m)$  for  $Q[x]/(\Phi_m(x))$  without much fear of ambiguity; we will do so from now on, the identification being

easily talk about cyclotomic fields being extensions of one another, or intersections or compositums; all of these things take place considering them as subfield of C. We now investigate some basic properties of cyclotomic fields. The first issue is whether or not they are all distinct; to determine this, we need to know which roots of unity lie in  $Q(\xi_m)$ . Note, for example, that if m is odd, then  $-\xi_m$  is a  $2m^{th}$  root of unity. We will show that this is the only way in which one can obtain any non-

 $\xi_m \mapsto x$ . One advantage of this is that one can

LEMMA 1.5 If m divides n, then  $Q(\xi_m)$  is contained in  $Q(\xi_n)$ 

 $m^{th}$  roots of unity.



PROOF. Since  $\xi^{n/m} = \xi_m$ , we have  $\xi_m \in Q(\xi_n)$ , so the result is clear

LEMMA 1.6 If m and n are relatively prime, then  $Q(\xi_{m}, \xi_{n}) = Q(\xi_{mm})$ 

and

$$Q(\xi_m) \cap Q(\xi_n) = Q$$

(Recall the  $Q(\xi_m,\xi_n)$  is the compositum of  $Q(\xi_m)$  and  $Q(\xi_n)$  )

PROOF. One checks easily that  $\xi_m \xi_n$  is a primitive  $mn^{th}$  root of unity, so that

$$Q(\xi_{mn}) \subseteq Q(\xi_{m}, \xi_{n})$$

$$[Q(\xi_m,\xi_n):Q] \leq [Q(\xi_m):Q][Q(\xi_n:Q]$$

 $= \varphi(m)\varphi(n) = \varphi(mn);$ 

Since  $Q(\xi_{mn}):Q=\varphi(mn)$ ; this implies that  $Q(\xi_{m},\xi_{n})=Q(\xi_{nm})$  We know that  $Q(\xi_{m},\xi_{n})$  has degree  $\varphi(mn)$  over Q, so we must have

$$[Q(\xi_m,\xi_n):Q(\xi_m)]=\varphi(n)$$

and

$$[Q(\xi_m,\xi_n):Q(\xi_m)]=\varphi(m)$$

$$[Q(\xi_m):Q(\xi_m)\cap Q(\xi_n)] \ge \varphi(m)$$
And thus that  $Q(\xi_m)\cap Q(\xi_n) = Q$ 

PROPOSITION 1.2 For any m and n

$$Q(\xi_m,\xi_n) = Q(\xi_{[m,n]})$$

And

$$Q(\xi_m) \cap Q(\xi_n) = Q(\xi_{(m,n)});$$

here [m,n] and (m,n) denote the least common multiple and the greatest common divisor of m and n, respectively.

PROOF. Write  $m = p_1^{e_1} .....p_k^{e_k}$  and  $p_1^{f_1} ....p_k^{f_k}$  where the  $p_i$  are distinct primes. (We allow  $e_i$  or  $f_i$  to be zero)

$$\begin{split} Q(\xi_{m}) &= Q(\xi_{p_{1}^{e_{1}}})Q(\xi_{p_{2}^{e_{2}}})...Q(\xi_{p_{k}^{e_{k}}}) \\ and \\ Q(\xi_{n}) &= Q(\xi_{p_{1}^{f_{1}}})Q(\xi_{p_{2}^{f_{2}}})...Q(\xi_{p_{k}^{f_{k}}}) \\ Thus \\ Q(\xi_{m}, \xi_{n}) &= Q(\xi_{p_{1}^{e_{1}}})......Q(\xi_{p_{2}^{e_{k}}})Q(\xi_{p_{1}^{f_{1}}})...Q(\xi_{p_{k}^{f_{k}}}) \\ &= Q(\xi_{p_{1}^{e_{1}}})Q(\xi_{p_{1}^{f_{1}}})...Q(\xi_{p_{k}^{e_{k}}})Q(\xi_{p_{k}^{f_{k}}}) \\ &= Q(\xi_{p_{1}^{max(e_{1},f_{1})}}).....Q(\xi_{p_{1}^{max(e_{k},f_{k})}}) \\ &= Q(\xi_{p_{1}^{max(e_{1},f_{1})}}......p_{1}^{max(e_{k},f_{k})}) \\ &= Q(\xi_{[m,n]}); \end{split}$$

An entirely similar computation shows that  $Q(\xi_m) \cap Q(\xi_n) = Q(\xi_{(m,n)})$ 

Mutual information measures the information transferred when  $x_i$  is sent and  $y_i$  is received, and is defined as

$$I(x_i, y_i) = \log_2 \frac{P(\frac{x_i}{y_i})}{P(x_i)} bits$$
 (1)

In a noise-free channel, **each**  $y_i$  is uniquely connected to the corresponding  $x_i$ , and so they constitute an input –output pair  $(x_i, y_i)$  for which

$$P(\frac{x_i}{y_j}) = 1 \text{ and } I(x_i, y_j) = \log_2 \frac{1}{P(x_i)} \text{ bits};$$

that is, the transferred information is equal to the self-information that corresponds to the input  $x_i$ . In a very noisy channel, the output  $y_i$  and input  $x_i$  would be completely uncorrelated, and so  $P(x_i/y_j) = P(x_i)$  and also  $I(x_i, y_j) = 0$ ; that is,

there is no transference of information. In general, a given channel will operate between these two extremes. The mutual information is defined between the input and the output of a given channel. An average of the calculation of the mutual information for all input-output pairs of a given channel is the average mutual information:

$$I(X,Y) = \sum_{i,j} P(x_i, y_j) I(x_i, y_j) = \sum_{i,j} P(x_i, y_j) \log_2 \left[ \frac{P(x_i / y_j)}{P(x_i)} \right]$$

bits per symbol. This calculation is done over the input and output alphabets. The average mutual information. The following expressions are useful for modifying the mutual information expression:



$$P(x_{i}, y_{j}) = P(\frac{x_{i}}{y_{j}})P(y_{j}) = P(\frac{y_{j}}{x_{i}})P(x_{i})$$

$$P(y_{j}) = \sum_{i} P(\frac{y_{j}}{x_{i}})P(x_{i})$$

$$P(x_{i}) = \sum_{i} P(\frac{x_{i}}{y_{j}})P(y_{j})$$
Then
$$I(X, Y) = \sum_{i,j} P(x_{i}, y_{j}) \log_{2} \left[\frac{1}{P(x_{i})}\right]$$

$$-\sum_{i,j} P(x_{i}, y_{j}) \log_{2} \left[\frac{1}{P(x_{i})}\right]$$

$$\sum_{i,j} P(x_{i}, y_{j}) \log_{2} \left[\frac{1}{P(x_{i})}\right]$$

$$= \sum_{i} \left[P(\frac{x_{i}}{y_{j}})P(y_{j})\right] \log_{2} \frac{1}{P(x_{i})}$$

$$\sum_{i} P(x_{i}) \log_{2} \frac{1}{P(x_{i})} = H(X)$$

$$I(X, Y) = H(X) - H(\frac{X}{Y})$$
Where
$$H(\frac{X}{Y}) = \sum_{i,j} P(x_{i}, y_{j}) \log_{2} \frac{1}{P(\frac{x_{i}}{y_{j}})}$$
 is

usually called the equivocation. In a sense, the equivocation can be seen as the information lost in the noisy channel, and is a function of the backward conditional probability. The observation of an output symbol  $y_i$  provides  $H(X) - H(X_V)$  bits of information. This difference is the mutual information of the channel. Mutual Information: Properties Since

$$P(\frac{x_i}{y_j})P(y_j) = P(\frac{y_j}{x_i})P(x_i)$$

The mutual information fits the condition

$$I(X,Y) = I(Y,X)$$

And by interchanging input and output it is also true

$$I(X,Y) = H(Y) - H(\frac{Y}{X})$$

Where

$$H(Y) = \sum_{i} P(y_i) \log_2 \frac{1}{P(y_i)}$$

This last entropy is usually called the noise entropy. Thus, the information transferred through the channel is the difference between the output entropy and the noise entropy. Alternatively, it can be said that the channel mutual information is the difference between the number of bits needed for determining a given input symbol before knowing the corresponding output symbol, and the number of bits needed for determining a given input symbol after knowing the corresponding output symbol

$$I(X,Y) = H(X) - H(X/Y)$$

As the channel mutual information expression is a difference between two quantities, it seems that this parameter can adopt negative values. However, and is spite of the fact that for some  $y_i, H(X/y_i)$  can be larger than H(X), this is not possible for the average value calculated over all the outputs:

$$\sum_{i,j} P(x_i, y_j) \log_2 \frac{P(x_i/y_j)}{P(x_i)} = \sum_{i,j} P(x_i, y_j) \log_2 \frac{P(x_i, y_j)}{P(x_i)P(y_j)}$$

$$-I(X,Y) = \sum_{i,j} P(x_i, y_j) \frac{P(x_i)P(y_j)}{P(x_i, y_j)} \le 0$$

Because this expression is of the form

$$\sum_{i=1}^{M} P_i \log_2(\frac{Q_i}{P_i}) \le 0$$

The above expression can be applied due to the factor  $P(x_i)P(y_i)$ , which is the product of two probabilities, so that it behaves as the quantity  $Q_i$ , which in this expression is a dummy variable that fits the condition  $\sum_{i} Q_{i} \leq 1$ . It can be concluded that the average mutual information is a nonnegative number. It can also be equal to zero, when the input and the output are independent of each other. A related entropy called the joint entropy is defined as

$$H(X,Y) = \sum_{i,j} P(x_i, y_j) \log_2 \frac{1}{P(x_i, y_j)}$$

$$= \sum_{i,j} P(x_i, y_j) \log_2 \frac{P(x_i)P(y_j)}{P(x_i, y_j)}$$

$$+ \sum_{i,j} P(x_i, y_j) \log_2 \frac{1}{P(x_i)P(y_j)}$$

**Theorem 1.5:** Entropies of the binary erasure channel (BEC) The BEC is defined with an alphabet



of two inputs and three outputs, with symbol probabilities.

 $P(x_1) = \alpha$  and  $P(x_2) = 1 - \alpha$ , and transition probabilities

$$P(\frac{y_3}{x_2}) = 1 - p \text{ and } P(\frac{y_2}{x_1}) = 0,$$
and  $P(\frac{y_3}{x_1}) = 0$ 
and  $P(\frac{y_1}{x_2}) = p$ 
and  $P(\frac{y_3}{x_2}) = 1 - p$ 

**Lemma 1.7.** Given an arbitrary restricted time-discrete, amplitude-continuous channel whose restrictions are determined by sets  $F_n$  and whose density functions exhibit no dependence on the state s, let n be a fixed positive integer, and p(x) an arbitrary probability density function on Euclidean n-space.  $p(y \mid x)$  for the density  $p_n(y_1,...,y_n \mid x_1,...x_n)$  and F for  $F_n$ . For any real number a, let

$$A = \left\{ (x, y) : \log \frac{p(y \mid x)}{p(y)} > a \right\} \tag{1}$$

Then for each positive integer u, there is a code  $(u, n, \lambda)$  such that

$$\lambda \le ue^{-a} + P\{(X,Y) \notin A\} + P\{X \notin F\}$$

Where

 $P\{(X,Y) \in A\} = \int_{A} \dots \int p(x,y) dx dy, \qquad p(x,y) = p(x)p(y \mid x)$ and

$$P\{X \in F\} = \int_{F} \dots \int_{F} p(x) dx$$

Proof: A sequence  $x^{(1)} \in F$  such that  $P\{Y \in A_{\perp} \mid X = x^{(1)}\} \ge 1 - \varepsilon$ 

where  $A_x = \{y: (x, y) \in A\};$ 

Choose the decoding set  $B_1$  to be  $A_{x^{(1)}}$ . Having chosen  $x^{(1)},\ldots,x^{(k-1)}$  and  $B_1,\ldots,B_{k-1}$ , select  $x^k\in F$  such that

$$P\left\{Y \in A_{x^{(k)}} - \bigcup_{i=1}^{k-1} B_i \mid X = x^{(k)}\right\} \ge 1 - \varepsilon;$$

Set  $B_k = A_{x^{(k)}} - \bigcup_{i=1}^{k-1} B_i$ , If the process does not terminate in a finite number of steps, then the sequences  $x^{(i)}$  and decoding sets  $B_i$ , i=1,2,...,u, form the desired code. Thus assume that the process terminates after t steps. (Conceivably t=0). We

will show  $t \ge u$  by showing that  $\varepsilon \le te^{-a} + P\{(X,Y) \notin A\} + P\{X \notin F\}$ . We proceed as follows.

$$B = \bigcup_{j=1}^{t} B_{j}. \quad (If \quad t = 0, take \quad B = \phi). \quad Then$$

$$P\{(X,Y) \in A\} = \int_{(x,y)\in A} p(x,y)dx dy$$

$$= \int_{x} p(x) \int_{y \in A_{x}} p(y \mid x) dy dx$$

$$= \int_{x} p(x) \int_{y \in B \cap A_{x}} p(y \mid x) dy dx + \int_{x} p(x)$$

# C. Algorithms

**Ideals.** Let A be a ring. Recall that an *ideal a* in A is a subset such that a is subgroup of A regarded as a group under addition;

$$a \in a, r \in A \Rightarrow ra \in A$$

The ideal generated by a subset S of A is the intersection of all ideals A containing a ----- it is easy to verify that this is in fact an ideal, and that it consist of all finite sums of the form  $\sum r_i s_i$  with

$$r_i \in A, s_i \in S$$
. When  $S = \{s_1, \dots, s_m\}$ , we shall

(2) write  $(s_1, ..., s_m)$  for the ideal it generates.

Let a and b be ideals in A. The  $\{a+b \mid a \in a, b \in b\}$  is an ideal, denoted by The ideal generated  $\{ab \mid a \in a, b \in b\}$  is denoted by ab. Note that  $ab \subset a \cap b$ . Clearly ab consists of all finite sums  $\sum a_i b_i$  with  $a_i \in a$  and  $b_i \in b$ , and if  $a = (a_1, ..., a_m)$  and  $b = (b_1, ..., b_n)$ ,  $ab = (a_1b_1,...,a_ib_i,...,a_mb_n)$ . Let a be an ideal of A. The set of cosets of a in A forms a ring A/aand  $a \mapsto a + a$  is a homomorphism  $\phi: A \mapsto A/a$ . The map  $b \mapsto \phi^{-1}(b)$  is a one to one correspondence between the ideals of A/aand the ideals of A containing a An ideal p if *prime* if  $p \neq A$  and  $ab \in p \Rightarrow a \in p$  or  $b \in p$ . Thus p is prime if and only if A/p is nonzero the and property  $b \neq 0 \Rightarrow a = 0$ , i.e., A/p is an integral domain. An ideal m is maximal if  $m \neq A$ and there does not exist an ideal n contained strictly between m and A. Thus m is maximal if and only if A/m has no proper nonzero ideals, and



so is a field. Note that m maximal  $\Rightarrow m$  prime. The ideals of  $A \times B$  are all of the form  $a \times b$ , with a and b ideals in A and B. To see this, note that if c is an ideal in  $A \times B$  and  $(a,b) \in c$ , then  $(a,0) = (a,b)(1,0) \in c$  and  $(0,b) = (a,b)(0,1) \in c$ . This shows that  $c = a \times b$  with  $a = \{a \mid (a,b) \in c \text{ some } b \in b\}$  and  $b = \{b \mid (a,b) \in c \text{ some } a \in a\}$ 

Let A be a ring. An A-algebra is a ring B together with a homomorphism  $i_R: A \to B$ . A homomorphism of A-algebra  $B \rightarrow C$  is a homomorphism of rings  $\varphi: B \to C$  such that  $\varphi(i_R(a)) = i_C(a)$  for all  $a \in A$ . An A-algebra B is said to be finitely generated ( or of finite-type over A) if there exist elements  $x_1,...,x_n \in B$  such that every element of B can be expressed as a polynomial in the  $x_i$  with coefficients in i(A), i.e., such that the homomorphism  $A[X_1,...,X_n] \rightarrow B$  $x_i$  is surjective. sending  $X_i$  to homomorphism  $A \rightarrow B$  is *finite*, and B is finitely generated as an A-module. Let k be a field, and let A be a k-algebra. If  $1 \neq 0$  in A, then the map  $k \rightarrow A$  is injective, we can identify k with its image, i.e., we can regard k as a subring of A. If 1=0 in a ring R, the R is the zero ring, i.e.,  $R = \{0\}$ . Polynomial rings. Let k be a field. A monomial in  $X_1,...,X_n$  is an expression of the form  $X_1^{a_1}...X_n^{a_n}, \qquad a_j \in N$  . The total degree of the monomial is  $\sum a_i$ . We sometimes abbreviate it by  $X^{\alpha}$ ,  $\alpha = (a_1, ..., a_n) \in \square^n$  The elements of the polynomial ring  $k[X_1,...,X_n]$  are finite sums  $\sum c_{a_1...a_n} X_1^{a_1}...X_n^{a_n}, \qquad c_{a_1...a_n} \in k, \quad a_i \in \square$ With the obvious notions of equality, addition and multiplication. Thus the monomials from basis for  $k[X_1,...,X_n]$  as a k-vector space. The ring  $k \mid X_1, ..., X_n$  is an integral domain, and the only units in it are the nonzero constant polynomials. A polynomial  $f(X_1,...,X_n)$  is irreducible if it is nonconstant and has only the obvious factorizations,

i.e.,  $f=gh\Longrightarrow g$  or h is constant. Division in k[X]. The division algorithm allows us to divide a nonzero polynomial into another: let f and g be polynomials in k[X] with  $g\ne 0$ ; then there exist unique polynomials  $q,r\in k[X]$  such that f=qg+r with either r=0 or  $\deg r<\deg g$ . Moreover, there is an algorithm for deciding whether  $f\in (g)$ , namely, find r and check whether it is zero. Moreover, the Euclidean algorithm allows to pass from finite set of generators for an ideal in k[X] to a single generator by successively replacing each pair of generators with their greatest common divisor.

(*Pure*) lexicographic ordering (lex). Here monomials are ordered by lexicographic(dictionary) order. More precisely, let  $\alpha=(a_1,...a_n)$  and  $\beta=(b_1,...b_n)$  be two elements of  $\square$  ; then  $\alpha>\beta$  and  $X^\alpha>X^\beta$  (lexicographic ordering) if, in the vector difference  $\alpha-\beta\in\square$ , the left most nonzero entry is positive. For example,

 $XY^2 > Y^3Z^4$ ;  $X^3Y^2Z^4 > X^3Y^2Z$ . Note that this isn't quite how the dictionary would order them: it would put XXXYYZZZZ after XXXYYZ. Graded reverse lexicographic order (grevlex). Here monomials are ordered by total degree, with ties broken by reverse lexicographic ordering. Thus,  $\alpha > \beta$  if  $\sum a_i > \sum b_i$ , or  $\sum a_i = \sum b_i$  and in  $\alpha - \beta$  the right most nonzero entry is negative. For example:

$$X^4Y^4Z^7 > X^5Y^5Z^4$$
 (total degree greater)  
 $XY^5Z^2 > X^4YZ^3$ ,  $X^5YZ > X^4YZ^2$ 

**Orderings on**  $k[X_1,...X_n]$ . Fix an ordering on the monomials in  $k[X_1,...X_n]$ . Then we can write an element f of  $k[X_1,...X_n]$  in a canonical fashion, by re-ordering its elements in decreasing order. For example, we would write

$$f = 4XY^{2}Z + 4Z^{2} - 5X^{3} + 7X^{2}Z^{2}$$
as
$$f = -5X^{3} + 7X^{2}Z^{2} + 4XY^{2}Z + 4Z^{2} \quad (lex)$$
or
$$f = 4XY^{2}Z + 7X^{2}Z^{2} - 5X^{3} + 4Z^{2} \quad (grevlex)$$



Let  $\sum a_{\alpha}X^{\alpha}\in k\big[X_{1},...,X_{n}\big]$  , in decreasing

$$f = a_{\alpha_0} X^{\alpha_0} +_{\alpha_1} X^{\alpha_1} + ..., \qquad \alpha_0 > \alpha_1 > ..., \quad \alpha_0 \neq 0 \text{ all } \alpha \text{ with } c_\alpha \neq 0.$$

Then we define.

- The *multidegree* of f to be multdeg(f)=  $\alpha_0$ ;
- The leading coefficient of f to be  $LC(f) = a_{\alpha}$ ;
- The leading monomial of f to be LM(f) =
- The leading term of f to be  $LT(f) = a_{\alpha} X^{\alpha_0}$

For the polynomial  $f = 4XY^2Z + ...$ , the multidegree is (1,2,1), the leading coefficient is 4, the leading monomial is  $XY^2Z$ , and the leading  $4XY^2Z$ . The division algorithm in  $k[X_1,...X_n]$ . Fix a monomial ordering in  $\square^2$ . Suppose given a polynomial f and an ordered set  $(g_1,...g_s)$  of polynomials; the division algorithm then constructs polynomials  $a_1,...a_s$  and r such that  $f = a_1 g_1 + ... + a_s g_s + r$  Where either r = 0 or no monomial in r is divisible by any of  $LT(g_1),...,LT(g_s)$ Step  $LT(g_1)|LT(f)$ , divide  $g_1$  into f to get  $f = a_1 g_1 + h, \qquad a_1 = \frac{LT(f)}{LT(g_1)} \in k[X_1, ..., X_n]$ 

If  $LT(g_1) | LT(h)$ , repeat the process until  $f = a_1 g_1 + f_1$  (different  $a_1$ ) with  $LT(f_1)$  not divisible by  $LT(g_1)$ . Now divide  $g_2$  into  $f_1$ , and so on, until  $f = a_1 g_1 + ... + a_s g_s + r_1$  $LT(r_1)$  not divisible by any  $LT(g_1),...LT(g_s)$ **Step 2:** Rewrite  $r_1 = LT(r_1) + r_2$ , and repeat Step  $r_2$ for f: $f = a_1 g_1 + ... + a_s g_s + LT(r_1) + r_3$  $a_i$ 's ) **Monomial ideals.** In general, an ideal awill contain a polynomial without containing the individual terms of the polynomial; for example, the ideal  $a = (Y^2 - X^3)$  contains  $Y^2 - X^3$  but not  $Y^2$  or  $X^3$ .

**DEFINITION 1.5**. An ideal a is monomial if  $\sum c_{\alpha} X^{\alpha} \in a \Rightarrow X^{\alpha} \in a$ 

PROPOSITION 1.3. Let a be a monomial ideal, and let  $A = \{ \alpha \mid X^{\alpha} \in a \}$ . Then A satisfies the condition  $\alpha \in A$ ,  $\beta \in \square^n \Rightarrow \alpha + \beta \in$ And a is the k-subspace of  $k[X_1,...,X_n]$ generated by the  $X^{\alpha}$ ,  $\alpha \in A$ . Conversely, of A is a subset of  $\square^n$  satisfying (\*), then the k-subspace a of  $k[X_1,...,X_n]$  generated by  $\{X^{\alpha} \mid \alpha \in A\}$ 

PROOF. It is clear from its definition that a monomial ideal a is the k -subspace of  $k[X_1,...,X_n]$ 

generated by the set of monomials it contains. If  $X^{\alpha} \in a$  and  $X^{\beta} \in k[X_1, ..., X_n]$ 

If a permutation is chosen uniformly and at random from the n! possible permutations in  $S_n$ , then the counts  $C_i^{(n)}$  of cycles of length j are dependent random variables. The joint distribution of  $C^{(n)} = (C_1^{(n)}, ..., C_n^{(n)})$  follows from Cauchy's formula, and is given by

$$P[C^{(n)} = c] = \frac{1}{n!}N(n,c) = 1\left\{\sum_{j=1}^{n} jc_{j} = n\right\} \prod_{j=1}^{n} \left(\frac{1}{j}\right)^{c_{j}} \frac{1}{c_{j}!}, \quad (1.1)$$

for  $c \in \square^n$ .

Lemma<sub>1.7</sub> For nonnegative integers

$$E\left(\prod_{j=1}^{n} (C_{j}^{(n)})^{[m_{j}]}\right) = \left(\prod_{j=1}^{n} \left(\frac{1}{j}\right)^{m_{j}}\right) 1\left\{\sum_{j=1}^{n} j m_{j} \le n\right\}$$
(1.4)

Proof. This can be established directly by cancellation of  $c_j^{[m_j]}/c_j^! = 1/(c_j - m_j)!$  when  $c_j \ge m_j$ , which occurs between the ingredients in Cauchy's formula and the falling factorials in the moments. Write  $m = \sum_{i} jm_{i}$ . Then, with the first sum indexed by  $c = (c_1, ..., c_n) \in \square_+^n$  and the last sum indexed by  $d = (d_1, ..., d_n) \in \square_+^n$  via the correspondence  $d_i = c_i - m_i$ , we have



$$\begin{split} E\bigg(\prod_{j=1}^{n}(C_{j}^{(n)})^{[m_{j}]}\bigg) &= \sum_{c}P[C^{(n)} = c] \prod_{j=1}^{n}(c_{j})^{[m_{j}]} \\ &= \sum_{c:c_{j} \geq m_{j} \ for \ all \ j} \mathbb{1}\bigg\{\sum_{j=1}^{n}jc_{j} = n\bigg\} \prod_{j=1}^{n}\frac{(c_{j})^{[m_{j}]}}{j^{c_{j}}c_{j}!} \\ &= \prod_{j=1}^{n}\frac{1}{j^{m_{j}}} \sum_{d}\mathbb{1}\bigg\{\sum_{j=1}^{n}jd_{j} = n - m\bigg\} \prod_{j=1}^{n}\frac{1}{j^{d_{j}}(d_{j})} \\ &= \sum_{j=1}^{n}\frac{1}{j^{m_{j}}} \sum_{d}\mathbb{1}\bigg\{\sum_{j=1}^{n}jd_{j} = n - m\bigg\} \prod_{j=1}^{n}\frac{1}{j^{d_{j}}(d_{j})} \\ &= \sum_{j=1}^{n}\frac{1}{j^{m_{j}}} \sum_{d}\mathbb{1}\bigg\{\sum_{j=1}^{n}jd_{j} = n - m\bigg\} \prod_{j=1}^{n}\frac{1}{j^{d_{j}}(d_{j})} \\ &= \sum_{j=1}^{n}\frac{1}{j^{m_{j}}} \sum_{d}\mathbb{1}\bigg\{\sum_{j=1}^{n}jd_{j} = n - m\bigg\} \prod_{j=1}^{n}\frac{1}{j^{d_{j}}(d_{j})} \\ &= \sum_{j=1}^{n}\frac{1}{j^{m_{j}}} \sum_{d}\mathbb{1}\bigg\{\sum_{j=1}^{n}jd_{j} = n - m\bigg\} \prod_{j=1}^{n}\frac{1}{j^{m_{j}}} \\ &= \sum_{j=1}^{n}\frac{1}{j^{m_{j}}} \sum_{d}\mathbb{1}\bigg\{\sum_{j=1}^{n}jd_{j} = n - m\bigg\} \prod_{j=1}^{n}\frac{1}{j^{m_{j}}} \\ &= \sum_{j=1}^{n}\frac{1}{j^{m_{j}}} \sum_{d}\mathbb{1}\bigg\{\sum_{j=1}^{n}jd_{j} = n - m\bigg\} \prod_{j=1}^{n}\frac{1}{j^{m_{j}}} \\ &= \sum_{j=1}^{n}\frac{1}{j^{m_{j}}} \sum_{d}\mathbb{1}\bigg\{\sum_{j=1}^{n}jd_{j} = n - m\bigg\} \prod_{j=1}^{n}\frac{1}{j^{m_{j}}} \\ &= \sum_{j=1}^{n}\frac{1}{j^{m_{j}}} \sum_{d}\mathbb{1}\bigg\{\sum_{j=1}^{n}jd_{j} = n - m\bigg\}$$

This last sum simplifies to the indicator  $1(m \le n)$ , corresponding to the fact that if  $n-m \ge 0$ , then  $d_i = 0$  for j > n - m, and a random permutation in  $S_{n-m}$  must have some cycle  $(d_1,...,d_{n-m})$ . The moments of  $C_i^{(n)}$ immediately as

$$E(C_i^{(n)})^{[r]} = j^{-r} 1 \{ jr \le n \}$$
 (1.2)

We note for future reference that (1.4) can also be written in the form

$$E\left(\prod_{j=1}^{n} (C_{j}^{(n)})^{[m_{j}]}\right) = E\left(\prod_{j=1}^{n} Z_{j}^{[m_{j}]}\right) 1\left\{\sum_{j=1}^{n} j m_{j} \le n\right\},\tag{1.3}$$

Where the  $Z_i$  are independent Poisson-distribution random variables that satisfy  $E(Z_i) = 1/j$ 

The marginal distribution of cycle counts provides a formula for the joint distribution of the cycle counts  $C_i^n$ , we find the distribution of  $C_i^n$  using a combinatorial approach combined with inclusion-exclusion formula.

**Lemma 1.8.** For  $1 \le j \le n$ ,

$$P[C_j^{(n)} = k] = \frac{j^{-k}}{k!} \sum_{l=0}^{\lfloor n/j \rfloor - k} (-1)^l \frac{j^{-l}}{l!}$$
 (1.1)

*Proof.* Consider the set I of all possible cycles of length j, formed with elements chosen from  $\{1,2,...n\}$ , so that  $|I| = n^{[j]/j}$ . For each  $\alpha \in I$ , consider the "property"  $G_{\alpha}$  of having lpha; that is,  $G_{\alpha}$  is the set of permutations  $\pi \in S_n$  such that  $\alpha$ is one of the cycles of  $\pi$ . We then have  $|G_{\alpha}| = (n-j)!$ , since the elements of  $\{1, 2, ..., n\}$ not in  $\alpha$  must be permuted among themselves. To use the inclusion-exclusion formula we need to calculate the term  $S_r$ , which is the sum of the probabilities of the r-fold intersection of properties, summing over all sets of r distinct properties. There are two cases to consider. If the r properties are indexed by r cycles having no elements in common,

then the intersection specifies how rj elements are moved by the permutation, and  $(n-rj)!1(rj \leq n)$ permutations the  $n^{[rj]}/(j^r r!)$ are intersection. There such intersections. For the other case, some two distinct  $= \prod_{i=1}^{n} \frac{1}{i^{m_i}} \sum_{d} 1 \left\{ \sum_{i=1}^{n} j d_i = n - m \right\} \prod_{i=1}^{n} \frac{1}{i^{d_i}(d_i)}$ permutation can have both these properties, and the r -fold intersection is empty. Thus

$$S_r = (n - rj)!1(rj \le n)$$

$$\times \frac{n^{[rj]}}{j^r r!} \frac{1}{n!} = 1 (rj \le n) \frac{1}{j^r r!}$$

Finally, the inclusion-exclusion series for the number of permutations having exactly properties is

$$\sum_{l\geq 0} (-1)^l \binom{k+l}{l} S_{k+l}$$

Which simplifies to (1.1) Returning to the original hat-check problem, we substitute j=1 in (1.1) to obtain the distribution of the number of fixed points of a random permutation. For k = 0, 1, ..., n,

$$P[C_1^{(n)} = k] = \frac{1}{k!} \sum_{l=0}^{n-k} (-1)^l \frac{1}{l!},$$
(1.2)

and the moments of  $C_1^{(n)}$  follow from (1.2) with j=1. In particular, for  $n \ge 2$ , the mean and variance of  $C_1^{(n)}$  are both equal to 1. The joint distribution of  $(C_1^{(n)},...,C_b^{(n)})$  for any  $1 \le b \le n$ has an expression similar to (1.7); this too can be inclusion-exclusion.  $c = (c_1, ..., c_b) \in \square_+^b$  with  $m = \sum_i i c_i$ ,

$$P[(C_1^{(n)},...,C_h^{(n)})=c]$$

$$= \left\{ \prod_{i=1}^{b} \left( \frac{1}{i} \right)^{c_i} \frac{1}{c_i!} \right\} \sum_{\substack{l \ge 0 \text{ with} \\ \sum |l_i \le n-m}} (-1)^{l_1 + \dots + l_b} \prod_{i=1}^{b} \left( \frac{1}{i} \right)^{l_i} \frac{1}{l_i!}$$
 (1.3)

The joint moments of the first b counts  $C_1^{(n)},...,C_h^{(n)}$  can be obtained directly from (1.2) and (1.3) by setting  $m_{h+1} = ... = m_n = 0$ 

# The limit distribution of cycle counts

It follows immediately from Lemma 1.2 that for each fixed  $n \to \infty$ 

$$P[C_j^{(n)} = k] \rightarrow \frac{j^{-k}}{k!} e^{-1/j}, \quad k = 0, 1, 2, ...,$$



So that  $C_j^{(n)}$  converges in distribution to a random variable  $Z_j$  having a Poisson distribution with mean 1/j; we use the notation  $C_j^{(n)} \rightarrow_d Z_j$  where  $Z_j \square P_o(1/j)$  to describe this. Infact, the limit random variables are independent.

**Theorem 1.6** The process of cycle counts converges in distribution to a Poisson process of  $\square$  with intensity  $j^{-1}$ . That is, as  $n \to \infty$ ,

$$(C_1^{(n)}, C_2^{(n)}, ...) \rightarrow_d (Z_1, Z_2, ...)$$
 (1.1)

Where the  $Z_j$ , j=1,2,..., are independent Poisson-distributed random variables with  $E(Z_j) = \frac{1}{i}$ 

*Proof.* To establish the converges in distribution one shows that for each fixed  $b \ge 1$ , as  $n \to \infty$ ,

$$P[(C_1^{(n)},...,C_b^{(n)})=c] \to P[(Z_1,...,Z_b)=c]$$

#### **Error rates**

The proof of Theorem says nothing about the rate of convergence. Elementary analysis can be used to estimate this rate when b=1. Using properties of alternating series with decreasing terms, for k=0,1,...,n,

$$\frac{1}{k!} \left( \frac{1}{(n-k+1)!} - \frac{1}{(n-k+2)!} \right) \le \left| P[C_1^{(n)} = k] - P[Z_1 = k] \right|$$

$$\le \frac{1}{k!(n-k+1)!}$$

It follows that

$$\frac{2^{n+1}}{(n+1)!} \frac{n}{n+2} \le \sum_{k=0}^{n} \left| P[C_1^{(n)} = k] - P[Z_1 = k] \right| \le \frac{2^{n+1} - 1}{(n+1)!}$$
 (1.11)

Since

$$P[Z_1 > n] = \frac{e^{-1}}{(n+1)!} (1 + \frac{1}{n+2} + \frac{1}{(n+2)(n+3)} + \dots) < \frac{1}{(n+1)!},$$

We see from (1.11) that the total variation distance between the distribution  $L(C_1^{(n)})$  of  $C_1^{(n)}$  and the distribution  $L(Z_1)$  of  $Z_1$ 

Establish the asymptotics of  $P[A_n(C^{(n)})]$  under conditions  $(A_0)$  and  $(B_{01})$ , where

$$A_n(C^{(n)}) = \bigcap_{1 \le i \le n} \bigcap_{\substack{r_i' + 1 \le j \le r_i}} \{C_{ij}^{(n)} = 0\},$$

and  $\zeta_i = (r_i / r_{id}) - 1 = O(i^{-g})$  as  $i \to \infty$ , for some g > 0. We start with the expression

$$P[A_n(C^{(n)})] = \frac{P[T_{0m}(Z') = n]}{P[T_{0m}(Z) = n]}$$

$$\prod_{\substack{1 \le i \le n \\ r, +1 \le i \le r}} \left\{ 1 - \frac{\theta}{ir_i} (1 + E_{i0}) \right\}$$
 (1.1)

$$P[T_{0n}(Z')=n]$$

$$= \frac{\theta d}{n} \exp \left\{ \sum_{i=1}^{n} [\log(1 + i^{-1}\theta d) - i^{-1}\theta d] \right\}$$

$$\left\{1 + O(n^{-1}\varphi_{\{1,2,7\}}(n))\right\} \tag{1.2}$$

and

$$P[T_{0n}(Z') = n]$$

$$= \frac{\theta d}{n} \exp \left\{ \sum_{i>1} [\log(1+i^{-1}\theta d) - i^{-1}\theta d] \right\}$$

$$\left\{1 + O(n^{-1}\varphi_{\{1,2,7\}}(n))\right\}$$
 (1.3)

Where  $\varphi_{\{1,2,7\}}(n)$  refers to the quantity derived from Z. It thus follows that  $P[A_n(C^{(n)})] \square Kn^{-\theta(1-d)}$  for a constant K, depending on Z and the  $r_i$  and computable explicitly from (1.1)-(1.3), if Conditions  $(A_0)$  and  $(B_{01})$  are satisfied and if  $\zeta_i^* = O(i^{-g})$  from some g>0, since, under these circumstances, both  $n^{-1}\varphi_{\{1,2,7\}}(n)$  and  $n^{-1}\varphi_{\{1,2,7\}}(n)$  tend to zero as  $n\to\infty$ . In particular, for polynomials and square free polynomials, the relative error in this asymptotic approximation is of order  $n^{-1}$  if g>1.

For  $0 \le b \le n/8$  and  $n \ge n_0$ , with  $n_0$   $d_{TV}(L(C[1,b]),L(Z[1,b]))$ 

$$\leq d_{TV}(L(C[1,b]), L(Z[1,b]))$$
  
 $\leq \varepsilon_{\{7,7\}}(n,b),$ 

Where  $\mathcal{E}_{\{7,7\}}(n,b) = O(b/n)$  under Conditions  $(A_0),(D_1)$  and  $(B_{11})$  Since, by the Conditioning Relation,



$$L(C[1,b]|T_{0b}(C)=l) = L(Z[1,b]|T_{0b}(Z)=l),$$

It follows by direct calculation that

$$d_{TV}(L(C[1,b]), L(Z[1,b]))$$

$$= d_{TV}(L(T_{0b}(C)), L(T_{0b}(Z)))$$

$$= \max_{A} \sum_{r \in A} P[T_{0b}(Z) = r]$$

$$\left\{1 - \frac{P[T_{bn}(Z) = n - r]}{P[T_{0n}(Z) = n]}\right\}$$
(1.4)

Suppressing the argument Z from now on, we thus obtain

$$\begin{split} &d_{TV}(L(C[1,b]),L(Z[1,b])) \\ &= \sum_{r \geq 0} P[T_{0b} = r] \left\{ 1 - \frac{P[T_{bn} = n - r]}{P[T_{0n} = n]} \right\}_{+} \\ &\leq \sum_{r > n/2} P[T_{0b} = r] + \sum_{r = 0}^{\lfloor n/2 \rfloor} \frac{P[T_{0b} = r]}{P[T_{0b} = n]} \\ &\times \left\{ \sum_{s = 0}^{n} P[T_{0b} = s](P[T_{bn} = n - s] - P[T_{bn} = n - r] \right\}_{+} \end{split}$$

$$\leq \sum_{r>n/2} P[T_{0b} = r] + \sum_{r=0}^{\lfloor n/2 \rfloor} P[T_{0b} = r]$$

$$\times \sum_{s=0}^{\lfloor n/2 \rfloor} P[T_{0b} = s] \frac{\left\{ P[T_{bn} = n - s] - P[T_{bn} = n - r] \right\}}{P[T_{0n} = n]}$$

$$+ \sum_{s=0}^{\lfloor n/2 \rfloor} P[T_{0b} = r] \sum_{s=0}^{n} P[T = s] P[T_{bn} = n - s] / P[T_{0n} = n]$$

The first sum is at most  $2n^{-1}ET_{0b}$ ; the third is bound by

$$\begin{split} &(\max_{n/2 < s \le n} P[T_{0b} = s]) / P[T_{0n} = n] \\ & \le \frac{2\varepsilon_{\{10.5(1)\}}(n/2, b)}{n} \frac{3n}{\theta P_{\theta}[0, 1]}, \\ & \frac{3n}{\theta P_{\theta}[0, 1]} 4n^{-2} \phi_{\{10.8\}}^{*}(n) \sum_{r=0}^{[n/2]} P[T_{0b} = r] \sum_{s=0}^{[n/2]} P[T_{0b} = s] \frac{1}{2} |r - s| \\ & \le \frac{12\phi_{\{10.8\}}^{*}(n)}{\theta P_{\theta}[0, 1]} \frac{ET_{0b}}{n} \end{split}$$

Hence we may take

$$\varepsilon_{\{7,7\}}(n,b) = 2n^{-1}ET_{0b}(Z)\left\{1 + \frac{6\phi_{\{10.8\}}^{*}(n)}{\theta P_{\theta}[0,1]}\right\}P$$

$$+ \frac{6}{\theta P_{\theta}[0,1]}\varepsilon_{\{10.5(1)\}}(n/2,b) \qquad (1.5)$$

Required order under Conditions  $(A_n),(D_1)$  and  $(B_{11})$ , if  $S(\infty) < \infty$ . If not,  $\phi_{\{10.8\}}^*(n)$  can be replaced by  $\phi_{\{10.11\}}^*(n)$  in the above, which has the required order, without the restriction on the  $r_i$  implied by  $S(\infty) < \infty$ . Examining the Conditions  $(A_0), (D_1)$  and  $(B_{11}),$ it is perhaps surprising to find that  $(B_{11})$  is required instead of just  $(B_{01})$ ; that is, that we should need  $\sum\nolimits_{l \geq 2} l \varepsilon_{il} = O(i^{-a_1}) \quad \text{to hold for some} \ \ a_1 > 1 \, . \, \, \text{A}$ first observation is that a similar problem arises with the rate of decay of  $\mathcal{E}_{i1}$  as well. For this reason,  $n_1$ is replaced by  $n_1$ . This makes it possible to replace condition  $(A_1)$  by the weaker pair of conditions  $(A_0)$  and  $(D_1)$  in the eventual assumptions needed for  $\mathcal{E}_{(7,7)}(n,b)$  to be of order O(b/n); the decay rate requirement of order  $i^{-1-\gamma}$  is shifted from  $\mathcal{E}_{i1}$  itself to its first difference. This is needed to obtain the right approximation error for the random mappings example. However, since all the classical applications make far more stringent assumptions about the  $\varepsilon_{i1}$ ,  $l \ge 2$ , than are made in  $(B_{11})$ . The critical point of the proof is seen where estimate of difference  $P[T_{bn}^{(m)} = s] - P[T_{bn}^{(m)} = s+1].$ The factor  $\mathcal{E}_{\{10,10\}}(n)$ , which should be small, contains a far tail element from  $n_1$  of the form  $\phi_1^{\theta}(n) + u_1^*(n)$ , which is only small if  $a_1 > 1$ , being otherwise of order  $O(n^{1-a_1+\delta})$  for any  $\delta > 0$ , since  $a_2 > 1$  is in any case assumed. For  $s \ge n/2$ , this gives rise to a contribution of order  $O(n^{-1-a_1+\delta})$  in the of the difference  $P[T_{bn} = s] - P[T_{bn} = s + 1],$  which, remainder of the proof, is translated into a contribution of order  $O(tn^{-1-a_1+\delta})$  for differences of the form  $P[T_{bn} = s] - P[T_{bn} = s+1]$ , finally



leading to a contribution of order  $bn^{-a_1+\delta}$  for any leading to a contribution of order  $bn^{-a_1+\delta}$  for any  $\mathcal{S} > 0 \text{ in } \mathcal{E}_{\{7,7\}}(n,b). \text{ Some improvement would } \Big| \sum_{[n/2]} \frac{P[T_{0b} = r]}{P[T_{0n} = n]}$ seem to be possible, defining the function g by  $g(w) = 1_{\{w=s\}} - 1_{\{w=s+t\}}, \quad \text{differences that are of } \times \left\{ \sum_{s=0}^{\lfloor n/2 \rfloor} P[T_{0b} = s] (P[T_{bn} = n - s] - P[T_{bn} = n - r] \right\}$ the form  $P[T_{bn} = s] - P[T_{bn} = s + t]$  can be directly estimated, at a cost of only a single contribution of the form  $\phi_1^{\theta}(n) + u_1^*(n)$ . Then, iterating the cycle, in which one estimate of a difference in point probabilities is improved to an

$$|P[T_{bn} = s] - P[T_{bn} = s + t]| = O(n^{-2}t + n^{-1-a_1+\delta})$$

a final error estimate in order  $O(bn^{-1} + n^{-a_1 + \delta})$ for any  $\delta > 0$ , to replace  $\mathcal{E}_{\{7.7\}}(n,b)$ . This would  $\leq \frac{6}{\theta n P_{a}[0.1]} ET_{0b} \mathcal{E}_{\{10.14\}}(n,b)$ 

but would still be coarser for small b.

With b and n as in the previous section, we wish to  $(\frac{3}{\theta n P_0[0.1]})$ , show that

$$\left| d_{TV}(L(C[1,b]), L(Z[1,b])) - \frac{1}{2}(n+1)^{-1} |1 - \theta| E |T_{0b} - ET_{0b}| \right| \le \varepsilon_{\{7,8\}}(n,b),$$

 $\varepsilon_{(7.8)}(n,b) = O(n^{-1}b[n^{-1}b + n^{-\beta_{12}+\delta}])$ for any  $\delta > 0$  under Conditions  $(A_0), (D_1)$  and  $(B_{12})$ , with  $\beta_{12}$ . The proof uses sharper estimates. As before, we begin with the formula

$$d_{TV}(L(C[1,b]), L(Z[1,b]))$$

$$= \sum_{r\geq 0} P[T_{0b} = r] \left\{ 1 - \frac{P[T_{bn} = n - r]}{P[T_{0n} = n]} \right\}$$

Now we observe that

$$\begin{split} & \left| \sum_{r \ge 0} P[T_{0b} = r] \left\{ 1 - \frac{P[T_{bn} = n - r]}{P[T_{0n} = n]} \right\}_{+} - \sum_{r = 0}^{[n/2]} \frac{P[T_{0b} = r]}{P[T_{0n} = n]} \right\} \\ & \times \left| \sum_{s = [n/2] + 1}^{n} P[T_{0b} = s] (P[T_{bn} = n - s] - P[T_{bn} = n - r]) \right| \\ & \le 4n^{-2} E T_{0b}^{2} + \left( \max_{n/2 < s \le n} P[T_{0b} = s] \right) / P[T_{0n} = n] \\ & + P[T_{0b} > n / 2] \\ & \le 8n^{-2} E T_{0b}^{2} + \frac{3\varepsilon_{\{10.5(2)\}}(n / 2, b)}{\theta P_{0}[0.1]}, \end{split}$$
(1.1)

We have

$$\sum_{r=0}^{\lfloor n/2 \rfloor} \frac{P[T_{0b} = r]}{P[T_{0n} = n]}$$

$$\times \left\{ \sum_{s=0}^{\lfloor n/2 \rfloor} P[T_{0b} = s] (P[T_{bn} = n - s] - P[T_{bn} = n - r] \right\}_{+}$$

$$- \left\{ \sum_{s=0}^{\lfloor n/2 \rfloor} P[T_{0b} = s] \frac{(s - r)(1 - \theta)}{n + 1} P[T_{0n} = n] \right\}_{+})$$

estimate of smaller order, a bound of the form 
$$\left| P[T_{bn} = s] - P[T_{bn} = s+t] \right| = O(n^{-2}t + n^{-1-a_1+\delta})$$
 
$$\leq \frac{1}{n^2 P[T_{0n} = n]} \sum_{r \geq 0} P[T_{0b} = r] \sum_{s \geq 0} P[T_{0b} = s] |s-r|$$
 for any  $\delta > 0$  could perhaps be attained, leading to  $\times \left\{ \varepsilon_{\{10.14\}}(n,b) + 2(r \vee s) |1-\theta| n^{-1} \left\{ K_0 \theta + 4 \phi_{\{10.8\}}^*(n) \right\} \right\}$  a final error estimate in order  $O(bn^{-1} + n^{-a_1+\delta})$  for any  $\delta > 0$ , to replace  $\varepsilon_{\{7.7\}}(n,b)$ . This would 
$$\leq \frac{6}{\theta n P_{\theta}[0,1]} ET_{0b} \varepsilon_{\{10.14\}}(n,b)$$
 be of the ideal order  $O(b/n)$  for large enough  $b$ , the ideal order  $O(b/n)$  for large enough  $b$ . With  $b$  and  $b$  as in the previous section, we wish to  $(\frac{3}{\theta n P_{\theta}[0,1]})$ ,  $(1.2)$ 

The approximation in (1.2) is further simplified by noting that

$$\sum_{r=0}^{[n/2]} P[T_{0b} = r] \left\{ \sum_{s=0}^{[n/2]} P[T_{0b} = s] \frac{(s-r)(1-\theta)}{n+1} \right\}_{+}$$

$$-\left\{\sum_{s=0} P[T_{0b} = s] \frac{(s-r)(1-\theta)}{n+1}\right\}_{+}$$

$$\leq \sum_{r=0}^{\lfloor n/2 \rfloor} P[T_{0b} = r] \sum_{s>\lfloor n/2 \rfloor} P[T_{0b} = s] \frac{(s-r)|1-\theta|}{n+1}$$

$$\leq |1-\theta|n^{-1}E(T_{0b}1\{T_{0b} > n/2\}) \leq 2|1-\theta|n^{-2}ET_{0b}^{2}, \tag{1.3}$$

and then by observing that

$$\sum_{r>\{n/2\}} P[T_{0b} = r] \left\{ \sum_{s\geq 0} P[T_{0b} = s] \frac{(s-r)(1-\theta)}{n+1} \right\}$$

$$\leq n^{-1} \left| 1 - \theta \right| (ET_{0b}P[T_{0b} > n/2] + E(T_{0b}1\{T_{0b} > n/2\}))$$

$$\leq 4 \left| 1 - \theta \right| n^{-2}ET_{0b}^{-2}$$
(1.4)

Combining the contributions of (1.2) –(1.3), we thus

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$$\left| d_{TV}(L(C[1,b]), L(Z[1,b])) - (n+1)^{-1} \sum_{r \ge 0} P[T_{0b} = r] \left\{ \sum_{s \ge 0} P[T_{0b} = s](s-r)(1-\theta) \right\}_{+} \right|$$

$$\le \varepsilon_{\{7.8\}}(n,b)$$

$$= \frac{3}{\theta P_{\theta}[0,1]} \left\{ \varepsilon_{\{10.5(2)\}}(n/2,b) + 2n^{-1} E T_{0b} \varepsilon_{\{10.14\}}(n,b) \right\}$$

$$+ 2n^{-2} E T_{0b}^{2} \left\{ 4 + 3 \left| 1 - \theta \right| + \frac{24 \left| 1 - \theta \right| \phi_{\{10.8\}}^{*}(n)}{\theta P_{\theta}[0,1]} \right\}$$

$$(1.5)$$

The quantity  $\mathcal{E}_{\{7.8\}}(n,b)$  is seen to be of the order claimed under Conditions  $(A_0),(D_1)$  and  $(B_{12})$ , provided that  $S(\infty)<\infty$ ; this supplementary condition can be removed if  $\phi_{\{10.8\}}^*(n)$  is replaced by  $\phi_{\{10.11\}}^*(n)$  in the definition of  $\mathcal{E}_{\{7.8\}}(n,b)$ , has the required order without the restriction on the  $r_i$  implied by assuming that  $S(\infty)<\infty$ . Finally, a direct calculation now shows that

$$\sum_{r\geq 0} P[T_{0b} = r] \left\{ \sum_{s\geq 0} P[T_{0b} = s](s-r)(1-\theta) \right\}_{+}$$

$$= \frac{1}{2} |1 - \theta| E |T_{0b} - ET_{0b}|$$

1.0. Consider  $O = (0,...,0) \in \square^n$ . For an arbitrary vector r, the coordinates of the point x = O + r are equal to the coordinates of the  $r: x = (x^1, ..., x^n)$  and  $r = (x^1, ..., x^n)$ . The vector r such as in the example is called the position vector or the radius vector of the point x . (Or, in greater detail: r is the radius-vector of x w.r.t an origin O). Points are frequently specified by their radiusvectors. This presupposes the choice of O as the "standard origin". Let us summarize. We have considered  $\square$  <sup>n</sup> and interpreted its elements in two ways: as points and as vectors. Hence we may say that we leading with the two copies of  $\square^n$ :  $\square^n =$ {points},  $\Box$   $^n = \{ \text{vectors} \}$ 

Operations with vectors: multiplication by a number, addition. Operations with points and vectors: adding a vector to a point (giving a point), subtracting two points (giving a vector).  $\square$  \*\* treated in this way is called an \*\*n-dimensional affine space\*. (An "abstract" affine space is a pair of sets, the set of points and the set of vectors so that the operations

as above are defined axiomatically). Notice that vectors in an affine space are also known as "free vectors". Intuitively, they are not fixed at points and "float freely" in space. From  $\square$  " considered as an affine space we can precede in two opposite directions:  $\square^n$  as an Euclidean space  $\Leftarrow \square^n$  as an affine space  $\Rightarrow \Box^n$  as a manifold. Going to the left means introducing some extra structure which will make the geometry richer. Going to the right means forgetting about part of the affine structure; going further in this direction will lead us to the so-called "smooth (or differentiable) manifolds". The theory of differential forms does not require any extra geometry. So our natural direction is to the right. The Euclidean structure, however, is useful for examples and applications. So let us say a few words about it:

**Remark 1.0.** Euclidean geometry. In  $\square^n$  considered as an affine space we can already do a good deal of geometry. For example, we can consider lines and planes, and quadric surfaces like an ellipsoid. However, we cannot discuss such things as "lengths", "angles" or "areas" and "volumes". To be able to do so, we have to introduce some more definitions, making  $\square^n$  a Euclidean space. Namely, we define the length of a vector  $a = (a^1, ..., a^n)$  to be

$$|a| := \sqrt{(a^1)^2 + \dots + (a^n)^2}$$
 (1)

After that we can also define distances between points as follows:

$$d(A,B) := \left| \overrightarrow{AB} \right| \tag{2}$$

One can check that the distance so defined possesses natural properties that we expect: is it always non-negative and equals zero only for coinciding points; the distance from A to B is the same as that from B to A (symmetry); also, for three points, A, B and C, we have  $d(A,B) \le d(A,C) + d(C,B)$  (the "triangle inequality"). To define angles, we first introduce the scalar product of two vectors

$$(a,b) := a^1b^1 + ... + a^nb^n$$
 (3)

Thus  $|a|=\sqrt{(a,a)}$ . The scalar product is also denote by dot: ab=(a,b), and hence is often referred to as the "dot product". Now, for nonzero vectors, we define the angle between them by the equality

$$\cos \alpha := \frac{(a,b)}{|a||b|} \tag{4}$$



The angle itself is defined up to an integral multiple of  $2\pi$ . For this definition to be consistent we have to ensure that the r.h.s. of (4) does not exceed 1 by the absolute value. This follows from the inequality

$$(a,b)^2 \le |a|^2 |b|^2$$
 (5)

known as the Cauchy–Bunyakovsky–Schwarz inequality (various combinations of these three names are applied in different books). One of the ways of proving (5) is to consider the scalar square of the linear combination a+tb, where  $t \in R$ . As  $(a+tb,a+tb) \ge 0$  is a quadratic polynomial in t which is never negative, its discriminant must be less or equal zero. Writing this explicitly yields (5). The triangle inequality for distances also follows from the inequality (5).

**Example 1.1.** Consider the function  $f(x) = x^i$  (the i-th coordinate). The linear function  $dx^i$  (the differential of  $x^i$ ) applied to an arbitrary vector h is simply  $h^i$ . From these examples follows that we can rewrite df as

$$df = \frac{\partial f}{\partial x^1} dx^1 + \dots + \frac{\partial f}{\partial x^n} dx^n, \tag{1}$$

which is the standard form. Once again: the partial derivatives in (1) are just the coefficients (depending on x);  $dx^1, dx^2,...$  are linear functions giving on an arbitrary vector h its coordinates  $h^1, h^2,...$ , respectively. Hence

$$df(x)(h) = \partial_{hf(x)} = \frac{\partial f}{\partial x^1} h^1 + \dots + \frac{\partial f}{\partial x^n} h^n, \quad (2)$$

**Theorem 1.7.** Suppose we have a parametrized curve  $t \mapsto x(t)$  passing through  $x_0 \in \square^n$  at  $t = t_0$  and with the velocity vector  $x(t_0) = \upsilon$  Then  $\frac{df(x(t))}{dt}(t_0) = \partial_{\upsilon} f(x_0) = df(x_0)(\upsilon)$ 

*Proof.* Indeed, consider a small increment of the parameter  $t:t_0\mapsto t_0+\Delta t$ , Where  $\Delta t\mapsto 0$ . On the other hand, we have  $f(x_0+h)-f(x_0)=df(x_0)(h)+\beta(h)|h|$  for

an arbitrary vector h, where  $\beta(h) \to 0$  when  $h \to 0$ . Combining it together, for the increment of f(x(t)) we obtain

$$f(x(t_0 + \Delta t) - f(x_0))$$

$$= df(x_0)(\upsilon \cdot \Delta t + \alpha(\Delta t) \Delta t)$$

$$+ \beta(\upsilon \cdot \Delta t + \alpha(\Delta t) \Delta t) \cdot |\upsilon \Delta t + \alpha(\Delta t) \Delta t|$$

$$= df(x_0)(\upsilon) \cdot \Delta t + \gamma(\Delta t) \Delta t$$

For a certain  $\gamma(\Delta t)$  such that  $\gamma(\Delta t) \to 0$  when  $\Delta t \to 0$  (we used the linearity of  $df(x_0)$ ). By the definition, this means that the derivative of f(x(t)) at  $t=t_0$  is exactly  $df(x_0)(\upsilon)$ . The statement of the theorem can be expressed by a simple formula:

$$\frac{df(x(t))}{dt} = \frac{\partial f}{\partial x^1} x^1 + \dots + \frac{\partial f}{\partial x^n} x^n \tag{2}$$

To calculate the value Of df at a point  $x_0$  on a given vector v one can take an arbitrary curve passing Through  $x_0$  at  $t_0$  with v as the velocity vector at  $t_0$  and calculate the usual derivative of f(x(t)) at  $t=t_0$ .

**Theorem 1.8.** For functions  $f,g:U\to\square$ ,  $U\subset\square^n$ ,

$$d(f+g) = df + dg \tag{1}$$

$$d(fg) = df \cdot g + f \cdot dg \tag{2}$$

Proof. Consider an arbitrary point  $x_0$  and an arbitrary vector v0 stretching from it. Let a curve v(t)1 be such that v(t)2 = v3 and v(t)3 = v4. Hence

$$d(f+g)(x_0)(v) = \frac{d}{dt}(f(x(t)) + g(x(t)))$$

at  $t = t_0$  and

$$d(fg)(x_0)(v) = \frac{d}{dt}(f(x(t))g(x(t)))$$

at  $t=t_0$  Formulae (1) and (2) then immediately follow from the corresponding formulae for the usual derivative Now, almost without change the theory generalizes to functions taking values in  $\square^m$  instead of  $\square$ . The only difference is that now the differential of a map  $F:U\to \square^m$  at a point x will be a linear function taking vectors in  $\square^n$  to



vectors in  $\square$   $^m$  (instead of  $\square$  ) . For an arbitrary vector  $h \in \square$   $^n$ .

$$F(x+h) = F(x) + dF(x)(h)$$

$$+\beta(h)|h| \qquad (3)$$
Where  $\beta(h) \to 0$  when  $h \to 0$ . We have  $dF = (dF^1, ..., dF^m)$  and  $dF = \frac{\partial F}{\partial x^1} dx^1 + ... + \frac{\partial F}{\partial x^n} dx^n$ 

$$= \begin{pmatrix} \frac{\partial F^1}{\partial x^1} ... \frac{\partial F^1}{\partial x^n} \\ ... & ... \\ \frac{\partial F^m}{\partial x^1} ... \frac{\partial F^m}{\partial x^n} \end{pmatrix} \begin{pmatrix} dx^1 \\ ... \\ dx^n \end{pmatrix}$$
(4)

In this matrix notation we have to write vectors as vector-columns.

**Theorem 1.9**. For an arbitrary parametrized curve x(t) in  $\square^n$ , the differential of a map  $F: U \to \square^m$  (where  $U \subset \square^n$ ) maps the velocity vector x(t) to the velocity vector of the curve F(x(t)) in  $\square^m$ :

$$\frac{dF(x(t))}{dt} = dF(x(t))(x(t)) \tag{1}$$

Proof. By the definition of the velocity vector,

$$x(t + \Delta t) = x(t) + x(t) \cdot \Delta t + \alpha(\Delta t) \Delta t \tag{2}$$

Where  $\alpha(\Delta t) \to 0$  when  $\Delta t \to 0$ . By the definition of the differential,

$$F(x+h) = F(x) + dF(x)(h) + \beta(h)|h$$
 (3)

Where  $\beta(h) \rightarrow 0$  when  $h \rightarrow 0$ , we obtain

$$F(x(t + \Delta t)) = F(x + \underbrace{x(t).\Delta t + \alpha(\Delta t)\Delta t}_{h})$$

$$= F(x) + dF(x)(x(t)\Delta t + \alpha(\Delta t)\Delta t) +$$

$$\beta(x(t)\Delta t + \alpha(\Delta t)\Delta t). \left| x(t)\Delta t + \alpha(\Delta t)\Delta t \right|$$

$$= F(x) + dF(x)(x(t)\Delta t + \gamma(\Delta t)\Delta t$$

For some  $\gamma(\Delta t) \to 0$  when  $\Delta t \to 0$ . This precisely means that  $dF(x)\dot{x(t)}$  is the velocity

vector of F(x). As every vector attached to a point can be viewed as the velocity vector of some curve passing through this point, this theorem gives a clear geometric picture of dF as a linear map on vectors.

**Theorem 1.10** Suppose we have two maps  $F:U\to V$  and  $G:V\to W$ , where  $U\subset \square^n, V\subset \square^m, W\subset \square^p$  (open domains). Let  $F:x\mapsto y=F(x)$ . Then the differential of the composite map  $GoF:U\to W$  is the composition of the differentials of F and G:

$$d(GoF)(x) = dG(y)odF(x)$$
 (4)

*Proof.* We can use the description of the differential Consider a curve x(t) in  $\square^n$  with the velocity vector x. Basically, we need to know to which vector in  $\square^p$  it is taken by d(GoF). the curve (GoF)(x(t) = G(F(x(t))). By the same theorem, it equals the image under dG of the Anycast Flow vector to the curve F(x(t)) in  $\square^m$ . Applying the theorem once again, we see that the velocity vector to the curve F(x(t)) is the image

under dF of the vector x(t). Hence  $d(GoF)(x) = dG(dF(x)) \quad \text{for an arbitrary}$  vector x.

**Corollary 1.0.** If we denote coordinates in  $\square$  <sup>n</sup> by  $(x^1,...,x^n)$  and in  $\square$  <sup>m</sup> by  $(y^1,...,y^m)$ , and write

$$dF = \frac{\partial F}{\partial x^{1}} dx^{1} + \dots + \frac{\partial F}{\partial x^{n}} dx^{n}$$
 (1)

$$dG = \frac{\partial G}{\partial y^{1}} dy^{1} + \dots + \frac{\partial G}{\partial y^{n}} dy^{n}, \qquad (2)$$

Then the chain rule can be expressed as follows:

$$d(GoF) = \frac{\partial G}{\partial y^{1}} dF^{1} + \dots + \frac{\partial G}{\partial y^{m}} dF^{m}, \tag{3}$$

Where  $dF^i$  are taken from (1). In other words, to get d(GoF) we have to substitute into (2) the expression for  $dy^i = dF^i$  from (3). This can also be expressed by the following matrix formula:



$$d(GoF) = \begin{pmatrix} \frac{\partial G^{1}}{\partial y^{1}} & \dots & \frac{\partial G^{1}}{\partial y^{m}} \\ \dots & \dots & \dots \\ \frac{\partial G^{p}}{\partial y^{1}} & \dots & \frac{\partial G^{p}}{\partial y^{m}} \end{pmatrix} \begin{pmatrix} \frac{\partial F^{1}}{\partial x^{1}} & \dots & \frac{\partial F^{1}}{\partial x^{n}} \\ \dots & \dots & \dots \\ \frac{\partial F^{m}}{\partial x^{1}} & \dots & \frac{\partial F^{m}}{\partial x^{n}} \end{pmatrix} \begin{pmatrix} dx^{1} \\ \dots \\ dx^{n} \end{pmatrix}$$
(4)

i.e., if dG and dF are expressed by matrices of partial derivatives, then d(GoF) is expressed by the product of these matrices. This is often written as

$$\begin{pmatrix}
\frac{\partial z^{1}}{\partial x^{1}} & \dots & \frac{\partial z^{1}}{\partial x^{n}} \\
\dots & \dots & \dots \\
\frac{\partial z^{p}}{\partial x^{1}} & \dots & \frac{\partial z^{p}}{\partial x^{n}}
\end{pmatrix} = \begin{pmatrix}
\frac{\partial z^{1}}{\partial y^{1}} & \dots & \frac{\partial z^{1}}{\partial y^{m}} \\
\dots & \dots & \dots \\
\frac{\partial z^{p}}{\partial y^{1}} & \dots & \frac{\partial z^{p}}{\partial y^{m}}
\end{pmatrix}$$

$$\begin{pmatrix}
\frac{\partial y^{1}}{\partial x^{1}} & \dots & \frac{\partial y^{1}}{\partial x^{n}} \\
\dots & \dots & \dots \\
\frac{\partial y^{m}}{\partial x^{1}} & \dots & \frac{\partial y^{m}}{\partial x^{n}}
\end{pmatrix}, (5)$$
Or
$$\frac{\partial z^{\mu}}{\partial x^{a}} = \sum_{i=1}^{m} \frac{\partial z^{\mu}}{\partial y^{i}} \frac{\partial y^{i}}{\partial x^{a}}, (6)$$

Where it is assumed that the dependence of  $y \in \square^m$  on  $x \in \square^n$  is given by the map F, the dependence of  $z \in \square^p$  on  $y \in \square^m$  is given by the map G, and the dependence of  $z \in \square^p$  on  $x \in \square^n$  is given by the composition GoF.

**Definition 1.6.** Consider an open domain  $U \subset \square^n$ . Consider also another copy of  $\square^n$ , denoted for distinction  $\square_y^n$ , with the standard coordinates  $(y^1...y^n)$ . A system of coordinates in the open domain U is given by a map  $F:V \to U$ , where  $V \subset \square_y^n$  is an open domain of  $\square_y^n$ , such that the following three conditions are satisfied:

- (1) F is smooth;
- (2) F is invertible;
- (3)  $F^{-1}: U \to V$  is also smooth

The coordinates of a point  $x \in U$  in this system are the standard coordinates of  $F^{-1}(x) \in \square_{y}^{n}$ 

In other words,

$$F:(y^1...,y^n) \mapsto x = x(y^1...,y^n)$$
 (1)

Here the variables  $(y^1..., y^n)$  are the "new" coordinates of the point x

**Example 1.2.** Consider a curve in  $\square$  <sup>2</sup> specified in polar coordinates as

$$x(t): r = r(t), \varphi = \varphi(t) \tag{1}$$

We can simply use the chain rule. The map  $t \mapsto x(t)$  can be considered as the composition of the maps  $t \mapsto (r(t), \varphi(t)), (r, \varphi) \mapsto x(r, \varphi)$ . Then, by the chain rule, we have

$$\dot{x} = \frac{dx}{dt} = \frac{\partial x}{\partial r}\frac{dr}{dt} + \frac{\partial x}{\partial \varphi}\frac{d\varphi}{dt} = \frac{\partial x}{\partial r}r + \frac{\partial x}{\partial \varphi}\varphi$$
 (2)

Here r and  $\varphi$  are scalar coefficients depending on t, whence the partial derivatives  $\frac{\partial x}{\partial r}, \frac{\partial x}{\partial \varphi}$  are vectors depending on point in  $\Box$  2. We can compare this with the formula in the "standard" coordinates:  $x = e_1 x + e_2 y$ . Consider

the vectors  $\frac{\partial x}{\partial r}$ ,  $\frac{\partial x}{\partial \varphi}$ . Explicitly we have

$$\frac{\partial x}{\partial r} = (\cos \varphi, \sin \varphi) \tag{3}$$

$$\frac{\partial x}{\partial \varphi} = (-r \sin \varphi, r \cos \varphi) \tag{4}$$

From where it follows that these vectors make a basis at all points except for the origin (where r=0). It is instructive to sketch a picture, drawing vectors corresponding to a point as starting from that point. Notice that  $\frac{\partial x}{\partial r}, \frac{\partial x}{\partial \varphi}$  are, respectively, the velocity vectors for the curves  $r\mapsto x(r,\varphi)$   $(\varphi=\varphi_0\ fixed)$  and  $\varphi\mapsto x(r,\varphi)$   $(r=r_0\ fixed)$ . We can conclude that for an arbitrary curve given in polar coordinates the velocity vector will have components  $(r,\varphi)$  if as a basis we take  $e_r := \frac{\partial x}{\partial r}, e_\varphi := \frac{\partial x}{\partial \varphi}$ :

$$x = e_r r + e_{\omega} \varphi \tag{5}$$

A characteristic feature of the basis  $e_r, e_\varphi$  is that it is not "constant" but depends on point. Vectors "stuck to points" when we consider curvilinear coordinates.



**Proposition 1.3.** The velocity vector has the same appearance in all coordinate systems.

**Proof.** Follows directly from the chain rule and the transformation law for the basis  $e_i$ . In particular,

the elements of the basis  $e_i = \frac{\partial x}{\partial x^i}$  (originally, a formal notation) can be understood directly as the velocity vectors of the coordinate lines  $x^i \mapsto x(x^1,...,x^n)$  (all coordinates but  $x^i$  are fixed). Since we now know how to handle velocities in arbitrary coordinates, the best way to treat the differential of a map  $F: \Box^n \to \Box^m$  is by its action on the velocity vectors. By definition, we set

$$dF(x_0): \frac{dx(t)}{dt}(t_0) \mapsto \frac{dF(x(t))}{dt}(t_0) \tag{1}$$

Now  $dF(x_0)$  is a linear map that takes vectors attached to a point  $x_0 \in \square^n$  to vectors attached to the point  $F(x) \in \square^m$ 

$$dF = \frac{\partial F}{\partial x^{1}} dx^{1} + \dots + \frac{\partial F}{\partial x^{n}} dx^{n}$$

$$(e_{1}, \dots, e_{m}) \begin{pmatrix} \frac{\partial F^{1}}{\partial x^{1}} \dots \frac{\partial F^{1}}{\partial x^{n}} \\ \dots \dots \dots \\ \frac{\partial F^{m}}{\partial x^{1}} \dots \frac{\partial F^{m}}{\partial x^{n}} \end{pmatrix} \begin{pmatrix} dx^{1} \\ \dots \\ dx^{n} \end{pmatrix}, \tag{2}$$

In particular, for the differential of a function we always have

$$df = \frac{\partial f}{\partial x^1} dx^1 + \dots + \frac{\partial f}{\partial x^n} dx^n, \tag{3}$$

Where  $x^i$  are arbitrary coordinates. The form of the differential does not change when we perform a change of coordinates.

**Example 1.3** Consider a 1-form in  $\Box$ <sup>2</sup> given in the standard coordinates:

A = -ydx + xdy In the polar coordinates we will have  $x = r\cos\varphi$ ,  $y = r\sin\varphi$ , hence  $dx = \cos\varphi dr - r\sin\varphi d\varphi$   $dy = \sin\varphi dr + r\cos\varphi d\varphi$  Substituting into A, we get  $A = -r\sin\varphi(\cos\varphi dr - r\sin\varphi d\varphi)$   $+r\cos\varphi(\sin\varphi dr + r\cos\varphi d\varphi)$ 

 $= r^2 (\sin^2 \varphi + \cos^2 \varphi) d\varphi = r^2 d\varphi$ 

Hence  $A = r^2 d\varphi$  is the formula for A in the polar coordinates. In particular, we see that this is again a 1-form, a linear combination of the differentials of coordinates with functions as coefficients. Secondly, in a more conceptual way, we can define a 1-form in a domain U as a linear function on vectors at every point of U:  $\omega(v) = \omega_1 v^1 + ... + \omega_n v^n, \qquad (1)$ 

If  $\upsilon = \sum e_i \upsilon^i$ , where  $e_i = \frac{\partial x}{\partial x^i}$ . Recall that the differentials of functions were defined as linear functions on vectors (at every point), and

$$dx^{i}(e_{j}) = dx^{i} \left(\frac{\partial x}{\partial x^{j}}\right) = \delta_{j}^{i}$$
 (2) at every point  $x$ .

**Theorem 1.9.** For arbitrary 1-form  $\omega$  and path  $\gamma$ , the integral  $\int_{\gamma} \omega$  does not change if we change parametrization of  $\gamma$  provide the orientation remains the same.

Proof: Consider 
$$\left\langle \omega(x(t)), \frac{dx}{dt} \right\rangle$$
 and  $\left\langle \omega(x(t(t'))), \frac{dx}{dt'} \right\rangle$  As  $\left\langle \omega(x(t(t'))), \frac{dx}{dt'} \right\rangle = \left\langle \omega(x(t(t'))), \frac{dx}{dt'} \right\rangle \cdot \frac{dt}{dt'}$ ,

Let p be a rational prime and let  $K = \square$  ( $\zeta_p$ ). We write  $\zeta$  for  $\zeta_p$  or this section. Recall that K has degree  $\varphi(p) = p-1$  over  $\square$ . We wish to show that  $O_K = \square$  [ $\zeta$ ]. Note that  $\zeta$  is a root of  $x^p-1$ , and thus is an algebraic integer; since  $O_K$  is a ring we have that  $\square$  [ $\zeta$ ]  $\subseteq O_K$ . We give a proof without assuming unique factorization of ideals. We begin with some norm and trace computations. Let j be an integer. If j is not divisible by p, then  $\zeta^j$  is a primitive  $p^{th}$  root of unity, and thus its conjugates are  $\zeta, \zeta^2, ..., \zeta^{p-1}$ . Therefore

$$Tr_{K/\square}(\zeta^j) = \zeta + \zeta^2 + ... + \zeta^{p-1} = \Phi_p(\zeta) - 1 = -1$$
If  $p$  does divide  $j$ , then  $\zeta^j = 1$ , so it has only the one conjugate 1, and



 $Tr_{K/\square}(\zeta^j) = p-1$  By linearity of the trace, we find that

$$Tr_{K/\square} (1-\zeta) = Tr_{K/\square} (1-\zeta^2) = \dots$$

$$=Tr_{K/\Box}(1-\zeta^{p-1})=p$$

We also need to compute the norm of  $1-\zeta$  . For this, we use the factorization

$$x^{p-1} + x^{p-2} + \dots + 1 = \Phi_p(x)$$

= 
$$(x - \zeta)(x - \zeta^2)...(x - \zeta^{p-1});$$

Plugging in x = 1 shows that

$$p = (1 - \zeta)(1 - \zeta^2)...(1 - \zeta^{p-1})$$

Since the  $(1-\zeta^j)$  are the conjugates of  $(1-\zeta)$ , this shows that  $N_{K/\square}(1-\zeta)=p$  The key result for determining the ring of integers  $O_K$  is the following.

# LEMMA 1.9

$$(1-\zeta)O_K \cap \Box = p\Box$$

*Proof.* We saw above that p is a multiple of  $(1-\zeta)$  in  $O_{K}$ , so the inclusion  $(1-\zeta)O_{\kappa} \cap \Box \supseteq p\Box$  is immediate. Suppose now that the inclusion is strict.  $(1-\zeta)O_{\kappa}\cap\Box$  is an ideal of  $\Box$  containing  $p\Box$ and  $p\Box$  is a maximal ideal of  $\Box$  , we must have  $(1-\zeta)O_{\kappa}\cap\Box=\Box$ Thus can write  $1 = \alpha(1 - \zeta)$ 

For some  $\alpha \in O_K$ . That is,  $1-\zeta$  is a unit in  $O_K$ .

COROLLARY 1.1 For any  $\alpha \in O_K$ ,  $Tr_{K/\square} ((1-\zeta)\alpha) \in p\square$ 

PROOF. We have

$$\begin{split} Tr_{K/\square} & ((1-\zeta)\alpha) = \sigma_1((1-\zeta)\alpha) + ... + \sigma_{p-1}((1-\zeta)\alpha) \\ & = \sigma_1(1-\zeta)\sigma_1(\alpha) + ... + \sigma_{p-1}(1-\zeta)\sigma_{p-1}(\alpha) \\ & = (1-\zeta)\sigma_1(\alpha) + ... + (1-\zeta^{p-1})\sigma_{p-1}(\alpha) \end{split}$$

Where the  $\sigma_i$  are the complex embeddings of K (which we are really viewing as automorphisms of K) with the usual ordering. Furthermore,  $1-\zeta^j$  is a multiple of  $1-\zeta$  in  $O_K$  for every  $j\neq 0$ . Thus

 $Tr_{K/\square}(\alpha(1-\zeta)) \in (1-\zeta)O_K$  Since the trace is also a rational integer.

PROPOSITION 1.4 Let p be a prime number and let  $K = |\Box(\zeta_p)|$  be the  $p^{th}$  cyclotomic field. Then  $O_K = \Box[\zeta_p] \cong \Box[x]/(\Phi_p(x));$  Thus  $1, \zeta_p, ..., \zeta_p^{p-2}$  is an integral basis for  $O_K$ .

PROOF. Let  $\alpha \in O_K$  and write

 $\alpha = a_0 + a_1 \zeta + \ldots + a_{p-2} \zeta^{p-2} \qquad \text{With} \quad a_i \in \square \ .$  Then

$$\alpha(1-\zeta) = a_0(1-\zeta) + a_1(\zeta - \zeta^2) + \dots + a_{p-2}(\zeta^{p-2} - \zeta^{p-1})$$

By the linearity of the trace and our above calculations we find that  $Tr_{K/\!\square}\left(\alpha(1-\zeta)\right)=pa_0$  We also have

 $Tr_{K/\square}(\alpha(1-\zeta)) \in p\square$ , so  $a_0 \in \square$  Next consider the algebraic integer

 $(\alpha-a_0)\zeta^{-1}=a_1+a_2\zeta+...+a_{p-2}\zeta^{p-3};$  This is an algebraic integer since  $\zeta^{-1}=\zeta^{p-1}$  is. The same argument as above shows that  $a_1\in \square$ , and continuing in this way we find that all of the  $a_i$  are in  $\square$ . This completes the proof.

Example 1.4 Let  $K = \square$ , then the local ring  $\square$  is simply the subring of  $\square$  of rational numbers with denominator relatively prime to p. Note that this ring  $\square_{(p)}$  is not the ring  $\square_p$  of padic integers; to get  $\Box_p$  one must complete  $\Box_{(p)}$ . The usefulness of  $O_{K,p}$  comes from the fact that it has a particularly simple ideal structure. Let a be any proper ideal of  $O_{K,p}$  and consider the ideal  $a \cap O_{\kappa}$  of  $O_{\kappa}$ . We claim  $a = (a \cap O_K)O_{K,p}$ ; That is, that a is generated by the elements of a in  $a \cap O_K$ . It is clear from the definition of an ideal that  $a \supseteq (a \cap O_K)O_{K,n}$ . To prove the other inclusion, let  $\alpha$  be any element of a. Then we can write  $\alpha = \beta / \gamma$  where  $\beta \in O_K$  and  $\gamma \notin p$ . In particular,  $\beta \in a$  (since  $\beta / \gamma \in a$  and a is an ideal), so  $\beta \in O_K$  and  $\gamma \notin p$ . so  $\beta \in a \cap O_K$ . Since  $1/\gamma \in O_{K,p}$ , this implies that  $\alpha = \beta / \gamma \in (a \cap O_K)O_{K,p}$ , as claimed. We can use this fact to determine all of the



ideals of  $O_{K,p}$ . Let a be any ideal of  $O_{K,p}$  and consider the ideal factorization of  $a \cap O_K$  in  $O_K$ . write it as  $a \cap O_K = p^n b$  For some n and some ideal b, relatively prime to p, we claim first that  $bO_{K,p} = O_{K,p}$ . We now find that

 $a=(a\cap O_K)O_{K,p}=p^nbO_{K,p}=p^nO_{K,p}$  Since  $bO_{K,p}$ . Thus every ideal of  $O_{K,p}$  has the form  $p^nO_{K,p}$  for some n; it follows immediately that  $O_{K,p}$  is noetherian. It is also now clear that  $p^nO_{K,p}$  is the unique non-zero prime ideal in  $O_{K,p}$ . Furthermore, the inclusion  $O_K\mapsto O_{K,p}/pO_{K,p}$  Since  $pO_{K,p}\cap O_K=p$ , this map is also surjection, since the residue class of  $\alpha/\beta\in O_{K,p}$  (with  $\alpha\in O_K$  and  $\beta\notin p$ ) is the image of  $\alpha\beta^{-1}$  in  $O_{K/p}$ , which makes sense since  $\beta$  is invertible in  $O_{K/p}$ . Thus the map is an isomorphism. In particular, it is now abundantly clear that every non-zero prime ideal of  $O_{K,p}$  is maximal. To show that  $O_{K,p}$  is a Dedekind domain, it remains to

in  $O_{K,p}$ ; write this polynomial as  $x^m + \frac{\alpha_{m-1}}{\beta_{m-1}} x^{m-1} + \ldots + \frac{\alpha_0}{\beta_0}$  With  $\alpha_i \in O_K$  and  $\beta_i \in O_{K-p}$ . Set  $\beta = \beta_0 \beta_1 \ldots \beta_{m-1}$ . Multiplying by  $\beta^m$  we find that  $\beta \gamma$  is the root of a monic polynomial with coefficients in  $O_K$ . Thus  $\beta \gamma \in O_K$ ; since  $\beta \notin p$ , we have  $\beta \gamma / \beta = \gamma \in O_{K,p}$ . Thus  $O_{K,p}$  is integrally close in K.

show that it is integrally closed in K. So let  $\gamma \in K$  be a root of a polynomial with coefficients

COROLLARY 1.2. Let K be a number field of degree n and let  $\alpha$  be in  $O_K$  then  $N_{K/\!\square}$   $(\alpha O_K) = \left| N_{K/\!\square} \left( \alpha \right) \right|$ 

PROOF. We assume a bit more Galois theory than usual for this proof. Assume first that  $K/\square$  is Galois. Let  $\sigma$  be an element of  $Gal(K/\square)$ . It is clear that  $\sigma(O_K)/\sigma(\alpha)\cong O_{K/\alpha}$ ; since  $\sigma(O_K)=O_K$ , this shows that

 $N_{K/\square}^{'}(\sigma(\alpha)O_K) = N_{K/\square}^{'}(\alpha O_K)$ . Taking the product over all  $\sigma \in Gal(K/\square)$ , we have  $N_{K/\square}^{'}(N_{K/\square}(\alpha)O_K) = N_{K/\square}^{'}(\alpha O_K)^n$  Since  $N_{K/\square}(\alpha)$  is a rational integer and  $O_K$  is a free  $\square$  -module of rank n,

 $O_{\scriptscriptstyle{K}}/N_{\scriptscriptstyle{K/\square}}(\alpha)O_{\scriptscriptstyle{K}}$  Will have order  $N_{\scriptscriptstyle{K/\square}}(\alpha)^n$ ; therefore

$$N_{K/\square}$$
  $(N_{K/\square}(\alpha)O_K) = N_{K/\square}(\alpha O_K)^n$ 

This completes the proof. In the general case, let L be the Galois closure of K and set [L:K] = m.